Fundamentals of POWER PLANT ENGINEERING

by

GEORGE E. REMP, B.S.M.E., M.S.E., M.E.

Assistant Professor of Mechanical Engineering University of Florida

Member, The American Society of Mechanical Engineers
Member, The National Society of Professional Engineers

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Carolyn Perkins Remp

and to my son

George E. Remp, Jr.

ERRATA AND REVISIONS

FUNDAMENTALS OF POWER PLANT ENGINEERING

ρĀ

GEORGE E. REMP

- Fage 9, first line after Table I: "load" should read "demand."
- Page 10, second line in solution of Example 1: "lcad" should read "demand."
- Page 11, equation in 3rd line from bottom of page: "5.3" should read "5.9."
- Page 37, last paragraph on page should read:

Make a block meter rate which will recover the fixed element in the first 8 kwh per customer per month; the customer element in the next 15 kwh per customer per month; and the energy and profit elements from the excess over 23 kwh per customer per month.

- Page 39, load table in Example 4: "11.2" in hours from 6 7 and 7 8 AM should read "12.2."
- Page 39, 15th line from bottom of page: Annual fuel cost "\$363" should read "\$3630."
- Page 39, 10th line from bottom of page: Total annual energy element "\$10803" should read "\$14070."
- Page 40, section E of Example 4 should read:
- E. Block meter rate:

First block:

rate =
$$\frac{$18730}{12(8000)}$$
 = \$0.195/kwh for first 8 kwh

Second block:

rate =
$$\frac{$27855}{12(15000)}$$
 = \$0.155/kwh for next 15 kwh

(continued next page)

Third block:

rate =
$$\frac{\$14070 + \$15750}{12[\frac{365(1653)}{12} - 23000]}$$
 = \\$0.091/kwh for excess over 23 kwh

Page 41, section B of Example 5: Total annual energy element "\$10803" should read "\$14070."

Page 41, section E of Example 5 should read:

E. Block meter rate:

First block:

rate =
$$\frac{$19148}{12(8000)}$$
 = \$0.199/kwh for first 8 kwh

Second block:

rate =
$$\frac{$28235}{12(15000)}$$
 = \$0.157/kwh for next 15 kwh

Third block:

rate =
$$\frac{$14070 + $15750}{12(27300)}$$
 = $$0.091/kwh$ for excess over 23 kwh.

Page 43, 10th lire from bottom of page: Machine shops "7 AM to 5 PM" should read "8 AM to 5 PM."

Page 44, last paragraph should read:

- (e) Set up a block meter rate which will recover the fixed element in the first 30, the customer element in the next 40 and the energy and profit elements in the excess over 70 kwh per customer per month.
- Page 46, Figure IV-2: Designations of points 5_a' and 5_b should be reversed.
- Page 98, solution of Example 9: Last term in denominator of expression for U, " $\frac{0.75}{0.652}[\frac{1}{3000}]$ " should read " $\frac{0.75}{0.652}[\frac{1}{1615}]$
- Page 115, 5th line from bottom of page: "on of the" should read "one of the."

- Page 136, solution of Example 18: Last term in equation for M, "C.025" should read "O.C38."
- Page 150, solution of Example 20: Numerator of equation for SH, "497" should read "4C7" and "0.24[535 - 70]" should read "0.24[535 - 70] [14.82]." Add the following note: See Example 24 for determination of $M_{\alpha\alpha} = 14.82$.
- Page 161, 5th line from bottom of page: "29(12)/4.5" should read "20(12)/4.5."
- Page 162, solution of Example 22-A beginning at the third line from the top of the page should read:

If the superheater is disposed in two of three boiler passes, each rass will be 20 ft wide by 6.67 ft deep. Then the area for gas flow between the superheater tubes will be (53-1)(4.5-2)(6.67)/12 = 72.3 sq ft.

The mass flow of gas between the tubes will be:

$$v_g = \frac{291000}{3600(72.3)} = 1.12 \text{ lbs/sec/sq ft}$$

and,
$$h_g = \frac{0.41(1.12)^{\circ 0.585} (2048)^{\circ 0.515}}{(2.0/12)^{\circ 0.415}} = 10.12$$

Then, the overall coefficient of heat transfer is:

$$0 = (0.90) \frac{1}{\frac{1}{10.12} + \frac{4.0}{2.0 + 1.67} \left[\frac{0.165}{12(30)} \right] + \frac{2.0}{1.67} \left[\frac{1}{236} \right]} = 8.65$$

Superheater area is:

heater area is:

$$A = \frac{50,600,000}{8.65(860)} = 6800 \text{ sq ft}$$

Each foot of superheater tube has an area of $2\pi/12 = 0.523$ sq ft. There will be 6800/0.523 = 13000 linear feet of tube, and the length of each element will be 13000/53 = 246 feet.

Page 186, solution of Example 26-A: Viscosity "z = 0.025" should read "z = 0.022" and Reynolds' number "r = 1,485,000" should read "r = 1.690,000."

Page 188, solution of Example 26-B, add at top of page:

$$\Delta h = 0.745 \frac{(0.041)(682)(2.91)^4}{3.548} = 49.5 \text{ ft of oil}$$

Page 207, Example 30, add the following:

Note: The above method of expressing entrance and exit losses is often used. Another method would be to express the entrance and exit losses as equivalent lengths of straigh pipe added to the fittings, etc., when calculating the friction losses.

Page 239, 12th line from bottom of page: "states in" should read "stated in."

Page 246, equation for draft produced by a stack should read:

h = 0.254 B H
$$(\frac{1}{T_a} - \frac{1}{T_g})$$
 - 1.49 X 10⁻⁹ $\frac{f H W}{\rho \Gamma_i} g^2$

where: W_{ρ} = weight of flue gas, lb per hr

"f = 0.0013" for steel stacks should read "f = 0.013"

"f = C.0018" for brick lined stacks should read

f = 0.018.

Page 248, last part of solution of Example 34 should read:

Then, by trial in the equation for stacks:

$$0.6 = C.00384 \text{ H} - 43.4 \frac{\text{H}}{D_{ij}}$$

H	160	170	180	190	200	210
$\Gamma_{f i}$	13.60	10.70	9.68	9.11	8.76	8.50
HD.	2180	1810	1740	1730	1750	1785
D./H	0.085	0.063	0.054	0.048	C.C44	0.041

It can be seen that the product HD, has reached a minimum at a stack height of 190 ft. This stack height by chance produces a diameter which falls within the stability range for the parameter D_i/H .

PREFACE

This book is the outgrowth of the author's experience in operating power plants and in teaching power plant engineering to students in mechanical, electrical and chemical engineering. The principal portion of the material is confined to steam power plants and is presented in what the author has found to be a logical sequence, beginning with a study of load curves and power plant economics and progressing through vapor cycles and a detailed examination of the fundamental principles of the various plant components. Chapters are added, as a matter of interest, on the subjects of the economics of load division and Diesel-electric, hydro-electric and gas turbine power plants. No attempt is made to treat the subject of power plant wiring, as this subject properly belongs to the field of the electrical engineer.

The presentation of the material in this book is based on the assumption that it will be presented as a one-semester course, and that the student is familiar with the fundamentals of engineering thermodynamics. The author prefers the project method of presentation of problems to the student because he has found that the method has the advantage of making it possible to integrate the problems into a whole which has meaning to the student. Examples are given in the text in sufficient quantity and detail to guide the student in the execution of the projects. Only one series of projects is given in the text as an aid to the instructor in preparing additional projects from materials out of his own experience.

In order to hew to the line of fundamentals, it has been found necessary to say little about the minor details of design and operation, because these only serve to obscure the job at hand, which is to give a broad picture of power plant engineering in a limited time. It is expected that the instructor will make free use of visual aids, such as motion pictures and plant inspection trips, to supplement and clarify the presentation in the text. A bibliography of selected references is given at the end of each chapter for the benefit of those students who would delve more deeply into this extremely important and intensely interesting subject.

It is a pleasure for the author to acknowledge with thanks the encouragement and assistance of Professors Newton C. Ebaugh, John A. Dent and Paul C. Hoffman of the Mechanical Engineering Department of the College of Engineering of the University of Florida; Mr. Robert B. Cowan, Engineer-Manager of the Utilities Department of the City of Jacksonville, Florida; and Mr. J. H. White, Vice-President of the Florida Power and Light Company of Miami, Florida. Thanks are also due to Mr. L. N. Rowley, Executive Editor of Power for permission to use many of the illustrations which originally appeared in that excellent magazine, and to Mr. L. K. Gaventa for his capable aid in making the many drawings necessary in a work of this type.

The author wishes to thank the many manufacturers and engineers who have contributed illustrations and information. An effort has been made to see that proper credit lines are attached.

George E. Remp

Gainesville, Florida July, 1949

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FUNDAMENTALS OF POWER PLANT ENGINEERING

CHAPTER I

IKTRODUCTION

- I. It is the function of the power plant engineer to specify the equipment for the production of power at the lowest possible cost per unit. This function requires the understanding of the fundamentals of power plant engineering from the standpoint of mechanical design as well as from the standpoint of economics. The word specify should be carefully noted. It is the function of the specialist in equipment design to supply machinery which will fit the specifications of the power plant engineer. Obviously, if the power plant engineer does not have a thorough knowledge of the possibilities and limitations of the machinery with which he is dealing, he cannot make workable recommendations.
- 2. Growth of Power Generation in the United States--Until Thomas A. Edison built the Pearl Street Station for the New York Edison Company in 1880, very little power was produced in the form of electricity. Most of the installed power machinery was in the form of steam engines of various specialized types in mills and mines. Some hydro power was utilized in mills located at the sites of the water wheels.

Before the turn of the century, the largest and most efficient steam power plants were found on ships. At about this time, however, the large central steam-electric station became a necessity and it was not long before these stations surpassed the marine power plant in both size and efficiency.

The growth of the power industry in the United States has been phenomenal. A study by the Twentieth Century Fund has shown conclusively that the rise in our standard of living has paralleled the rise in the utilization of power. Figures I-1 through I-3 show some important data on the growth of the power industry in the United States and the increase in our standard of living during the same period, with an extension of the electric production and installed capacity curves to 1960 to correspond with the Twentieth Century Fund's predictions.

3. Improvement in Thermal Efficiency of Steam Power Plants--Examination of the data on growth of the power industry is not sufficient to show the contributions of the engineer. The engineer's contribution has been the continual improvement in thermal efficiency of the power plant. Figure I-4 shows this improvement in terms of British thermal units

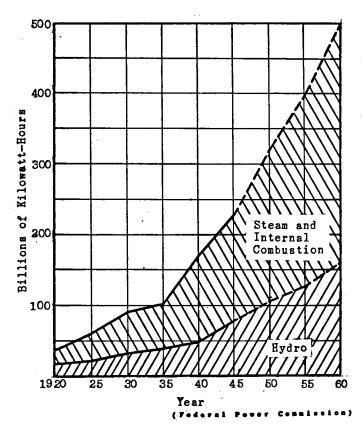


Fig. 1-1. Production of electric energy by type of prime mover, 1920 - 1960.

per kilowatt-hour for plants in the United States. It will be seen that the rate of improvement has apparently reached a minimum, and truly it seems that it is impossible to further improve our steam power cycles. The end of the engineer's contribution is not yet in sight, however. Binary vapor cycles have been proven practicable and thermal efficiency improvements up to 50 per cent will be seen in the not-too-distant future. Power from nuclear fission may someday become economical when cheaper sources of energy become exhausted or so scarce as to be more valuable as raw materials for chemical syntheses.

4. Progress in Hydro and Internal Combustion Power--Fig-

ure I-1 showed that the larger part of the nation's power production has been obtained from steam power plants, and it is likely that such a relation will continue so long as fuels are easily obtainable. Hydro plants at present provide slightly more than half as much power as steam plants, and they are not likely to improve their relative position very much in the future due to the decreasing availability of suitable dam sites. A good example of maximum utilization of hydro power possibilities is to be seen in the installations of the Tennessee Valley Authority which has 26 dams in the more than 1000 miles of the Tennessee River and its tributaries. The TVA, as well as any other well designed hydro system, makes use of steam power plants as auxiliary power sources during the annual dry season.

In Figure I-1, internal combustion engine plants have been included with steam plants because the power produced by the internal combustion engine plants is somewhat less than one per cent of the total production. It must not be concluded, however, that the internal combustion engine plant

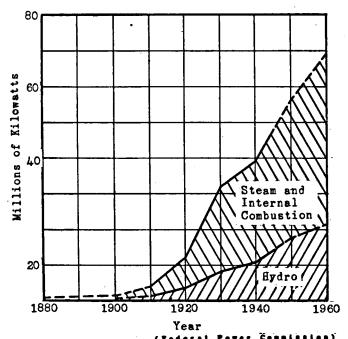


Fig. I-2. Installed capacity of generating plants.

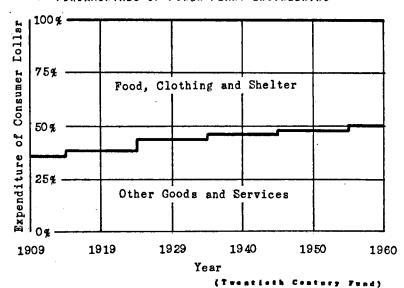


Fig. I-3. Standard of living in the United States. 1909 - 1960.

is insignificant, because there are locations where conditions are such that steam power plants would be impractical. As a general rule, where the total load is less than about 7,500 kilowatts the Diesel-generator plant will be found to be most economical.

5. Power Systems -- Due to the fact that larger power plants can produce power more cheaply than small plants, we have the central station rather than individual power plants for each consumer. The problem of supplying power to the consumer from the central station gives rise to a rather complicated system of transmission and distribution lines. Figure I-5 shows the elements of a power system which includes a power plant and a primary and secondary distribution system. The secondary distribution system is always existent in any power system, but the primary distribution (sometimes called transmission) system may or may not be present depending upon the distance between the power plant and the area which it is to serve. The complexity of the secondary distribution system depends upon the types and sizes of the loads served. Figure I-6 shows a schematic wiring diagram for a typical power system with the various voltages used in present practice. The voltages selected for a particular system will depend largely on the distance:

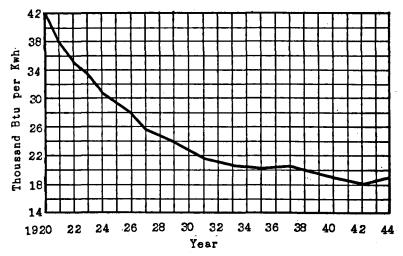


Fig. 1-4. Plant heat rate, Btu per kvh, for plants in the United States. 1920 - 1944.

to be covered and the type of load to be supplied.

- 6. Public Utility Organization 1--At the present time in the United States there are three classes of utility businesses:
 - 1. Private utilities.
 - 2. Municipal utilities.
 - 3. Federal utilities.

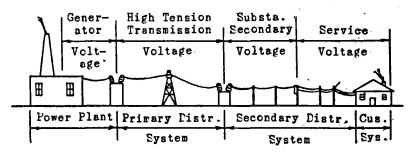


Fig. I-5. A typical power system.

 $^{^{2}}$ "Public utility" is the generic term for utilities of all classes.

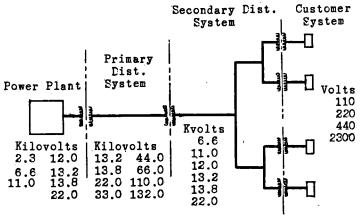


Fig. I-6. Wiring diagram for a typical power system.

Private utilities are almost always incorporated and are usually a part of a very complex organization called a holding company. Briefly, the holding company owns all or a majority of the common stock of a number of subsidiary power companies, known as operating companies. The common stock of the holding company is usually held by the public, but a few individuals may hold sufficient of the common stock to exert effective control of the whole holding company system. Ordinarily the common stock in any corporation amounts to only 40 to 50 per cent of the capitalization, therefore it is possible by pyramiding through sub-holding companies for a few individuals to control huge enterprises. This is not to say that the holding comrany system is undesirable from the theoretical point of view, because it is possible to effect great economies in equipment purchasing and in the furnishing of engineering services with an organization of this type. The Public Utilities Holding Company Act of 1934 was intended to break up these large aggregations of power com-ranies, and it has had some success to date with several which had been in particularly vulnerable positions due to mismanagement.

Municipal utilities are operated by municipal governing bodies for the benefit of the citizens of the municipality. Financing is obtained by the sale of either bonds or revenue certificates. Revenue certificates encumber only the future income of the system and usually do not require a property owners referendum for their authorization. Municipal utility financing lately has been largely accomplished with revenue certificates.

Federal utilities do not usually sell power directly to the consumers but wholesale it to municipalities, private utilities and rural electrification cooperatives. State governments are compensated for loss of tax revenues by payments in lieu of taxes which would have been collected from the former owners of lands which have been permanently flooded by storage reservoirs which are necessary for most hydroelectric plants. These payments are sometimes claimed to be inadequate by the States which receive them.

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CHAPTER II

LOAD CURVES

- 7. Before an attempt is made to design any type of power plant, whether a new installation (quite rare) or an extension to existing facilities, it is necessary to make an exhaustive analysis of the load which the plant is to serve. This analysis must take into account the various types of loads, their probable maximum demands and their geographic location in relation to the plant. Other items for consideration would include a decision as to the relative priority for maintenance of service during an emergency and provision for future growth of load.
- 8. Demand and Diversity -- Consideration of the problem of providing adequate plant capacity for a given situation will show that in very few cases will it be necessary to provide capacity to serve every electrical device in the system at the same time. Engineers have developed two statistical factors to deal with this fact: the demand factor and the diversity factor.

Demand factor expresses as a fraction the probability that any one customer will require full rated power for any particular class of service at any given moment, or during any given period of time. The periods of time are not standardized, but usually are taken as fifteen or thirty minutes. Expressed as an equation, the demand factor is the maximum load demand in kilowatts during a given period for a particular class of service divided by the total rated load in kilowatts for the same class of service, or

$f_d = \frac{\text{actual maximum demand in kw}}{\text{total connected load in kw}}$

Diversity factor expresses as a fraction the probability that any number of customers will require the maximum demand for a particular class of service at the same moment. Expressed as an equation, the diversity factor is the ratio of the maximum demand in kilowatts for a class of service, measured at the point of supply, to the sum of the individual customers' maximum demands in kilowatts for the same class of service, or

 $f_d^T = \frac{\text{maximum demand in kw at point of supply}}{\Sigma \text{ individual maximum demands in kw}}$

Table I gives the generally accepted demand and diversity factors for a number of classes of service.

TABLE I

DEMAND FACTORS

Residence Lighting	0.50
Commercial Lighting .	
General commercial lighting	0.60
Restaurants, stores, offices	0.70
Theaters	0.60
Schools and churches	0.55
Hotels	0.50
Hospitals	0.45
Warehouses	0.40
General Power Service	
Less than 10 hp	0.75
10 to 20 hp	0.65
20 to 100 hp	0.55
Over 100 hp	0.50

DIVERSITY FACTORS

	Between Cust.	Between Trans.	Between Fdrs.	Between Substa.
Residence Lighting	0.30	0.75	0.85	0.90
Commercial Lighting	0.70	0.75	0.85	.0.90
General Power	0.70	0.75	0.85	0.90

Example 1 shows the use of demand and diversity factors in the determination of the capacity of the various components of an elementary power system such as is shown in Figure II-1.

Example 1.

A rower plant supplies energy to a town through a primary distribution system consisting of a transmission line and three substations. Substation A serves a residential district through two feeders and five transformers; Substation B serves a commercial district through two feeders and five transformers; and substation C serves an industrial district through three feeders and four transformers. In-

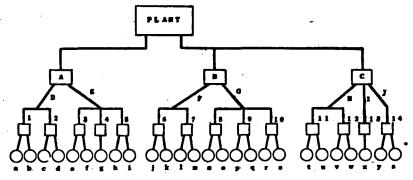


Fig. II-1. Elementary power system.

dividual customer connected loads are as follows (Refer to Figure II-1):

8.	3	kw	j.	15	kw	(theater)	t,	10	hр
b.			k.	20	kw	(restaurant)	u.	50	hp
c.	5	kw	1.	17	kw	(store)	٧.	35	hp
d.	4	kw	m.	25	kw	(warehouse)	W.	2	hp
e.	6	kw	n.	30	kw	(school)	X.	100	hp
f.			0.	26	,kw	(hospital)	у.	40	hp
g.	5	kw	p.	15	kw	(office)	z.	60	hp
h.			q.	18	kw	(store)			•
i.	8	kw	r.	35	kw	(store)			
			8.	40	kw	(hotel)			

Determine the rating of each transformer, feeder, substation, transmission line section; and the probable maximum demand on the power plant.

Solution:

1. Determine the probable maximum kilowatt demand for each customer by applying the demandfactor appropriate for each case, as follows:

```
3(0.5)=1.5 \text{ kw} j. 15(0.60)=9.0 \text{ kw} t.
                                                     10(0.65)(0.746)
а.
                                                     \div 0.80 = 6.1 kw
b. 2(0.5)=1.0 \text{ kw} k. 20(0.70)=14.0 \text{ kw}
                                                     50(0.55)(0.746)
                                                u.
                                                     \div 0.80 = 25.7 kw
c.
     5(C.5)=2.5 \text{ kw} 1. 17(0.70)=11.9 \text{ kw}
                                                     25(0,55)(0,746)
                                                v.
                                                     \div 0.80 = 18.0 kw
d. 4(0.5)=2.0 \text{ kw} m. 25(0.40)=10.0 \text{ kw} w.
                                                      2(0.75)(0.746)
                                                     \div 0.80 = 1.4 kw
```

```
e. 6(0.5)=3.0 \text{ kw} n. 30(0.55)=16.5 \text{ kw}
                                                x. 100(0.55)(0.746)
                                                     \div 0.80 = \xi1.3 kw
    7(0.5)=3.5 \text{ kw} o. 26(0.45)=11.7 \text{ kw}
                                                     40(0.55)(C.746)
f.
                                                ӯ.
                                                     \div 0.80 = 20.5 kw
g. E(C.5)=2.5 \text{ kw} p. 1F(0.70)=10.5 \text{ kw}
                                                     60(0.55)(0.746)
                                                z.
                                                     + 0.80 = 30.8 \text{ kw}
h. 10(0.5)=5.0 kw q. 18(0.70)=12.6 kw
                                                 Note: Motor eff-
i. 8(0.5)=4.0 \text{ kw} r. 35(0.70)=24.5 \text{ kw}
                                                 iciency estimated at
                                                 80 per cent.
```

2. Determine the kilowatt rating of each transformer by applying the appropriate diversity factor for consumers, as follows:

s. 40(0.50) = 20.0 kw

```
1. (1.5 + 1.0)(0.30) = 0.75 kw 8. (16.5 + 11.7)(0.70) = 19.8 kw 2. (2.5 + 2.0)(0.30) = 1.35 kw 9. (10.5 + 12.6)(0.70) = 16.2 kw 2. (2.0 + 3.5)(0.30) = 1.95 kw 10. (24.5 + 20.0)(0.70) = 31.2 kw 4. (2.5)(no \ div.) = 2.50 kw 11. (6.1 + 25.7)(0.70) = 22.2 kw 5. (5.0 + 4.0)(0.20) = 2.70 kw 12. (18.0 + 1.4)(0.70) = 12.6 kw 6. (9.0 + 14.0)(0.70) = 16.1 kw 13. (51.3)(no \ div.) = 51.3 kw 7. (11.9 + 10.0)(0.70) = 15.3 kw 14. (20.5 + 30.8)(0.70) = 25.9 kw
```

3. Determine the kilowatt rating of each feeder by applying the diversity factor for transformers, as follows:

```
D. (0.75 + 1.35)(0.75) = 1.58 kw

E. (1.95 + 2.50 + 2.70)(0.75) = 5.36 kw

F. (16.1 + 15.2)(0.75) = 23.6 kw

G. (19.8 + 16.2 + 21.2)(0.75) = 50.4 kw

H. (22.2 + 12.6)(0.75) = 26.8 kw

I. (51.2)(\text{no div.}) = 51.3 kw

J. (25.9)(\text{no div.}) = 35.9 kw
```

4. Petermine the kilowatt rating of each substation by applying the diversity factor for feeders, as follows:

```
A. (1.58 + 5.26)(0.85) = 5.9 kw
B. (23.6 + 50.4)(0.85) = 62 kw
C. (26.8 + 51.3 + 25.9)(0.85) = 97 kw
```

F. Petermire the probable maximum demand on the power plant by applying the diversity factor for substations, as follows:

$$(5.9 + 63 + 97)(0.90) = 149 \text{ kw}$$

It will be obvious from the preceeding analysis that the rating of any component of the system will never be less

than the probable maximum demand of the largest component served by it. For example, transformer 1 would of necessity have a rating of 1.5 km in spite of the mathematics which indicates that the rating is 0.75 km. However, as the size of the system increases, the behavior of the system approaches the statistical ideal and these irrationalities disappear.

9. Load Curves -- The foregoing discussion of demand and diversity factors indicates that a constant load cannot be expected for any actual power plant. This indication is found to be true, and furthermore, it has also been found to be true that systems serving approximately the same classes of loads anywhere will have approximately the same type of load curves. For example, industrial plant load will show flat - topped curves with sharp drops at the time of shift changes and at meal times. Utility plant loads will show a minimum during the night; a minor and somewhat flat - topped peak during the morning; a sharp drop at noon; and a sharp peak in the early evening. These characteristics are shown

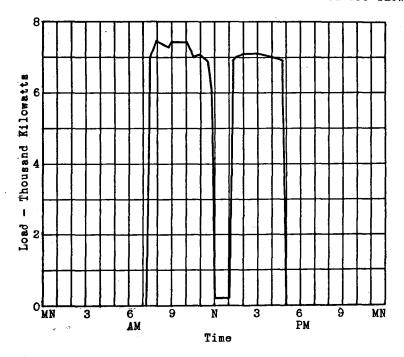


Fig. II-2. Typical daily load curve for industrial plant.

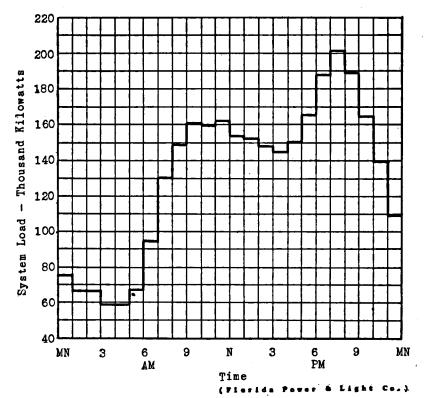


Fig. II-3. Typical daily load curve for public utility.

in Figures II-2 and II-3. Load curves of these types can easily be synthesized if the character and size of the loads of various classes along with the time of their demands are known.

Other types of load curves can be drawn in addition to the daily load curves. For example, the daily maximum loads may be plotted to give an annual peak load curve. This type of curve is useful in deciding upon the best time during the year for scheduling major plant maintenance operations. The yearly maximum loads may be plotted to give a load growth curve. This type of curve will give information as to when it will be necessary to increase plant capacity. These two types of curves are used most frequently with the public utility plant because electric power is the primary product. The industrial plant will make use of similar curves with units of production, such as tons of steel, for ordinates

rather than kilowatts of load. In the ordinary industrial plant power is not a salable commodity. There are cases, however, where industrial plants have arranged to sell power to public utilities, and vice versa, when the peaks in the two types of operations do not coincide. Figures II-4 and II-5 show typical annual peak and load growth curves for a public utility.

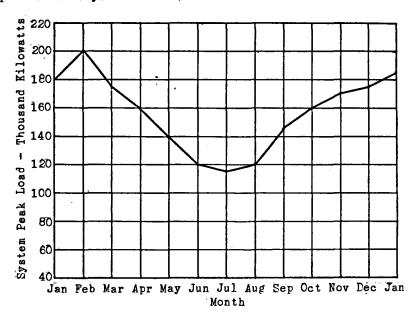


Fig. II-4. Annual peak load curve for public utility.

The daily lcad curve or the annual peak load curve may be plotted in a form known as the load duration curve. This curve is plotted with time units as abscissa and load as ord-unate, such that any point on the curve gives the total time for which the particular lcad was in demand. Figure II-6 shows a typical daily load duration curve for a public utility. It will be noted that since power output is the product of load and time, the area under the load duration curve is the same as under the load curve.

Example 2 shows the method of synthesis of a daily load curve for the elementary power system of Example 1.

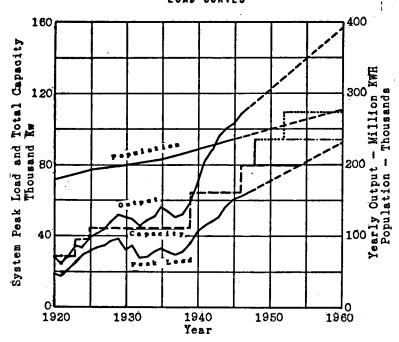


Fig. II-5. Load growth curve for public utility,

Example 2.

The various loads in the town of Example 1 are to be expected at the following times:

Residential load:					
10 o/o			l de		•
100 o/o	6	AM	to	9	AM
	4	PM	to	8	FM
Commercial load:					
Theater, j	11	AM	to	11	PM
Restaurant, k	. 6	AM	to	9	AM
	11	AM	to	2	PM
	5	PM	to	9	PM
Stores, 1, q, r		AM			
Offices, p	9	AM	to	12	N
.					
Schools, n	9	AM			
	1	PM	to	4	PM

Hospitals, o	
10 0/0	all day
100 o/o	9 AM to 5 FM
Warehouses, m	6 AM to 6 PM
Hotels, s	
10 o/o	all day
100 0/0	7 AM to 10 FM
Industrial load:	
25 o/o	all day
100 o/o	8 AM to 12 F
	1 PM to 5 PM

Using the demand and diversity data from Example 1, set up a load table and plct an expected daily load curve.

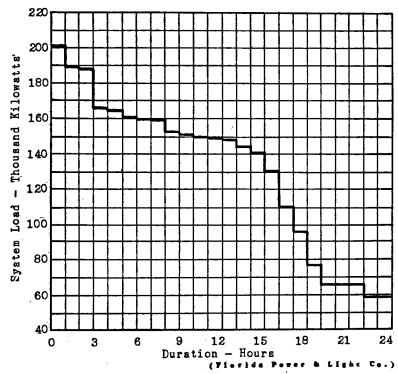


Fig. II-6. Daily load duration curve for public utility.

Solution:

LOAD TABLE

· · · · · · · · · · · · · · · · · · ·	٥.	_	e	f'd	21	fid	e t	kw max.
Service	Co lo	ad	f _d	cons.	fd trans	. fdrs.	fd sub.	den.
. 8	3	kw	0.50	0.30	0.75	0.85	0.90	0.26
ъ	2	kw	41	n		**	•	0.17
c	5	kw	11	п	**	11	. 17	0.43
đ	4	kw	#1		11	11	**	0.34
е		kw	**		**	**	n er	0.52
f		kw	#	H	n	17 32	77	0.60
g h		kw	11 Ti	1.00	# #1	. #		1.44
h		kw	#1 #1	0.30	**	,,	"	0.86
i	8	kw						0.69
j	_	kw	0.60	0.70	0.75	0.85	0.90	3.61
k		kw	0.70	"	#	#1 	••	5.62
1	17	kw	0.70	"	H 	11		4.78
30		kw	0.40	11 11	. #	11 [*]	#	4.01
n	30		0.55		11 11		,	6.63
Ö	26		0.45		"	,,	**	4.70
p		kw	0.70	n-	,, n	" "		4.21
q		kw	C.70		,,	. 11	. "	5.06
r	-	kw	0.70		n	. n		9.85
S	40	kw	0.50		"			8.04
ŧ	10	hp	0.65		•	H	n	2.43
u		hp	0.55		**.	n	**	10.28
v	35	hp	0.55	."	n	n		7.21
W		hp	0.75		n	**	'n	0.56
x	100		0.55	1.00	1.00			39.30
У	40		0.55		**	H	#	11.00
z	60	hр	0.55				" .	16.50
Ser	٧.	M	N-1AM	1-2AM	2-3AM	3—4AM	4-5AM	5-6AM
Res			0.53	C.53	0.53	0.53	0.53	0.53
ţ	i							•
1								
1,0								
Ţ								
1			0.47	0.47	0.47	0.47	0.47	0.47
	2		V.4/	·	O 141	V.31		0,47
	p 3		0.80	0.80	0.80	0.80	0.80	0.80
Inc	ł.		21.70	21.70	21.70	21.70	21.70	21.70
To	tal		23.50	23.50	23.50	23.50	23.50	23.50

Serv.	6-7AM	7-8 AM	8-9AM	9-10 AM	10-11AM	11-12N
Res.	5.31	5.31	5.31	0.53	0.53	0.53
j k l,q,r m	5.62	5.62	5.62	19.69 4.21	4.21	3.61 5.62 19.69 4.21
n O	0.47	0.47	0.47	6.63 4.70		6.63 4.70
p	4.01	4.01	4.01	4.01		4.01
8	0.80	8.04	8.04	8.04	8.04	8.04
Ind.	21.70	21.70	86.78	86.78	86.78	86.78
Total	37.91	37.91	110.23	134.59	134.59	143.82
Serv.	12-1PM	1-2PM	2-3PM	3-4PM	4-5PM	5-6PM
Res.	C.53	0.53	0.53	0.53	5.31	5.31
j	3.61	3.61	3.61	3.61	3.61	3.61
k	5.62	5.62				5.62
1,q,r	19.69	19.69	19.69	19.69	19.69	19.69
m	4.21	4.21	4.21	4.21	4.21	4.21
n		6.63	6.63	6.63		
. 0	4.70	4.70	4.70	4.70	4.70	0.47
p		4.01	4.01	4.01	4.01	4.01
S	8.04	8.04	8.04	8.04	8.04	8.04
Ind.	21.70	86.78	86.78	86.78	86.78	21.70
Total	68.20	143.82	138.20	138.20	136.35	72.66
Serv.	6-7 PM	7-8PM	8-9PM	9-10PM	10-11PM	11-MN
Res.	5.31	5.31	0.53	0.53	0.53	0.53
j	3.61	3.61	3.61	3.61	3.61	
k	5.62	5.62	5.62			
l,q,r			•			•
'n						
n		0.45	0.47	0.477	0.47	0.477
0	0.47	0.47	0.47	0.47	0.47	0.47
P	9 04	8.04	8.04	8.04	0.80	0.80
S	8.04	0.04	0,04	0.04	0.00	0.00
Ind.		21.70	21.70	21.70	21.70	21.70
Total	44,75	44,75	44.75	34.35	27.11	23.50

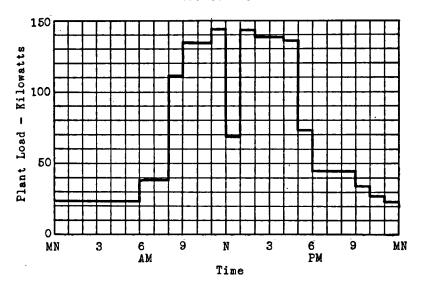


Fig. II-7. Load curve for elementary power system.

10. Factors Used in Operation Analysis -- Due to the variability of the load on power systems, it is seldom that a plant operates at its ultimate capacity continuously. In order to express the power output of a plant in terms that are easily analyzed, several factors have come into use.

Load factor is the average load during any specified time divided by the peak load of any specified duration. The average load may be taken over a day, month or year, and the peak load duration may be taken as fifteen or thirty minutes or one hour. For the most common case of daily load factor:

f₁ = total generation for day in kwh

24[maximum demand in kw sustained for one hour]

Capacity factor is the total power production for any specified time divided by the product of the installed capacity times the length of the specified period. For the most common case of daily capacity factor:

 $f_c = \frac{\text{total generation for day in kwh}}{24[\text{installed capacity in kw}]}$

Use factor is the total power production for any given .

time divided by the summation of the products of the hours operated times the installed capacity for each machine. For the most common case of daily use factor:

$$f_w = \frac{\text{total generation for day in kwh}}{\Sigma[(\text{hours operated})(\text{capacity of machine in kw})]}$$

II. Determination of Size of Equipment--The problem of variable load and the necessity for insuring continuity of service are the two factors which have the greatest influence in the selection of equipment for a power plant. It has been seen that the plant must have a capacity sufficient to carry the maximum demand, but it is almost never the case where only one generating unit is selected to meet this demand. It is much more usual to select two or more units for the installation. Examination of Figure II-8 will show why

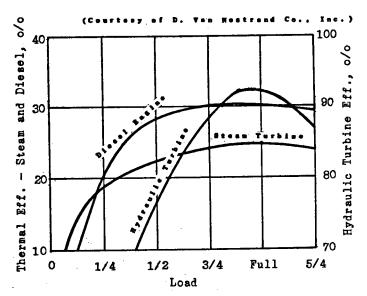


Fig. II-8. Efficiency vs load, engines and turbines.

this is the case. Power generating equipment design involves many compromises, especially when the equipment is expected to operate with variable load, and the usual result is that the thermal efficiency of the machine is at its maximum between three-fourths load and full load, with the efficiency falling as the load increases or decreases beyond this range.

It can be seen, then, that for maximum operating efficiency the machine should operate at or near its rated load. This means that the usual plant will have more than one machine, and they may very well be of different sizes so that maximum efficiency could be approached at all loads. Such an ideal case is never the most economical, however, when the complete picture is considered. It is a fact that as the size of the machine increases the thermal efficiency at full load increases and the cost per kilowatt of installed capacity decreases. Economies in investment can also be made by installing duplicate machines wherever possible.

Continuity of service and ability to meet unusual conditions require that:

- In order to insure against breakdown in equipment, reserve idle capacity must be provided equal to the capacity of the largest machine in the plant or system.
- 2. In order to insure against inability to meet sudden unusual demands, spinning or active reserve capacity must be provided, usually to the extent of about 25 per cent of the demand at any time.

The following criteria are useful in the examination of a proposed design, or will be of help in selecting equipment to fit a particular situation:

- The total plant investment per kilowatt of capacity must be at the minumum, or it must be demonstrated that the sum of the operating costs and fixed charges are at the minimum.
- Capacity and use factors for the plant must be at the maximum consistent with adequate idle and spinning reserves and with leeway for growth of load.

Similar considerations apply to the selection of boiler and other equipment in the plant. Each major item of equipment should be subjected to a critical analysis in order to determine the most economical combination.

Example 3-A shows the method of selection of machines for the installation considered in the elementary power system of Example 1 and Example 3-B shows the method of selection of machine sizes to fit an extension of a load growth curve.

Example 3-A.

The town of Examples 1 and 2 is to be served by a Diesel-electric plant. Available generator sizes are: 25 kw, 50 kw, 75 kw and 100 kw. Select a unit or units which will give the maximum use and capacity factors consistent with required reserves. Set up an operating schedule for the plant.

Solution:

Since the peak load is almost 150 kw, provision for adequate spinning reserve would dictate an assured plant capacity of at least 185 kw to care for present needs. This capacity would necessitate the use of at least two machines in addition to a spare machine for assuring plant capacity. Possible combinations are: (a) 3 - 100 kw (one spare); (b) 5 - 50 kw (one spare); (c) 1 - 25 kw, 1 - 50 kw and 2 - 100 kw (one spare); and (d) 2 - 50 kw and 2 - 75 kw (one spare). Capacity factors for these combinations would be:

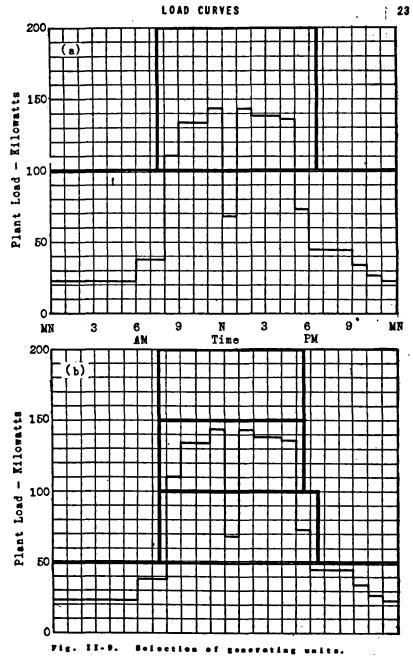
(a)
$$f_e = \frac{1652.7}{24(300)} = 0.229$$

(b) and (d) $f_e = \frac{1652.7}{24(250)} = 0.275$
(c) $f_e = \frac{1652.7}{24(275)} = 0.250$

On the basis of capacity factors alone, (a) could be eliminated from further consideration, although some lee-way would be provided for future growth of load. However, (b) offers the same leeway with a higher capacity factor, and (c) and (d) offer high capacity factors with no leeway for growth. From this analysis, then, (c) and (d) are eliminated and the next step is to investigate the use factors. Load factors are determined after setting up operating schedules as in Figure II-9. These schedules with use factors are:

$$f_u = \frac{1652.7}{24(100) + 11(100)} = 0.472$$





$$f_{\mu} = \frac{1652.7}{24(50) + 11(50) + 20(50)} = 0.602$$

Then, from the standpoints of maximum use and capacity factors and leeway for growth, combination (b) consisting of 5-50 kw machines would be selected. Additional advantages would be realized in the purchase and maintenance of identical units.

Example 3-B.

The power plant whose load growth curve is shown in Figure II-5 had a net assured capacity (total plant capacity less the capacity of the largest machine in the plant) of 49,000 kw in 1945. At this time the peak load was 60.000 kw which necessitated operation with practically no reserve capacity. An order for a 15,000 kw machine had been placed in 1942, but, due to wartime conditions, the plant extention was not placed into operation until 1946. Another 15,000 km machine was ordered in 1945 and was scheduled for operation before the 1948 peak load. In 1948 the net assured capacity of the plant was 79,000 kw and the peak load was 66,000 kw. At this time it was decided to plan for future load growth. The load growth curve was projected and it appeared that the peak load would equal the plant net assured capacity in 1954. It was decided to order another 15,000 kw unit to operate initially in 1952, and indications were that the plant net assured capacity would then be sufficient until about 1960. It was also decided that the situation would be resurveyed in 1955 with the view of retiring some of the older machines which had been operating since 1920.

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CHAPTER III

ECONOMICS OF POWER PRODUCTION

12. Power Plant Income -- The public utility derives its income directly from the customers it serves, while the industrial power plant usually receives its operating funds from the revenues of the company of which it is a part.

It can be seen, after some thought, that the public utility company's charge, or rate, for electrical service to its customers must include an element to pay for the labor, materials and fuel, an element to pay for the cost of the necessary machinery, and an element which represents the investors' profit. Furthermore, since industrial customers usually supply a higher type of load than domestic customers, the industrial customers are usually accorded a more favorable rate than the domestic customers. These considerations, when coupled with the inevitability of daily and seasonal demand fluctuations, result in a somewhat complex rate structure for a public utility.

The industrial power plant is not usually faced with the problem of setting a rate for its power. The present method of cost accounting in industry takes only the average annual cost of power per unit of production for the determination of the power element in the cost of the manufactured product. This may not be a fair method of accounting if the industry suffers from seasonal peaks, but, on the other hand, it is only the net annual profit for the industry which is significant.

In spite of the tremendous growth of the manufacturing industries in this country, the domestic customer supplies the greater part of the public utilities' revenue. Figure III-1 shows the growth of load and decrease in cost per kilo-watt-hour for domestic customers. The engineer's contribution can be seen clearly in these curves. Increasing efficiency, as seen in Figure I-4, has made lower rates possible, which in turn has made it possible for customers to increase their use of electricity. It should be noted that the trend of rates has been continually downward in spite of the fact of creeping inflation which has been reducing the real value of money.

13. Public Utilities as Monopolies--It does not take careful consideration to arrive at the conclusion that a service such as electric power can be supplied most economic-

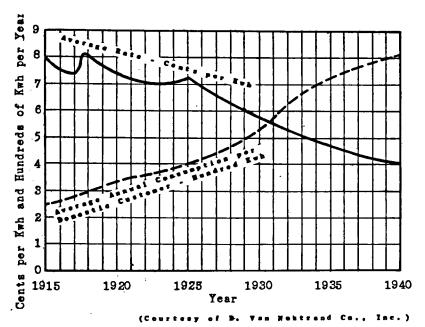


Fig. III-1. Coat and use of domestic electric service.

ally by only one company operating within a given area. Any other arrangement would be wasteful and even inconvenient. Consider, for example, the case of a large city which has two telephone companies competing in the same area. Obviously, the customer is being made to pay for two systems where one would be sufficient. The public utility, then, takes on the character of a monopoly, but abuses are prevented, in the case of private ownership, by the necessity of obtaining a renewal of its permit to do business (or franchise) at intervals of time. Franchise granting agencies always reserve the right to purchase the utility at the end of the franchise period, or to grant a new franchise. Further insurance against abuses is provided by regulation by various public utility commissions, local, State or Federal.

The municipal utility has no such restrictions upon it, and the rates charged may or may not include an element of profit. Most cities which own and operate their own power systems do include an element of profit which is intended to make it possible to charge lower ad valorem property taxes. This method of collecting taxes for municipal operating expenses is based on the theory that taxes will receive

the least opposition from the public when they are spread out in small payments attached to some widely used commodity.

- 14. Rates -- Any rate for power should be examined with the following criteria in mind:
 - Fairness—Rates should distribute costs in accordance with the variability of the customer's demand.
 - Simplicity—Rates, especially domestic, should be as simple as possible in order to avoid antagonism from the public.
 - Cost—The outlay for metering equipment and accounting machinery should not be so great as to increase costs to a greater amount than can be paid for out of increased revenues from more complicated rate structures.

Every rate charged the consumer of electrical energy contains the following elements:

- Fixed element is the element of cost which is present in the system operation whether or not energy is produced.
- Energy element is the element of cost which is proportional to the amount of energy produced.
- 3. Variable load element is the element of cost which is due to the character of the load served.
- 4. Customer element is the element of cost which is proportional to the number of customers.
- Investors' profit is the element of cost which goes to reimburse the investor for the use of his capital.
- 15. The Fixed Element--The fixed element of the cost of producing power is the sum of the following factors:
 - 1. Capital cost of the power plant.
 - a. Real estate.
 - b. Building and machinery.
 - c. Cost of construction.
 - d. Engineering fees..
 - Capital cost of primary distribution or transmission system.

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FUNDAMENTALS OF POWER PLANT ENGINEERING

- a. Cost of easements or right-of-way.
- b. Cost of line and substation equipment.
- c. Cost of construction.
- d. Engineering fees.
- 3. Interest, taxes and insurance.
- 4. Depreciation of buildings and equipment.
- 5. Sinking fund contributions for bond retirement.1
- 6. Cost of management.
- Capitalized maintenance, or the cost of general maintenance required whether or not power is produced.

Estimating the capital cost of a power plant is a rather difficult problem due to variations in such factors as location, availability of labor and materials and size of plant. Table II in connection with Figure III-2 may be useful in making a first approximation. The lower figures in Table II apply to larger plants.

In the final analysis, the only way to prepare an accurate cost estimate is from manufacturers' and contractors' bids after the general layout of the plant has been determined.

If the plant is not situated within very close proximity to the area which it is to serve, a primary distribution system will be necessary. This system should be laid out to carry the expected load of the plant, therefore its cost will be proportional to the caracity of the plant and to the distance over which the lines extend. It is almost impossible to give any figures as to cost per mile for transmission systems because of the many factors, such as type of terrain and size of conductors and carrying capacity, that enter into the cost. However, for purposes of estimating very

Municipal utilities are required by State lems to include a sinking fund for band rutirement in the fixed element of the cost. Sometimes a depreciation reserve is also included to provide for replacement of equipment when it becomes obselete, but more often a small reserve is maintained for minor replacements only, and when major improvements are required bonds or revenue certificates are sold.

The private utility always includes a depreciation reserve in the fixed element of the cost, but it is expected to provide a sinking fund for bond retirement out of the profit element. This is usually explained as "preventing the practice of asking the customer pay for the capital required to provide him with service."

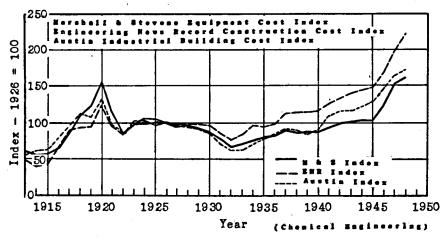


Fig. III-2. Building and equipment cost index.

TABLE II

APPROXIMATE POWER PLANT CAPITAL COSTS dollars per kilowatt capacity (base year 1932)

Turbo-generator plants	80 to 150
Engine-generator plants	70 to 150
Piesel-generator plants	130 to 250
Hydro-generator plants	100 to 300

roughly, the cost of the primary distribution system may be taken as ranging between \$1500 and \$5000 per mile (based on 1928), the lower figure applying to low voltages and light loads in relatively flat and open country.

Since it is usual to finance part or all of the cost of construction by the issuance of bonds or revenue certificates, interest on these securities is part of the cost of operation. Interest rates vary according to the money pclicies of the Federal Government. At the present time, interest rates are low, usually around three per cent, but in the past they have been as high as six per cent.

If the utility is privately owned, ad valorem property taxes and occupational licence fees must be paid. These

usually amount to from two to three per cent of the investment annually. In some States municipally owned utilities must pay the State occupational licence fee.

All utilities maintain insurance on their equipment and liability insurance for protection against claims for damages. The cost of insurance amounts to from one to two per cent of the investment annually.

In order to provide a fund for eventual replacement of worn-out or obsolete equipment, a retirement reserve is accumulated over a period of years. The length of time for the accumulation of the reserve depends upon the estimated life of the equipment. Table III shows the legally approved retirement periods for various types of equipment. The reserve may be accumulated in two different ways: by the sinking fund method or by the straight line method, although the straight line method is most usual. In either case, the sum accumulated will be equal to the original cost of the equitment less a salvage value. For a number of years business organizations have been attempting to have income tax laws amended to allow retirement reserves to be accumulated on the basis of present replacement cost, but they have had no success up to the time of this writing. Inadequate retirement reserves have necessitated the sale of additional stocks and bonds to obtain capital for needed modernization.

The sinking fund for retirement of bonds or obsolete equipment is similar to an annuity, and, as a matter of fact, annuity tables can be used for computing the annual payment. The sinking fund is accumulated by making an annual payment to an account in a bank, which draws interest at the current rate, or by purchasing negotiable securities which draw interest. The total of the payments plus the interest earned over the retirement period will equal the total sinking fund reserve. The annual payment can be calculated from the following equation:

annual payment = {P - S}{r + [(1 + r)^n - 1]}

where: P = original cost

S = salvage value

r = rate of interest earned

(compounded annually)

n = number of years

Since the accumulation of the plant retirement reserve is not a very accurate process at best, it is more usual to use the straight line method of degreciation and to treat

TABLE III PROBABLE LIFE OF POWER PLANT COMPONENTS

Component	Υe	818	
Boilers, fire-tube	10	to	15
Boilers, water-tube		20	
Breechings and ducts	10	to	30
Buildings	20	to	30
Cables and feeders	15	to	25
Coal and ash machinery	10	to	20
Condensers		20	
Cranes		30	
Economizers and air heaters		15	
Electric generators		.20	
Electric motors		20	
Engines, steam	15	to	35
Engines, Diesel		15	
Feed-water heaters		20	
Fuel oil handling equipment		20	
Fipe and pipe covering	15	to	25
Pumps	15	to	20
Stacks, brick or concrete		30	
Stacks, steel	12	to	15
Stokers and other coal burning equipment		20	
Superheaters		20	
Switchboards and equipment		20	
Transformers		15	
Turbines, steam	15	to	20
Turbines, hydro		30	

any interest earned as income. The annual payment can be calculated from the following equation, with symbols as above:

annual payment =
$$\frac{(P-S)}{n}$$
.

Unlike the plant retirement reserve, the amount necessary for bond retirement is known exactly. For this reason, the sinking fund is universally used for bond retirement. The annual payment can be calculated from the previous sinking fund equation (leaving out the salvage item). It is necessary to recalculate the annual payment from time to time as interest rates change.

The cost of management is included in the fixed charges, and usually management is taken to include all administrative officers down to and including the plant superintendent.

Usually about ten per cent of the total maintenance for the system is included in the fixed charges. This so-called capitalized maintenance is assumed to be the portion of the maintenance which is independent of the amount of power produced.

- 16. The Energy Element -- The energy element of the cost of producing power is the sum of the following factors:
 - 1. Cost of fuel.
 - 2. Cost of labor.
 - 3. Cost of water.
 - a. Boiler feed makeup water.
 - b. Condensing water.
 - . Service water.
 - 4. Lubricating oil and operating supplies.
 - 5. Maintenance.

The greatest portion of the energy element is due to the cost of fuel, and for this reason it will be found that highly skilled labor in the power plant will always operate the plant in such a way as to save more than enough fuel to pay the difference in salary over unskilled labor. Due to the fact that the labor force required is proportional to the capacity of the plant rather than to plant output, employment in the public utility business is very stable and attractive to those who would make a career in the business.

Since the steam power plant operates best with pure water, and natural waters are very seldom pure enough, the cost of preparation of the water for use as well as the cost of the water itself, in cases where it must be purchased or pumped, must be considered.

A not inconsiderable item in operating costs is that of supplies of various sorts. These would include forms and charts for record keeping, cleaning and wiping sundries, and all other expendable items which might be needed. Lubricating oil is not a large item of expense in the well operated turbc-generator plant, since with proper care a lubricating cil charge will last almost indefinitely and requires only a small amount of makeup at infrequent intervals.

Maintenance costs include the cost of repair and of replacement of parts which wear cut. Properly supervised plants have a schedule of continuous maintenance designed to keep the machinery in the best possible condition at all times. All but the smallest plants employ a maintenance force in addition to the plant operators. The cost of maintenance is usually estimated at about one per cent of the investment annually.

- 17. The Variable Load Element -- This element of the cost of production of power is usually included with the energy element because of the difficulty in assessing its effect. Consideration would have to be made of the reserve capacity required and the status of the plants in the system with respect to the type of loading, that is, base load, standby or swinging.
- 18. The Customer Element -- The customer element in the cost of producing power is the sum of the following factors:
 - Capital cost of the secondary distribution system.

 - a. Cost of easements or right-of-way.b. Cost of line and substation equipment.
 - c. Cost of construction.
 - d. Engineering fees.
 - 2. Interest, taxes and insurance.
 - Depreciation of equipment.
 - Sinking fund contributions for bond retirement.
 - 5. Cost of franchise.
 - 6. Publicity.
 - a. Public relations.
 - b. Advertising.
 - 7. Cost of collecting revenue.
 - a. Meter reading.
 - b. Billing and collecting.
 - c. Accounting.
 - 8. Maintenance.

Most of the factors appearing in the customer element have been discussed under the fixed element. In the matter of depreciation and sinking fund, the remarks in the footnote on page 28 also apply.

Investment in the secondary distribution system is roughly proportional to the number of customers served, and for purposes of estimation may be taken as approximately \$50 per customer (based on 1928).

The cost of publicity is usually lumped together with the cost of the franchise and may be taken as between one and two dollars per year per customer, depending upon the amount of advertising considered necessary. The cost of the franchise itself is usually nominal, and is usually considered to be in the nature of an occupational licence tax.

The cost of collecting revenue may vary between \$1.50 and \$3 per customer per year (based on 1932), the smaller figure applying to the larger systems.

19. Investors' Profit -- The profit element in the cost of producing power may or may not be present, according to the type of utility operation. It is inherently present in the private utility system, and it may be present in the municipal utility operation, as previously pointed out, when it is desired to reduce property taxes. It is usually absent in the Federal utility system, and, as a matter of fact, the Federal utility may suffer a deficit without danger of bank-rupcy.

The profit on electric utility operations is usually estimated at about ten per cent of the investment annually before income taxes and bond sinking fund deductions are made.

20. Types of Rates -- There have been a great number of methods of calculating rates for electrical power used in the past, but of them only the following remain to any considerable extent:

Block Meter Rate—In this type of rate a certain specified price per unit is charged for all or part of a block of units, and reduced prices per unit are charged for succeeding blocks. Figure III-3 shows a rate schedule based on this system. The block meter rate has the advantage of encouraging the use of electricity, providing the initial blocks are not too large, and is almost universally used for residential service,

ECONOMICS OF POWER PRODUCTION

Rate Schedule RS-1

RESIDENTIAL SERVICE—ELECTRIC

APPLICATION:

For service for all domestic purposes in private residences and individually metered apartment units.

LIMITATION OF SERVICE:

All residential service required on the premises by Customer shall be supplied through one meter. Stand-by or resale service not permitted hereunder.

CHARACTER OF SERVICE:

Single phase, 115/230 volts, 3 wire or 115 volts, 2 wire depending upon Customer's requirements. Three phase, 230 volt service may be furnished but only under special arrangements.

MONTHLY RATE.

\$1.00 which entitles Customer to use 14 kwh,

4.5# per kwh for the next 36 kwh,

2.5¢ per kwh for the next 130 kwh,

1.5¢ per kwh for all additional kwh.

Tax Adjustment,—Plus the applicable proportionate part of any taxes and assessments imposed by any governmental authority in excess of those in effect January 1, 1937, which are assessed on the basis of meters or customers or the price of or revenues from electric energy or service sold or the volume of energy generated or purchased for sale or sold.

Inflation Adjustment,—In order to take care of extreme increases in costs in the event of inflation there shall be added to the above rate ½ of 1% for each 1% increase in cost of oil over \$2.00 a barrel (or \$2.00 per 6,300,000 Btu, if fuel used is other than oil) based on the average cost of fuel used in Company's plants during the next preceding month.

Excess Capacity Charge: For each kw of Demand in excess of 10, add \$1.75. The kw of Demand is the average kw supplied during the 15-minute period of maximum use during the current month.

RULES AND REGULATIONS:

Service under this schedule is subject to orders of governmental bodies having jurisdiction and to the currently effective "General Rules and Regulations for Electric Service" on file in Company's office, which are hereby made a part hereof. A copy of the present effective "General Rules and Regulations for Electric Service" is attached hereto. In case of confilict between any provision of this schedule and said "General Rules and Regulations for Electric Service" the provision
of this schedule shall smally.

FLORIDA POWER & LIGHT COMPANY

Effective June 1, 1946

Fig. III-3. Residential rate schedule.

Block Hopkinson Demand Rate—In this type of rate a demand charge based on the maximum demand during any stated period is added to a block meter rate. Figure III-4 shows a rate schedule based on this system. It is named after Dr. John Hopkinson, an Englishman who did much to develop the theory of rate making. This rate has the advantage of encouraging commercial and industrial customers to develop a high load factor. It is almost never used for residential service because of the necessity for demand meters.

Room Rate—In this type of rate the block meter rate is modified to include the number of rooms in the residence. For example: 9¢ per kwh for the first three kwh per room; 6-1/2¢ per kwh for the next three kwh per room; and 3-1/2¢ per kwh for the excess over six kwh per room.

Rate Schedule IR

INDUSTRIAL SERVICE—ELECTRIC

APPLICATION:

For power service for industrial, manufacturing or processing purposes. The use of energy for lighting purposes with and about the buildings, structures and premises housing or enclosing the power driven and operated machinery and equipment used in or in connection with such industrial, manufacturing or processing purposes and incidental to the use of power supplied hereunder for such purposes is permissible under this Schedule. For such lighting use, Customer will provide and maintain all transformers and auxiliary appurtenances.

CHARACTER OF SERVICE:

Single or three phase, 60 cycles and at any evailable standard voltage.

LIMITATION OF SERVICE:

All service required on premises by Customer shall be furnished through one meter. Not applicable to service in stores, hotels, office buildings, theatres, apartment houses, restaurants, clubs, or in any other commercial establishments, nor to stand-by or resale service.

MONTHLY RATE:

\$1.50 per kw for the first 400 kw of Demand, \$1.00 per kw for all additional kw of Demand,

2.6¢ per kwh for the first 1,700 kwh, 1.3¢ per kwh for the next 4,300 kwh, 1.0¢ per kwh for the next 34,000 kwh, 0.8¢ per kwh for the next 40,000 kwh, 0.6¢ per kwh for all additional kwh.

For each kw of the highest Excess On-Peak Demand established during the 12 months ending with the current month, add to the 2.6¢ block: 100 kwh per kw for the first 50 kw, 80 kwh per kw for the next 100 kw, and 70 kwh per kw for all additional kw.

First,—Minus or plus .0183¢ per kwh used for each 1¢ below 12¢ or above 18¢, respectively, in Company's cost per million British Thermal Units of fuel. For this purpose there shall be employed the weighted average delivered cost (including freight, storage and handling costs) as shown by Company's books, of all fuel used during the next preceding month in the generating stations interconnected with Company's main transmission system.

Second,—Plus the applicable proportionate part of any taxes and assessments imposed by any governmental sutherity in excess of those in effect January I, 1937, which are assessed on the basis of meters or customers or the price of or revenues from electric energy or service sold or the volume of energy generated or purchased for sale or sold.

Minimum: \$1.50 per kw of the highest Demand established during the 12 months ending with the current month, but not more than an amount sufficient to make the total charges for such 12 months equal to \$20.00 per kw of such Demand, and not less than the charge for the currently effective Demand.

The foregoing rate is not and not subject to any municipal, county or other discount,

DEMAND:

The kw as determined from Company's demand meter for the 15-minute period of Customer's greatest use during the month, adjusted for power factor, but not less than 15 kw.

ON-PEAK DEMAND:

The kw as determined from Company's demand meter for the 15-minute period of Customer's greatest use during the On-Peak Hours of the current month, adjusted for power factor.

The On-Peak Hours shall be from 5 to 9 p.m. each day, excluding Sundays, Christmas and New Year's, from December 1 to March 31 inclusive. Company reserves the right to change the monthly and hourly periods after giving 30 days' notice, but the On-Peak Hours of each day shall always be consecutive.

EXCESS ON-PEAK DEMAND:

The number of kw by which the On-Peak Demand exceeds 20% of the average of the three highest Demands determined during the next preceding calendar months of April to November, inclusive.

The Excess On-Peak Demand effective for each month of the first contract year will be mutually agreed upon by Company and Customer, and used until actual operating data are available.

POWER FACTOR:

If Customer's power factor shall average less than 85% lagging during any month, then Company may adjust the readings taken to determine the Demand by multiplying the kw obtained through such readings by 85 and by dividing the result by the average power factor actually established during the current month, Such adjusted readings shall be used in determining the Demand.

Not less than one year. In the event this schedule is a part of an agreement for a term of less than 5 years, 5% shall be added to the amount computed under the above Monthly Rate, before Adjustments, and to the above Minimum.

RULES AND RECULATIONS:

service under this schedule is subject to orders of governmental bodies having jurisdiction and to the currently effective "General Rules and Regulations for Electric Service" on file in Company's office, which are hereby made a part bereof. A copy of the present effective "General Rules and Regulations for Electric Service" is attached hereto. In case of conflict between any provision of this schedule and said "General Rules and Regulations for Electric Service" the provision of this schedule shall apply.

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The computations involved are somewhat cumbersome, but the rate is used to a great extent in large cities where the majority of the population lives in apartments.

Example 4 shows the method of computing rates for the elementary power system of Example 1, based on the assumption that the system is privately owned, and Example 5 shows the calculations for rate making for the same power system based on the assumption that the system is owned by a municipality.

Example 4.

The power plant of Examples 1, 2 and 3-A contains five 50 kw Diesel-generators which cost \$250 per kw, installed, and have a salvage value of 10 o/o at the end of 15 years. The primary distribution system cost \$20000 and has a salvage value of 20 o/o at the end of 12 years. Management cost is \$6000 per year and taxes and insurance together are 7 o/o. Depreciation of the investment is on the straight line basis. Sixty per cent of the investment is raised by the sale of bonds which mature in 20 years and bear interest at 3.5 o/o. The bonds are to be retired out of profits and the remaining 40 o/o of the investment is raised by the sale of common stocks.

The engine heat rates (Btu/kwh net output) are: full load, 11800; 0.75 load, 12000; 0.5 load, 12900; 0.25 load, 14850. Fuel oil having 6,300,000 Btu/bbl and costing \$3.00 per bbl is used; maintenance \$600 per year; labor \$800 per month; supplies \$200 per year; and cost of water \$100 per year.

The system has 1000 customers and the cost of collection is \$3.00 per customer per year. The secondary distribution system cost \$75000 and has a salvage value of 15 o/o at the end of 12 years; labor \$1000 per month; maintenance \$800 per year; and taxes and insurance together are 7 o/o. The depreciation and financing are managed in the same way as for the plant and primary distribution system.

The annual profit is expected to be 10 o/o of the total investment, before income taxes are deducted. The income tax rate is 30 o/o. Find the amount of money available for common stock dividends after setting up a sinking fund for bond retirement. The sinking fund earns interest at 2 o/o.

Make a block meter rate which will recover the fixed element in the first 8 kwh per customer per month; the customer element in the next 15 kwh per customer per month; and

the energy and profit elements from the excess over 30 kwh per customer per month.

Solution:

A. Fixed element:

Investment in plant = 250(\$250) Investment in primary distribution system	£	62500 20000
Annual depreciation of plant = $0.90 \frac{62500}{15}$	S	3 7 50
Ann. depr. of pri. dist. sys.= $0.80 \frac{20000}{12}$		1333
Taxes and insurance = 0.07(82500)		5 7 75
Interest = $0.035(0.60)(82500)$		1732
Capitalized maintenance = (0.10)(1400)		140
Management		6000
Total annual fixed element	æ	18730

£ 18730

B. Energy element:

In Figure III-5 a the engine heat rates are plotted vs kw output. Since all units are identical, the load will be divided equally among the units operating at any particular time. Figure III-5 b shows the plant heat rate vs plant load in kw. The following table can then be calculated:

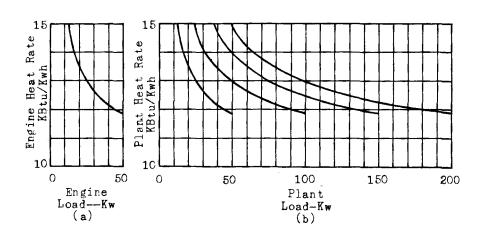


Fig. III-5. Engine heat rates for elementary power system.

\$27855

Time AM	kwh	HR K	Heat KBtu	Time PM	kwh	ĦR К	Heat KBtu
MN- 1 1 - 2 2 - 3 3 - 4 4 - 5	23.5 23.5 23.5 23.5 23.5	13.1 13.1 13.1 13.1 13.1	308 308 308 308 308	N - 1 $1 - 2$ $2 - 3$ $3 - 4$ $4 - 5$	68.2 143.8 138.2 138.2 136.4	13.7 12.3 12.4 12.4 12.4	935 1770 1715 1715 1690
5 - 6 6 - 7	23.5	13.1	308	5 - 6 6 - 7	72.7	13.6 { 13.0 13.2	968
7 - 8	37.9	12.2	425 591	7 - 8	44.8	12.0 12.0	565 533
8 - 9 9 -10 10-11 11- N	110.2 134.6 134.6 143.8	20.0 12.8 12.4 12.4 12.3	1410 1670 1670 1770	8 - 9 9 -10 10-11 11-MN	44.8 34.4 27.1 23.5	12.0 12.3 12.8 13.1	533 424 347 308

Total Btu/day = 20,887,000

Total annual customer element

Cost of fuel/day =
$$\frac{20,887,000}{6,300,000}$$
(\$3.00) = \$9.95

Annual fuel cost = $365(\$9.95)$	2 363.0
Labor = 12(\$800)	9600
Water	100
Suprlies	2 0 C
Maintenance = 0.90(\$600)	540

Total annual energy element. \$14970

C. Customer element:

Investment in sec. dist. sys. =	\$7 500 0
Depr. of sec. dist. sys. = $9.85 \frac{75000}{12}$	\$ 5310
Taxes and insurance = 0.07(75000) Interest = 0.035(0.60)(75000) Maintenance = 0.90(800) Labor = 12(1000) Cost of collection = 1000(\$3.00)	5250 1575 720 12000 3000

D. Profit element:

Total investment	2157500
Profit = 0.10(157500) Less income tax = 0.30(15750)	\$15750 472 5
Profit after taxes Less sinking fund cont.=	\$11025
$0.60(157500) \frac{0.02}{(1+0.02)^{2.0}-1}$	3880
Available for dividends	E 714 5
Dividend rate = $\frac{7145(100)}{0.40(157500)}$	11.3 o/o

E. Block meter rate:

First block:

rate =
$$\frac{$18730}{12(10000)}$$
 = \$0.156/kwh for first 10 kwh

Second block:

rate =
$$\frac{$27855}{12(20000)}$$
 = \$0.116/kwh for next 20 kwh

Third block:

rate =
$$\frac{\$10803 + \$15750}{12[\frac{365(1653)}{12} - 30000]} = \$0.109/\text{kwh for excess over 30 kwh}$$

Example 5.

This example is the same as Example 4 except for the method of financing, which is typical of municipal practice. All of the capital is raised by the sale of revenue certificates which mature in 20 years and bear interest at 3.5 o/o and the sinking fund earns interest at 2 o/c. Profit of 10 o/o on the investment goes to defray town operating expenses. No taxes are paid, and insurance is 2 o/o.

Solution:

A. Fixed element:

ECONOMICS OF POWER PRODUCTION

	Investment in plant = 250(\$250) Investment in primary distribution system	\$ 62500 20000
	Annual depreciation of plant Annual depr. of pri. dist. sys. Insurance = 0.02(82500) Interest = 0.035(82500) Capitalized maintenance Management Sinking fund cont. = 0.0411(82500)	\$ 3750 1333 1650 2885 140 6000 3390
	Total annual fixed element	2 19148
В.	Energy element:	
	Total annual energy element	3 10803
c.	Customer element:	
	Investment in secondary distribution system	\$75000
	Depreciation of sec. dist. sys. Insurance = 0.02(75000) Interest = 0.035(75000) Maintenance Labor Cost of collection Sinking fund cont. = 0.0411(75000)	5310 1500 2625 720 12000 3000 3080
	Total annualcustomer element	E28235
D.	Profit element:	
	Frofit = 0.10(157500)	\$15750

E. Block meter rate:

First block:

rate =
$$\frac{$19148}{12(10000)}$$
 = \$0.159/kwh for first 10 kwh

Second Block:

rate =
$$\frac{$28235}{12(20000)}$$
 = \$0.118/kwh for next 20 kwh

Third block:

rate =
$$\frac{$10803 + $15750}{12(20300)}$$
 = $\frac{$0.109/\text{kwh}}{\text{over 30 kwh}}$

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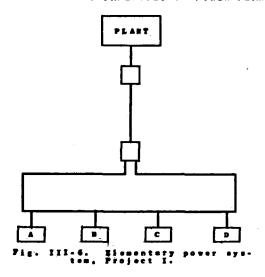
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PROJECT I

1. A certain town of 20,000 population is served by a private utility and has the equivalent of 6,000 customers, divided among three substations. Substation A serves a residential district; substation B serves a commercial district; and substation C serves an industrial district. Street lighting is supplied from substation D. The system (incomplete) is shown in Figure III-6. Connected loads and hours of use are as follows:

Class of Service	Connected Load kilowatts	Hours of Use
Substation A:		
Residences	7500	6 to 8 AM
		5 to 11 PM
Apartments	2500	6 to 8 AM
		5 to 11 PM
Substation B:		
Public buildings	250	9 AM to 5 PM
Churches	140	10 AM to 1 PM
		6 to 9 PM
Hospitals	4 50	all day
Theaters	250	2 to 11 PM
Laundries	100	7 AM to 5 PM
Retail stores	450	9 AM to 6 PM
Offices	300	9 AM to 5 PM
Newspapers	150	9 AM to 8 PM
Substation C:	•	
Factories	3000	7 AM to 5 PM
Machire shops	500	8 AM to 5 FM
Substation D:		-
Street lighting	500	6 PM to 6 AM

- (a) Find the probable maximum demand for each class of service.
- (b) Find the probable maximum demand on each substation.
- (c) Find the probable maximum demand on the power plant and plot a typical daily load curve for the plant.
- (d) Turbo-generator sizes available are: 500 kw, 750 kw, 1000 kw, 1500 kw, 2000 kw, 2500 kw and 3000 kw. Choose a unit or units which will give the maximum use and caracity factors



consistent with required reserves and leeway for growth, and set up an operating schedule for the plant.

(a) Estimate the cost of the plant and transmission system, assuming that the transmission system consists of approximately five miles of 13.2 kv line in flat country. Management cost is \$10000 per year and taxes and insurance together are 7 o/o. Depreciation of the investment is on the straight line basis. Sixty per cent of the investment is fin-

anced by the sale of bonds which mature in twenty years and bear interest at 3 o/o. The remainder of the investment is financed by the sale of common stock. What is the annual fixed cost?

- (b) If the turbo-generator heat rates (Btu per kwh net output) are estimated as: full load, 17500; 3/4 load, 17850; 1/2 load, 18650; and 1/4 load, 20500; fuel oil having 6,325,000 Btu per bbl, costing \$1.50 per bbl is ned; labor, \$2,000 per month; supplies, \$3,000 per year; and cost of water, \$1,000 per year, what is the annual energy element? (Assume that the typical day in part one is the average day.)
- (c) Estimate the cost of the secondary distribution system and the annual customer element if labor costs \$5,000 per month and taxes and insurance together are 7 o/o. The depreciation and financing of the investment is in the same manner as with the plant and primary distribution system.
- (d) If the annual gross profit is to be ten per cent of the investment, how much will be left for common stock dividends after setting up a sinking fund for bond retirement and paying income taxes? The sinking fund earns interest at two per cent and the income tax rate is 38 o/o.
- (e) Set up a block meter rate which will recover the fixed element in the first ten, the customer element in the next 25 and the energy and profit elements in the excess over 35 kwh per customer per month.

CHAPTER IV

VAPOR POWER CYCLES

21. Carnot Cycle--A cycle composed of completely reversible processes was proposed by Sadi Carnot, a French engineer, in 1824. The Carnot cycle is accepted as being the most efficient between any two temperature limits. The efficiency of the Carnot cycle can only be approached, however, because all real processes are irreversible. Figure IV-1 shows the Carnot cycle on the T-S and H-S planes for saturated steam and for superheated steam, both produced from saturated water.

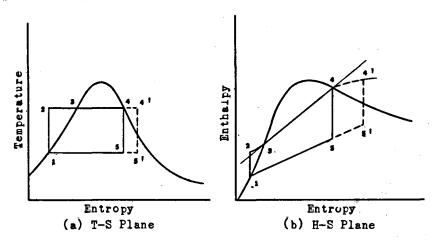


Fig. IV-1. Carnot cycle.

It will be noted that the Carnot cycle consists of two reversible isothermal and two reversible adiabatic processes. Examination of the cycle on the H-S plane will show that the characteristics of real vapors, such as steam, will not allow a practical realization of the Carnot cycle for the reason that the isothermal additions of heat from 2 to 3 and from 4 to 4' require that the pressure must decrease as the heat is being added.

I Since these figures are not to scale, considerable distortion exists, especially in the liquid region.

22. Rankine Cycle--The simplest vapor power cycle which has been proposed was conceived by Prof. W. J. M. Rankine, of the University of Glasgow, in 1849. This cycle is shown on the T-S and H-S planes in Figure IV-2, and will be seen to be a modification of the Carnot cycle to meet the requirements of real vapors. The expansion and compression lines

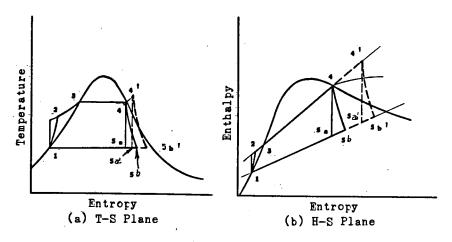


Fig. IV-2. Rankine cycle,

are shown to follow paths of increasing entropy because of the irreversibility of real processes. The ideal Rankine cycle would have these two processes follow paths of constant entropy.

A schematic diagram of the Rankine cycle is shown in Figure IV-3. The process 1-2 takes place in the pump and is one of subcooling in which the pressure is increased to that prevailing in the boiler. The process 2-3 takes place in the boiler and represents the constant pressure heating of water to the saturation temperature. The process 3-4 takes place in the boiler, also, and represents the production of steam from water at constant pressure. The steam at state 4 may be either wet or superheated. The process 4-5 represents the expansion in the engine, during which process work is done. The process 5-1 takes place in the condenser and is a constant pressure rejection of heat. It should be remembered that within the region of vapor mixtures (under the saturation line) a constant pressure process is identical with a constant temperature process.

The cycle represented in Figure IV-3 is closed, in that the same working fluid is used over and over. Often in

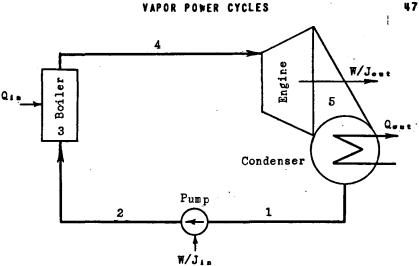
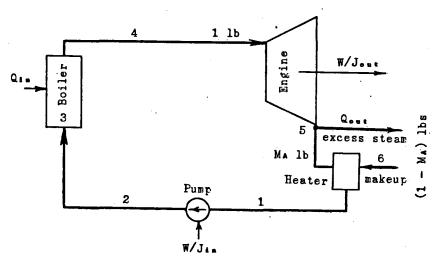


Fig. IV-3. Rankine cycle.

simple plants the cycle is open, that is, the condenser is absent and the steam is exhausted to the atmosphere at state 5. Sometimes part of the steam exhausted at state 5 in an open cycle is used to heat the feed-water going to the pump. This arrangement is shown in Figure IV-4.



The general energy equation of steady flow can be applied to each piece of equipment in the cycle and the various heat and work quantities determined, most conveniently on the basis of one pound of steam to the throttle of the engine.

For the closed Rankine cycle:

Pump:
$$1$$
 $W/J_{in} = H_2 - H_1$

Boiler:
$$Q_{in} = H_4 - H_2 = H_4 - H_1 - (H_2 - H_1)$$

Engine:
$$W/J_{out} = H_4 - H_{5b}$$

Condenser:
$$Q_{out} = H_{so} - H_1$$

The efficiency of the cycle will be the net output divided by the input, or

$$e_{t_c} = \frac{F_4 - F_{5,b} - (F_2 - F_1)}{F_4 - F_1 - (F_2 - F_1)}$$

For the open Rankine cycle with feed-water heating:

Pump:
$$W/J_{in} = H_2 - H_1$$

Boiler:
$$Q_{in} = H_4 - H_1 - (H_2 - H_1)$$

Engine:
$$W/J_{out} = H_4 - H_{sb}$$

The preceding relations are the same as for the closed Rankine cycle, but the fact that part of the exhaust steam is used for feed-water heating makes it necessary to introduce a weight term, M_A , to represent the amount of steam used for feed-water heating and to make an energy balance for the feed-water heater, as follows:

$$M_{A} (H_{6b} - H_{1}) = (1 - M_{A})(H_{1} - H_{6})$$

Por pressure rises in the pump of less than 300 lbs per sq in. the pump work terms may be neglected without appreciable error. For pressure rises greater than 300 lbs per sq in. the pump work can be determined from Table 4 of Thermodynamic Properties of Steem, by Keenen and Keyes.

$$M_{A} = \frac{H_1 - H_6}{H_{50} - H_6}$$

H, would ordinarily be unknown, but, from the conditions under which the cycle operates, it can be no greater than the enthalpy of saturated water at the exhaust pressure (atmospheric usually). Then as before,

$$e_{t_c} = \frac{F_4 - F_{5b} - (F_2 - F_1)}{F_4 - F_1 - (F_2 - F_1)}$$

This same sort of analysis could be applied to the open. Rankine cycle without feed-water heating. The advantage of using part of the exhaust steam for feed-water heating can be seen if the expression for thermal efficiency is examined. With feed-water heating H₁ will be greater, thereby causing a decrease in the value of the denominator with a resulting increase in efficiency. In addition, less makeup water would be required.

23. Engine Efficiency--The ratio of the actual work energy removed from a pound of steam in the engine to the amount that could be removed in an adiabatic expansion is known by several names: engine efficiency, Rankine cycle efficiency ratio, and Rankine cycle ratio. The preferred name, however, is engine efficiency. The mathematical relation (referring to Figure IV-2) is:

$$e_{e} = \frac{H_{4} - H_{5}_{h}}{H_{4} - H_{5}_{a}}$$

The engine efficiency can be determined directly for steam engines because it is possible to follow the path of the steam expansion with an engine indicator and to determine the power delivered to the piston. In this case,

$$e_{ig} = \frac{2545}{S_{ig}(H_{4} - H_{5g})}$$

where: S; = steam rate, lbs/indicated hp-hr

In the case of the steam turbine, it is difficult to determine the engine efficiency because the path of the expansion cannot be followed completely and the mechanical efficiency cannot be determined easily. Therefore, the term

everall engine efficiency, which is the product of engine efficiency and mechanical efficiency (and generator efficiency, if electrical output is measured) is employed. In this case,

$$e_{s_0} = \frac{2545}{S_{s_0}(H_4 - H_{s_0})}$$

where: S, = steam rate, lbs/shaft hp-hr

It should be pointed out here that the steam rate has significance in the above equations only for engines or turbines operating on the Rankine cycle where all of the steam passes completely through the engine or turbine. In other cases the steam rate is of value only in comparing similar cycles or for checking day to day performance of a particular machine.

24. Reheating Cycle -- This cycle was proposed by Ferranti as a means of improving the efficiency of the Rankine cycle for real engines. It is a characteristic of real engines and turbines that the engine efficiency is greater in the region of superheat than in the region of mixtures. The ideal reheating cycle is intended to cause a greater portion of the expansion to take place in the region of superheat where the enthalpy drop between any two pressures is greater than the enthalpy drop in the region of mixtures between the same two pressures. This difference is slight, however, and results in very insignificant gains in thermal efficiency for the ideal cycle. The opposite is true for the real reheating cycle, and very significant gains are made as a result of the improved engine efficiency. Figure IV-5 shows the reheating cycle on the T-S and H-S planes, and Figure IV-6 shows a schematic diagram of the cycle.

The reheating of the steam may be done in one of three ways: (a) Using steam from the boiler in a heat exchanger; (b) using a separate reheating furnace; or (c) using a reheater built into the boiler gas passages. The reheating cycle utilizing the first method of reheating has very little advantage, if any, over the Rankine cycle due to thermal losses in the reheating system. The second method has thermodynamic advantages over the first method, but requires considerable additional investment in fuel burning equipment, etc. The third method has both thermodynamic and economic advantages over the first method.

The cycle can be analyzed as follows:

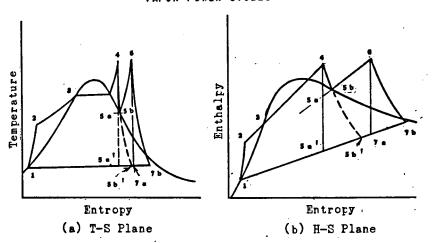


Fig. IV-S. Reheating cycle.

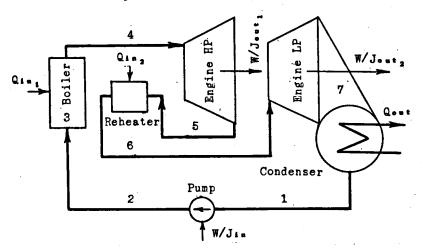


Fig. 17-6. Reheating cycle.

Pump:
$$W/J_{in} = P_{p} - H_{1}$$

Boiler: $Q_{in_{1}} = H_{4} - H_{1} - (H_{2} - H_{1})$
Engine H.P.: $W/J_{out_{1}} = H_{4} - H_{50}$
Reheater: $Q_{in_{2}} = H_{6} - H_{50}$

Engine L.P.:
$$W/J_{out_2} = H_6 - H_{7b}$$

Condenser: $Q_{out} = H_{7b} - H_1$
 $= H_4 - H_{5b} + H_6 - H_{7b} - (H_2 - H_1)$

$$\mathbf{e}_{t_{c}} = \frac{\mathbf{H}_{4} - \mathbf{H}_{5,b} + \mathbf{H}_{6} - \mathbf{H}_{7,b} - (\mathbf{E}_{2} - \mathbf{H}_{1})}{\mathbf{H}_{4} - \mathbf{H}_{1} + \mathbf{H}_{6} - \mathbf{H}_{5,b} - (\mathbf{E}_{2} - \mathbf{H}_{1})}$$

25. Regenerative Cycle-This cycle was proposed by Frof. J. H. Cotterill in 1890, and depends for its improvement in thermal efficiency over the Rankine cycle upon the use of part of the steam from the engine, after it has done some work, to heat feed-water going to the boiler. The regenerative cycle is similar to the open Rankine cycle with feed-water heating except for the addition of the condenser to carry the expansion down to lower temperatures and to close the cycle. Figure IV-7 shows the regenerative cycle on the T-S and H-S planes, and Figure IV-8 (a) shows a schematic diagram of the cycle with two mixing (or open) heaters to correspond with Figure IV-7. Figure IV-8 (b) shows the modification of the schematic diagram and the lower part of the H-S plane that is necessary when the low pressure open heater is replaced by a closed (or surface) heater.

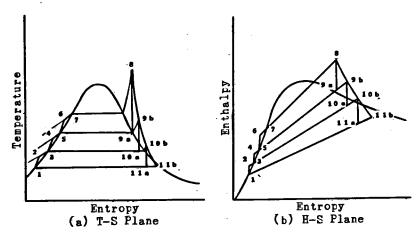


Fig. IV-7. Regenerative cycle.

R. L. Reynolds has given considerable data on the performance of regenerative cycles with various numbers of feedwater heaters, and Figure IV-9 shows some of his data which are used for the determination of the best final feed-water temperature. It will be noted that the curves in Figure IV-9 (a) are drawn for a standard throttle temperature of

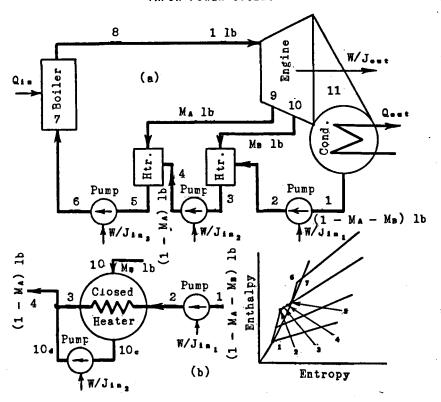


Fig. IV-S. Regenerative cycle.

750 F and a standard condenser vacuum of 29 inches of mercury. The effect of departure from these standards can be computed with the aid of the correction curves in Figures IV-9 (b) and IV-9 (c). It is customary to select the extraction pressures for feed-water heating so that the temperature rises in each heater are approximately equal. Ordinarily only one open heater is used in the mcdern regenerative cycle, so placed that the pressure in the heater is slightly above atmospheric at the minimum normal load.

The cycle can be analyzed as follows:

A. For two open heaters:

Pumps: $W/J_{in_2} = (H_2 - H_1)(1 - M_A - M_B)$

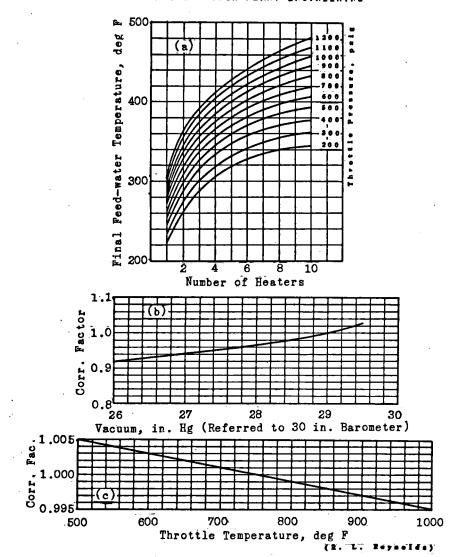


Fig., IV-9. Regenerative cycle feed-water temperatures.

$$W/J_{in_2} = (H_4 - H_5)(1 - M_4)$$

 $W/J_{in_3} = (H_6 - H_5)$

Boiler:
$$Q_{in} = E_{8} - H_{8} - (E_{6} - H_{8})$$
Engine:
$$W/J_{out} = H_{8} - H_{9b} + (1 - M_{A})(H_{9b} - H_{10b}) + (1 - M_{A} - M_{B})(H_{10b} - E_{11b})$$
Condenser:
$$Q_{out} = (1 - M_{A} - M_{B})(E_{11b} - H_{1})$$

The extraction quantities can be determined by the use of energy balances for each feed-water heater, as follows:

$$M_A[H_{90} - H_5] = [1 - M_A][H_5 - H_5 - (H_4 - H_5)]$$

 $M_R[H_{100} - H_5] = [1 - M_A - M_R][H_5 - H_1 - (H_2 - H_1)]$

The thermal efficiency is:

$$\mathbf{e}_{t_c} = \frac{\mathbf{W}/\mathbf{J}_{out} - \mathbf{W}/\mathbf{J}_{in_1} - \mathbf{W}/\mathbf{J}_{in_2} - \mathbf{W}/\mathbf{J}_{in_2}}{\mathbf{Q}_{in}}$$

B. For one high pressure open heater and one low pressure closed heater:

Pumps:
$$W/J_{iw_1} = (H_2 - H_1)(1 - M_R - M_B)$$

 $W/J_{in_2} = (H_{10d} - H_{10c})(M_B)$
 $W/J_{in_3} = (H_6 - H_5)$
Boiler: $Q_{in} = H_6 - H_5 - (H_6 - H_5)$
Engine: $W/J_{out} = H_6 - H_{00} + (1 - M_A)(H_{00} - H_{100}) + (1 - M_A - M_B)(H_{100} - H_{110})$
Condenser: $Q_{out} = (1 - M_A - M_B)(H_{110} - H_1)$

Extraction quantities:

$$H_{A}[H_{00} - H_{0}] = [1 - M_{A}][H_{0} - H_{4}]$$

$$H_{A} = \frac{[1 - M_{A} - M_{B}][H_{0}] + [M_{B}][H_{10c} + (H_{10d} - H_{10c}!)]}{[1 - M_{A}]}$$

$$M_{B}[H_{10c} - H_{10c}] = [1 - M_{A} - M_{B}][H_{0} - H_{1} - (H_{2} - H_{1})]$$

The addition of more stages of feed-water heating merely makes the mathematics more complicated. An energy balance for each feed-water heater and for each junction point where closed heater drip pumps discharge into the main condensate line will give sufficient equations to solve for all of the unknowns.

Modifications of the cycle are sometimes made in the minor details, such as disposal of the closed heater drips. Where closed heaters are placed in series, the drips may be cascaded counterflow to the condensate with a slight loss in thermodynamic efficiency, but with saving in installation and maintenance costs due to the elimination of several pumps. It is usual, however, to have at least one heater drip pump, located at the closed heater nearest to the condenser. Figure IV-10 shows the heat balance diagram for a modern high pressure regenerative cycle.

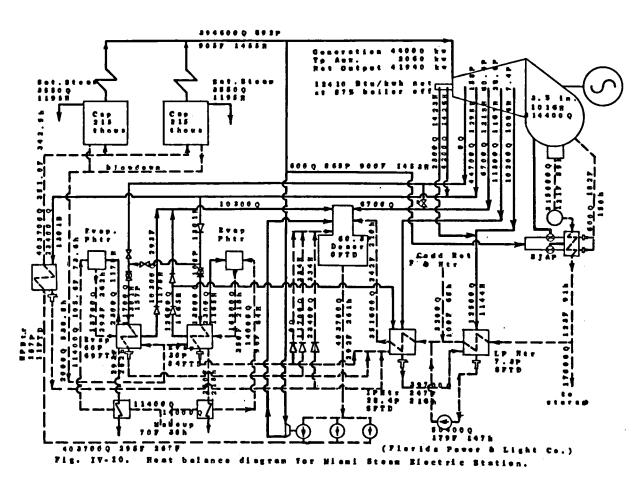
- 26. Reheating-Regenerative Cycle-The simple reheating cycle is no longer used in central station practice because greater efficiency gains can be realized by combining it with the regenerative cycle. In the last few years there have been several large stations built on this cycle using flue gas reheating. The reheating-regenerative cycle is generally considered to provide the ultimate improvement in efficiency over the Rankine cycle where only a single working fluid is concerned. The cycle analysis is made in the same manner as given for the two separate cycles. Figure IV-11 shows the heat balance diagram for a modern high pressure reheating-regenerative cycle.
- 27. Field Cycle--This cycle was proposed by J. F. Field, of the British Electric Authority, in 1948. The cycle eliminates the boiler and depends upon vapor compression to bring the steam back to the saturation condition before superheating. The cycle is shown on the T-S and H-S planes in Figure IV-12, and a schematic diagram is shown in Figure IV-13. The unavailable energy is rejected in a small condenser called a "calorifier," and the condensate is returned to the system through a spray nozzle. Field has also shown modifications of the cycle which include reheating and regeneration.

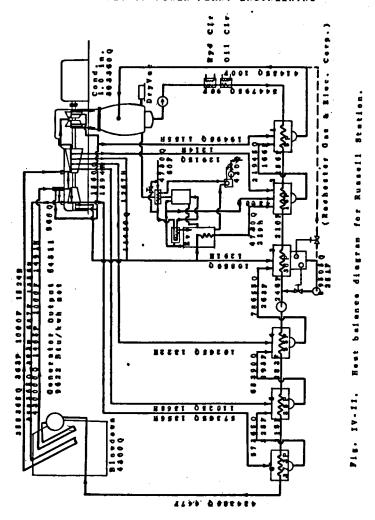
The simple cycle can be analyzed as follows:

Pump: $W/J_{in_1} = .(M)(H_6 - H_6)$

Compressor: $W/J_{in_2} = H_2 - H_{i_0}$

Superheater: $Q_{in} = H_s - H_s$





Engine:

$$W/J_{out} = H_s - H_{4b}$$

Calorifier flow:

$$[1 - M][H_{4b} - H_{1b}] = M[H_{1b} - H_{5} - (H_{6} - H_{5})]$$

$$e_{t_c} = \frac{E_s - E_{4b} - E_{2} - E_{1b} - M(E_s - E_s)}{E_s - E_s}$$

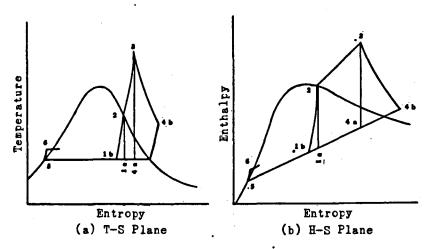


Fig. IV-12. Field cycle.

The theoretical advantages of this cycle are such as to make the efficiency much greater than that of the reheating-regenerative cycle because only a small portion of the steam flowing is condensed to water. The practical disadvantage, however, is the large amount of the output of the engine which is returned to the compressor. The cycle is similar to the gas turbine cycle in this respect (see Chapter XIX), and for all practical purposes is a closed gas turbine cycle using steam instead of air as the working fluid.

28. Superposed Cycle-The superposed (sometimes called "topping") cycle was developed during the depression of the 1930's to allow the addition of capacity without the necessity for building complete new plants. The cycle is a compound arrangement in which new high pressure boilers, replacing obsolete low pressure boilers, supply steam to a new high pressure turbo-generator which exhausts to the existing low pressure turbo-generators. The low pressure turbo-generators are rehabilitated to give additional useful life. This type of additional capacity can be installed for about one-half the cost of a complete new plant of the same capacity. The upper portion of the cycle is shown on the H-S plane in Figure IV-14, and a schematic diagram is shown in Figure IV-15. To date, superposition has been practiced only on regenerative cycles.

The cycle can be analyzed in the usual manner, remembering that the condition of the steam at state 10, (the exhaust of the high pressure turbine) must be identical with

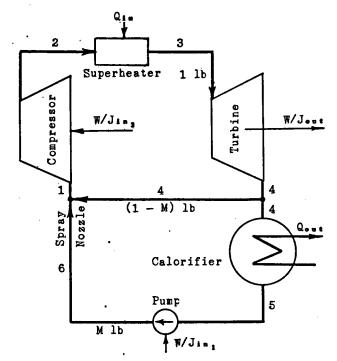


Fig. IV-13. Field cycle.

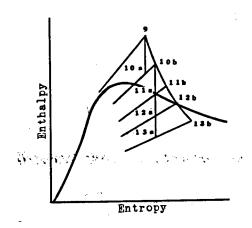


Fig. IV- 14. Superposed cycle

the specified conditions for the throttle of the old low pressure turbines. It will be seen that the high pressure turbine must pass large weights of steam, but the initial and final specific volumes are small enough to make the actual physical size of the high pressure turbine rather small in comparison with that of the low pressure turbines.

Perhaps the most important calculations in the superposed cycle have to do with the determination of the additional power produced by the high pressure turbine. The heat balance of the regenerative cycle upon which the high pressure system is superimposed is already known, and it is only necessary to consider the effect of the superposition upon the complete system, as follows:

Additional pump work:

$$W/J_{in_{add}} = (H_e - H_s)_{p_g} - (H_e - H_s)_{p_{10}}$$

Additional engine work:

$$W/J_{out_1} = H_0 - H_{100}$$

State 10, is known, but in order to determine state 9 the engine efficiency of the superposed turbine and the approximate amount of power required from it must be known. In general, a superposed turbine produces about one-third as much power as the low pressure turbines which it supplies with steam. Then, if M is the total steam flow in pounds per hour required for the low pressure turbines, and kw is the output of the low pressure turbines,

$$e_{e} = \frac{P_{e} - P_{1C_{e}}}{P_{e} - P_{1C_{e}}} = \frac{3413 (kw)}{3M (P_{e} - P_{1C_{e}})}$$

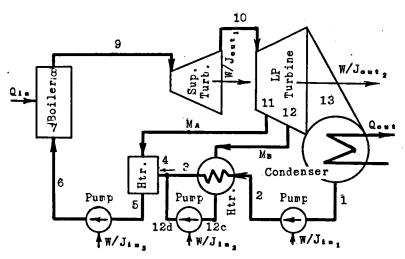
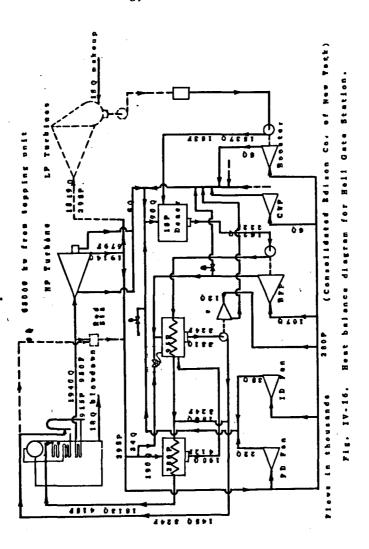


Fig. IV-15. Superpassd evelo

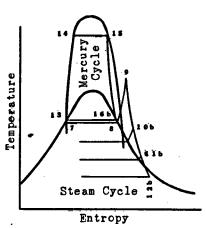
The two enthalpies, H, and H_{10a}, will serve to locate the pressure and temperature to the throttle of the superposed turbine, as a first approximation. A standard temperature and pressure nearest to the calculated requirement is selected and then the heat balance is adjusted accordingly. If steam is extracted from the superposed turbine for feed-water heating, the heat balance and throttle flow



are further adjusted. Figure IV-16 shows the heat balance diagram for a modern superposed installation.

29. Binary Cycles -- Some years ago it was realized that progress in increasing the efficiency of vapor power cycles was being retarded by the thermodynamic disadvantages of steam in the higher temperature ranges. That is, extremely high pressures are required for high initial temperatures so that the condition of the steam at the end of expansion in the engine will be within a range that can be economically handled. This limitation led to the development of the mercury-steam binary cycle. Mercury has the advantage of requiring much lower pressures for higher temperatures than does steam.

The mercury-steam binary cycle utilizes the latent heat of condensation of the mercury to generate steam. Figure IV-17 shows the cycle on the T-S plane, and Figure IV-18 shows a schematic diagram of



the cycle. Figure IV-19 is a Mollier diagram for mercury vapor.

The cycle can be analyzed in two parts on the basis of one pound of each fluid, and then combined into the binary cycle, as follows:

$$\frac{[\text{lbs Hg}]}{[\text{lbs H}_20]} = \frac{\text{H}_8 - \text{H}_5 - (\text{H}_6 - \text{H}_5)}{\text{H}_{16}_b - \text{H}_{18}}$$

$$W/J_{net_c} = W/J_{H_20_{net}} + W/J_{Hg} \frac{[\text{lbs Hg}]}{[\text{lbs H}_20]}$$

$$Q_{in_c} = Q_{in_1} + Q_{in_2} \frac{[\text{lbs Hg}]}{[\text{lbs H}_20]}$$

$$e_{t_c} = \frac{W/J_{net}}{Q_{in_c}}c$$

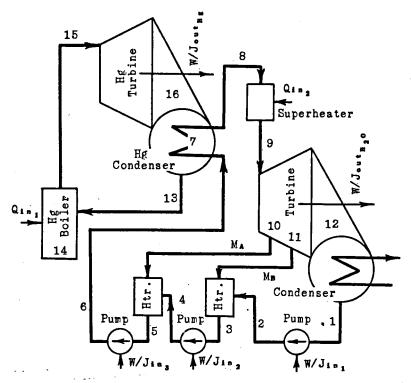


Fig. IV-18. Mercury-steam binary cycle.

An interesting feature of the mercury cycle is the absence of a mercury pump. The mercury turbire and condenser is placed on an elevation above the mercury boiler, and the difference in density between liquid and vapor is sufficient to feed the boiler. One problem that has slowed down the acceptance of the mercury-steam binary cycle is that of sealing in the highly toxic mercury vapors. However, this difficulty has been overcome satisfactorily and a complete plant is presently being built in New England.

Other binary cycles have been proposed which use steam as the upper fluid followed by sulphur dioxide, petroleum ether, butane or propane. The disadvantages of these cycles lie in the undesirable properties of steam at higher temperatures and in the toxicity and inflammability of some of the secondary fluids proposed. The advantage, however, is in the fact that both engines exhaust at pressures above atmospheric, thus eliminating the necessity for a vacuum pump

on the condenser. There is little, if any, gain in thermal efficiency over the straight steam cycle.

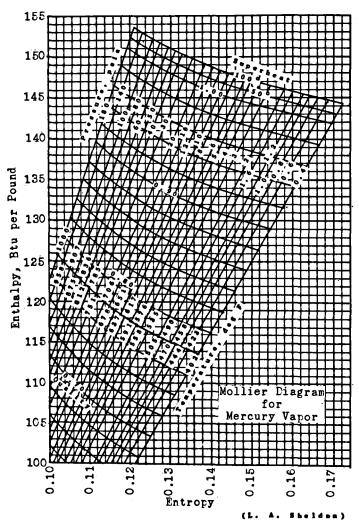


Fig. IV-19. Mollier diagram for mercury wangs.

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CHAPTER V

PRIME MOVERS

30. The only prime movers used in vapor power cycles are engines or turbines. The engine is a reciprocating device which utilizes both pressure and expansion energy of the steam, while the turbine is a rotating machine which utilizes only the expansion energy of the steam.

Steam engines in sizes below 500 hp are more efficient than steam turbines and find their greatest application where variable speed and exhaust steam for process use is required. Steam turbines have replaced steam engines in sizes over 500 hp, principally because of their compactness and higher efficiency.

31. Steem Engines.-The steam engine became a practical reality in the 1600's, and after improvements by Newcomen and Watt, and many others, reached the peak of its growth and use in the early 1900's. Engine sizes reached 25,000 hp on a single shaft with four cylinder triple expansion machines for driving battleships. As previously mentioned, however, the steam engine is finding its field limited to small sizes. It appears that the steam engine on the railroads is also doomed to obsolescence. The American Locomotive Company, which built steam locomotives for 100 years, recently announced that its entire production in the future would be confined to Diesel-electric equipment.

The principal types of engine remaining in use today are divided into two general classifications, counterflow and uniflow, depending upon the location of the main exhaust ports with relation to the stroke of the piston. The classifications are further subdivided as to valve types, whether slide, rotary or poppet. It can be said, in general, that the uniflow engine is more efficient than the counterflow engine because of reduced thermal losses resulting from the avoidance of alternate heating and cooling of the cylinder. Rotary and poppet valve engines are more efficient than slide valve engines. The rotary and poppet valves reduce losses by avoiding the throttling of steam at the intake and exhaust. Figures V-1 and V-2 show two common types of steam engine.

The mechanical efficiency of a steam engine can be determined directly if the output horsepower (usually called brake horsepower because some form of brake or dynamometer is used for its measurement) and input horsepower (called indicated horsepower because an engine indicator is used

for its determination) are known. Figure V-3 shows one type of brake used to determine engine output, and Figure V-4 shows an engine indicator with a diagram (or card) drawn by it. If the constants of the brake, engine and indicator are known, the following equations will give the required information:

$$Bhp = \frac{2\pi FLN}{33COO}$$

where: F = net force exerted by brake, lbs.

L = length of brake arm, ft.

N = speed of engine, rpm.

$$Ihp = \frac{P_{m_h} LA_h N}{33000} + \frac{P_{m_c} LA_c N}{33000}$$

where: P_{m_h} and P_{m_c} = mean effective pressures exerted on the head and crank ends of the piston, lbs /sq in.

L = length of piston stroke, ft.

Ah and Ac = effective piston area of head and crank ends, sq in.

N = speed of engine, rpm.

$$P_m = \frac{AdS}{Ld}$$

where: A_d = area of indicator diagram, sq in. L_d = length of indicator diagram, in. S = scale of indicator spring, lbs/sq in./in.

$$e_{m} = \frac{Bhp}{Ihp}$$

The mechanical efficiency of steam engines is high, and the shape of the efficiency vs load curve depends primarily upon the valve type, as can be seen in Figure V-5. Engire efficiencies are moderately high and depend primarily upon the flow direction within the cylinder, as shown in Figure V-6.

a

w i c h

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32. Estimation of Steam Engine Dimensions--In preliminary calculations for engine plants it is necessary to estimate the dimensions of the engine. This estimation is based on empirical data as to piston speeds and departure from the ideal expansion in the engine.

The approximate work obtained from the engine can be obtained from consideration of the ideal indicator diagram with the application of a diagram factor to allow for the effects of compression and the departure of the expansion from the adiabatic. Figure V-7 shows the ideal indicator diagram with a realized diagram superimposed. From the ideal diagram:

Net area = net work

$$= P_1V_1 + P_1V_1\log_{e}\frac{V_2}{V_1} - P_2V_2$$

V₂/V₃ is called the expansion ratio, R. It can be seen from the disgram that the larger the value of R, the larger the proportion of the total work done by the expansion of the steam. One method of governing the output of a steam engine is

to vary the expansion ratio. For purposes of estimation for full load. R is usually taken as being between three and four. Making the substitution for R,

$$\frac{\text{Net work}}{V_2} = P_m = [(P_1/R)(1 + \log_a R) - P_a]F_d$$

where: $P_m = \text{mean effective pressure, } 1\text{bs/sq in.}$

P₁ and P₂ = throttle and exhaust pressure, respectively, lbs/sq in abs.

R = expansion ratio

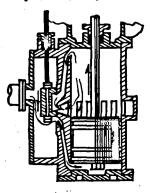
 F_d = a diagram factor depending on the type of engine (Table IV)

The engine dimensions are affected by the rotational speed and the allowable piston speed, as shown in Table IV.

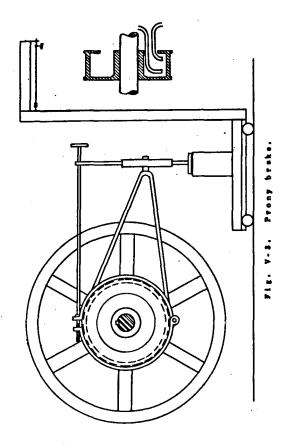
Example 6 shows the method of estimating steam engine dimensions by use of the preceeding information.

Example 6.

Determine the approximate dimensions of a single cylinder double-acting uniflow steam engine to develop 250 bhp at 200 rpm.







Steam to the throttle is at 150 psig and saturated. Exhaust is at 3 in. Hg abs.

Solution:

From Figure V-5, the mechanical efficiency is found to be about 90 o/o, and from Table IV, the piston speed and diagram factor are estimated to be 750 fpm and 0.80 respectively. Taking the expansion ratio as 4,

$$P_m = [\frac{164.7}{4}(1 + \log_e 4) - 1.47][0.80] = 77.5 \text{ psi}$$

Length of stroke = $\frac{750}{2(209)} = 1.875 \text{ ft} = 22.5 \text{ in.}$

$$I_{hp} = \frac{250}{0.90} = 278 = \frac{77.5(1.875)(200)(2)A}{33000}$$

Counterflow engines

Non-condensing

from which, A = 158 sq in. and d = 14.2 in.

The required engine dimensions are 221 in. X 141 in. X 200 rpm.

TABLE IV

DESIGN DATA FOR STEAM ENGINES

200 to 350 rpm

0.75 to 0.85

0.70 to 0.80

ROTATIONAL SPEEDS

Uniflow engines	100 to 300 rpm
PISTON SPEEDS	
Counterflow engines	600 to 450 fpm
Uniflow engines	900 to 600 fpm
DIAGRAM FACTORS	
Counterflow engines	0.70 to 0.85
Uniflow engines	01.0 00 0,60
Condensing	0.75 +0 0.95

Figure V-8 shows the approximate cost (based on 1936) of modern steam engine-generator sets.

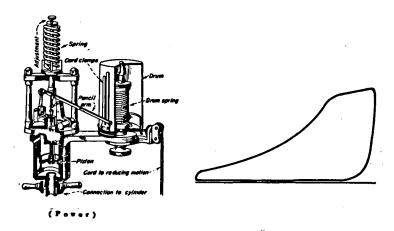
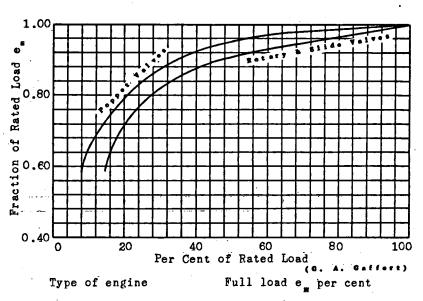


Fig. V-4. Engine indicator with diagram.



Four rotary valves 90 to 92 Uniflow 90 to 91 Slide valves (D or piston) 88 to 90

Fig. V-5. Mechanical efficiency of steam engines.

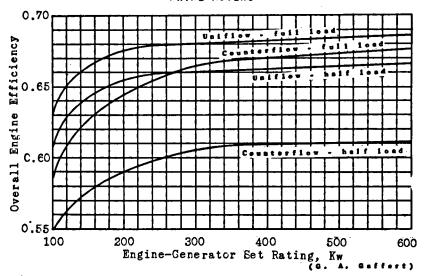


Fig. V. 6. Engine efficiency of steam engines.

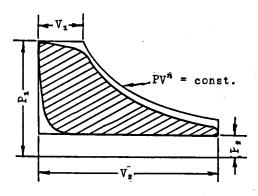


Fig. V-7. Ideal indicator diagram.

33. Steam Turbines -- The steam turbine as we know it today has been developed from the machine invented by DeLaval in 1882 to run his cream separator. DeLaval's machine was of the single stage impulse type having an extremely high rotational speed. C. A. Parsons, an Englishman, invented the multistage reaction turbine; Rateau, a Frenchman, the pressure compounded impulse turbine; and C. G. Curtis, an American, the velocity compounded impulse turbine, all of which make it possible to reduce rotational speeds to reasonable levels.

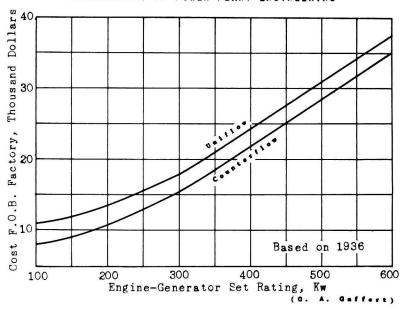


Fig. V-8. Cost of steam engine-generator sets.

The first commercial turbo-generator installations appeared around 1900, and were of the vertical type to conserve floor space. The largest vertical turbine, a 20,000



Fig. V-9. DeLav al impulse stage kw unit, was built in 1914, but shortly after its initial operation the horizontal turbine was introduced in order to eliminate troubles which had been experienced with thrust bearings in the vertical machines. By 1930 50,000 kw single cylinder machines had been built, but further increases in size required multi-cylinder cross or tandem compound arrangements. largest cross compound unit in existence is a three cylinder 1800 rpm machine at State Line Station at Hammond, Indiana. This machine has a rated capacity of 208,000 kw, and consists of one high pressure and two low pressure cylinders with separate generators and steam reheating between high and low pressure cylinders. Manufacturers now are confining their production of compound machines to the tandem type in capacities of from 40,000 to 100,000 kw and with speeds of 3600 rpm. The 3600 rpm machines are attractive because of their saving in space over the 1800 rpm machines.

All of the basic types of turbine are built in this country, and are more or less characteristic for the various

manufacturers. Allis-Chalmers produces the Parsons reaction turbine; Westing-house produces a combination machine, usually with one Curtis velocity compounded stage followed by Parsons stages; and General Electric makes what they call the Curtis turbine, but which at present is really a Rateau pressure compounded turbine, sometimes with an initial Curtis stage. DeLaval still builds the single impulse stage machine with reduction gearing. Figures V-9 through V-13 show the various types of staging, with pressure and velocity relations for the steam.



(Pewer)
Fig. V-10, Curtis
velocity compounded staging.

34. Engine Efficiencies and Willans' Line for Steam Turbines-Steam turbines for power plants are divided into two classifications: auxiliary or mechanical drive turbines, and turbines for generator drive.

In sizes from five to 2,000 hp and are used for driving fans, pumps, etc. They have from one to three velocity compounded stages and exhaust at a back pressure, ordinarily to the open heater in the feed-water system. They are not very efficient due to the small ratio of blade speed to jet velocity, but where process ateam or steam for feed-water heating is required they can often be justified for continuous service. In other cases they are used for emergency service. Figure V-14 shows the variation in overall engine efficiency (including mechanical



Fig. V-11. Parson reaction staging.

efficiency) for typical mechanical drive turbines.

Two ines for generator drive are made in sizes from 500 kw capacity upwards to a maximum of about 100,000 kw. All major manufacturers have adopted the A.S.M.E.-A.I.E.E. Standards for Turbo-Generators which are shown in Table V.

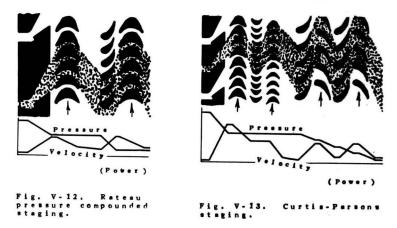


TABLE V

AIFE-ASME-STANDARDIZED TURBINE GENERATOR UNITS

3600 rpm, 60 cycles, 3 phase

	S	Single-casing, Single-flow Turbine				Tandem-compound, Double-flow Turbine			
		ir-cooled Hydrogen-cooled Generator							
	11,500 H	Kw	15,000 Kw	20,000 Kw	30,000 Kw	40,0	00 Kw	60,000	Kw
Throttle pressure, lb gage	600		850	850	850	850	1250	850	1250
Throttle temperature, deg F.			900	900	900	900	950	900	950
Exhaust pressure, in. Hg. abs			1.5	1.5	1.5		1.5		1.5
Turbine capability, kw		1	16,500	22,000	33,000	44,	000	66,0	00
Number of extraction opening		- 1	4	4	5		5	1	5
	lst 175		175	175	175		175		75
	2nd . 235	- 1	235	235	235		235		35
	3rd 285	- 1	285	285	285		285		85
	th. 350	- 1	350	350	350		350		50
service, deg F	5th —		_	_	410		410	1 4	10
Generator kva								l	
Air-cooled generator Hydrogen-cooled generator		-1	-	_	_		_	-	-
0.5 lb hydrogen pressure		lı	7,647	23,529	35,294	47.	058	70,5	88
15.0 lb hydrogen pressure					40,588		117	81,1	
Power factor	0.	85	0.85			٠,,	0.85	,-	0.85
Short-circuit ratio			0.8	0.8	0.8		0.8	L	0.8

Figures V-15 and V-16 show the approximate overall engine efficiencies (including mechanical and generator efficiencies) to be expected from small turbo-generators, both condensing and non-condensing.

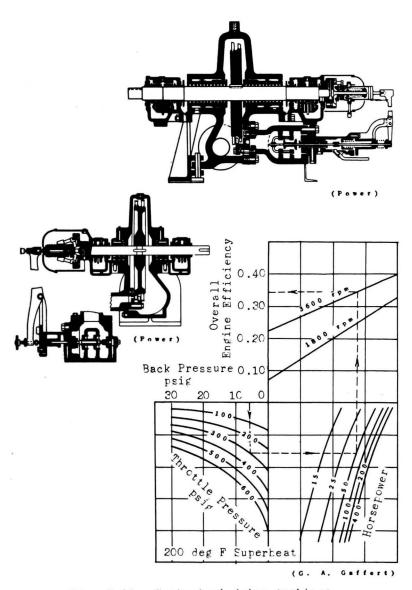


Fig. V-14. Mechanical drive turbines.

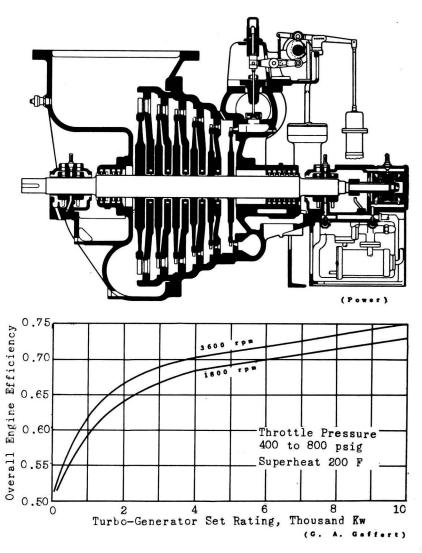


Fig. V-15. Small condensing turbo-venerators.

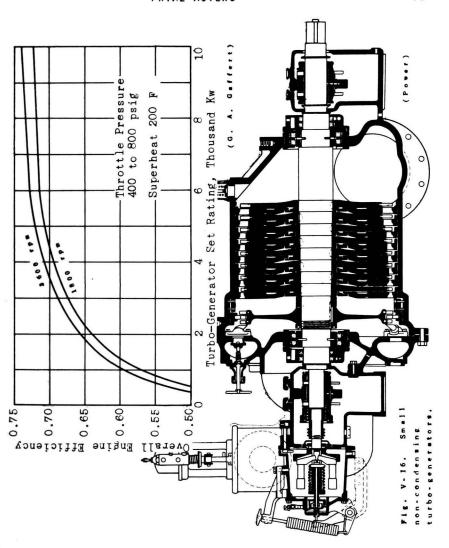
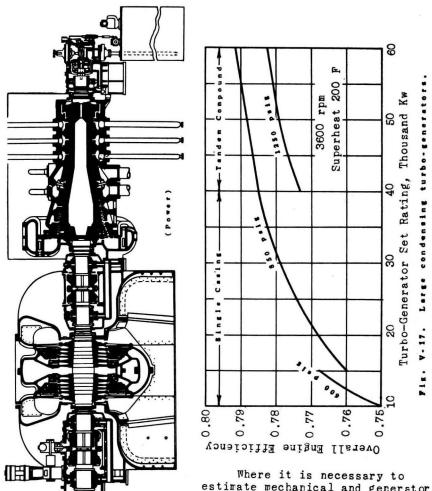


Figure V-17 shows overall engine efficiency data for large condensing turbo-generators; Figure V-18 shows data for correcting engine efficiencies for part load operation; and Figure V-19 shows some turbo-generator cost data (based on 1945).



Where it is necessary to estimate mechanical and generator efficiencies in order to predict the turbine expansion line, the

following empirical formulae will be useful:

For 1800 rpm machines: $e_m = 1.00 - \frac{0.026(Rating)}{\sqrt{Rated \, kw/1000} \, (Load)}$

$$e_g = 0.9885 - \frac{0.0685(Rating)}{\sqrt{Rated \, kw/1000 \, (Load)}}$$

$$e_m = 1.00 - \frac{0.04(Rating)}{\sqrt{Rated \ kw/1000} \ (Load)}$$

$$e_g = 0.96 - \frac{0.019(Rating)}{\sqrt[9]{Rated kw/1000}}$$
 (Load)

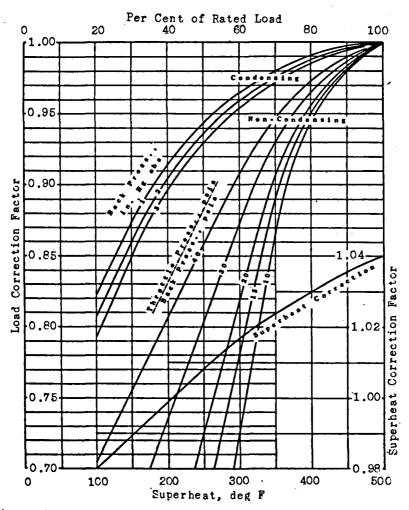


Fig. V-18. Engine efficiency correction fectors for turbines,

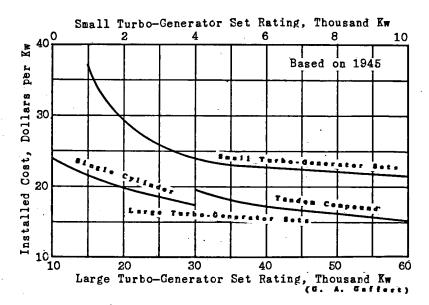
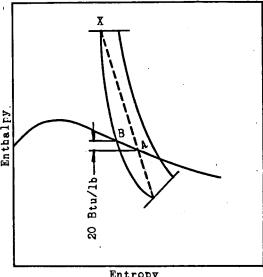


Fig. V-19. Turbo-generator cost data.

The likeness of a throttle-governed turbine to an orifice having the back pressure less than the critical makes it possible to predict extraction pressures at any partial load if they are known for any one load. It can be assumed, with very slight error, that the back pressure remains constant, and that the first stage and extraction absolute pressures will vary directly with load. With this relation it is possible to estimate part-load performance of a turbine, providing the expansion lines are known. The full load expansion line can be estimated quite closely, if the initial and final conditions are known, by referring to Figure V-20. Connect the initial and final steam conditions on the Mollier diagram by a straight line; move along the saturation line to point B which is 20 Btu per pound greater than point A, the intersection of the straight line with the saturation line. Connect the initial and final conditions with a smooth curve through point B so that the slope decreases gradually from the initial to the final conditions. The extraction conditions can then be located at the intersections of the pressure lines with the expansion lines. For part-load expansion lines, the curve will be of the same shape from the initial condition but with point X displaced to the right. A ship curve is excellent for drawing expansion lines.

The curve of total steam flow vs output for a turbine is a straight line (called Willans' line), and can be estimated from engine, mechanical and generator efficiency data. The slope of the line is constant only so long as a single throttle valve is employed. As additional valves open the slope becomes steeper. Example 7 shows the method of estimating Willans' line and the expansion lines for a single valve turbine.



Entropy

Example 7.

Fig. V-20. Esti Estimating turbine

 Δ 2500 kw turbo-generator operates at 3600 rpm with steam to the throttle at 400 psig and 750 F and exhausts at 2 in. Hg abs. Estimate the non-extraction Willans' line.

Solution:

From Figure V-15, the full load overall engine efficiency is estimated to be 0.68. From Figure V-18, the superheat correction factor is found to be 1.017 (superheat 205F) and the following table is made:

Load	Eng. eff. full load	supht.	load .corr.	corr. eng. eff.
Full	0.68	1.017	1.000	0.69
0.75	0.68	1.017	0.975	0:67
0.50	0.68	1.017	ó.940	0.65
0.25	0.68	1.017	0.820	0.57

The expansion lines are drawn as in Figure V-21, and from these lines the work output per pound of steam can be determined as: full load, 320 Btu; 0.75 load, 298 Btu; 0.50 load, 274 Btu; and 0.25 load, 216 Btu. From these work out-

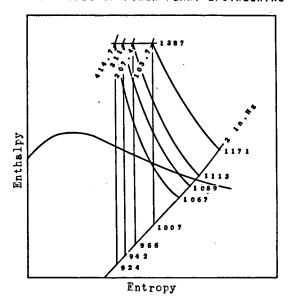


Fig. V-21. Turbine expansion lines for Example 7.

puts, the steam flows to the throttle at the various loads can be computed from the relation:

Throttle flow, lbs/hr =
$$\frac{3413(2500)(1cad frac.)}{Work, Btu/lb}$$

Flows from this relation are: Full load, 26700 lbs/hr; 0.75 load, 21450 lbs/hr; 0.50 load, 15600 lbs/hr; and 0.25 load, 9880 lbs/hr. Willans' line is plotted in Figure V-22.

Note: The expansion lines in Figure V-21 are "apparent" or based on the overall engine efficiencies. For the true expansion lines which give the actual enthalpies of the steam to the condenser, see Example 8.

Example 8 shows the method of calculating the approximate heat balance for a regenerative cycle using the steam turbine of Example 7.

Example 8.

The turbo-generator of Example 7 is placed in a regenerative cycle. One closed heater is used with the extraction point in the turbire located so that the extraction pressure

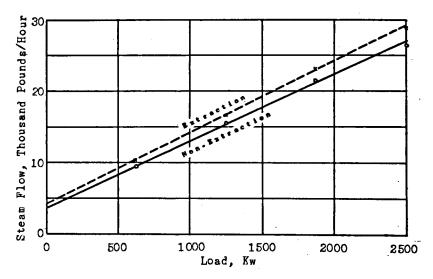


Fig. V-22. Willens' lines for Examples 7 and 8.

is 5 psia at 0.25 load. One open heater is placed in the cycle so that the extraction pressure is 16 psia at 0.25 load. The arrangement is similar to that in Figure IV-8(6). Neglect pump work. Make a heat balance and estimate the extraction Willans' line.

Solution:

The expansion lines and engine efficiencies of Example 7 were on the overall basis. In order to estimate the extraction performance it is necessary to estimate the mechanical and generator efficiencies and to replot the expansion lines.

$$e_m = 1.00 - \left[\frac{0.04(\text{rating})}{\sqrt{2500/1000} \text{ (load)}} \right]$$

$$e_g = 0.96 - \left[\frac{0.019(\text{rating})}{\sqrt[3]{2500/1000} \text{ (load)}} \right]$$
Load overall mech. gen. internal eng. eff. eff. eff. eng. eff.

Full C.69 0.975 0.946 0.749
0.75 C.67 0.966 0.941 0.737

Load	overall eng. eff.	mech. eff.	gen. eff.	internal eng. eff.
0.50	0.65	0.949	0.932	0.735
0.25	0.57	0.899	0.904	0.702

The expansion lines are plotted in Figure V-23 and the extraction enthalpies determined, remembering that the extraction pressures are directly proportional to load. The following table is set up to aid in the calculation of extraction quantities:

Load	H,	H ₁₀ c	H _s	H s	He	H'96	H ₁₀	H ₁₁₀
Full	69.0	196.2	190.2	266.4	1387	1224	1150	1040
0.75	69.0	181.1	175.1	247.5	1387	1240	1169	1059
0.50	69.0	161.2	155.2	222.6	1387	1245	1175	1077
0.25	69.0	130.1	124.1	184.4	1387	1262	1195	1120

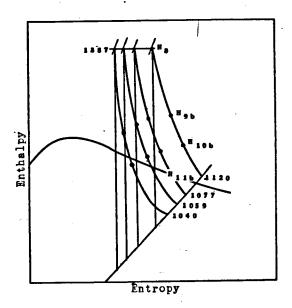


Fig. V-28. Turbine expansion lines for Example 8

It should be noted that it was assumed in setting up the preceding table that H₃ is 6 Btu/lb less than H₁₀ due to the use of a closed feed-water heater. Also, in order to

avoid the necessity of solving for H_4 , the drip from the closed heater is assumed to be pumped directly to the open heater.

The extraction quantities, M_A and M_B , can be found by use of the equations in Section 25, as follows:

Full load:

$$M_A(1224 - 266.4) = (1 - M_A - M_B)(266.4 - 190.2) + M_B(266.4 - 196.2)$$

$$M_B(1150 - 196.2) = (1 - M_A - M_B)(190.2 - 69.0)$$

Work output to the turbine shaft per pound of steam to the throttle is then calculated, as follows:

Full load:

$$W/J = (1387 - 1224) + 0.927(1224 - 1150) + 0.823(1150 - 1040)$$

Throttle flows are:

W	3413(2500)	_	00756	
Full load:	322.0(0.975)(0.946)	_	20750	lbs/hr
0.75 load			23100	
0.50 load	•		16500	
0.25 load			10260	

Extraction and condenser flows are:

Load	open htr. lbs/hr	closed htr. lbs/hr	conden- ser lbs/hr
Full	2100	3000	23700
0.75	1570	2290	19240

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Load	open htr. lbs/hr	closed htr. lbs/hr	conden- ser lbs/hr
0.50	1027	1225	14298
0.25	545	481	9234

The Willans' line for extraction operation is plotted in Figure V-22.

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CHAPTER VI

COMPENSERS. FEED-WATER HEATERS AND EVAPORATORS

35. Condensers--The purpose of the condenser in a vapor cycle is primarily to lower the temperature of the prime mover exhaust so that the thermal efficiency of the cycle can be increased. Due to the nature of steam, operation of the condenser at a temperature less than 212 F requires that the pressure within the condenser be less than atmospheric. Operation of condensers at sub-atmospheric pressures requires the use of vacuum pumps to remove the air which unavoidably leaks into the condenser and which, if not removed, would make it impossible to maintain the low exhaust temperature.

Condensers may be divided into two general classes: mixing condensers and surface condensers. The most important type is the surface condenser which allows the reduction of exhaust temperature while at the same time allowing the recovery of the condensate. The mixing condenser is able only to lower the exhaust temperature without recovering the condensate.

36. Mixing Condensers -- Mixing condensers may be either of the jet type or of the barometric type. The jet type depends upon the aspirating action of a jet of water to condense the steam and to remove the air from the condenser. A typical jet condenser is shown in Figure VI-1. The use of the jet for aspirating air requires more pump power for a given vacuum, but a separate vacuum pump is not required. The barometric condenser utilizes a leg of water to provide the seal for the condenser rather than power from a pump as in the jet condenser. The barometric condenser is always provided with a vacuum pump for air removal. A typical barometric condenser is shown in Figure VI-2. With 70 F cooling water, about the best vacuur which can be maintained with a mixing condenser is 28 inches of mercury. Due to the fact that the volume of the condenser is small in relation to the amount of steam and water handled, the mixing process does not reach equilibrium and the temperature of the mixture leaving the condenser will be about 5 F less than the saturation temperature corresponding to the pressure in the condenser. following equations can be written for the mixing type condenser, considering it a steady flow device with no shaft work:

 $M_{x}(H_{m}-H_{x})=M_{v}(H_{v}-H_{m})$

where: M = weight of condensing water, lbs/hr.

M, = weight of steam, lbs/hr.

H = enthalpy of mixture leaving the

condenser, Btu/lb.

H_ν = enthalpy of steam entering the condenser, Btu/lb.

H. = enthalpy of cooling water, Btu/lb.

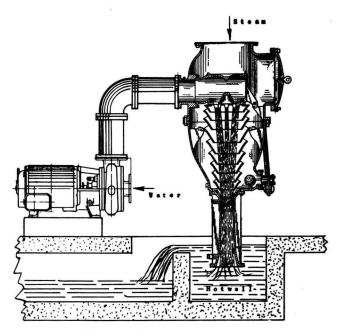


Fig. VI-1. Jet condenser.

37. Surface Condensers-The surface condenser has the advantage of making it possible to save the condensed steam for use as boiler feed, but keeping the condensing steam separate from the cooling water requires that the heat be transferred through a dividing wall. This necessity complicates the design of condensers by making it necessary to take into account the resistance of the dividing wall to heat flow. Two methods of calculating condenser size are available: calculating from the known and estimated resistances to heat flow

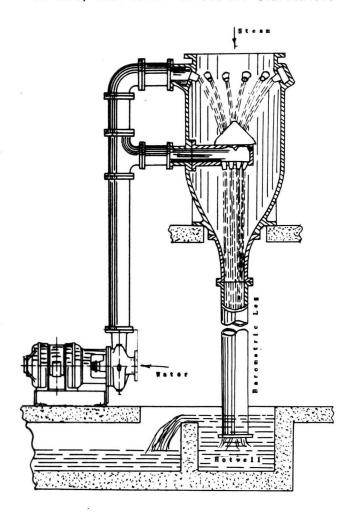
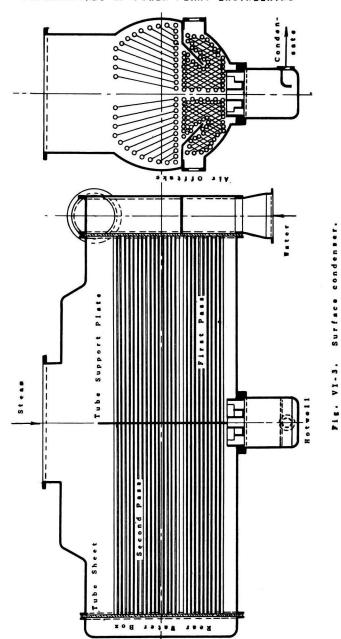


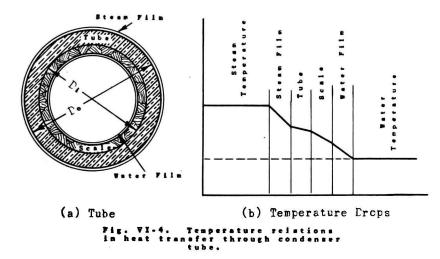
Fig. VI-2. Becometric condenser.

of the various elements in the heat flow path; and estimating from empirical data of the Heat Excharge Institute.

Surface condensers will be made in as many shapes and designs as there are manufacturers, the only similarity being in the existence of water boxes, tubes, a shell and a hotwell. Figure VI-3 shows a typical surface condenser with its various parts.



38. Heat Transfer in Surface Condensers--Heat transfer in surface condensers is accomplieded entirely by conduction, and consideration of the process will reveal that the heat must pass first through a film of water formed by the condensing steam, then in succession through the tube, scale from compounds and dirt deposited by the cooling water and finally through the boundary water film on the cooling water side of the tube. Figure VI-4 shows the heat paths in relation to a conventional condenser tube in which the tube conveys the cooling water and is surrounded by the steam.



The outside and inside diameters of the tube are the only definitely measureable distances in the heat path. It can be seen that the heat flow path is similar to series resistances in an electrical circuit, that is:

If the individual resistances to heat flow are defined in terms of the thickness and area of each element,

$$R_t = \frac{1}{K_t S_t}$$
, etc.

where: R = resistance,(hr)(deg F)(sq ft)/Btu

l = thickness in feet

K = thermal conductivity,

Btu/(hr)(sq.ft)(deg F)(ft)

S = area, sq ft

and if 1/R is called the overall heat transfer coefficient U, and S allowed to be one square foot,

$$U = \frac{1}{\frac{1}{K_{v}} + \frac{1}{K_{t}} + \frac{1}{K_{s}} + \frac{1}{K_{u}}}$$

As previously mentioned, the only readily determined dimensions are the inside and outside diameters of the tube. Furthermore, the area of condensing surface is always measured on the outside diameter of the tube. For these reasons all areas are corrected to the outside area of the tube, so that if all elements in the heat flow path except the tube are considered to have no thickness,

$$U = \frac{1}{\frac{1}{K_{v}} + \frac{(2D_{o})}{D_{o} + 0} \frac{(1t)}{D_{o}(K_{t})} + \frac{D_{o}}{D_{o}(K_{s})} \frac{1}{K_{s}} + \frac{D_{o}}{D_{o}(K_{s})} \frac{1}{K_{s}}}$$

where: D_o and D_i are the outside and inside diameters of the tube

It is not possible to determine 1/K for any resistance except the tube wall, so empirical formulae based on rational concepts of heat flow are employed. Substitution of 1/h is made for the vapor film, scale and water film, so that,

$$U = \frac{1}{\frac{1}{h_{v}} + \frac{2D_{o}}{D_{o} + D_{i}} \frac{(1_{i})}{(K_{i})} + \frac{D_{o}}{D_{i}} \frac{1}{h_{s}} + \frac{D_{o}}{D_{i}} \frac{1}{h_{s}}}$$

The value of h for the vapor film is dependent principally on the amount of air dissolved in the film. Nusselt and others have found that this value varies between 1500 and 3000, and in the absence of better data a value of 2000

may be used.

Table VI shows approximate values of K for several materials which are commonly used for heat transfer surfaces.

TABLE VI '

THERMAL CONFUCTIVITY OF METALS Bu per hr(sq ft)(deg F)(ft)

Pure copper	220
Aluminum	118
Brasses (Admiralty, Muntz, etc.)	58
Steel	30

The amount of scale present in tubes is highly variable, depending upon the type of water and the season of the year, among other factors. For purposes of estimation, h for the scale may be taken as ranging between 4000 for favorable conditions and 2000 for unfavorable conditions, with an average of about 3000.

The value of h for the water film may be calculated from Nusselt's equation, with water velocity between 6 and 9 feet per second, as follows:

$$h_{R} = \frac{93.6(1 + 0.01 t)V^{0.8}}{V^{0.12}}$$

Table VII gives the most useful size constants for tubes of various thicknesses and diameters.

After having found the value of U for the condenser, the total surface required may be found from the relation,

$$A = \frac{Q}{Ut_{\bullet}}$$

where: A = total area, sq ft Q = heat transferred, Btu/hr U = overall heat transfer
 coefficient
t_m = mean temperature difference
 between steam and water, deg F

TABLE VII
STANFARE CONFENSER TUBE DATA

OD in.	Thi BWG	ickness in.	ID in.	Outside surface sq ft/ft	Inside sectional area sq ft
5/8	16	C.065	0.495	0.1636	0.00134
	18	0.049	0.527	0.1636	0.00152
	20	0.035	0.555	0.1636	0.00168
3/4	14	0.083	0.584	0.1963	0.00186
	16	0.065	0.620	0.1962	0.00210
	18	0.049	0.652	0.1963	0.00232
7/8	14	0.083	0.709	0.2291	0.00274
	16	0.065	0.745	0.2291	0.00303
	18	0.049	0.777	0.2291	0.00329
1	14	0.083	0.834	0.2618	0.00379
	16	0.065	0.870	0.2618	0.00413
	18	0.049	0.902	0.2618	0.00444

The value of the mean temperature difference is somewhat different from the arithmetic mean temperature difference due to the fact that the temperature rise of the water is not a straight line function of the distance traversed by it. Figure VI-5 illustrates this point. It can be shown that

$$t_{m} = \frac{t_{m_0} - t_{m_i}}{\log s_{t_v} - t_{m_i}}$$

where: t_m = mean temperature difference between steam and water, deg F t_v = temperature of steam, deg F
t_z and t_z = inlet and outlet cooling
water temperatures, deg F

It is necessary to have a temperature difference between the steam and the outlet cooling water in order to maintain heat flow. This difference in temperature is called the terminal difference, and is usually from 5 to 10 F for surface condensers, the smaller applying to two pass and the larger to single pass designs. Refrig-

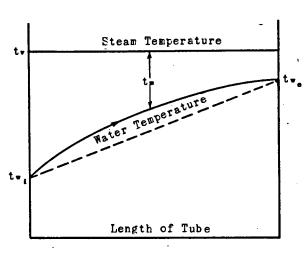


Fig. VI-5. Temperature relations in heat trenufer through condenser tube.

eration or reheating (according to the design of the condenser) is encountered with the water leaving the hotwell. This action will have to be taken into account in the computation of the heat transferred, but it must be ignored in the computation of the mean temperature difference because of the smallness of the effect of removal of sensible heat in the condenser. In well designed condensers refrigeration will not exceed 5 F, and some designs will actually reheat as much as 2 F.

Example 9 shows the method of calculating the principal dimensions of a surface condenser for the regenerative cycle of Example 8.

Example 9.

Estimate the principal dimensions of a surface condenser to serve the turbine of Example 8. Inlet cooling water temperature 72 F, terminal difference 8 F. Use 3/4" X 18 BWG aluminum brass tubes with a water velocity of 8 fps. Assume h = 3000.

Solution:

From Example 8, the condenser flow at full load is 23700 lbs per hour, and the heat removed in the condenser is (1040-69)=971 Btu/lb. The temperature of the steam in the condenser is 101 F.

$$Q = 23700(971) = 23,000,000 Btu/hr$$

$$t_{m} = \frac{(101 - 8) - 72}{\log_{e} \frac{101 - 72}{101 - (101 - 8)}} = \frac{21}{\log_{e} \frac{29}{8}} = 16.3 \text{ F}$$

$$h_{R} = \frac{93.6[1 + 0.01(82.5)][8]^{\circ ..8}}{[0.652/12]^{\circ ..2}} = 1615$$

$$U = \frac{1}{\frac{1}{2000} + [\frac{1.5}{0.75 + 0.652}][\frac{0.049}{12(58)}] + \frac{0.75}{0.652}[\frac{1}{3000}] + \frac{0.75}{0.652}[\frac{1}{4615}]}$$

$$v = \frac{1}{0.00167} = 599$$

$$A = \frac{23,000,000}{599(16.3)} = 2360 \text{ sq ft}$$

Water required = $\frac{23,000,000}{21(60)(8.33)}$ = 2190 gal/min = 4.88 cu ft/sec

Flow area =
$$\frac{4.88}{8}$$
 = 0.61 sq ft

Number of tubes = $\frac{0.61}{0.00232}$ = 264

Length of tubes =
$$\frac{2360}{264(0.1963)}$$
 = 45.5 ft

The condenser would be too long for a single pass, so two passes, each 22!-9 long, are specified.

The principal dimensions of the condenser are:

Area 2360 sq ft
Tube material 3/4" X 18 BWG Alum. brass
Number of passes 2
Active tube length 22'-9"
Heat duty 23 million Btu/hr

39. Commercial Heat Transfer Standards for Surface Condensers-Several of the large manufacturers of heat transfer equipment have cooperated in setting up the Heat Exchange Institute with the purpose of standardizing design data for this type of equipment. Figure VI-6 shows the curves in use for surface condensers. These curves are drawn for tubes

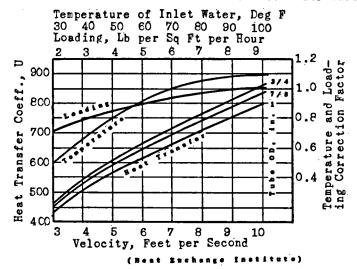


Fig. VI-6. Heat Exchange Institute design data for surface condensers.

with no scale on the water side. In use, the overall heat transfer coefficient from the curves is reduced by a so-called "cleanliness factor," which is usually taken as about 0.80. Two other correction factors, for loading and inlet water temperature, are used in addition to the cleanliness factor. In commercial practice loading ranges from 7 to 10 pounds of steam per square foot per hour. Example 10 shows the method of application of the Heat Exchange Institute data to the determination of the dimensions of a surface condenser for the same duty as in Example 9.

Example 10.

Estimate the principal dimensions of the condenser of Example 9 by means of the Heat Exchange Institute data. Take the clearliness factor as 0.85.

Solution:

From Figure VI-6, U = 765, temperature corr. = 1.01, loading corr. = 1.0.

Corrected U = 765(1.01)(1.0)(0.85) = 657

$$A = \frac{23,000,000}{657(16.3)} = 2150 \text{ sq ft}$$

Cooling water flow and number of tubes would be the same as in Example 9.

Length of tubes =
$$\frac{2150}{264(0.1963)(2)}$$
 = 20.7 ft

The principal dimensions of the condenser are:

Area 2150 sq ft.
Tube material 3/4" X 18 BWG Alum. brass
Number of passes 2
Active tube length 20'-9"
Heat duty 23 million Btu/hr

These calculations show that the method of Example 9 is more conservative than the method of the Heat Exchange Institute.

- 40. Feed-Water Heaters.—The primary function of feed-water heaters is to utilize extracted or exhaust steam from prime movers to increase the temperature of water before feeding it to boilers. Feed-water heaters may be divided into two general classes: open (or mixing) and closed (or surface) heaters. Feed-water heaters are counterparts of condensers, the difference being in their smaller size and higher operating temperature.
- 41. Open Heaters--The name "open heater" often leads to a misconception of the operation of this type of heater. Actually, an open heater can operate at any pressure, not necessarily atmospheric. All modern regenerative cycles include one open heater in the feed-water circuit, so designed as to act as a scrubber for deaerating the feed-water before it enters the boiler. Deaeration is necessary to remove oxygen and other gases which are harmful to boilers and piping, and which have entered the system by leaking into the condenser. There are a number of types of deaerating open heaters in use, but the most common is the tray type shown in Figure VI-7. It is quite common to provide integral or connected

storage for deaerated feed-water, as shown in Figure VI-2. This storage is usually about a two or three minute supply at full load. Often additional surge tanks are connected through overflows and surge pumps with the water storage section of the deaerating heater to provide additional emergency storage for unusual conditions.

The same considerations of heat balance apply to open heaters as to mixing condensers. Well proportioned deaerating heaters will produce water with almost immeasureable quantities of dissolved gases and

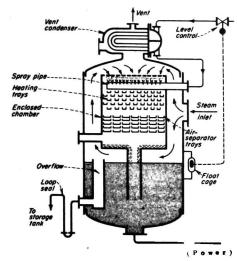


Fig. VI-7. Tray type deaerating

with practically zero terminal difference.

42. Closed Heaters--Due to the fact that the water is separated from the steam in a closed heater, it is possible to operate them at high water pressures in relation to the steam pressure. Figure VI-9 shows a section of a typical closed feed-water heater.

The same considerations of heat transfer apply to close feed-water heaters as to surface condensers. Because feed-water heaters operate with relatively clean tubes, the value of h for scale can be taken to be about 6000. Following are the terminal differences to be expected in good heater designs:

Temperature of	Terminal
exit feed-water	difference
deg F	deg F
86 - 230	5
230 - 300	10
300 - 400	15
400 - 525	20

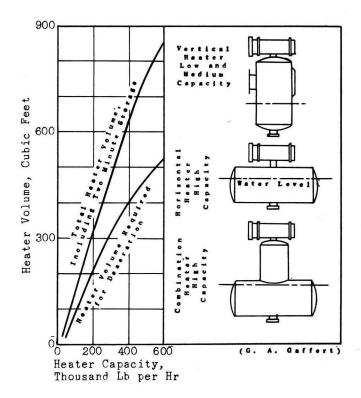


Fig. VI-8. Design data for deserating feed-water heaters

Refrigeration in the hotwell of horizontal heaters will be of about the same magnitude as for surface condensers, but will be from 1C to 15 F for vertical heaters. Water velocities are somewhat smaller than for surface condensers,

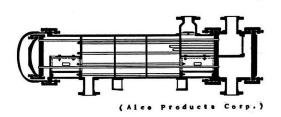
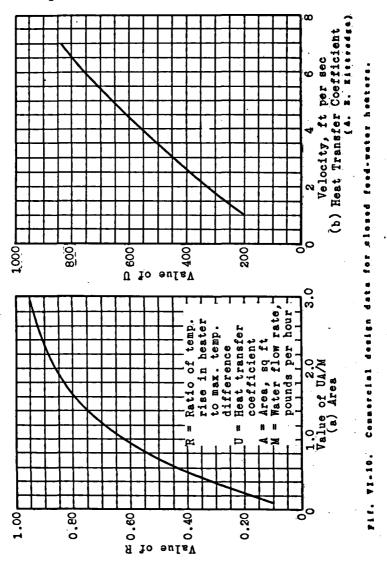


Fig. VI-9. Closed feed-water heater

ranging from 3 to 8
feet per second. The
drainage from closed
heaters is usually
flashed to the next
lower heater in the
circuit, while the
drainage from the
first closed heater
may be either pumped
to the condensate
line before the

deaerating heater or flashed to the hotwell of the condenser. The heater air offtakes are connected to the condenser through flow limiting orifices.

Commercial closed heaters are designed using Kittredge's data from Figure VI-10.



Examples 11 and 12 show the method of computing the principal dimensions of a closed feed-water for the regenerative cycle of Example 8.

Example II.

Specify the principal dimensions of a six-pass closed feed-water heater to serve the regenerative cycle of Example 8. Terminal difference 6 F. Use 5/8" X 16 BWG Admiralty brass tubes. Water velocity 6 ft/sec and $h_s = 6000$.

Solution:

From Example 8, the extracted steam to the closed heater at full load is 3000 lbs/hr and the heat given up by the steam is (1150 - 196.2) = 953.8 Btu/lb. The temperature of the steam in the heater is 228 F; water temperature to the heater 101 F and from the heater 222 F.

$$t_{m} = \frac{222 - 101}{\log \frac{228 - 101}{e^{228} - 222}} = \frac{121}{\log \frac{127}{e^{6}}} = 39.7 \text{ F}$$

$$t_{m} = \frac{93.6[1 + 0.01(161.5)][6] \cdot \cdot \cdot \cdot \cdot}{[0.495/12] \cdot \cdot \cdot \cdot \cdot} = 1945$$

$$U = \frac{1}{2000} + [\frac{1.25}{0.625 + 0.495}] [\frac{0.065}{12(58)}] + \frac{0.625}{0.495} [\frac{1}{6000}] + \frac{0.625}{0.495} [\frac{1}{1945}]$$

$$U = 684$$

$$A = \frac{2,860,000}{684(39.7)} = 105.5 \text{ sq ft}$$

$$\text{Water flow = condenser flow = 23700 lbs/hr}$$

$$= \frac{23700}{3600} (0.0164) = 0.108 \text{ cu ft/sec}$$

$$\text{Flow area} = \frac{0.108}{6} = 0.018 \text{ sq ft}$$

$$\text{Number of tubes} = \frac{0.018}{0.00134} = 14 \text{ per pass}$$

$$\text{Length of tubes} = \frac{105.5}{14(0.1636)(6)} = 7.68 \text{ ft}$$

The principal dimensions of the feed-water heater are:

Area 105.5 sq ft
Tube material 5/8" X 16 BWG Adm. brass
Number of passes 6
Active tube length 7'-8"
Heat duty 2.86 million Btu/hr

Example 12.

Estimate the area required for the closed feed-water heater of Example 11 by means of the Kittredge data.

Solution:

Using Figure VI-10, $R = \frac{222 - 101}{228 - 101} = 0.953$, $\frac{UA}{M} = 3.0$ and U = 760. From Example 1C, M = 23700 lbs/hr, then,

$$A = \frac{23700(3.0)}{760} = 107 \text{ sq ft.}$$

This area is quite close to the area determined by the method of Example 11.

43. Vacuum Pumps--As previously mentioned, a condenser must be supplied with some sort of air removal apparatus. The usual equipment consists of steam ejectors with suitable coolers, although some plants make

use of mechanical air purps.

Figures VI-11 and VI-12 show some details of the steam ejector type of vacuum pump, and Figure VI-13 shows some operating characteristics. The size of the vacuum equipment will depend upon the air leakage to be expected,

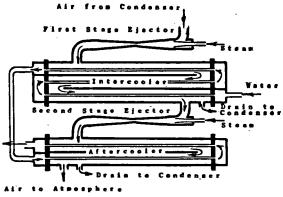
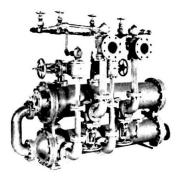


Fig. VI-11. Steam jet vacuum pump



(Allis-Chalmers Co.)

Fig. VI-12. Steam jet vacuum pump.

which in turn depends upon the size and type of condenser. Figure VI-14 shows the approximate maximum air leakage to be expected.

**H*. Water Cooling Equipment --Plants must often be located where an abundant source of condenser cooling water is not available. In these cases it is necessary to provide a spray pond or cooling tower and to use a closed circuit for the cooling water. Both spray ponds and cooling towers depend upon the evaporation into the air of a part of the water to be cooled.

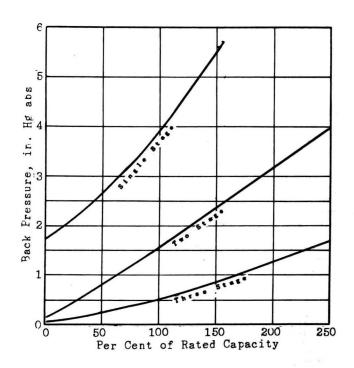


Fig. VI-13. Ejector operating characteristics.

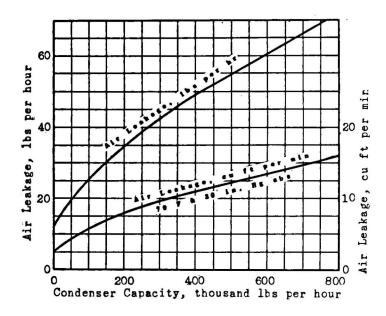


Fig. VI-14. Maximum expected condenser mir lenkage.

Spray ponds are wasteful of ground space because they depend upon natural air motion for their effective operation. On the other hand, they are relatively irexpensive to operate because they only require enough power to produce a pressure of about 10 pounds per square inch at the spray nozzles. Figure VI-15 shows a spray pond with the nozzles arranged in

clusters, each cluster having a capacity of about 200 gallons per minute.. It can be estimated that a spray pond will require about one square foot of ground area for each 250 pounds per hour of water to be cooled. The depth of the pond need not be greater than three feet. Figure

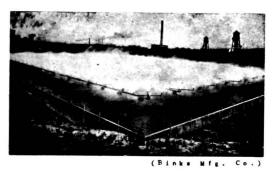


Fig. VI-15. Spray pond.

VI-16 shows some typical spray rond performance data.

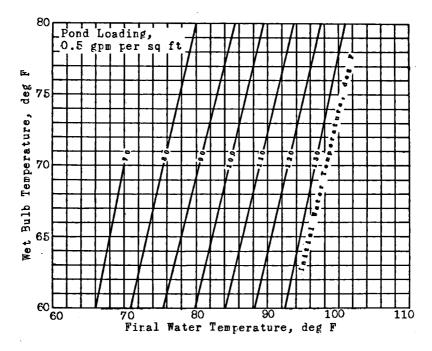


Fig. VI-16. Spray pond performance.

Cooling toxens of the forced or induced draft types are much more economical of ground space than the spray pond because they do not depend upon natural air motion for their cooling effectiveness. In either type the water is broken into drops by being allowed to impinge upon wooden packing in the tower. The air flows through the falling water drops. The tower requires sufficient pump power to raise the water to the distributing troughs at the top, and sufficient fan power to move the required amount of air. Figures VI-17 and VI-18 show forced and induced draft towers, and Figure VI-19 shows some typical performance data.

45. Evaporators -- Since regenerative cycles operate with the same water continuously and leakage is slight, it has been found advantageous to use evaporators to produce distilled water for the small amount of makeup required. Makeup requirements vary from about two to five per cent, depending upon a number of factors such as the amount of boiler blowdown and auxiliary steam unrecovered. The makeup water might be introduced directly to the boiler as raw or treated

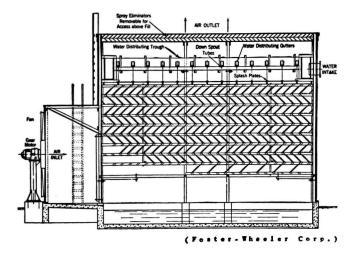


Fig. VI-17. Forced draft cooling tower.

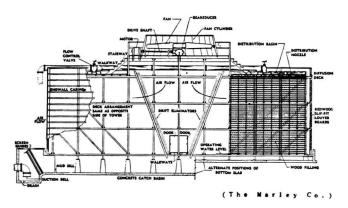


Fig. VI-18. Induced draft cooling tower.

water, but it has been found to be much easier to clean a small evaporator than to clean a large boiler.

Evaporators are most often fitted into the feed-water and steam cycle so that they use steam from one of the high pressure extraction nozzles in the turbine and use condensate to condense the vapor produced. Figure VI-20 shows one of the methods of arrangement of an evaporator set in a regen-

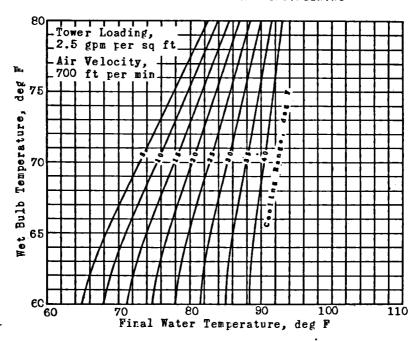


Fig. VI-19. Forced and induced draft

erative cycle. It will be noted that the only heat loss from the system is due to the necessary blowdown from the evaporator shell. Figure VI-21 shows one of the usual types of evaporator.

In process industries which perform evaporation on a product, multiple effect evaporator systems are used because of their economy of operation. Figure VI-22 shows a typical quadruple effect system with a finishing pan for the production of citrus molasses. It will be noted that the vapor from a preceding effect is used as a source of heat for successive evaporations. The evaporators are of the long tube type with vapor separating heads. In this type of evaporator the liquid flows inside the tubes. In the first four effects the circulation is natural, and in the finishing pan a pump forces circulation. Figure VI-23 shows long tube types of evaporator.

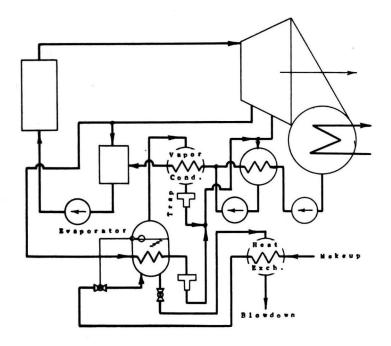


Fig. VI-20. Evaporator in regenerative cycle.

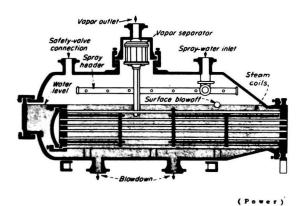


Fig. VI-21. Shell and tube evaporator.

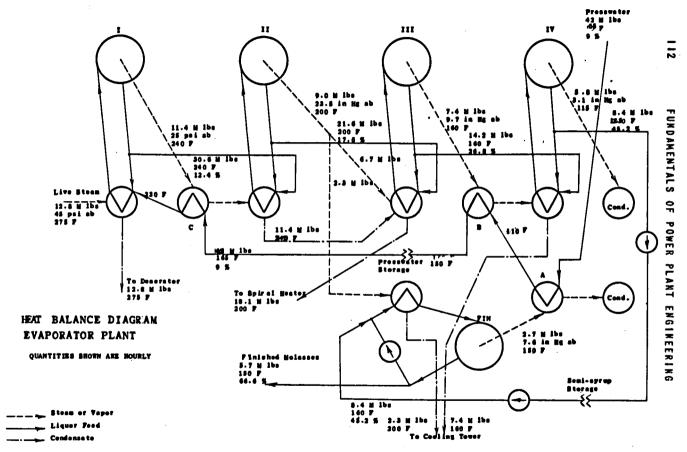
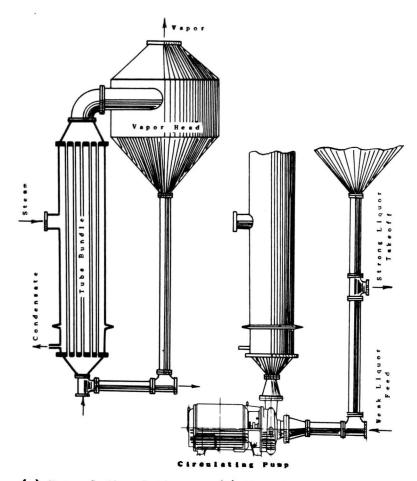


Fig. VI-22. Quadruple effect evaporator system with findshing pan for production of citrus molasses.



(a) Natural Circulation

(b) Forced Circulation

Fig. VI-23. Long tube evenorators

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PROJECT II

1. The power plant of Project I is to be designed around three turbo-generator units. One of the units will operate continuously, one unit will be operated to handle peak loads and the remaining unit will be a spare. The plant will operate on the regenerative cycle with one closed feed-water heater and one evaporator vapor condenser per. machine. There will be one evaporator and one deaerating feed-water heater common to all machines. Boiler blowdown is fed to the evaporator shell. The flow diagram is shown in Figure VI-24.

It is expected that boiler and evaporator blowdown will each be five per cent and makeup will be two per cent. The open heater and evaporator will be supplied with steam from extraction points in the turbines so located that at full load the water temperature leaving the open heater will be at the most economical value. The closed heaters will be supplied with steam from extraction points in the turbines so located that at full load half of the feed-water temperature rise will be supplied by the open heater. Throttle conditions are: 350 psig, 725 F. Back pressure is 2.5 in. Hg abs. Refrigeration in the condenser hotwell is 2 F and terminal difference in the open heater is zero. Make heat balances for the plant at full and half load (two machines operating).

- 2. Specify the principal dimensions for a surface condenser to serve one of the turbines of part one. The average temperature of the inlet cooling water is 85 F. Use 3/4" OD X 18 BWG Admiralty metal tubes with a water velocity of 7 ft per sec.
- 3. Specify the area of an induced draft cooling tower to serve the condenser of part two. Design wet bulb temperature is 78 F.
- 4. Specify the principal dimensions of a closed feed-water heater to serve one of the turbines of part one. Use 5/8" OD X 16 BWG Admiralty metal tubes with a water velocity of 5 ft per sec. Maximum tube length is six feet.
- 5. Specify the principal dimensions of the deaerating feed-water heater for the regenerative cycle of part one.

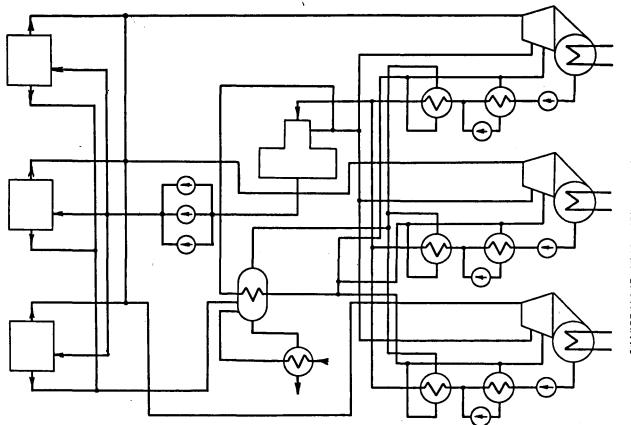


Fig. VI-24. Flow diagram for power plant of elementary power system, Project II.

CHAPTER VII

FUELS AND COMBUSTION

- 46. The variety of fuels used for the production of power is large, and the type used in any particular locality is always dependent upon economic factors. The most import-fuels for commercial power production, in order of import-ance, are: coal, natural gas and fuel oil. Process industries often make use of cellulosic wastes as fuel..
- 47. Coal--Coal is a product of fermentation of vegetable matter laid down in prehistoric times. In this country, coal is found in almost every State in various phases of the evolutionary process. The most important coals for power production are the bituminous and semi-bituminous types, but some use is made of lignite and anthracite which are respectively younger and older than bituminous coal.

Coal is made up of moisture, volatile matter, fixed carbon and ash. The proportions of moisture and volatile matter decrease as the age of the coal deposit increases. Volatile matter consists of hydrocarbons which have a greater heating value than carbon on a weight basis, but which are more difficult to burn in small furnace volumes. Table VIII shows the analysis of some of the more important coals, both in the proximate form (fixed carbon, volatile, ash and moisture) and in the ultimate form.

- 48. Fuel Oil -- The fuel oil ordinarily used in the production of electrical power is residuum from petroleum refining operations. Petroleum is found in large areas throughthe United States and South America, and is also a product of the fermentation of vegetable matter laid down in past ages. Its principal difference from coal is the almost complete absence of fixed carbon and ash. Table IX shows some analyses of typical fuel oils. The most important sources of fuel oil for the United States are the Pacific and Gulf coasts and Venezuela.
- 49. Watural Gas -- Natural gas is principally a by-product from petroleum wells, although much gas is removed from the earth's crust through wells which are not connected with petroleum operations. Natural gas is a product of the fermentation of vegetable matter to produce coal and petroleum. Table X shows some analyses for natural gas.

TABLE VIII

ANALYSES OF TYPICAL AMERICAN COALS

	Proximate			Ultimate - mois. incl. S H C N O						
	H.O	Vol.	FC	Ash	S	H	С	N	0	HHV
Ark. Semi-anth.							77.29			13406
Penna. Anth.	2.19	5.67	86,24	5.90	0.57	2.70	86.37	0.91	3.55	13828
Penna. Semi-bit.	2.95	21.29	66.92	8.84	1.35	4.74	78.51	1.19	5.37	13901
W. Va. Semi-bit.							84.02		6.03	14724
Penna. Bit.	2.44	28.44	60,68	8.44	1.32	5.07	76.91	1.31	6.95	13732
	2.53	26.94	59.48	11.05	0.79	4.80	74.44	1.59	7.32	13286
Ill. Bit.	13.31	33.62	41.34	11.73	3.75	5.19	59.07	0.95	19.31	10548
Ind. Bit.	9.69	38.59	41.04	10.68	4.79	5,39	62.36	1,28	15.50	11412
N. Dak. Lig.	36.18	29.77	25.35	8.70	0.68	6,76	39.45	0.59	43,82	6700
Texas Br. Lig.	13.04	42.75	29.00	14.85	1.04	5.51	52.06	0.95	25.33	9358
				(0.	S. Bu	reas 1	of Him	es)		

TABLE IX

ANALYSES OF TYPICAL FUEL OILS,

	ָט:	ltimate -	- mois.	API			
	. C	H	S	N	Ο,	60 F	HHV
Mexican	84.02	10.06	4.93	0.09	0,00	11.9	18510
Mid-cont.	85.62	11.98	0.35	0.50	0.60	27.1	19376
Calif.	84.67	12.36	1.16	0.90	0.91	16,7	18835
Texas	83.26	12.41	1.80	1.50	1.03	21.2	19654

TABLE X

ANALYSES OF TYPICAL NATURAL GASES

		Volumetric - std. cond. 60 F, 30 in. Hg CH4 C.H4 C.H4 CO H. O. N.							HHV
Findlay, O.	72.18	3.00 0.35	0.00	1.00		0.80 0.39	0.00 3.41		920

(Parious isources)

50. Combustion--The process of combustion of fuels is one of rapid oxidation of the carbon, hydrogen and sulphur which it contains. In order to calculate the amount of air, which is the source of oxygen, for combustion, it is necessary to know the ultimate analysis of the fuel. Since the combustion process takes place too rapidly to reach equilibrium, it is always necessary to supply more than the theoretical amount of air. The air supplied in relation to the theoretical air is known as the total air, while that supplied above the theoretical air is called the excess air. Both total air and excess air are expressed as percentages by weight of the theoretical air. Examples 13 and 14 show the method of making combustion calculations for each of the principal types of fuels.

Example 13.

Determine the weight of air required for theoretically perfect combustion and the flue gas analysis for a fuel with the following analysis by weight: C = 0.8339; H = 0.0428; O = 0.0283; N = 0.0103; S = 0.0064; moisture = 0.025; and ash = 0.0533. Also, determine the weight of air required and the flue gas analysis for combustion with 30 o/o excess air.

Solution:

The combustion reactions are:

$$C + O_2 \longrightarrow CO_2$$
, $O_2 = 2.67 C$, $CO_2 = 3.67 C$
 $H_2 + \frac{1}{2} O_2 \longrightarrow H_2O$, $O_2 = 8 H$, $H_2O = 9 H$
 $S + O_2 \longrightarrow SO_2$, $O_2 = S$, $SO_2 = 2 S$
 N_2 from air = 3.31 O_2 from air

Theoretically perfect combustion: Basis, one lb fuel.

Const.	weight	Og' req.	prod.	wght. prod.			anal. by vol. wet	
С	0.8339	2.230	CO2	3.060	44_0	.0696	17.70	18.78
H	0.0428	0.342	H ₂ O	0.385	18 (0.0214	5.80	
Ō	0.0283	-0.028						
N	0.0103		Nz'	C.010	28 (0.0004	76.50	81.22
2	0.0064	0.006	.20°.	0.012	64 (0.0002		
moist.	0.0250		H ₂ O	0.025	18 (0.0014		
ash	0.0533			·				

weight	O ₂ req.	wght. prod.	mols,
1.0000	2.550	3.492	0.0930
N, air	8.440	8.440	0.3010
Totals/lb	10.990 air	11.932 gas	

For theoretically perfect combustion, 10.99 lbs of air are required and 11.932 lbs of gas are produced per lb of fuel burned. The dry (Orsat) analysis of the flue gas would be: CO₂ = 18.78 o/o and N₂ = 81.22 o/o.

Combustion with 30 o/o excess air: Basis, one 1b fuel.

$O_{2'}$ req. O_{2} suppl. $N_{2'}$ air		2.550 3.320 11.000	Exc	e prod., cess O ₂ , with exc		11.932 0.770 2.560
Air suppl	•	14.320	Gas	s produce	ed	15.262
	Prod.	wght. prod.	mol. wt.	mols.	anal. by vol. wet	
	CO,	3.060	44	0.0696	13.70	14.30
	H ₂ O	0.410	18	0.0228	4.48	
	0,	0.770	32	0.0240	4.71	4.93
•	N ₂	11.010	28	0.3930	77.21	80.72
	SO.	0.012	64	0.0002		
		15.262 gas		0.5096		

For combustion with 30 o/o excess air, 14.32 lbs of air are required and 15.262 lbs of gas are produced per lb of fuel burned. The dry (Orsat) analysis of the flue gas would be: $CO_2 = 14.30$ o/o, $O_2 = 4.93$ o/o and $N_2 = 80.72$ o/o.

Example 14.

Determine the weight of air required for theoretically perfect combustion and the flue gas analysis for a gaseous fuel with the following analysis by volume: $CO_2 = 0.004$; $CH_4 = 0.921$; $C_2H_6 = 0.041$; and $N_2 = 0.034$. Also, determine the amount of air required and the flue gas analysis for

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combustion with 20 o/o excess air.

Solution:

Combustion reactions are:

$$CH_4 + 2 O_2 - CO_2 + 2 H_2O$$
, $O_2 = 4 CH_4$, $CO_2 = 2.75 CH_4$, $H_2O = 2.25 CH_4$

$$C_2H_6 + 3\frac{1}{2} O_2 - 2 CO_2 + 3 H_2O, O_2 = 3.73 C_2H_6, CO_2 = 2.94 C_2H_6, H_2O = 1.79 C_2H_6$$

No from air = 3.31 Op from air

Theoretically perfect combustion: Basis, one lb fuel.

Const. anal. mol. lbs. anal.
$$O_2$$
 Products by vol. wt. by wt. req. CO_2 H_2O N_2

Total air 16.02 2.59 2.07 12.36 17.02 Mol. wt. 44 18 28 gas mols. 0.06 0.12 0.44 0.62

Anal. by. vol. wet.o/o 9.60 18.7 71.70 Anal. by vol. dry o/o 11.80 88.20

From the preceeding analysis it is seen that one pound of this particular fuel gas requires 16.02 lbs of air for theoretically perfect combustion and produces 17.02 lbs of wet flue gas. The dry (Orsat) analysis of the flue gas would be: $CO_2 = 11.80$ o/o and $N_2 = 88.20$ o/o.

Combustion with 20 o/o excess air: Basis, one lb fuel.

Ogr req	3.72	Gas prod., theo.	17.02
O. suppl.	4.46	Excess O2	0.74
Ng from air	14.76	N_2 with exc. $O_{2'}$	2.46
Air supplied	19.22	Gas produced	20.22

Prod.	wght. prod.	mol.	mols	anal. by vol. wet	anal. by vol. dry o/o
CO,	2.59	44	0.0589	8.10	9.63
H.O	2.07	18	0.1150	15.82	•
0.,	0.74	32	0.0231	3.08	3.78
N.	14.82	28	0.5300	73.00	86.59
	20.22 gas		0.7270		

For combustion with 20 o/6 excess air, 19.22 lbs of air are required and 20.22 lbs of flue gas are produced per lb of fuel. The dry (Orsat) analysis of the flue gas would be: $10_2 = 9.63$ o/o, $0_2 = 3.78$ o/o and $N_2 = 86.59$ o/o.

Often it is necessary to determine combustion conditions from a fuel analysis and an analysis of the flue gas. Examples 15 and 16 show the method followed for each of the types of fuels.

Example 15.

Fuel with an ultimate analysis of: C = 0.83; H = 0.127; O = 0.012; N = 0.016; S = 0.004; moisture = 0.01; and ash = 0.001 is burned in a boiler furnace. The flue gas has the following volumetric analysis on the dry basis: $CO_2 = 13.7$ o/o; $O_2 = 2.6$ o/o; CO = 0.4 o/o; and $N_2 = 83.3$ o/o find the theoretical air, actual air and flue gas produced in lbs per lb of fuel fired.

Solution:

1. The theoretical air can be found from the formula:

$$M_{st} = 11.53 \text{ C}_{bursed} + 34.36[H - \frac{0}{8}] + 4.32 \text{ S lbs/lb fuel}$$

$$M_{st} = 11.53(0.83) + 34.36[0.127 - \frac{0.012}{8}] + 4.32(0.004)$$

$$M_{st} = 13.91 \text{ lbs/lb fuel}$$

2. The dry flue gas produced can be determined with the aid of a carbon balance, as follows:

$$M_{gd} = \frac{44 \text{ CO}_2 + 32 \text{ O}_2 + 28(N_2 + C0)}{12(CO_2 + C0)} [C_{purmed} + \frac{3}{2}S] + \frac{2}{3}S \text{ lbs/lb}$$

$$M_{gd} = \frac{44(13.7) + 32(2.6) + 28(83.3 + 0.4)}{12(13.7 + 0.4)} [0.83 + (0.375)0.004] + \\ +0.625(0.004)$$

 $M_{pd} = 19.93$ lbs/lb fuel

Note: Sulphur dioxide in the flue gas is analyzed as carbon dioxide by the Orsat apparatus. If all of carbon in the fuel is not burned a correction is made (see Example 17).

3. The wet flue gas produced can be found from a weight balance, as follows:

$$M_{g\pi} = M_{gd} + 9 H + moisture lbs/lb fuel$$

$$M_{pq} = 19.93 + 9(0.127) + 0.01 = 21.08 lbs/lb fuel$$

4. The actual air supplied can be found from a weight balance, as follows:

$$M_{aa} = M_{gd} + 8[H - \frac{O}{8}] - C_{burned} - S - N lbs/lb fuel$$

$$M_{aa} = 19.93 + 8[C.127 - \frac{0.012}{8}] - 0.83 - 0.004 - 0.016$$

 $M_{\alpha\alpha} = 20.08$ lbs/lb fuel

5. Excess air can be found from the relation:

Excess air o/o =
$$\frac{M_{aa} - M_{at}}{M_{at}}$$
 (100)

Excess air
$$o/o = \frac{20.08 - 13.91}{13.91}(100) = 44.3 o/o$$

Example 16.

A fuel gas having an analysis of CH₄ = 77.5 o/o, C_2H_6 = 16.0 o/o, CO_2 = 6.3 o/o, N_2 = 0.2 o/o is burned in a boiler furnace. The flue gas has the following analysis on the dry basis: CO_2 = 9.0 o/o, O_2 = 3.0 o/o, CO = 0.2 o/o, N_2 = 87.8 o/o. Find the theoretical air, actual air and flue gas in lbs per lb of fuel fired and the percentage of excess air supplied.

Solution:

The first step in the solution is the conversion of the volumetric analysis to a gravimetric analysis.

Const.	anal. by vol.	mol. wt.	lbs.	anal. by wt.
CH.	0.775	16	12.40	0.619
C ₂ H _e	0.1€0	30	4.80	0.239
CO.	0.063	44	2.77	C.138
N ₂	0.002	28	0.06	0.004
	1.000		20.03	1.000

1. Theoretical air:

$$M_{at} = 17.25 \text{ CH}_4 + 16.10 \text{ C}_2\text{H}_6 + 14.8 \text{ C}_2\text{H}_4 + 34.5 \text{ H}_2 \text{ lbs/lb}$$

$$M_{at} = 17.25(0.619) + 16.10(0.239) = 14.53 lbs/lb fuel$$

$$M_{gd} = \frac{44 \text{ CO}_2 + 32 \text{ O}_2 + 28(\text{N}_2 + \text{CO})}{12(\text{CO}_2 + \text{CO})} [0.75 \text{ CH}_4 + 0.8 \text{ C}_2\text{H}_6] + 0.857 \text{ C}_2\text{H}_4] + \text{CO}_2f + \text{N}_2f}$$

$$M_{gd} = \frac{44(9.0)+32(3.0)+28(87.8+0.2)}{12(9.0+0.2)}[0.75(0.619)+0.8(0.239)]$$
+0.138+0.004

$$M_{pd} = 17.64 \text{ lbs/lb fuel}$$

3. Wet flue gas:

$$M_{gx} = M_{gd} + 2.25 \text{ CH}_4 + 1.8 \text{ C}_2\text{H}_6 + 1.29 \text{ C}_2\text{H}_4 + 9 \text{ H}_2$$

$$M_{gg} = 17.64 + 2.25(0.619) + 1.8(0.239) = 19.46 lbs/lb fuel$$

4. Actual air supplied:

$$M_{aa} = M_{gd} + 1.25 \text{ CH}_4 + 0.80 \text{ C}_2 \text{H}_a + 0.29 \text{ C}_2 \text{H}_4 + 8 \text{ H}_2 - N_2 - \text{CO}_2$$

$$M_{aa} = 17.64 + 1.25(0.619) + 0.80(0.239) - 0.002 - 0.138$$

 $M_{aa} = 18.47$ lbs/lb fuel

5. Excess air:

Excess air =
$$\frac{18.47 - 14.53}{14.53}$$
(100) = 27.1 o/o

51. Heating Value of Fuels-One of the most important properties of any fuel is its heating value per unit of weight or volume. The higher heating value (HHV) is the value which would be obtained if the products of combustion were cooled to the temperature of the original fuel and air. The lower heating value (LHV) is the higher heating value less the latent heat of vaporization of the moisture formed in the combustion process and present in the fuel originally. It is customary to base efficiencies on the higher heating values, although efficiencies would appear to be higher if they were based on the lower heating value.

The heating values of fuels can be determined approximately by calculation if the ultimate analysis, proximate analysis (for coal) or specific gravity (for fuel oil) is known.

For solid or liquid fuels with the ultimate analysis known:

HHV = 14600 C + 62000 (H
$$-\frac{0}{8}$$
) + 4050 S
where: HHV = higher heating value, Btu/lb
C, H, O and S = carbon, hydrogen,
oxygen and sulphur, lbs/lb fuel

For coals with the proximate analysis known, the heating value can be determined from Table XI. It will be noted that the proximate analysis must be converted to the combustible basis before the table is used, and the result converted to the as fired basis. These conversions may be made as follows:

Converting proximate analysis as fired to combustible basis, lbs/lb fuel:

Volatile matter =
$$\frac{\text{Volatile matter as fired}}{1 - \text{moisture as fired} - \text{ash as fired}}$$

Fixed carbon =
$$\frac{\text{Fixed carbon as fired}}{1 - \text{moisture as fired} - \text{ash as fired}}$$

Converting heating value on combustible basis to as fired basis, Btu/lb:

HHV = (HV on comb. basis)(1 - mois. as fired - ash as fired)

TABLE XI
HEATING VALUE OF COAL BASED ON PROXIMATE ANALYSIS

States	Per cent Volati on Combustible Basis	le Btu/lb on Combustible Basis
Pa., Ohio, W.Va., Md.,	0 - 16	HHV = 14550 + 7810 V
Va., Ky., Ga., Tenn., Ala., Ind., Iowa, Neb.,	16 - 36	HHV = 16160 - 2250 V
Kan., Mo., Okla., Ark., and Texas	36 - up	HHV = 18750 - 9440 V
Ill. and Mich.	all	HHV = 16062 - 3830 V

where: HHV = higher heating value V = decimal fraction volatile on combustible basis

(F. C. Evans)

For heavy fuel oils with the specific gravity known:

 $AHV = 18650 + 40(A.P.I.^{0} - 10)$

where: A.P.I. = specific gravity
on the American Petroleum
Institute scale, or
A.P.I. = 141.5 - (131.5/sp.gr. 50)

For gaseous fuels with the volumetric analysis known:

HHV = 323 CO + 328 H₂ + 1012 CH₄ + 1762 C₂H₆ + 1641 C₂H₄

where: HHV = higher heating value, Btu/cu ft

at 60 F and 30 in. Hg

CO, H₂, CH₄, C₂H₈ and C₂H₄ = carbon monoxide, hydrogen, methane, ethane and ethylene, volume fractions as fired.

The preceding equations are only approximate, and for the most precise determination of heating value it will be necessary to employ a calorimeter of a suitable type.



Fig. VII-1. Parr bomb calorimeter.

tance wire.

For heating value determinations on sclid or heavy liquid fuels, it is customary to employ one of the constant volume "bomb" types of calorimeter. The Parr bomb calcrimeter is shown in Figure VII -1, and consists of a heavy steel contairer into which the fuel sample is placed with sufficient pure oxygen to insure complete combustion.. The bomb is immersed in a known weight of water whose temperature rise is determined after igniting the fuel sample with an electric resis-

For heating value determinations on gaseous or light

liquid fuels, the steady flow calcrimeter is used. The Sargent calcrimeter shown in Figure VII-2 is of this type and consists of a water jacket with flues for absorbing heat from the products of combustion of the fuel. If the products of combustion leave the calcrimeter at the same temperature as the incoming fuel and air, the higher heating value will be determined. The heating value is calculated from the known weight of water heated through a known temperature rise by a known volume or weight of fuel.

52. Flue Gas Analysis-Analysis of the products of
combustion is most conveniently made with an Crsat type

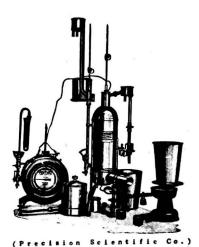


Fig. VII-2. Sargent

of apparatus which absorbs the various flue gas constituents, one at a time, by means of chemical solutions. Figure VII-3 shows a common portable form of the apparatus.

The Orsat analysis is given on the dry basis because the sample remains saturated with moisture throughout the test procedure. The ordinary form of the apparatus measures volumetric or mol fractions of carbon dioxide, oxygen and carbon monoxide. It is assumed that the remaining gas is nitrogen. Due to the fact that the reagent for carbon dioxide will absorb all acid gases, the sulphurcus gases in the products of

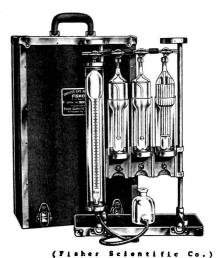


Fig. VII-3. Orsat gas analysis apparatus.

combustion will be analyzed as carbon dioxide, but the error is slight for ordinary fuels.

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CHAPTER VIII

STEAM GENERATORS

53. Steam generator is the preferred name for equipment which produces steam from water by means of fuel. Steam generators are divided into two general classifications: fire -tube and water-tube, depending upon whether the water surrounds the tubes or is contained within the tubes. It might be said here that the term boiler is used somewhat flexibly, but strictly speaking should only be applied to the convection steam generating surface in a steam generator, while the term isteam generator includes the boiler and all auxiliary heat recovery apparatus.

Steam generators in small sizes are rated in terms of boiler horsepower, which amounts to a heat absorption of 33,480 Btu per hour per boiler horsepower, or 34.5 pounds of steam per hour per boiler horsepower from and at 212 F. One boiler horsepower is also approximately equal to ten square feet of convection heating surface. Large steam generators are rated in terms of the actual weight of steam produced - per hour at definite outlet conditions with feed-water at a definite temperature.

54. Steam Generator Efficiency--Steam generator efficiency is the quotient of the heat output of the steam generator divided by the heat input from the fuel. That is:

$$e_{sg} = \frac{M_{v}(H_{v} - h_{f_{1}}) + M_{b}(h_{f_{1}} - h_{f_{1}})}{M_{f}(HHV)}$$

where: M = steam flow, lbs/hr

M = blowdown rate, lbs/hr

M_f = fuel feed rate, lbs/hr or cu ft/hr H_y = enthalpy of steam at

boiler outlet, 3tu/lb

h ; = enthalpy of boiler feed, Btu/lb h ; = enthalpy of saturated water

at boiler pressure, Btu/lb

HHV = higher heating value of fuel, Btu/lb or Btu/cu ft

Quite often it is impossible to measure accurately the quantities of fuel, steam and blowdown, making it necessary to determine the steam generator efficiency by the method of losses. This method requires that the flue gas analysis and ultimate analysis of the fuel, as well as combustion air temperature, fuel temperature and flue gas temperature be known. The various losses per pound of fuel can then be calculated and the difference between the heating value of the fuel and the losses considered as the heat absorbed by the steam generator. The losses are:

- (a) Due to sensible heat in the dry flue gases ..
- (b) Due to sensible and latent heat in the vapor formed from the moisture in the fuel.
- (c) Due to sensible and latent heat in the vapor formed from the combustion of hydrogen in the fuel.
- (d) Due to carbon monoxide resulting from the incomplete combustion of carbon.
- (e) Due to unburned carbon in the ash pit refuse and in fly ash.
- (f) Due to radiation from the boiler setting.
- (g) Due to unaccounted for losses, such as sensible heat in the ash pit refuse, unburned hydrocarbons, and moisture in the air used for combustion. (This item can be estimated at from C.1 to 0.2 per cent.)

The above losses might be regrouped into two general classifications, as follows:

- A. Combustion losses due to failure to develop the full heating value of the fuel. These items would include items (d) and (e) and part of (g).
- B. Conversion losses due to failure to transmit all of the heat developed to the steam. These losses would include items (a), (b), (c) and (f) and part of (g).

Example 17 shows the method of determination of steam generator efficiency by the direct method and by the method of losses for solid or liquid fuels, and Example 18 shows the same procedure for gaseous fuels.

Example 17.

The following data were recorded for a test on a boiler

rated at 200,000 lbs per hour and equipped with a completely water-cooled furnace:

Average steam temperature	900	F
Average steam pressure	910	psig
Steam generated in 8 hours	1,600,000	lbs
Average feed-water temperature	300	
Heating value of coal as fired	14,400	Btu/lb
Fuel fired in 8 hours	74	tons
Heating value of refuse	5,100	Btu/lb
Refuse collected in 8 hours	5.9	
Average flue gas temperature	250	F
Average combustion air temperature	110	F
Average fuel temperature	70	F

Flue gas analysis by volume	Fuel analysis lb/lb as fired
$CO_2 = 12.8 \text{ o/o}$	C = 0.8152
$\theta_2 = 6.4 \text{ o/o}$	H = 0.0447
$C\bar{O} = C.2 \text{ o/o}$	0 = 0.0495
$N_2 = 80.6 \text{o/o}$	N = 0.0100
	S = 0.0063
	ash = 0.0521
•	mois. = 0.0222

Make a heat balance for the boiler:

Solution:

A. Carbon burned:

$$C_{burned} = C - [\frac{\text{wt. of refuse(heating value of refuse)}}{14600(\text{wt. of fuel})}]$$

$$C_{burned} = 0.8152 - [\frac{5.9(5100)}{14600(74)}] = 0.7873 \text{ lb/lb fuel}$$

B. Dry gas: (See Example 15)

$$M_{gd} = \frac{44(12.8) + 32(6.4) + 28(80.6+0.2)}{12(12.8+0.2)} [0.7873 + 0.375(0.0063)] + 0.625(0.0063)$$

= 15.30 lbs/lb fuel

C. Heat absorbed by boiler:

$$Q_1 = \frac{1,600,000(1450.9 - 269.6)}{74(2000)} = 12,780 \text{ Btu/lb fuel}$$

D. Heat loss due to dry flue gas:

$$Q_{z'} = 0.24(M_{pd})$$
 (temp. of flue gas - temp. of comb. air)

$$2. = 0.24(15.30)(250 - 110) = 515 Btu/lb fuel$$

E. Heat loss due to moisture in fuel:

When temperature of flue gas < 575 F:

$$Q_s = [mois. in fuel][1083 - temp. of fuel + 0.46(temp. of flue gas)]$$

When temperature of flue gas > 575 F:

$$Q_s = [mois. in fuel][1066 - temp. of fuel + 0.50(temp. of flue gas)]$$

$$Q_s = 0.0222[1089 - 70 + 0.46(250)] = 25 Btu/lb fuel$$

F. Heat loss due to moisture from combustion of R:

When temperature of flue gas < 575 F:

$$Q_4 = [9 \text{ H}][1089 - \text{temp. of fuel} + 0.46(\text{temp. of flue gas})]$$

When temperature of flue gas > 575 F:

$$Q_4 = [9 \text{ H}][1066 - \text{temp of fuel} + 0.50(\text{temp. of flue gas})]$$

$$Q_4 = 9[0.0447][1089 - 70 + 0.46(250)] = 456 Btu/lb fuel$$

G. Heat loss due to carbon monoxide in flue gas:

$$Q_s = \frac{CO}{CO_2 + CO} (10160) (C_{burned})$$

$$Q_s = \frac{0.2}{12.8 + 0.2}$$
 (10160)(0.7873) = 123 Btu/lb fuel

H. Heat loss due to combustible in refuse:

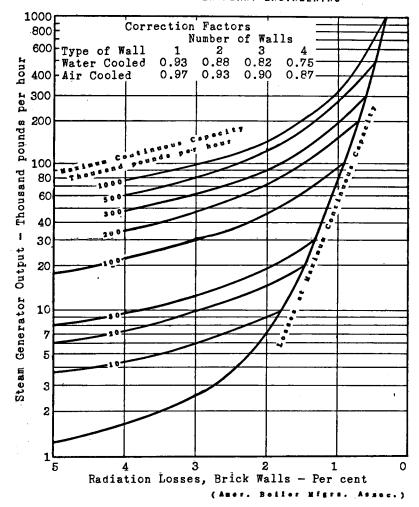


Fig. VIII-1. A.B. W. A. boiler radiation loss chart,

$$Q_e = \frac{5.9}{74}(5100) = 407 \text{ Btu/lb fuel}$$

I. Radiation losses from boiler setting:

From Figure VIII-1, radiation losses for a 200,000 lb/hr boiler at full load are 0.75 o/o for brick walls, with a

correction factor of 0.75 for four water-cooled walls.

 $Q_7 = 0.0075(0.75)(14400) = 81 Btu/lb fuel$

J. Unaccounted for losses:

$$Q_e$$
 = Heating value of fuel - $[Q_1 + Q_2 + Q_3 + Q_4' + Q_5 + Q_6 + Q_7]$

Q_e = 14400 - [12780+515+25+456+123+407+81] = 13 Btu/lb fuel Recapitulation: Basis, one lb fuel as fired.

Heat	absorbed	bу	boiler .		12780	Btu/1b	88.7 o	/ o
Heat	loss due	to	dry flue gas		515		3.6	
Heat	loss due	to	moisture in fuel		25		0.2	
Heat	loss due	to	mois. from comb.	of	H 456		3.2	
Heat	loss due	to	carbon monoxide		123		8.0	
Heat	loss due	to	comb. in refuse		407		2.8	
Heat	loss due	to	radiation		81	·	0.6	
Unac	counted for	or	losses		13		0.1	
Heating value of fuel			14400	Btu/lb	100 .0 o	/ o		

Example 18.

The following data were recorded for a test on a boiler rated at 25,000 lbs per hour and equipped with plain brick furnace walls:

Average steam pressure	215	psig
Average steam temperature	540	F
Steam generated in 4 hours	86,000	lbs
Average feed-water temperature	260	
Heating value of gas (std.cond.)	1,060	Btu/cu ft
Fuel fired in 4 hours, at		
70 F and 3 psig	89,400	cu ft
Average flue gas temperature	350	F
Average combustion air temp.	100	F

Flue gas analysis	Fuel analysis
by volume	by volume
$CO_{\bullet} = 9.3 \text{ o/o}$	$CH_4 = 0.840$
$0_2 = 2.5 \text{o/o}$	$C_2H_a \approx 0.120$
CO = 0.3 o/o	$N_{\bullet} = 0.025$
$N_{\bullet} = 87.9 \text{ o/o}$	1.9 0.020

Make a heat balance for the boiler.

Folution:

The volumetric fuel analysis is first changed to a gravimetric analysis:

Const.	anal. by vol.	mol. wt.	lbs.	anal. by wt.
CH ₄	0.840	16	13.45	0.759
C ₂ H ₆	0.120	30	3.60	0.203
N 2	0.025	28	0.70	0.038
	1.000		17.75	1.000

A. Dry gas: (See Example 16)

$$\mathbf{M}_{gd} = \frac{44(9.3)+32(2.5)+28(87.9+0.3)}{12(9.3+0.3)} [0.75(0.759)+0.8(0.203)] +0.038$$

 $M_{pd} = 18.82 \text{ lbs/lb fuel}$

B. Heat absorbed by boiler: (See Example 17)

From the calculations converting the fuel analysis to the gravimetric basis, the apparent molecular weight is 17.75, from which R = 1544/17.75 = 87.0. The weight of fuel fired is:

$$M_f = \frac{(3+14.7)(144)(89400)}{87(530)} = 4950 \text{ lbs}$$

The weight of one cubic foot of fuel at standard conditions is:

$$\rho = \frac{30(0.491)(144)}{87(520)} = 0.0469 \text{ lb/cu ft}$$

from which the heating value of the fuel is:

$$HHV = \frac{1060}{0.0469} = 22600 Btu/lb$$

$$Q_1 = \frac{86,000(1287.8 - 228.6)}{4950} = 18400 \text{ Btu/lb fuel}$$

- C. Heat loss due to dry flue gas: (See Example 17) $Q_s = 0.24(18.82)(350 100) = 1130 \text{ Btu/lb fuel}$
- D. Heat loss due to moisture from combustion of R: When temperature of flue gas < 575 F:

$$Q_4 = [2.25 \text{ CH}_4 + 1.8 \text{ C}_2\text{H}_6 + 1.29 \text{ C}_2\text{H}_4 + 9 \text{ H}_2][1089 - \text{temp.}$$
of fuel + 0.46(temp. of flue gas)]

When temperature of flue gas > 575 F:

$$Q_4 = [2.25 \text{ CH}_4 + 1.8 \text{ C}_2\text{H}_6 + 1.29 \text{ C}_2\text{H}_4 + 9 \text{ H}_2][1066 - \text{temp.}$$
of fuel + 0.50(temp. of flue gas)]

$$Q_4 = [2.25(0.759) + 1.8(0.203)][1089 - 70 + 0.46(350)]$$

 $Q_{\bullet} = 2440 \text{ Btu/lb fuel}$

E. Heat loss due to carbon monoxide in flue gas:

$$Q_s = \frac{C0}{CO_{e^+} + CO} (10160) (0.75 \text{ CH}_4 + 0.8 \text{ C}_2\text{H}_6 + 1.857 \text{ C}_2\text{H}_4)$$

$$Q_s = \frac{0.3}{9.3 + C.3} (10160)(0.75[0.759] + 0.8[0.203]) = 232 \text{ Btu/lb}$$

F. Radiation losses from boiler setting:

From Figure VIII-1, radiation losses for a 25,000 lb/hr boiler operating at 21,500 lbs/hr are 1.5 o/o.

$$Q_7 = C.015(22600) = 339 \text{ Btu/lb fuel}$$

G. Unaccounted for losses:

 $Q_e = 22600 - [18400+i130+2440+232+339] = 59 Btu/lb fuel$ Recapitulation: Basis, one lb fuel as fired.

Heat	absorbed	bу	boiler	18400: Btu/lb	81.5 o/o
Heat	loss due	to	dry flue gas	1130	5.0
Heat	loss due	to	mois. from comb. of	H 2440	10.8
Heat	loss due	to	carbon monoxide	232	1.0
Rest	loss due	t.o	radiation	3.39	1.5

Unaccounted for losses

59

0.2

Heating value of fuel

22600 Btu/lb 100.0 o/o

55. Fire-Tube Boilers--Fire-tube boilers (or steam generators) are built in sizes up to about 500 horsepower and for pressures up to about 200 psig. The principal types are: horizontal return tubular; Scotch marine; vertical tubular; and firebox tubular.

Horizontal return tubular boilers consist of a drum with longitudinal flues set over a brick firebox, as shown in Figure VIII-2. They are moderately efficient and are suitable for use with any kind of fuel.

Scotch marine boilers are so named because they were first used in steam powered ships built in Scotland. They have been adapted to stationary power plants and are most popular for packaged units tecause the furnace is internal to the boiler. Figure VIII-3 shows a modern Scotch marine package unit.

Vertical tubular bailers are built in small sizes up to about 50 boiler horsepower and are popular for laundries and small food processing plants because of the small floor space requirement. Due to the fact that part of the flues are above the water line, the steam is delivered with a small amount of superheat. Figure VIII-4 shows a boiler of this type.

Firebox tubular boilers are made in a number of different designs, one of which is shown in Figure VIII-5. Their advantage is the ability to put more heating surface in a given total boiler volume than the other fire-tube types.

Fire-tube boilers are not made for high pressures because of the difficulty of staying the flat tube sheets against bulging, and because tubes are relatively less resistant to collapse from external pressure than they are to rupture from internal pressure.

56. Water-Tube Boilers--Water-tube boilers are built in capacities from about 1CC boiler horsepower to more than one million pounds of steam per hour, with pressures ranging from 15 to 3500 psig. The variety of water-tube boilers is very large, but the general types are: straight tube and bent tube.

Straight tube bailers require the use of headers for the tube ends. The main disadvantage is that the many hand-

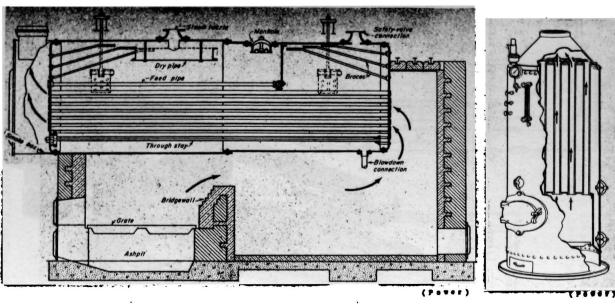


Fig. VIII-2. Horisontal return tubular boiler.

Fig. VIII-4. Vertical tubular boiler.

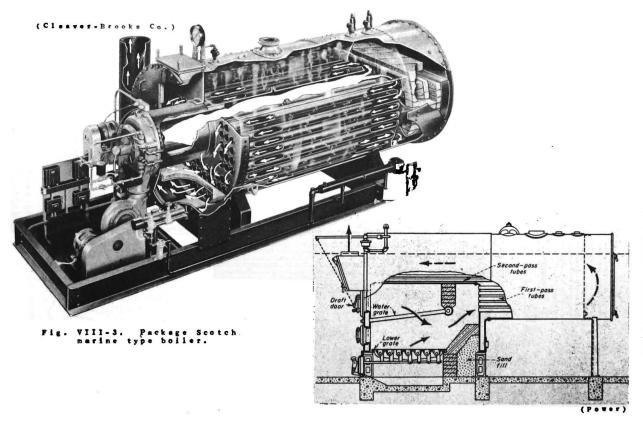
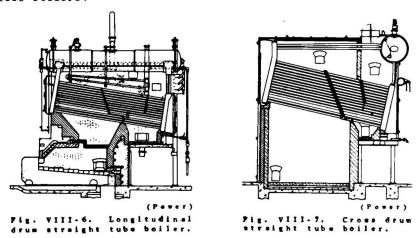


Fig. VIII-5. Firebox tubular boiler.

hole covers which are necessary have a tendency to leak. The main advantages are that all of the steam generating tubes are alike and they are easy to clean, when necessary. Figures VIII-6 and VIII-7 show two types of straight tube boilers.



Bent tube boilers were designed to eliminate the tube headers and handhole covers which gave trouble in the straight tube boilers in the higher pressure ranges. Figures VIII-8 through VIII-11 show some modern bent tube boilers of various sizes and arrangements and Figure VIII-12 shows boiler cost data (based on 1946).

57. Heat Transfer in Boilers -- The major portion of the heat transfer in boilers takes place by convection from the flue gases and by conduction through the tubes. Consideration of the mechanisms of heat transfer in boilers will show that, while they are similar to those in surface condensers, some of the factors are not easily determined. For instance, in most boilers the circulation of water is by natural forces set up by the difference in density between steam and water, and as a consequence it is not easy to determine the velocity of the water passing over the boiler surface. In a similar manner to the effect of velocity on water film coefficients, the velocity of the flue gas over the surface of the tubes affects the gas film coefficient. For these reasons, manufacturers of boilers employ empirical formulae to a great extent in calculating the heating surface required.

One of the best known empirical formulae for estimation

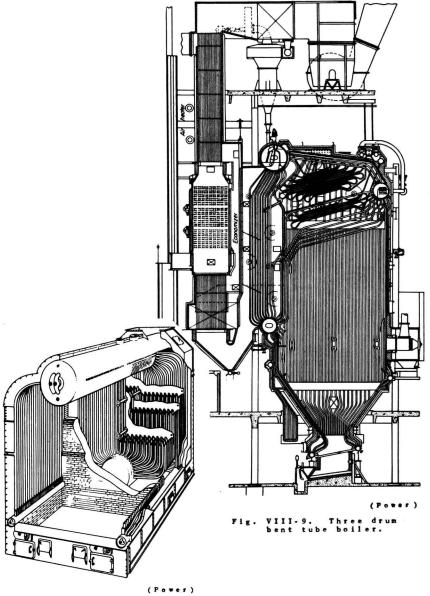


Fig. VIII.8. Longitudinal drum bent tube boiler.

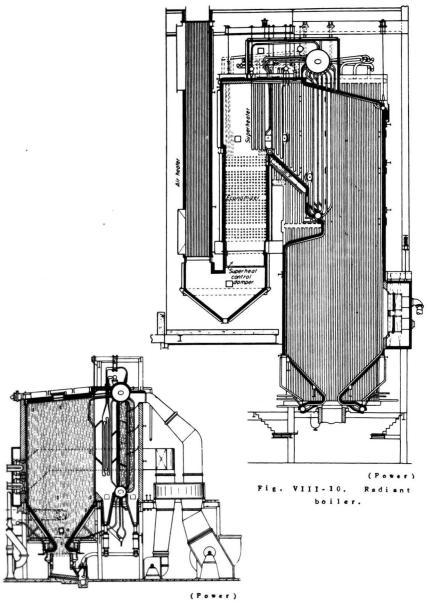


Fig. VIII-11. Two drum

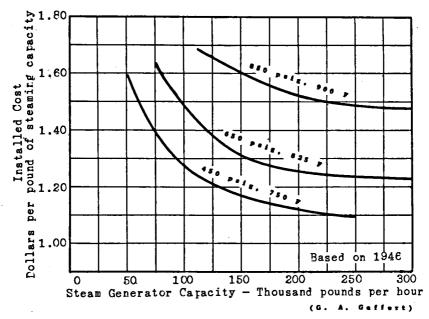


Fig. VIII-12. Boiler cost dats.

of overall heat transfer coefficients in water-tube boilers is the Babcock and Wilcox equation:

U = 2.00 + 0.0014 G

where: G = mass flow of gas, lbs/hr per sq ft
of projected flow area transverse
to the tubes.

Example 19 shows the method of estimating the surface required for a water-tube boiler.

Example 19.

A boiler using fuel oil of the analysis of Example 15 with a heating value of 18500 Btu/lt and with the same flue gas analysis is to produce 20,000 pounds of steam per hour at 150 psig and saturated from feed-water at 212 F. Temperature of flue gas 500 F, temperature of combustion air 80 F, temperature of combustion 2500 F, and temperature of fuel 180 F. The boiler is to be of the three-pass straight

tube sectional header type with 3 in. OD tubes spaced § in. on centers. There will be 24 tubes per bank, and the tubes will be 12 ft long. Petermine the number of banks required, neglecting the area of the sectional headers and the boiler drum. The furnace has plain brick walls. Assume that no refuse is formed.

Solution:

The heat absorbed by the boiler per pound of fuel fired can be determined by the method of losses, allowing about 1 o/o for unaccounted for.

$$Q_2 = 0.24(19.95)(500 - 80) =$$
 2010 Btu/lb
 $Q_3 = 0.01(1089 - 180 + 0.46[500]) =$ 11
 $Q_4 = 9(0.1270)(1089 - 180 + 0.46[500]) =$ 1300
 $Q_5 = \frac{0.4}{13.7 + 0.4}(10160)(0.83) =$ 239
 $Q_7 = 0.015(18500) =$ 278
 $Q_6 = 0.01(18500) =$ 185
Total losses 4023
 $Q_1 = 18500 - 4023 =$ 14477 Btu/lb

The weight of fuel fired per hour is:

$$M_f = \frac{20,000(1195.6 - 180.1)}{14477} = 1405 \text{ lbs/hr}$$

The heat transferred is:

$$Q = 20,000(1195.6 - 180.1) = 20,300,000 Btu/hr$$

The gas flow is:

$$M_{gw} = 21.08(1405) = 29600 \text{ lbs/hr}$$

The transverse gas flow area between tubes is:

$$A_f = \frac{(24-1)(5-3)12}{3(12)} = 15.35 \text{ sq ft}$$

from which,
$$G = \frac{29600}{15.35} = 1930 \text{ lbs/hr/sq ft}$$

and from the Babcock and Wilcox equation,

$$U = 2.00 + 0.0014(1930) = 4.70$$

The mean temperature difference is:

$$t_m = \frac{2500 - 500}{\log \frac{366 - 2500}{6366 - 500}} = 724 \text{ F}$$

and the boiler area required is:

$$A = \frac{20,300,000}{4.70(724)} = 5970 \text{ sq ft}$$

Each bank of tubes contains $12(\frac{3\pi}{12})(24) = 226 \text{ sq ft}$

therefore, the number of banks of tubes required is:

$$\frac{5970}{226} = 26 \text{ banks of tubes.}$$

58. Water-Cooled Furnaces--The water-cooled furnace was introduced as a means of reducing maintenance on furnace refractories. Acceptance was slow because of difficulties with water circulation, but a modern steam generator is almost all water-cooled furnace with only enough boiler surface to reduce the gas temperature to an economical point for use in economizers or air heaters. Reference to the illustrations of water-tube boilers will show the application of furnace cooling.

Heat transfer to furnace cooling surface is almost entirely by radiation and the heat transfer coefficients are extremely high. The amount of radiant heating surface required in a furnace is a function of several variables, such as the volume of the furnace, the temperature of the flue gas leaving the furnace, the sensible heat in the flue gas and the arrangement of the cooling surface.

Table XII gives some generally accepted data on values of heat release per unit of furnace volume.

The temperature of the flue gas leaving the furnace should always be at least 200 F and preferably 250 F less than the ash fusion temperature of the fuel used in order to avoid tightly adhering deposits in the convection sections of the steam generator. The result is that with fuels with very low

ash fusion temperatures (fuel oil has an ash fusion temperature of about 1600 F) the steam generator would necessarily be mostly furnace.

The sensible heat in the flue gas can be calculated from the following relation:

Sensible heat/lb fuel = LHV - h' - h'' + h'''

where: LHV = lower heating value of fuel

h! = heat lost by radiation and

convection from the boiler

setting

h'' = heat lost due to incomplete combustion and due to combustible in refuse

h!!! = heat gained by use of preheated air

TABLE XII
FURNACE HEAT RELEASE

KBtv per cubic foot per hour

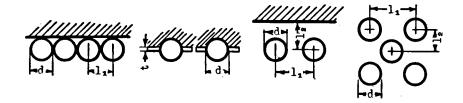
Fuel	All refrac-	Air	Water
	tory	cooled	cooled
Coal Fulverized	12 - 18	15 - 20	18 - 25
Stokers	15 - 25	25 - 40	30 - 45
Grates	25 - 40	35 - 45	40 - 50
Gas and Oil	20 - 35	25 - 40	30 - 50

Lower values in a range are for smaller boilers.

The arrangement of the cooling surface affects the effective area as shown in Figure VIII-13. The effective area is related to the projected area on the furnace envelope by the equation:

where: S, = effective radiant cooling surface, sq ft

s = a slagging factor (0.90 to 0.95) $S_p = projected$ tube area, sq ft $\delta = an$ arrangement factor



$$\delta = [\pi/2][1_1/d] \qquad \delta = \pi/2 + 6t \quad \delta = [\pi/2][1_1/d] \qquad \delta = 2$$
up to $1_1/d = 1.5$

$$(e_{ourtes}, bf p. Ven Hostrand Co., Inc.)$$

Fig. VIII-13. Arrangement of radiant heat transfer surface.

Figure VIII-14 shows some empirical data which ties together the preceeding factors in the estimation of furnace cooling area and Example 20 shows the method of estimation of the principal dimensions of a water-cooled furnace.

Example 20.

Determine for the boiler of Example 17:

- 1. The furnace volume required for full load operation using pulverized coal.
- 2. The amount of effective radiant furnace cooling surface required for a gas temperature of 1950 F entering the superheater (250 F less than the ash fusion temperature). Assume that the preheated air temperature is 535 F.
- 3. The number and length of 2½ in. OD furnace tubes, spaced on three inch centers. The boiler screen consists of three banks of 2½ in. OD tubes spaced on nine inch centers. The plan dimensions of the furnace are 20 ft X 20 ft.

Solution:

1. From Table XII, a heat release value of 22,000 Btu per cubic foot per hour is selected. The total heat re-

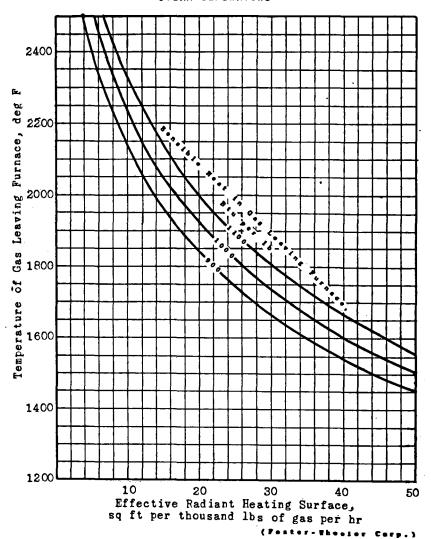


Fig. VIII-14. Furnace cooling surface design curvés.

leased in the furnace is:

$$Q = \frac{74}{8}(2000)(14400) = 266,000,000 \text{ Btu/hr}$$

from which the furnace volume is:

$$\nabla_f = \frac{266,000,000}{22000} = 12100 \text{ cu ft}$$

2. The sensible heat per pound of flue gas is:

$$SH = \frac{LHV - h' - h'' + h'''}{M_{gx}} + 0.24[535 - 70]$$

$$SH = \frac{14400 - [9(0.0447) + 0.0222][1066] - 81 - [123 + 497]}{15.30 + 9(0.0447) + 0.0222}$$

SH = 856 Btu/lb flue gas

Then, from Figure VIII-14, the area of effective radiant surface per thousand pounds of gas per hour is 14 sq ft. The total area is:

$$S_r = \frac{14(15.72)(2000)(74/8)}{1000} = 4070 \text{ sq ft}$$

3. The boiler screen will have 20 ft tubes, and there will be 20(12)/9=26 tubes per bank. Assuming a slagging factor of 0.90 and $\delta=2$, the effective radiant area of the boiler screen will be:

$$S_r = 0.90(2)(2.25/12)(20)(26)(3) = 525 \text{ sq ft}$$

The furnace walls will have an effective radiant surface of 4070-525=3545 sq ft. The length of the furnace wall tubes will be approximately the same as the height of the furnace, or 30.2 ft. Assuming a slagging factor of 0.90 and $\delta = (\pi/2)(3/2.25) = 2.09$, the projected area of the furnace wall tubes will be:

$$S_p = \frac{3545}{0.90(2.09)} = 1885 \text{ sq ft}$$

The furnace walls can each have 20(12)/3 = 80 tubes, each 30.2 ft long, for a total of:

$$S_{p_{malls}} = 4(80)(30.2)(2.25/12) = 1810 \text{ sq ft}$$

This leaves 75 sq ft of projected area for the furnace floor, and since each floor tube would have a projected area of 20(2.25)/12 = 3.75 sq ft, there would be 75/3.75 = 20 floor tubes.

59. Furnace Refractories and Insulation--In order to minize heat losses from furnaces, it is necessary to supply some sort of insulating material, and in order to prevent fire damage to insulation it is necessary to supply refractory.

Refractories are made from materials which have good resistance to damage from direct flame impingement. They are usually supplied in brick form, although plastic refractories are available. The standard refractory brick is $2\frac{1}{2}$ X $4\frac{1}{3}$ X 9°, and other sizes are available in multiples and submultiples of these dimensions as well as in special shapes. Figure VIII-15 shows some heat conductivity data for various refractory materials.

Insulation is made of material which has good resistance to heat flow but which is not necessarily resistant to direct flame impingement. Several companies make insulating refractories which are very useful. Figure VIII-16 shows some heat conductivity data for various insulating materials.

Heat flows through the furnace wall by conduction and leaves the outer wall by radiation and convection. It is usually assumed that the heat lost by convection is about 50 per cent of that lost by radiation, so that:

 $Q = 1.5 CA(T_1^4 - T_2^4)$

where: Q = heat loss, Btu/hr

C = a radiation coefficient

A = area of radiating surface, so ft

 T_1 = temperature of radiating

surface, deg R

T₂ = ambient temperature, deg R

The preceeding expression will be recognized as a modification of the Stefan-Boltzman relation, and values of the radiation coefficient can be found from Table XIII.

It is obvious that the heat lost from the wall surface must be equal to the heat flowing through the wall, so that the determination of the thickness of insulation required resolves itself into a trial and error process. Ordinarily it will be known how much structural brick is necessary on the outside of the furnace wall and how much refractory is used inside the furnace (from $2\frac{1}{2}$ " to $4\frac{1}{2}$ " for backing water walls and from 9" to $13\frac{1}{2}$ " for plain walls). Example 21

shows the method of procedure.

TABLE XIII

RADIATION COEFFICIENTS

Bare brick work	16.3 X 10-10
Unpainted steel	16.3 X 10 ⁻¹⁰
Asbestos cloth	16.0 X 10 ⁻¹⁰
Aluminum paint	11.0 X 10 ⁻¹⁰
Flat black paint	16.4 X 1C-10
Glossy black paint	14.0 X 10 ⁻¹⁰

Example 21.

Determine the thickness of insulation required for a furnace wall which consists of Armstrong insulation sandwiched between eight inches of hard burned red brick and nine inches of kaolin refractory brick. The furnace temperature is 2300 F and ambient temperature is 95 F. Desirable outside wall temperature is 175 F.

Solution:

The heat transferred through the wall is:

 $Q = 1.5[16.3 \times 10^{-1}][(635)^4 - (555)^4]$ Q = 164 Btu/hr/sq ft

Temperature drop through the wall is 2300 - 175 = 2125F. Temperature drops through the individual sections of the wall will be determined by trial.

Assume mean temperature of red brick = 300 F and mean temperature of kaolin brick = 2150 F. From Figure VIII-15, the values of k at these temperatures will be 6.5 for red brick and 15.0 for kaolin brick.

Then, for red brick, $\Delta t = Qd/k = 164(8)/6.5 = 202$ F, and for kaolin brick, $\Delta t = 164(9)/15.0 = 98.5$ F.

Recalculation of mean temperatures gives for red brick, 0.5[175 + (175 + 202)] = 276 F, and for kaolin brick, 0.5[2300 + (2300 - 98.5)] = 2251 F. At these mean temperatures, k for red brick is 6.4 and for kaolin brick is 15.5.

Then, for red brick, $\Delta t = 164(8)/6.4 = 205$ F, and for kaolin brick, $\Delta t = 164(9)/15.5 = 95.3$ F.

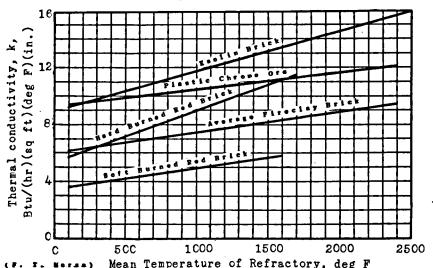


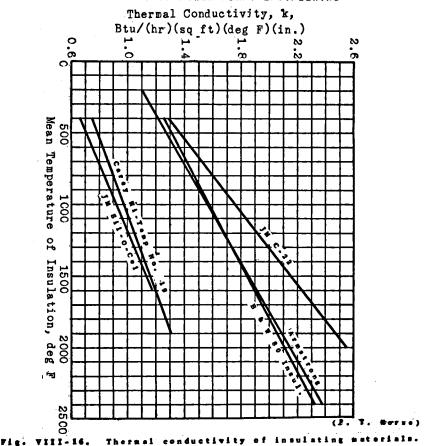
Fig. VIII-15. Thermal conductivity of refractory materials.

Recalculation of mean temperatures gives for red brick, 0.5[175 + (175 + 205)] = 278 F, and for kaolin brick, 0.5[2300 + (2300 - 95.3)] = 2257 F. These mean temperatures are reasonably close to the previous calculations so that further trials are not considered necessary.

The temperature drop for the insulation will be $\Delta t = (2300-95.3)-(175+295)=1825$ F, and the mean temperature is 0.5(2205+380)=1292 F. At this mean temperature, k for the insulation is 1.7, and the insulation thickness is $d = k\Delta t/Q = 1.7(1825)/164 = 18.9$ inches.

Example 21 shows clearly why insulated refractory and brick furnaces are seldom used in modern boiler installations. Better arrangements are water-cooled or air-cooled walls.

In the water-cooled wall the inside wall temperature is reduced almost to the saturation temperature of the steam in the boiler, thus necessitating less refractory and insulation. As previously mentioned, water walls are backed with up to $4\frac{1}{7}$ of refractory, and in addition a layer of Superex block and 3" of rockwool (conductivities of these materials can be found from Figure X-13) with an outside casing of sheet metal are usually used to complete the wall. The effect of the metal casing can be ignored with respect to the



amount of insulation required, but the exterior finish (cclor of paint) must be considered in determination of the heat loss from the surface.

In the air-cooled wall the refractory and structural brick are separated by an air space which serves as a conduit for combustion air. Thus, most of the heat passing through the refractory is salvaged and returned to the furnace.

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CHAPTER IX

SUPERHEATERS, REHEATERS, ECONOMIZERS AND AIR HEATERS

60. Superheaters -- The economies to be realized with superheated steam in a vapor power cycle have been set forth previously under the discussion of the various cycles. Superheaters are always made up of tubular elements and they are classified as to their arrangement in the boiler, that is, pendant or drainable, convection or radiant.

Pendant superheaters are hung from headers which are located at the top of the boiler. This is the simplest arrangement, but they are difficult to clean. Figure IX-1 shows a pendant superheater in a bent tube boiler.

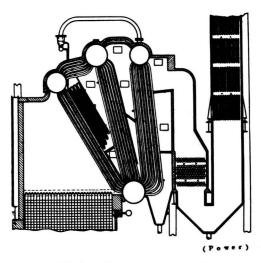


Fig. IX-1. Pendant superheater in bent tube boiler.

Drainable superheaters are mounted in such a way that condensation in the elements can drain back to the headers. This type of superheater is more difficult to place in the boiler because it requires that all of the element slope toward the headers. but they are easy to Figure IX-2 clean. shows a drainable superheater in a straight tube boiler.

Convection superheaters are so named because they receive all or a major portion of their

heat by convection from the flue gas. They have a rising temperature characteristic because of increase in heat transfer coefficients with increasing gas velocities. For this reason, modern boilers which contain convection superheaters alone have some means of bypassing part of the flue gas around part of the superheater to effect temperature control in the steam from the superheater. The temperature control damper can be seen in Figure VIII-10.

Radiant superheaters receive all or most of their heat through radiation because they are mounted where they can "see" the fire in the furnace. They have a drooping temperature characterístic because the temperature of the fire in the furnace remains essentially constant throughout the load range. Radiant superheaters are often placed in series with convection superheaters, and proper proportioning of the areas in each will result in a nearly constant steam temperature characteristic over the

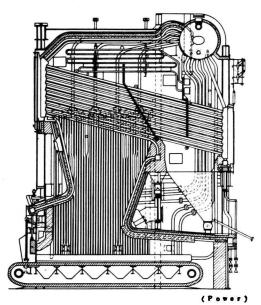


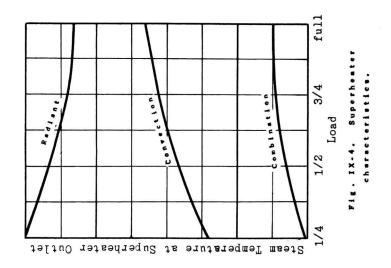
Fig. IX-2. Drainable superheater in straight tube boiler.

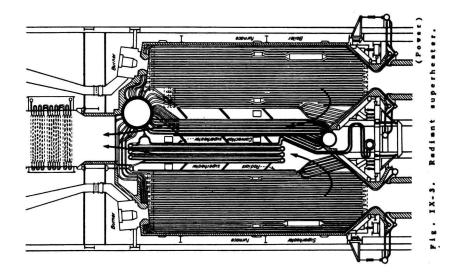
normal boiler operating range. Figure IX-3 shows a radiant superheater mounted in series with a convection superheater, and Figure IX-4 shows the temperature characteristics.

61. Heat Transfer in Superheaters--Heat transfer in a convection superheater involves a decreasing flue gas temperature and an increasing steam temperature. Wherever possible the surfaces are arranged so that the gases enter the superheater at the same end at which the steam leaves. This counterflow principle results in more uniform temperature differences and higher steam outlet temperatures in comparison with parallel flow arrangements. Figure IX-5 shows a comparison of the temperatures existing in the two arrangements.

The logarithmic mean temperature difference for the counterflow case can be found from the relation:

$$t_{m} = \frac{(t_{f_{i}} - t_{v_{i}}) - (t_{f_{i}} - t_{v_{i}})}{\log \frac{t_{f_{i}} - t_{v_{i}}}{et_{f_{i}} - t_{v_{i}}}}$$





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where: t = log mean temperature difference, deg F t and t = inlet and outlet flue gas temperatures, deg F t and t = inlet and outlet steam temperatures, deg F

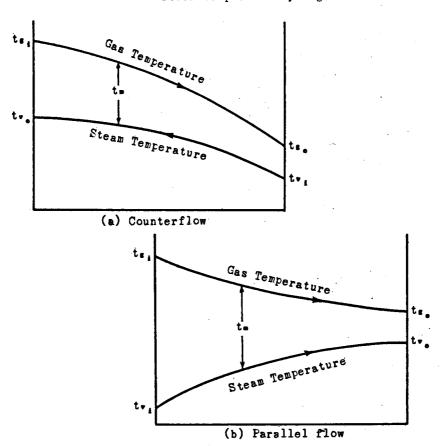


Fig. IX-5. Temperature relations in heat transfer and for the parallel flow case from the relation:

ì

$$t_{m} = \frac{(t_{g_{i}} - t_{v_{i}}) - (t_{g_{0}} - t_{v_{0}})}{\log_{e} \frac{t_{g_{i}} - t_{v_{i}}}{t_{g_{0}} - t_{v_{0}}}}$$

The overall heat transfer coefficient for a convection superheater is calculated in a manner similar to that for other heat transfer apparatus. There is no reliable empirical equation for the overall heat transfer coefficient as there is for boilers, so that is is necessary to estimate film coefficients and combine them according to the usual manner with the addition of a slagging factor, as follows:

$$U = s \frac{1}{\frac{1}{h_{g}} + \frac{2\Gamma}{D_{o} + oD_{i}} \frac{1}{K_{t}} + \frac{D_{o}}{D_{i}} \frac{1}{h_{v}}}$$

The gas film coefficient can be estimated from the equation:

$$h_g = \frac{0.41 \, v_g^{0.585} \, T^{0.315}}{\Gamma_0^{0.415}}$$

where: v = mass flow of gas, lbs/sec
per sq ft of gas flow area
T = arithmetic mean gas temperature, deg R
D = outside diameter of tube, ft

The steam film coefficient can be approximated from the following equation:

$$h_{\nu} = 8.0 \, v_{\nu}^{0.8}$$

where: v, = mass flow of steam, lbs/sec per sq ft of steam flow area

The Nusselt equation for radiant superheaters is used for the determination of the overall heat transfer coefficient, as follows:

$$U = \frac{0.0333 \, k^{0.214} \, (G \, c_p)^{0.786}}{D_i^{0.214}}$$

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Example 22-A shows the method of calculating the dimensions of a convection superheater and Example 22-B shows the procedure for a radiant superheater.

Example 22-A.

Determine the number and length of tubes required for a convection superheater for the boiler of Examples 17 and 20. The tubes are 2 in. OD X 8 BWG and are spaced on 4½ in. centers. Assume counterflow and a slagging factor of 0.90. Dispose of the superheater in two of the three boiler passes.

Solution:

$$M_{gw} = (74/8)(2000)(15.72) = 291,000 lbs/hr$$
 $Q = 200,000(1450.9 - 1197.8) = 50,600,000 Btu/hr$

$$\Delta t_{flwe} g_{gs} = \frac{50,600,000}{291000(0.24)} = 725 F$$

Temperature	of	gas at superheater inlet	1950	F
Temperature	of	gas at superheater outlet	1225	F
Temperature	οf	steam at superheater inlet	535	F
Temperature	of	steam at superheater outlet	900	F

$$t_{m} = \frac{(1950 - 900) - (1225 - 535)}{\log_{e} 1225 - 535} = 860 \text{ F}$$

Since the boiler is 20 ft wide, there will be 20(12)/4.5 = 53 tubes in the superheater. Each tube has a flow area of $[\pi/4][2.0 - 2(0.165)]^2/144 = 0.0152$ sq ft, and the total steam flow area is 53(0.0152) = 0.806 sq ft.

The mass flow of steam through the tubes will be:

$$v_{\nu} = \frac{200000}{3600(0.806)} = 69.0 \text{ lbs/sec/sq ft}$$

and,
$$h_{\nu} = 8.0(69.0)^{\circ.8} = 236$$

If the superheater is disposed in two of three boiler passes, each pass will be 20 ft wide by 6.67 ft deep. Then the area for gas flow between the superheater tubes will be (53-1)(4.5-2)(6.67)/12=23 sq ft.

The mass flow of gas between the tubes will be:

$$v_g = \frac{291000}{3600(23)} = 3.52 \text{ lbs/sec/sq ft}$$

and,
$$h_g' = \frac{0.41(3.52)^{\circ.888} (2048)^{\circ.818}}{(2.0/12)^{\circ.415}} = 19.8$$

Then: the overall heat transfer coefficient is:

$$U = (0.90) \frac{1}{\frac{1}{19.8} + \frac{4.0}{2.0 + 1.67} \left[\frac{0.165}{12(30)} \right] + \frac{2.0}{1.67} \left[\frac{1}{236} \right]} = 16.1$$

Superheater area is:

$$A = \frac{50,600,000}{16.1(860)} = 3666 \text{ sq ft}$$

Each foot of superheater tube has an area of $2\pi/12 =$ 0.523 sq ft. There will be 3666/0.523 = 7000 linear feet of tube, and the length of each element will be 7000/53 = 132 feet.

Example 22-B.

The boiler of Examples 17 and 20 is to be fitted with a combination convection-radiant superheater. Half of the temperature rise in the superheated steam is to be supplied by the radiant section which is in series with and follows the convection section. The radiant superheater is to replace part of the rear furnace wall cooling surface. The tubes will be 2½ in. OD X 8 BWG spaced on 3 in. centers with a length of 5 feet. Find the number of tubes in the radiant superheater.

Solution:

Steam temperature at rad. suphtr. inlet
Steam temperature at rad. suphtr. outlet
900 F
Temperature in furnace
1950 F

$$t_m = \frac{900 - 718}{\log \frac{1950 - 718}{1950 - 900}} = 1134 \text{ F}$$

The heat transferred in the radiant superheater is:

Q = 200,000(1450.9 - 1342.1) = 21,800,000 Btu/hr

Assume, as a first approximation, that U=400 based on effective radiant surface. Then the effective radiant surface required would be 21800000/[(400)(1134)]=48 sq ft. One linear foot of 2.25 in. OD tube, in this particular wall arrangement, has an effective radiant surface of $0.90(\pi/2)(3.0/2.25)(2.25/12)=0.354$ sq ft. Then the linear length of tube required would be 48/0.354=136 ft. If each tube is 5 ft long, there would be 136/5=27 tubes in the radiant superheater.

It is necessary to check the assumed value of $\overline{\mathbf{U}}$ by use of the Nusselt equation, as follows:

The steam flow area per tube is $[\pi/4][2.25-2(0.165)]/144$ = 0.02 sq ft, and the total steam flow area is 27(0.02) = 0.54 sq ft. The mass flow of steam is G = 200000/0.54 = 370000 lbs/hr/sq ft. The conductivity of the steam is k = 0.0131 + 0.000029(809) = 0.0365 and the mean specific heat is 0.62, then the overall heat transfer coefficient is:

$$U = 0.0333 \left[\frac{0.0365}{1.92/12} \right]^{0.214} [370000(0.62)]^{0.784} = 404$$

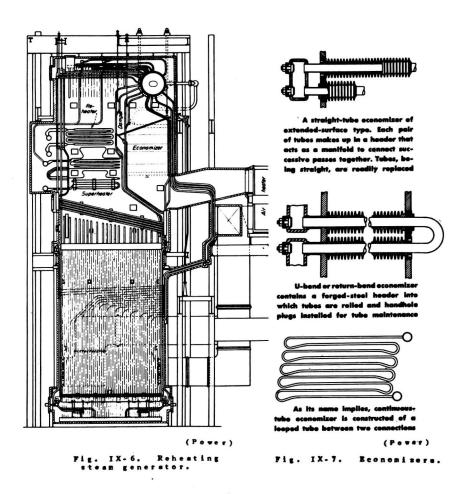
This value of \boldsymbol{U} checks very closely with the assumed value.

- 62. Reheaters--All reheaters which are presently being built use flue gas as the heat source. For this reason, the reheater may be considered to be the same as a superheater. Figure IX-6 shows a modern reheating steam generator.
- 63. Economizers -- An economizer is essentially a feed-water heater employing flue gas as the heat source. Its primary purpose is to lower the temperature of the flue gas leaving the steam generator in order to increase the efficiency. Economizers are made in several styles, some of

which are shown in Figure IX-7.

64. Heat Transfer in Economizers -- Heat transfer in an economizer is from flue gas at decreasing temperature to water at increasing temperature. The overall heat transfer coefficient may be approximated from the same Babcock and Wilcox equation used for boilers.

Example 23 shows the method of estimating economizer surface.



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Example 23.

Determine the number and length of tubes required for an economizer for the boiler of Examples 17 and 20. The tubes are 2 in. OD X 8 BWG and are spaced on 3 in. centers. Tubes run lengthwise in the economizer casing whose plan dimensions are 12 ft X 18 ft. The water flows counterflow to the gas. Gas inlet temperature is 1225 F, water inlet temperature is 300 F and water outlet temperature 485 F.

Solution:

The heat transferred in the economizer is:

$$Q = 20000C(470.2 - 269.6) = 40,100,000 Btv/hr$$

The weight of wet flue gas is 291000 lbs/hr, so that the flue gas temperature drop is 40100000/[291000(0.24)] = 575 F. Then the flue gas temperature leaving the economizer is 1225 - 575 = 650 F.

$$t_m = \frac{(1225 - 485) - (650 - 300)}{\log_e \frac{1225 - 485}{650 - 300}} = 520 \text{ F}$$

Since the economizer casing is 12 ft wide, there will be 12(12)/3 = 48 tubes giving a gas flow area of (48-1)(3-2)(18)/12 = 70.5 sq ft and a gas mass flow G = 291000/70.5 = 4130 lbs/hr/sq ft.

Then, the overall heat transfer coefficient is:

$$U = 2.0 + 0.0014(4130) = 7.78$$

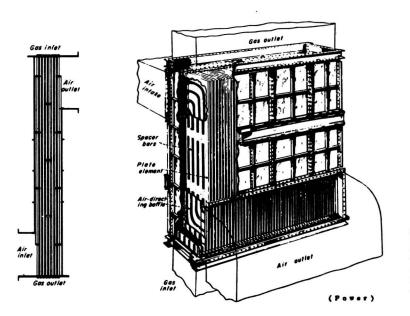
and,
$$A = \frac{40,100,000}{7.78(520)} = 9300 \text{ sq ft}$$

Each linear foot of economizer tube has a surface area of 0.523 sq ft, so that there will be 9900/0.523 = 18950 linear feet of tube. Then, the length of each tube will be 18950/48 = 395 feet.

65. Air Heaters--Air heaters are also devices for salvaging heat from the flue gas in order to improve overall steam generator efficiency. They are of two general types: recuperative, either tubular or plate; and regenerative.

Recuperative air keaters depend upon the flow of heat from the flue gas through a dividing wall to the air. Ord-

inarily, in the tubular type the air flows around the outside of the tubes and the flue gas passes through the tubes. In the plate type the gas and air flow through alternate passages. Figure IX-8 shows the tubular and plate types of air heater.



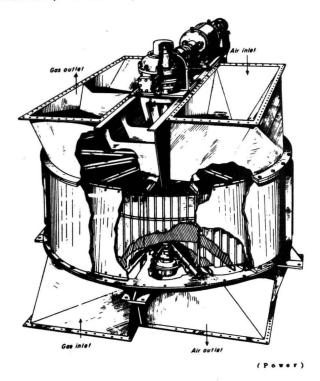
(a) Tubular Type

(b) Plate Type

Pie. IX.8. Recuperative air heaters.

Regenerative air heaters depend upon the transfer of heat from gas to air by the alternate heating and cooling of a mass of metal or other heat storing material. Figure IX-9 shows the principal type of regenerative air heater in use at the present time. This type uses corrugated steel sheets for the heat storage material.

66. Heat Transfer in Air Heaters--Heat transfer in air heaters is a similar mechanism to those previously discussed. For tubular air heaters where the gas flows through the tubes and the air flows across the tubes, the overall heat transfer coefficient can be estimated from the equation:



wis ry. 9. Regenerative air heater.

U = 0.45 + 0.00045 G

where: G = arithmetic mean mass flow of air and gas, lbs/hr/sq ft

For plate type air heaters the overall coefficient of heat transfer can be approximated from the following equation:

II = 0.28 V0.75

where: V = arithmetic mean velocity
of gas and air, ft/sec

Regenerative air heaters do not depend upon the transfer of heat through a dividing wall for their action. The temperature relations are somewhat complicated, but, for purposes of estimation, it can be assumed that (with the rotor turning at the usual speed of three revolutions per minute) the weight of metal in the rotor will be:

$$W_{m} = \frac{M_{g}^{2}}{93.8}$$

where: W_m = weight of metal, lbs M_g = gas flow, lbs/hr

and, if 20 USG steel sheets are used, the area would be (one side of the sheet):

$$A = \frac{M_{e}}{140.7}$$

where: A = area (one side) of 20 U.S. gage steel sheets, square feet

Example 24 shows the method of estimation of the dimensions of a tubular air heater.

Example 24.

Determine the number and length of tubes and number of passes on the air side required for a tubular air heater for the boiler of Examples 17 and 20. The tubes are $1\frac{1}{2}$ in. OD X 14 BWG and are spaced on $2\frac{1}{2}$ in. centers. The gas flows through the tubes in a single pass and the air flows across the tubes. Plan dimensions of the air heater casing are 12 ft X 18 ft and the tubes run vertically. The air inlet is 6 ft X 12 ft and placed so that counterflow is obtained.

Solution:

From previous examples:

Air	temperature	entering air heater	110 F	ř
Gas	temperature	entering air heater	650 F	P
Gas	temperature	leaving air heater	250 F	7

Heat transferred in the air heater:

Q = 291,000(0.24)(650 - 250) = 28,000,000 Btu/hr

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Weight of air heated:

$$M_{aa} = [74/8][2000][15.30+8(0.0447-\frac{0.0495}{8})-0.7873-0.0063$$

 $M_{aa} = 274,000 \text{ lbs/hr}$

Temperature rise in the air is 28000000/[274000(0.24)] = 425 F and the temperature of the air leaving the air heater is 110 + 425 = 535 F.

$$t_m = \frac{(650 - 425) - (250 - 110)}{\log_{e} \frac{650 - 425}{250 - 110}} = 179 \text{ F}$$

Since the air heater tube sheet is 12 ft X 18 ft, there will be [12(12)/2.5][18(12)/2.5] = 5000 tubes in the air heater. The gas flow area is $5000[\pi/4][1.5 - 2(0.083)]^2/144 = 48.5$ sq ft, and the mass flow of gas is $G_g = 291000/48.5 = 6000$ lbs/hr/sq ft.

Since the air enters the short side of the air heater, the air passes between 58 tubes in the first row. The air flow area per pass is (58-1)(2.5-1.5)(6)/12=28.5 sq ft, and the mass flow of air is $G_{\alpha}=274000/28.5=9600$ lbs/hr per sq ft.

The overall coefficient of heat transfer is:

$$0 = 0.45 + 0.00045(\frac{9600 + 6000}{2}) = 3.96$$

and,
$$A = \frac{28,000,000}{3.96(179)} = 39,500 \text{ sq ft.}$$

Each $1\frac{1}{2}$ in. OD tube has an area of $1.5\pi/12 = 0.393$ sq ft per linear foot, so the total length of tube required would be 39500/0.393 = 100,500 ft, and since there are 5000 tubes, the tubes would be 100500/5000 = 20.1 feet long. Each air pass is 6 feet high, so there would be 2 passes on the air side.

67. Apportionment of Heating Surface--The relative amounts of heating surface provided in a complete steam generator installation is a compromise between the economic factors and the practical necessities. For example, air heater surface is the cheapest, but combustion considerations limit air temperatures to about 600 F for pulverized coal, oil or gas, and to about 350 F for stokers. In order to avoid corrosion

difficulties, the minimum gas outlet temperature for air heaters should be about 250 F. It is usual to limit the water temperature leaving the economizer to about 5C F less than the saturation temperature in the boiler in order to avoid steaming in the economizer.

Equations can be set up to give the most economical temperatures at the various locations after the superheater in the steam generator. The temperature of the gas leaving the furnace is fixed by the characteristics of the fuel; the temperature of the water entering the economizer is fixed by the most efficient operation of the regenerative cycle; the temperatures of the gas and air leaving the air heater are fixed as previously mentioned; and the temperature of the air entering the air heater is fixed by ambient conditions. The boiler screen is ordinarily two or three rows deep and receives the major portion of its heat by radiation, so that it may be included as part of the furnace cooling surface. Then, the temperature of the gas entering the superheater will be fixed as the temperature of the gas leaving the furnace. The temperature of the gas leaving the superheater will be determined by the necessities of heat transfer in the superheater. The boiler, economizer and air heater are then subject to economic analysis as follows (according to Helander):

Let: C = annual charges per square foot of additional surface, dollars

U = heat transfer coefficient for surface
Cf = annual saving per Btu of heat recovered per
hour. dollars

value of Btu X hours service per yearvoverall steam generator efficiency

Ma, Mg and Mm = air, gas and water flows, lbs/hr

tg = gas temperatures, deg F

ta = air temperatures, deg F

tm = water temperatures, deg F

ty = saturation temperature in boiler, deg F

a, b, e, i and o are subscripts referring to air heater, boiler, economizer, in and out

To determine the most economical temperature of the outlet gas from an economizer when the inlet gas and inlet water temperatures are fixed:

SUPERHEATERS, REHEATERS, ECONOMIZERS AND AIR HEATERS 171

$$\frac{C_e}{U_eC_f} = \frac{(t_{goe} - t_{mie})(t_{gie} - t_{moe})}{(t_{gie} - t_{mie})}$$

$$\frac{0.24 \text{ Mg}}{\text{cpm Mm}} = \frac{(\text{tmoe} - \text{tmie})}{(\text{tgie} - \text{tgoe})} = e$$

To determine the most economical temperature of the outlet gas from an air heater when the inlet gas and inlet air temperatures are fixed:

$$\frac{C_a}{U_a C_f} = \frac{(t_{goa} - t_{aia})(t_{gia} - t_{aoa})}{(t_{gia} - t_{aia})}$$

$$\frac{M_g}{M_a} = \frac{(t_{aoa} - t_{aia})}{(t_{gia} - t_{goa})} = a$$

To determine the most economical boiler outlet gas temperature and water temperature leaving the economizer in the case of a boiler-economizer combination when the economizer gas outlet temperature and inlet water temperature are fixed:

$$\frac{C_e U_b}{U_e C_b} = \frac{(t_{gie} - t_{noe})}{(t_{gie} - t_v)}$$

To determine the most economical boiler outlet gas temperature and air temperature leaving the air heater in the case of a boiler-air heater combination when the air heater gas outlet temperature and entering air temperature are fixed:

$$\frac{C_aU_b}{U_aC_b} = \frac{(t_{gia} - t_{aoa})}{(t_{gia} - t_v)}$$

To determine the most econorical boiler and economizer outlet gas temperatures, water temperature leaving the economizer and air temperature leaving the air heater in the case of a boiler-economizer-air heater combination when the air heater gas outlet temperature, entering air temperature and water inlet temperature are fixed:

$$\frac{C_bU_b}{C_cU_b} = \frac{(t_{gis} - t_v)}{(t_{gis} - t_{ros})}$$

$$\frac{U_eC_a}{C_eU_a} = \frac{(t_{gie} - t_{nie})(t_{gia} - t_{aoa})}{(t_{gie} - t_{noe})(t_{gia} - t_{nie})}$$

The following notes will be helpful in coming to a decision as to the proper combination of boiler and heat recovery apparatus:

- 1. If the water cutlet temperature in a boiler-economizer combination is not at least 50 F less than the saturation temperature in the boiler, try adding an air heater to the combination.
- 2. If the air cutlet temperature from the air heater in a boiler-air heater combination is greater than 6CC F for pulverized coal, oil or gas, or 350 F for stokers, try adding an economizer to the combination.

Table XIV gives some mass flows and heat transfer coefficients to be expected in commercial practice. After making an estimate from the table, the final values of economical temperatures can be arrived at by successive approximations.

TABLE XIV

GAS MASS FLOWS AND COEFFICIENTS OF HEAT TRANSFER FOR

STEAM GENERATOR COMPONENTS

	G	U		
	Lbs/hr/sq ft	Btu/hr/sq ft/deg		
Boiler	1000 to 5000	3 to 8		
Superheater	2000 to 9000	10 to 25		
Economizer	2000 to 9000	6.5 to 9		
Air Heater	2000 to 9000	2.5 to 4		

Example 25 shows the method of application of the preceding equations to the determination of the most economical division of heat transfer surface.

Example 25.

Determine the most economical temperatures for the boil-

er of Examples 17 and 20, given the following additional data:

Annual fixed charges	15 o/o
Cost of additional boiler surface	\$10.00 /sq ft
Cost of additional econ. surface	£ 4.00 / sq ft
Cost of additional air htr. surface	£ 1.00 /sq ft
Boiler heat transfer coefficient	4.7
Econ. heat transfer coefficient	7.8
Air htr. heat transfer coefficient	4.0
Cost of fuel	\$ 8.00 /ton
Hours operated per year	8000

Solution:

$$C_f = \frac{\$8.00(8000)}{2000(14400)(0.887)} = 0.00251$$

$$C_b = 10(0.15) = 1.50$$

$$C_e = 4(0.15) = 0.60$$

$$C_c = 1(0.15) = 0.15$$

1. Assuming a boiler-economizer combination:

The temperature of the water entering the economizer is greater than the temperature specified for the flue gas outlet temperature, therefore this combination would be impossible.

2. Assuming a boiler-air heater combination:

$$t_{goa} = 250 \text{ f} t_{aia} = 110 \text{ f} t_{v} = 535 \text{ f}$$

$$\frac{0.15(4.7)}{4.0(1.50)} = 0.1175 = \frac{(t_{gia} - t_{aoa})}{(t_{gia} - 535)}$$

$$a = \frac{291000}{274000} = 1.06 = \frac{(t_{aoa} - 110)}{(t_{gia} - 250)}$$

from which, $t_{gia} = 1230 \text{ F}$ and $t_{aoa} = 1145 \text{ F}$.

These temperatures indicate that no boiler surface could be justified, but the temperature of the air leaving the air heater is too high for satisfactory operation.

3. Assuming an economizer-air heater combination:

$$\frac{7.8(0.15)}{0.60(4.0)} = 0.488 = \frac{(1225 - 300)(t_{gia} - t_{aoa})}{(1225 - t_{aoa})(t_{gia} - 300)}$$
[A]

$$\frac{\mathbf{t}_{xoe} - 300}{1225 - \mathbf{t}_{gia}} = 0.322 \qquad \frac{\mathbf{t}_{xoa} + 110}{\mathbf{t}_{gia} - 250} = 1.06$$

Assume for a first approximation that $t_{zos} = 485$ F, then, $t_{gis} = 650$ F and $t_{zos} = 535$ F. Substitution of these temperatures in equation [A] gives a ratio of 0.414 instead of the required value of 0.488.

Assume for a second approximation that $t_{aos} = 500 \text{ F}$, then, $t_{cia} = 602 \text{ F}$ and $t_{aoa} = 483 \text{ F}$. Substitution of these temperatures in equation [A] gives a ratio of 0.503.

Interpolation gives $t_{\text{mos}} = 498$ F, however, this temperature is too close to the saturation temperature in the boiler so the first approximation will be taken as most satisfactory.

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CHAPTER X

PIPING AND DUCT-WORK

- 68. Power plants require conduit for the transportation of fuel, air, gas, water and steam. Conduit for air and gas are called ducts, and for other materials are called pipe. The variety of conduit is large, but the laws of fluid flow apply to all.
- 69. Piping--Calculations for piping involve the determination of pipe size, pressure class and pressure drop.

Pipe size is determined from the fluid velocity which will produce a pressure drop which is not excessive. Table XV gives practical velocities for various fluids and vapors.

TABLE XV
PIPE LINE VELOCITIES

	ft per sec
Low press. sat. steam	65 - 100
High press. sat. steam	100 - 170
High press. suphtd. steam	170 - 250
Atmos. exhaust steam	130 - 200
Low press. exhaust steam	330 - 400
Boiler feed-water	8 - 12
Pump suction, water	4 - 7
General service water	5 - 10
Fuel oil	1 - 4

Application of the continuity equation will give the solution for pipe size, as follows:

$$A = \frac{Mv}{3600 \text{ V}} \qquad d_i = 0.4 \sqrt{Mv/\pi V}$$

where: M = flow, lbs/hr

v = specific volume, cu ft/lb

V = velccity, ft/sec

d: = inside diameter of pipe, in.

A = flow area, sq ft

the

is determined

1000 F

TABLE XVI

(Crane Co.)

		·		
Dimensions (of Welded	and Seam	less Steel Pine	

Nominal	Out-	Ī	Nominal Wall Thicknesses for Schedule Numbers								
Pipe Size	side Diam.	Schedule 10	Schedule 20	Schedule 30	Schedule 40	Schedule 60	Schedule 80	Schedule 100	Schedule 120	Schedule 140	Schedule 160
16	.405 .540				.068 .088		.095 .119				
),),	.675				091		126	****		::::	
13	.840 1.050			• • • • •	.109 .113		.147		••••		.187
1	1.315				.133		.179		• • • • •		.250
11/2	1.660 1.900				.140		.191 .200	. ::::	::::	1 ::::	.250 .281
2 21/2	2.375 2.875				.154 .203		.218 .276				.343
3	3.500			::::	.216		.300				.437
33/2	4.00 4.500	::::			.226 .237	::::	.318 .337	::::	.437	l :::;	.53i
5	5.563 6.625			••••	. 258 . 280	• • • • • •	.375 .432		.500 .562		.625 .718
.8	8.625	::::	250	277	.322	.406	.500	.593	.718	.812	.906
10 12	10.75 12.75		.250 250	.307 .330	.365 .406	.500 .562	.593 .687	.718 .843	.843 1.000	1.000 1.125	1.125
14 O.D. 16 O.D.	14.00	.250	.312	.375 .375	.437 .500	.593	.750 .843	1.031	1.062	1.250	1.406 1.562
18 O.D.	18.00	.250	.312	.437	.562	.718	.937	1.156	1.343	1.562	1.750
20 O.D. 24 O.D.	20 00 24 00	.250 .250	.375 .375	.500 .562	.593 .687	.812 .937	1.031	1.250	1.500 1.750	1750	

Dimensions of Welded Wrought-Iron Pipe

Nominal	Outside	Nominal Wall Thicknesses for Schedule Numbers					
Pipe Size	Diameter	Schedule 10	Schedule 20	Schedule 30	Schedule 40	Schedule 60	Schedule 80
1/8	.405	1			.070		.098
14	.540		• • • •		.090		.122
	.675				.093		.129
1 3	.840 1.050				.111		.151 .157
74	1.315	1 ::::	::::		136		.183
i ₁₄	1.660	1		1	.143		.195
11/2	1.900				.148		.204

P = working pressure in pipe,
lbs/sq in. gage

TABLE XVI continued

,	1 2.375 1	 	ı ı	.158	۱ ا	.223
21/2	2 875	 1		.208		.282
3 2	3,500	 		. 221		.306
31/2	4.000	 		.231	••••	.325
4	4.500	 		. 242		.344
5	5.563	 		.263		.383
6	6.625	 	*:::	.286		.441
8	8.625	 	.283	.329	****	510
10	10.75	 • • • • •	.313 336	.372	.510	
13	1 12 75 1	1	1 339	1		

Note: Bold face numbers in Scheduses 40 and 80 correspond to standard and extra strong weights respectively.

A.P.I. and A.S.T.M. Specifications for Pipe, With Tensile Strengths
Called for and the Corresponding Uses for Which Each Material Is Intended

Title	Tensile Strength Lbs. per Sq. In.	Scope
Line Pipe Specifications (A.P.I. 5-L)	l'urnace welded: Bessemer, 50,000 min Open-hearth, Class I, 45,000 min Open-hearth, Class II, 48,000 min Wrought iron, 42,000 min Seamless or electric welded: Open-hearth, Grade B, 60,000 min Open-hearth, Grade B, 60,000 min Open-hearth, Grade C, 75,000 min Seamless or furnace welded: Open-hearth iron, 42,000 min	Pipe for line pipe purposes, to convey gas, water, or oil. Sizes covered, ½ inch nominal size to 24-inch outside diameter. Couplings for threaded line pipe are of special design. Threads on pipe and in couplings are subject to official gage limits.
Seamless Carbon-Molybdenum Alloy-Steel Pipe for Service at Temperatures from 750 to 1000 F. (A.S.T.M. A206	55,000 min	Seamless carbon—½ per cent molybdcnum alloy steel pipe intended for service at metal temperatures from 750 to 1000 F. Suitable for bending, flanging, and similar forming operations, and for fusion welding.
Seamless Alloy-Steel Pipe for Service at Temperatures From 750 to 1100 F (A.S.T.M. A158	Ferritic, 60,000 min Austenitic, 75,000 min	Seamless alloy-steel pipe intended for service at metal temperatures from 750 to 1100 F. Several classes of materials that have been rather extensively used are included. Choice from the respective steels should be made on the basis of requirements of design, service conditions, and the physical properties.

for the terials. gives lat the for the 73 present . of ultimate allowable lbs/sq data for time, a. grades in. working stress (10 for the various pipe sched-along with minimum tensile is of steel and wrought iron tensile ť 15 strength) per cent

Table XVI ules made strengths piping material

Pressure drop is determined by the application of the pipe friction formula, as follows:

$$\Delta P = 0.00518 \frac{f L V^2}{V d_i} \qquad \Delta h = 0.745 \frac{f L V^2}{d_i}$$

where: ΔP = pressure drop, lbs/sq in.

 Δh = pressure drop, ft of fluid

f = a friction factor

L = equivalent length of pipe, ft

V = average fluid velocity, ft/sec

v = specific volume of fluid, cu ft/lb

d; = inside diameter of pipe, in.

The equivalent length of pipe is equal to the sum of the length of straight pipe plus an allowance for valves and fittings. Figure X-1 gives the equivalent pipe length allowances.

The friction factor is a function of Reynolds' number, which is in turn a function of the size of the pipe, and the velocity, density and viscosity of the fluid. That is,

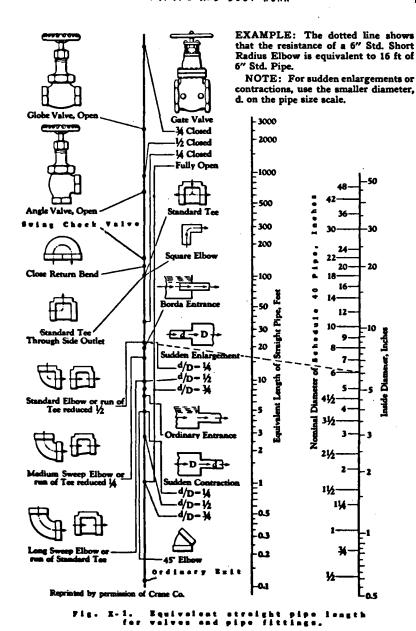
$$r = 122.5 \frac{d_i V}{z v}$$

where: r = Reynolds' number
z = viscosity of fluid, centipoises

Figures X-2 and X-3 show curves of viscosity for water, steam and oils, and specific gravity for oils. Figure X-4 shows the relation of friction factor to Reynolds' number.

Pipe fittings are specified by pressure class and material. Figures X-5 and X-6 and Table XVII show some piping details and standard types of fittings. Figure X-7 shows some of the various types of valves used in power plants.

Examples 26-A and 26-B show the method of calculation of pipe data based on the preceeding information.



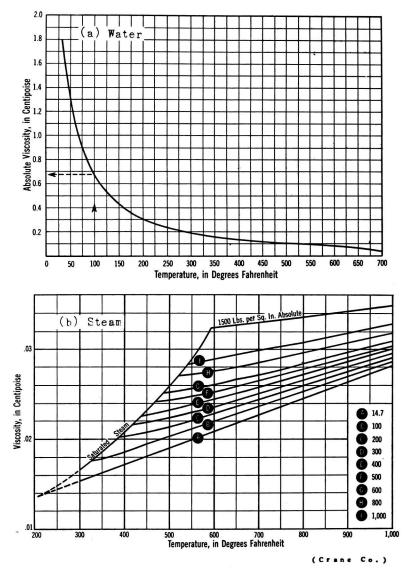
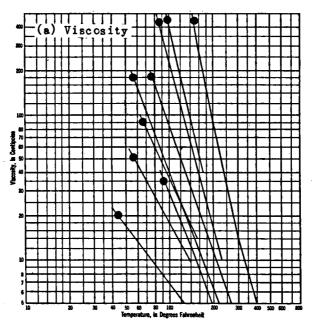


Fig. X-2. Viscosity of water and steam.



(1) Midcontinent residuum, S=0.97; (2) California heavy crude, S=0.976; (3) Residuum, S=0.94; (4) Eastern light lubricating oil, SAE 30; (5) Midcontinent light lubricating oil, S=0.90; (6) Olive oil; (7) Eastern light lubricating oil, SAE 10; (8) California light crude, S=0.91; (9) Midcontinent crude, S=0.85.

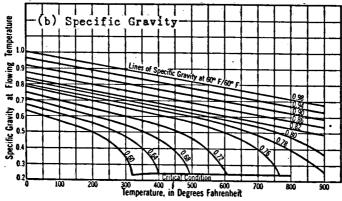
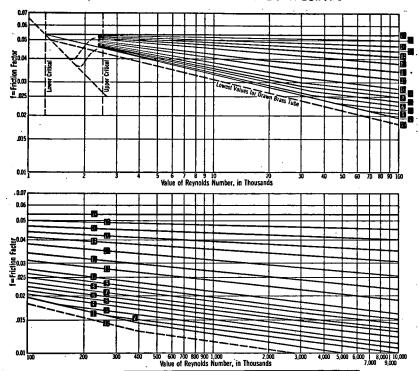


Fig. X-3. Characteristics of petroleum oils.

(Crase Co.)



Curve	Rough-	Diameter of Pipe, in Inches					
No.	ness %	Туре А	Type B	Type C	Type D	Type E	Type F
1 2	0.2 0.45 0.81	0.35 up	72 48 to 66 14 to 42	30	48 to 96	96	220
3 4 5	1.35		6 to 12 4 to 5	10 to 24 6 to 8	20 to 48 12 to 16	42 to 96 24 to 36	84 to 204 48 to 72
6 7	3.0 3.8 4.8		2 to 3 1½ 1 to 1¼	3 to 5 2½ 1½ to 2	5 to 10 3 to 4 2 to 21/2	10 to 20 6 to 8 4 to 5	20 to 42 16 to 18 10 to 14
9 10 ·	6.0 7.2		1/2	11/4	11/4	3	8 5
11 12 13	10.5 14.5 19.0		*/ ₂	3/4 1/2			3
14 15	24.0 28.0	0.125		*			
16 17 18	31.5 34.0 37.5	0.0625		1/4			

Type A: Drawn tubing, and brass tin, lead, and glass pipe.

Type B: Clean steel and wrought iron pipe.

Type C: Clean galvanized pipe.

In drawn tubing, the actual inside diameter is given.

Type D: Best cast iron pipe, cement, and light riveted sheet ducts.

Type E: Average cast iron pipe and rough formed concrete ducts.

Type F: First class brick, heavy riveted, and spiral riveted ducts.

In pipe, the nominal size of standard weight is given.

(Cress Co.)

Fig. X-4. Friction factor with relation to Reynolds' number.

WORKING PRESSURES, BANDED FITTINGS

150 pounds steam 300 pounds cold water, oil, or gas, non-shock

Dimensions, in Inches



TABLE

IIAX

continued

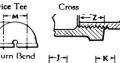






























1/8 | 11/16 | 1/2 | 1 | 11/16 | 13/16 |

1/4 | 13/16 | 3/4 | 13/16 | 5/8 | 15/16 |

31/2 37/16 23/8 51/8 23/8 33/4

41/2 31/16 63/3

51/8 | 37/16

Size A B C D E F G H



 $\frac{3}{8}$ | $\frac{15}{16}$ | $\frac{13}{16}$ | $\frac{17}{16}$ | $\frac{11}{16}$ | $\frac{11}{16}$ | $\frac{21}{8}$ | $\frac{17}{16}$ | $\frac{3}{4}$ | $\frac{13}{16}$ | $\frac{11}{8}$ | $\frac{3}{8}$

 $\frac{1}{2}$ | $\frac{1}{8}$ | $\frac{7}{8}$ | $\frac{15}{8}$ | $\frac{13}{16}$ | $\frac{13}{16}$ | $\frac{27}{16}$ | $\frac{111}{16}$ | $\frac{7}{8}$ | $\frac{15}{16}$ | $\frac{11}{4}$ | $\frac{1}{2}$

 $\frac{3}{4}$ | $\frac{15}{16}$ | $\frac{1}{1}$ | $\frac{17}{8}$ | $\frac{15}{16}$ | $\frac{15}{16}$ | $\frac{1213}{16}$ | $\frac{21}{16}$ | $\frac{11}{16}$ | $\frac{11}{2}$ | $\frac{17}{16}$ | $\frac{9}{16}$

1 | $1\frac{1}{2}$ | $1\frac{1}{8}$ | $2\frac{1}{8}$ | $1\frac{1}{16}$ | $1\frac{1}{2}$ | $3\frac{3}{8}$ | $2\frac{7}{16}$ | $1\frac{3}{16}$ | $1\frac{11}{16}$ | $1\frac{11}{16}$ | $1\frac{11}{16}$ | $1\frac{11}{16}$

 $1\frac{1}{4} + \frac{13}{4} + \frac{15}{16} + \frac{27}{16} + \frac{11}{4} + \frac{111}{16} + \frac{41}{16} + \frac{215}{16} + \frac{11}{4} + \frac{115}{16} + \frac{21}{16} + \frac{11}{16}$

 $1\frac{1}{2}$ $|1\frac{15}{16}|$ $|1\frac{7}{16}|$ $|2\frac{11}{16}|$ $|1\frac{3}{8}|$ $|1\frac{7}{8}|$ $|4\frac{1}{2}|$ $|3\frac{5}{16}|$ $|1\frac{5}{16}|$ $|2\frac{1}{8}|$ $|2\frac{5}{16}|$ $|1\frac{1}{16}|$

 $2 \quad | \quad 2\frac{1}{4} \quad | \quad 1^{11}/_{16} \quad | \quad 3\frac{1}{4} \quad | \quad 1^{11}/_{16} \quad | \quad 2\frac{1}{4} \quad | \quad 5^{7}/_{16} \quad | \quad 4 \quad | \quad 1^{7}/_{16} \quad | \quad 2\frac{1}{2} \quad | \quad 2^{13}/_{16} \quad | \quad 3\frac{1}{4} \quad | \quad$

21/2 |211/16|115/16|313/16| 17/8 | 25/16 | 61/4 |411/16|111/16| 27/8 | 31/4 | 15/16

 $| 3\frac{1}{8} | 2\frac{3}{16} | 4\frac{1}{2} | 2\frac{1}{16} | 3 | 7\frac{1}{4} | 5\frac{9}{16} | 1\frac{13}{16} | 3\frac{3}{16} | 3\frac{11}{16} | 1$

 $| 3\frac{3}{4} | 2\frac{5}{8} | 5^{11}/_{16} | 2\frac{1}{4} | 3\frac{3}{4} | 8\frac{7}{8} | 6^{15}/_{16} | 2^{1}/_{16} | 3^{11}/_{16} | 4\frac{3}{8} | 1\frac{1}{8}$

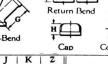


9/16 | 15/16 |

5/8 | 11/16 | 1

25/16 41/4 51/8 | 11/4





1/4

115/16 37/18 4 11/16 1 17/8 21/4

29/16 43/8 | 15/16 | 2 | 3 | 37/8 |



Size M N Size M N

Close Pattern | Open Pattern

 $\frac{1}{2}$ | 1 | 13/4 | $\frac{1}{2}$ | 11/2 | 17/8

3/4 | 11/4 | 23/16 | 3/4 | 2 | 21/4 |

1 | 11/2 | 21/2 | 1 | 21/2 | 25/8 |

11/4 | 13/4 | 213/16 | 11/4 | 3 | 33/16

11/2 23/16 33/16 11/2 31/2 35/8

2 | 25/8 | 31/8 || 2 | 4 | 43/8

MediumPattern | 21/2 | 41/2 | 41 5/16

1/2 | 11/4 | 15/8 | 3

3/4 | 11/2 | 115/16 | 4

11/4 21/4 213/16

11/2 | 21/2 | 33/16 ||



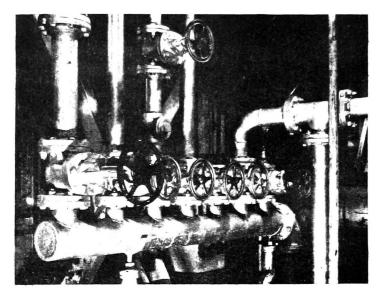
6 611/16

1111

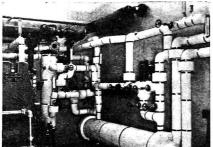
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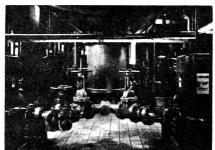


IPING AND DUCT-WORK









(Crane Co.)

Fig. X-6. Piping details.

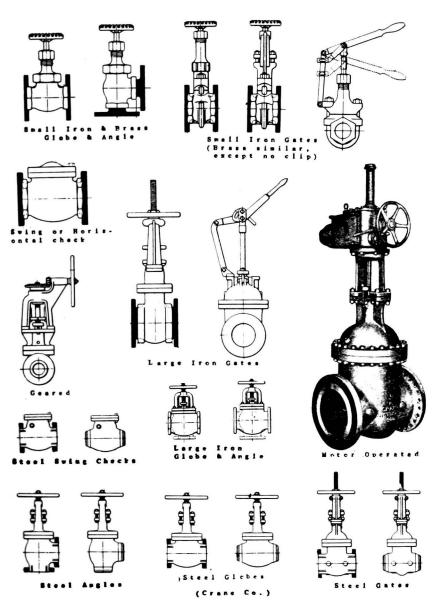


Fig. X.7 Power plant valve types

Example 26-A.

Determine the nearest standard pipe size and pressure drop for a steam line (seamless Grade A steel pipe) carrying 30,000 lbs/hr at 215 psig and 560 F. The line contains 100 ft of straight pipe, 2 gate valves, 1 globe valve, 1 swing check valve and 3-90° elbows.

Solution:

The steam is at a moderately high pressure and is super heated, so a velocity of 180 ft/sec is selected. The required inside diameter is:

$$d_i = 0.4 \left[\frac{30000(2.531)}{180 \pi} \right]^{0.5} = 4.65 \text{ in.}$$

The pipe schedule number (allowing a factor of safety of about 10) is:

$$SN = \frac{1000(215)}{4800} = 45$$

Schedule 40 pipe would be safe enough. The inside diameter of 5 in. Schedule 40 pipe is 5.05 in. which is close enough to the required 4.65 in. The true velocity of the steam is:

$$V = \frac{0.16(30000)(2.531)}{\pi (5.05)^2} = 152 \text{ ft/sec}$$

In order to determine the friction factor, Reynolds' number is calculated (finding the viscosity, z, from Figure X-2) as follows:

$$r = \frac{122.5(5.05)(152)}{0.025(2.531)} = 1,485,000$$

From Figure X-4, the friction factor is 0.016.

The total equivalent length of pipe is found with the aid of Figure X-1, as follows:

Straight pipe 2 Gate valves 2(2.8) =	100 ft 5.6 ft
1 Globe valve	140 ft
1 Swing check valve	31 ft
3 90° Elbows 3(14)	42 ft

Total equivalent length

318.6 ft

The pressure drop is:

$$\Delta P = 0.00518 \frac{(0.016)(318.6)(152)^2}{2.5.2(5.05)} = 47.7 \text{ psi}$$

Example 26-6.

Petermine the nearest standard pipe size and pressure drop for a seamless Grade A steel pipe carrying 40,000 lbs/hr of residual fuel oil at a pressure of 200 psig and a temperature of 200 F. The specific gravity of the oil at 60°/60° is 0.94. The pipe line contains 500 ft of straight pipe, 6-90° elbows, 2 gate valves, 1 swing check valve, and 1 globe valve.

Solution:

From Figure X-3 b, the specific gravity of the oil at $200^{\circ}/60^{\circ}$ is 0.89, from which the specific volume v is 1/[62.4(0.89)] = 0.018 cu ft/lb, and taking V = 2 ft/sec,

$$d_i = 0.4 \left[\frac{40000(0.018)}{2\pi} \right]^{0.5} = 3.5 \text{ in.}$$

3½ ir. Schedule 40 pipe has an inside diameter of 3.548 in. The actual velocity would be:

$$V = \frac{0.16(40000)(0.018)}{\pi(3.548)^2} = 2.91 \text{ ft/sec}$$

Reynolds' number is calculated using the viscosity, z, from Figure X-3 a, as follows:

$$\mathbf{r} = \frac{122.5(3.548)(2.91)}{17(0.018)} = 4130$$

From Figure X-4, the friction factor is 0.041.

The total equivalent length of pipe is:

Straight pipe 6 90° Elbows 6(9.5)	500 ft 57
2 Gate valves 2(2.0) 1 Swirg check valve 1 Globe valve	4 21 100
Total equivalent length	682 ft

The pressure drop is:

$$\Delta h = 0.745 \frac{(0.041)(682)(2.91)^2}{3.548} = 49.5 \text{ ft. of oil}$$

70. Friction in Condensers and Feed-Water Meaters--Friction losses in condensers and feed-water heaters may be calculated from the pipe friction formula, but it is more usual to use the curves shown in Figure X-8 for the losses in the tubes and Figure X-9 for the losses in the water boxes.

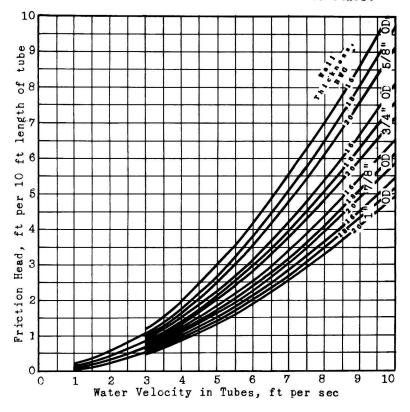


Fig. X-8. Head loss in condenser and feed-water heater tubes

Examples 27-A and 27-B show the method of calculation of head losses in the condenser and feed-water heater of Examples 9 and 11.

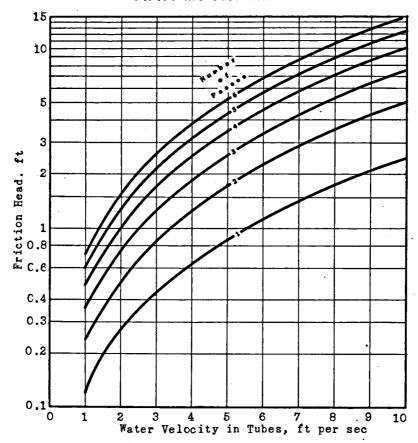


Fig. X-9. Read loss in condenser and feed-water heate:

Example 27-A.

Determine the head loss in the tubes and water boxes of the condenser of Example 9.

Solution:

From Figure X-8, the head loss in 3/4 in. X 18 BWG tubes at a velocity of 8 ft/sec is 5.1 ft per 10 ft length of tube. The water box head loss for a two-pass condenser at a velocity of 8 ft/sec, from Figure X-9, is 3.6 ft. Then the total head loss is:

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FUNDAMENTALS OF POWER PLANT ENGINEERING

Tubes 5.1(45.5/10)23.3 ft Water boxes 3.6 Total head loss

26.9 ft

Example 27-B.

Determine the head loss in the tubes and water boxes of the feed-water heater of Example 11.

Solution:

From Figure X-8, the head loss in 5/8 in. X 16 BWG tubes at a velocity of 6 ft/sec is 4.2 ft per 10 ft length of tube. The water box head loss for a six-pass heater at a velocity of 6 ft/sec, from Figure X-9, is 6.9 ft. Then the total head loss is:

Tubes	4.2[7.68(6)/1C]	19.4 ft
Water	boxes	6.9
Total	head loss	26.3 ft

71. Duct-Work-- The method of calculating the dimensions of ducts for gas or air are very similar to those employed for piping. Ducts usually operate at maximum pressures of the order of several inches of water, so that ordinarily if the duct is strong enough to support itself without buckling, it will be strong enough to withstand the pressure.

Duct size is determined by substituting the allowable velocity in the continuity equation, as for piping. Ducts, however, are usually rectangular in cross section rather than round. Table XVIII gives reasonable velocities in ducts. The higher velocities apply to higher static pressures.

TABLE XVIII

PUCT VELOCITIES

	ft per sec
Air ducts	3C - £O
Gas ducts	20 - 40
Stacks	20 - 40

Pressure drop (or draft loss) is usually measured in irches of water, and is given by the equation:

$$\Delta p = 0.00298 \frac{c f L V^2}{R}$$

where: $\Delta p = draft loss$, in. water

p = density of gas, lbs/cu ft

R = hydraulic radius of duct, ft

= area/perimeter

Reynolds' number is as for piping, except that the smallest dimension of the duct (if rectangular) is substituted for the inside diameter of the pipe. Viscosity in centipoises for various gases can be found from the following equations:

Flue gas: $z = 0.0153 + 2.5 \times 10^{-8} t$

Air: $z = 0.0165 + 2.5 \times 10^{-6} t$

where: t = temperature, deg F

Figure X-1C shows densities for air and flue gas.

Length allowances for bends may be found from Table XIX after converting rectangular duct sizes to equivalent round duct diameter by the equation:

$$d_e = 1.35 \frac{(a \ b)^{0.427}}{(a + b)^{0.255}}$$

where: d = equivalent round duct diameter, in.

a and b = rectangular duct
 dimensions, in.

Figure X-11 shows some typical ducts, and Example 28 shows some sample duct calculations.

Example 28.

Determire the draft loss in the flue gas duct between the economizer and air heater for the boiler of Example 17. The duct has a straight length of 20 ft and has 2-90° bends with a radius of 8 ft. The maximum width of duct is 12 ft and the maximum gas velocity is 20 ft/sec.

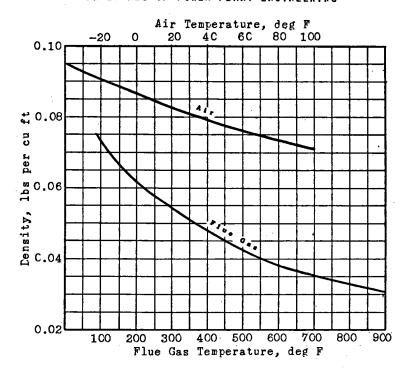


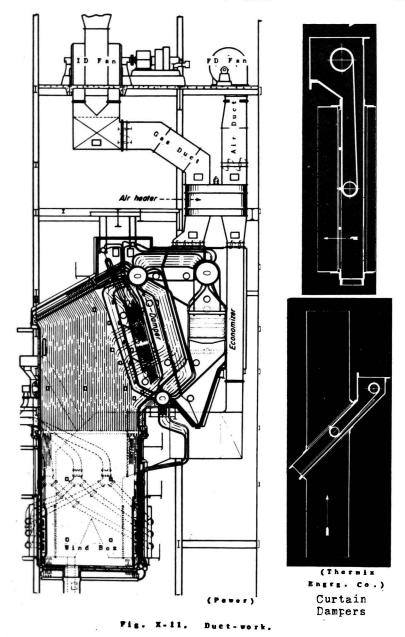
Fig. X-10. Densities of air and flue gas.

TABLE XIX

EQUIVALENT LENGTHS OF PUCT BENDS

Diameters of Straight Duct

Fight angle hends	
Center line radius 2 to 8 dia.	10
Center line radius 1 dia.	20
Mitred right angle elbow	50



Solution:

From Example 23, the gas temperature is 650 F, and from Figure X-10, the density of the gas is 0.0375 lbs/cu ft, which is equivalent to a specific volume of 26.7 cu ft/lb. The required area of duct is:

$$A = \frac{291000(26.7)}{3600(20)} = 108 \text{ sq ft}$$

The duct dimensions are then 9 ft X 12 ft.

The viscosity of the gas is $z = 0.0153 + 2.5 \times 10^{-6} (650) = 0.03155$ centipoises. Reynolds' number is:

$$\mathbf{r} = \frac{122.5(9)(12)(20)}{0.03155(26.7)} = 314,000$$

and the friction factor is found to be 0.0152 (Curve 1, Figure X-4).

The equivalent duct diameter is:

and the radius of the bends is 8(12)/83 = 1.15 dia., for which the allowance is 20 diameters of duct each. The total equivalent duct length is:

Total equiv. length 297 ft

The draft loss in the duct is:

$$\Delta p = 0.00298 \frac{(0.0375)(0.0152)(297)(20)^2}{108/[2(12) + 2(9)]} = 0.078 \text{ in.}$$

72. Pipe and Duct Insulation—Most pipes and ducts which carry liquids or gases whose temperature is very much different from the ambient temperature require insulation of some kird to prevent heat transfer to or from the surroundings. The most common insulation materials for pipe and ducts are asbestos, magnesia, mineral wool and diatomite. These materials are made in the form of blocks, moulded circular sections, blankets or plastic for application with a

trowel. Figure X-12 shows some of the various forms of insulation and Figure X-13 gives thermal conductivities for a variety of materials with the approximate temperature ranges for best utilization.

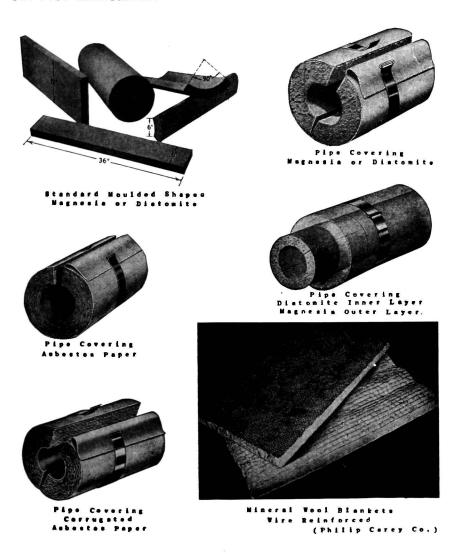
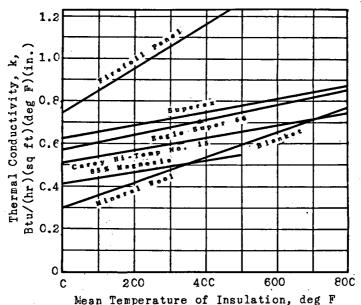


Fig. X-12. Insulating materials.



(F. T. No

Fig. X-13. Thermal conductivities of insulating materials.

Heat transfer through insulation follows the same laws as previously discussed. It is not usual, however, to make an exhaustive analysis of the economic factors involved, but to use an economic selection chart such as is shown in Figure X-14. Example 29 shows the use of the chart.

Example 29.

Find the most economical thickness of 85 o/o magnesia pipe covering for the steam line of Example 26-A. Ambient temperature is 90 F; cost of insulation 24 cents/bd ft; fixed charges 20 o/o; used 8000 hrs/yr; fuel costs \$2.00 per bbl of 6,300,000 Btu.

Solution:

From Figure X-13, the conductivity of 85 o/o magnesia at a mean temperature of 325 F is 0.53. The value of the heat in the fuel is \$0.318 per million Btu. Entering Figure X-14 at 8000 hours and following the dotted lines, the most economical thickness of 85 o/o magnesia insulation for a 5 in. pipe is determined as 2 in.

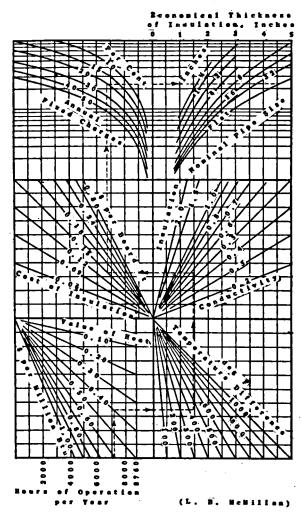


Fig. X-14. Selection chart for insulation thickness.

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Walker, J. H., and Crocker, S., Piping Handbook, McGraw-Hill, New York, 1939.

PROJECT III

- 1. The power plant of Projects I and II is to contain three steam generators. One will be operated continuously, one will be operated to handle peak loads and one will be a spare. Each steam generator will have a capacity sufficient to handle one turbo-generator at full load. The fuel oil has a higher heating value of 18,500 Btu per lb as fired and the analysis is as follows: Carbon = 0.851, hydrogen = 0.11, oxygen = C.C22, nitrogen = 0.002, sulphur = C.013, and moisture = 0.002. The expected flue gas analysis is: Carbon dioxide = 14.0 o/o, carbon monoxide = 0 o/o, and oxygen = 3.0 o/o. Expected steam generator efficiency is 86.5 o/o. Temperatures expected are: Combustion air, 100 F; flue gas to stack, 300 F; fuel oil at burners, 175 F; and gas entering superheater, 1500 F.
- (a) Make a heat balance for the steam generator and calculate the weight of fuel fired per hour.
- (b) Estimate the furnace volume required and specify the furnace dimensions.
- (c) Estimate the effective radiant surface in the furnace and the number of linear feet of $2\frac{1}{4}$ OD tubes spaced on four inch centers in the furnace walls and boiler screen (assume that the boiler screen consists of three staggered rows).
 - (d) Estimate the superheater surface required and calculate the number and length of 2" OD X 12 BWG tubes required (assume that the superheater is placed in one boiler pass).
 - (e) Estivate the most economical temperatures for the remainder of the steam generator. Assume that boiler surface costs \$8.00, economizer surface \$3.50 and air heater surface \$0.90 per incremental square foot with annual fixed charges of 12 o/o.
 - (f) If boiler surface is justified, estimate the amount necessary and the number of rows and banks of $2\frac{1}{4}$ OD X 12 BWG tubes required.
 - (g) If economizer surface is justified, estimate the amount necessary and the number and length of 1^{1}_{8} X 12 BWG tubes required.
 - (h) If air heater surface is justified, estimate the amount necessary and the number and length of $1\frac{1}{2}$ OD X 16 BWG tubes required for one of the recuperative type with the gas

through the tubes. Calculate the number of gas passes necessary if the maximum tube length is 20 ft.

- 2. Specify the pipe sizes and calculate the pressure drops for the following lines in the power plant of Projects I and II:
- (a) Circulating water: Suction line, 50 ft with three 90° elbows. Pump to condenser, 6 ft with one 90° elbow and one gate valve. Condenser discharge, 100 ft with four 90° elbows.
- (b) Condensate: Suction line, 10 ft with two 90° elbows and one gate valve. Discharge line, 50 ft with eight 90° elbows, six gate valves and one swing check valve. Assume vapor condenser tube velocity is 5 ft per sec through two 6 ft passes.
 - (c) Boiler feed: Suction line, 40 ft with three 90° elbows and one gate valve. Discharge line, 75 ft with eight 90° elbows, two gate valves, two swing check valves and one globe valve.
 - (d) Main turbine lead (manifold to throttle): Length, 50 ft with the equivalent of 3-90° elbows and one gate valve.
 - (e) Extraction to closed heater: Length, 15 ft with two 90° elbows, one gate valve and one swing check valve.
 - 3. Specify the size and estimate the draft loss for the following ducts for one of the steam generators of part one:
 - (a) Air heater to windbox (if necessary): Length, 50 ft with two 90° benese:
 - (b) Economizer to air heater (if necessary): Length, 20 ft with two 90° bends.
 - (c) Air heater, economizer or boiler to induced draft fan: Two ducts in parallel, each with length of 20 ft with one 90° bend and one 45° bend.
 - 4. Specify the insulation thickness to be used in the following locations in the power plant of Projects I and II (fixed charges 15 o/o and ambient temperature 90 F):
 - (a) Main turbine lead: 85 o/o magnesia at \$C.21 per bd ft.
 - (b) Freheated air duct: Mineral wool at \$C.05 per bd ft.

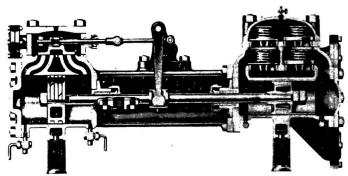
CHAPTER XI

WATER HANDLING EQUIPMENT

73. Water handling equipment consists of pumps and water softeners. Pumps may be either reciprocating or rotating; and water softeners may be either precipitating or base exchanging.

74. Types of Pumps--Pumps are used to move liquids, usually through pipe lines, and at the same time to increase the pressure (head) of the liquid.

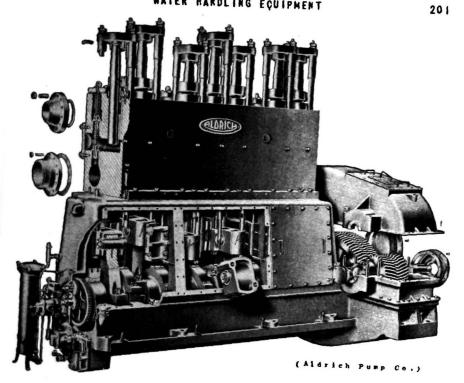
Reciprocating pumps may be simplex or multiplex; steam driven or motor driven; or of the plunger or piston type. Figures X-1 and X-2 show two of the various types of reciprocating pumps. Reciprocating pumps have the advantage of being able to operate at extremely high discharge heads or extremely low suction (inlet) heads; and to operate at variable delivery with good efficiency.



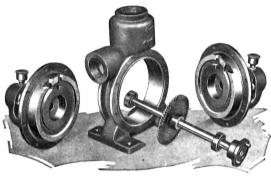
(Varren Steam Pump Co.)

Fig. XI-1. Duplex steem driven pump.

Rotating pumps may be centrifugal, turbine, screw or gear; steam driven or motor driven; single stage or multistage. Figures XI-3 through XI-7 show a number of rotating pumps of various types. Rotating pumps have the advantage of simplicity and being adapted to direct drive by high speed turbines or motors. They have the disadvantage of being unable to start pumping at negative suction pressures without being primed. This disadvantage is not serious, however, because it is always possible to find a type of pump which can be installed with a flooded suction.

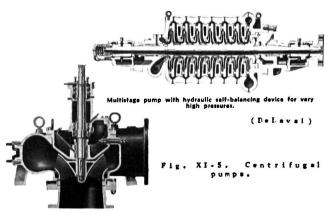




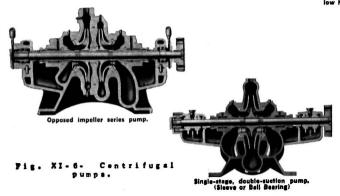


(Aurora Pump Co.)

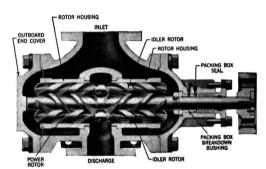
Fig. XI-4. Turbine pump.



Mixed flow pump for large capacity against low heads at high speed.



(DeLavel)



(DeLaval Steam Turbine Co.) Fig. XI-7. Screw pump. 75. Head and Work Relations in Pumps--Consideration of a pump as a work transferring device will show that the work input to the pump is equal to the work output plus the losses, or:

whp =
$$e_p(shp)$$

where: whp and shp = water horsepower
and shaft horsepower
e p = pump efficiency

Water horsepower is the power required to move the water, or:

$$whp = \frac{M_{\pi}H_{t}}{60 \times 33000}$$

where: M_z = water flow, lbs/hr
H_t = total head on pump,
ft of fluid

The total head on the pump is the sum of the friction losses in the pipe, the increase in velocity head through the pump, the entry losses and leaving losses for the system, the difference in pressure between inlet and discharge and the difference in static heads, or:

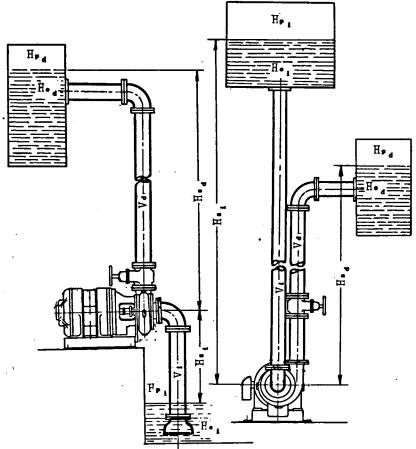
$$\mathbf{H}_{t} = \mathbf{H}_{f} + \frac{\mathbf{V}_{d}^{2} - \mathbf{V}_{i}^{2}}{2g} + \mathbf{H}_{s_{d}} - \mathbf{H}_{s_{i}} + \mathbf{H}_{p_{d}} - \mathbf{H}_{p_{i}} + \mathbf{H}_{e_{d}} + \mathbf{H}_{e_{i}}$$

where: H_f = friction in piping, ft V = velocity, ft/sec H_s = static head, ft H_p = pressure head, ft H_g = entry and exit losses, ft

Figure XI-8 shows pump and pire systems with the various heads in relation to the above equation. It should be noted that the static head on the inlet side of hot water pumps should always be great enough to prevent boiling in the eye of the pump impeller due to pressure reduction below the saturation pressure corresponding to the temperature of the water. An empirical equation which gives the static head at the pump suction is as follows:

 $h = 0.32 \times 10^{-7} t^{3.6}$

where: h = static head on pump suction, ft.
 t = water temperature, deg F.



(a) Negative suction head (b) Positive suction head

Fig. XI-8. Head relations in oums systems.

76. Pump Efficiencies--In general, large slow speed pumps have higher efficiencies than small high speed pumps. Table XX shows the ranges of speed and efficiency to be

expected for various types of pumps for power plant service.

TABLE XX

CENTRIFUGAL PUMP DATA

	Cap.	, grn	Head,	ft	Speed,	rpm	Eff., o/o
Boiler feed	10 t	400	300 to	6000	1200 to	2600	70
Condensate	10 t	4C0	50 to	300	1200 to	3600	70
Condenser	200 t	5CCCC	√10 to	£0	120 to	450	80
General	5 t	100	10 to	300	1200 to	2600	75

77. Laws of Similarity for Centrifugal Pumps-The performance of centrifugal pumps follows certain definite patterns and laws. Figure XI-9 shows typical performance curves for two types of centrifugal pump. It will be noted that the head capacity curve has a falling characteristic and the horsepower curve reaches a maximum and then falls. The "self-limiting" power characteristic is of value in preventing the overloading of the driver in case the pump should operate against zero head.

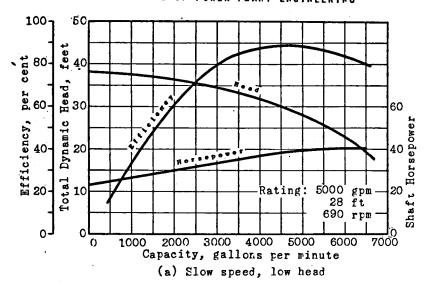
The following equation expresses the laws of similarity for centrifugal pumps.

$$\frac{A_1}{A_2} = \frac{[B_1]^X}{[B_2]^X} \frac{[\rho_1]^Y}{[\rho_2]^Y}$$

		X				Y		-	
A =	Q	M	P	H	Q	M	P	H	= A
B = N	1	1 .	3	2	0	1	1	1	D const
B = D	2	2	5	2	• 0	1	1	1	N const

where: Q = capacity, volume N = speed P = power M = capacity, weight
$$\rho$$
 = density H = head D = diameter

It will be seen that, for any given pump, the quantity of water delivered varies directly with the speed, the head with the square of the speed and the power with the cube of the speed. The equation will be valuable in predicting the effect of changes of speed or impeller diameter for a particular pump.



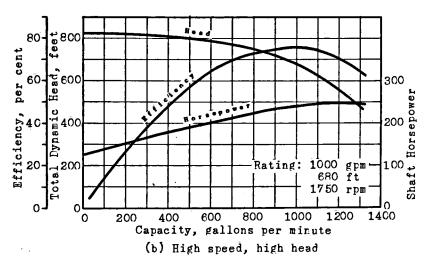


Fig. XI-9. Centrifugal pump performance curves.

Example 30 shows the method of computing the specifications for a centrifugal pump, and Example 31 shows the method of estimating the effect of changes in pump speed or impeller diameter.

Example 30.

Specify the capacity in gallons per minute and total dynamic head in feet for a centrifugal boiler feed pump to operate under the following conditions: Flow 28800 lbs/hr at 297 F; static lift 20 ft; friction losses 2 ft in suction line and 20 ft in discharge line; inlet velocity 5 ft/sec and outlet velocity 10 ft/sec; entrance and exit losses one-half of the velocity head; pressure in boiler 400 psia and in open heater 64 psia.

Solution:

The capacity of the pump is:

$$Cap. = 28800(0.01742)(7.48)/60 = .62.5 gpm$$

The total dynamic head is:

Total dynamic head

886.8 ft

Example 31-A.

A centrifugal pump with an impeller diameter of 12 in. cperates at 3600 rpm and delivers 100 gpm at 300 ft head. What would be the capacity and head if (a) the impeller diameter were reduced to 10 inches, or (b) the speed were reduced to 1800 rpm?

Solution:

(a) At constant speed, the capacity and head are proportional to the square of the diameter.

$$\frac{100}{Q_2} = \frac{(12)^2}{(10)^2}$$

$$\frac{300}{H_2} = \frac{(12)^2}{(10)^2}$$

$$Q_2 = 69.5 \text{ gpm}$$

$$H_2 = 208 \text{ ft}$$

(b) At constant diameter, the capacity is proportional to the speed and the head is proportional to the square of the speed.

$$\frac{100}{Q_2} = \frac{3600}{1800}$$

$$\frac{300}{H_2} = \frac{(3600)^2}{(1800)^2}$$

$$Q_2 = 50 \text{ gpm}$$

$$H_2 = 75 \text{ ft}$$

Example 31-B.

A three stage boiler feed pump with 17 inch diameter impellers is rated at 1000 gpm at 675 ft TDH with a motor speed of 1750 rpm. The characteristic curves are shown as solid lines in Figure XI-10. It was found when the pump was placed into operation that the boiler feed system head loss had been overestimated, and instead of being 675 ft at 1000 gpm was actually 600 ft. It was proposed to correct the pump characteristics by turning down the outside diameter of the impeller. Calculate the required impeller diameter and plct the new characteristic curves for the pump.

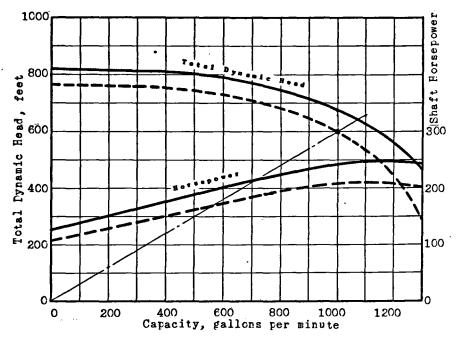


Fig. XI-10. Effect of reduced impelier disseter on

Solution:

An empirical method for determination of the new impeller diameter is as follows:

- Praw a line from the origin through the new specification point to intersect the original head-capacity curve.
- Read the head and capacity at the point of intersection and solve for the impeller diameter from both head and capacity relations in the pump equation.
- 3. Obtain the root mean square of the two solutions for impeller diameter.

This method can be illustrated, as follows:

- 1. The line from the origin through the new specification point intersects the original head-capacity curve at 640 ft and 1070 gpm.
 - 2. Solving for impeller diameter,

$$\frac{1070}{1000} = \frac{(17)^2}{(\Gamma_2)^2} \qquad \frac{640}{600} = \frac{(17)^2}{(\Gamma_2)^2}$$

$$\Gamma_2 = 16.40 \text{ in.} \qquad D_2 = 16.45 \text{ in.}$$

3. The root mean square of these two solutions is:

$$\Gamma_2 = \sqrt{\left[\frac{(16.40)^2 + (16.45)^2}{2}\right]} = 16.425 \text{ in.}$$

The new head curve can be plotted by applying the pump equation at several points on the original curve, as follows:

$$\frac{1200}{Q_2} = \frac{555}{P_2} = \frac{(17.000)^2}{(16.425)^2} = 1.07$$

$$\frac{Q_2}{H_2} = \frac{1120 \text{ gpm}}{519 \text{ ft}}$$

$$\frac{800}{Q_2} = \frac{745}{H_2} = 1.07$$

$$Q_2 = 748 \text{ gpm}$$

$$H_2 = 696 \text{ ft}$$

etc.

The new power curve can be plotted in a similar manner, as follows:

$$Q_1 = 1200 \text{ gpm}$$
 $\frac{248}{P_2} = \frac{(17.000)^5}{(16.425)^5} = 1.182$ $P_2 = 210 \text{ hp}$
 $Q_1 = 800 \text{ gpm}$ $\frac{224}{P_2} = 1.182$ $P_2 = 190 \text{ hp}$
 $Q_2 = 748 \text{ gpm}$ $\frac{224}{P_2} = 1.182$

etc.

The new characteristic curves are shown as dotted lines on Figure XI-10.

78. Water Treatment--All natural waters contain quantities of dissolved salts which make the water undesirable for use in steam generators. Calcium and magnesium salts precipitate in the boiler to form scales or sludges which adversely affect the heat transfer coefficients. Sodium salts are soluble at all temperatures, but excessive concentrations of dissolved sodium salts lead to foaming which has the bad effect of causing deposits in superheaters and turbines. It is the purpose of water treatment to either remove the undesirable salts before they reach the boiler or to produce conditions within the boiler which are such as to cause precipitation of the undesirable salts in forms which will not produce scale and which can be easily removed from the boiler by blowdown.

Calcium and magnesium bicarbonates are soluble at ord-inary temperatures, but when water containing them is boiled the bicarbonates decompose to carbonates and precipitate. The fact that bicarbonates can be removed by boiling gives them the name temperary hardness (hardness is a measure of the soap destroying power of water). Non-carbonate salts of calcium and magnesium produce what is termed permanent hardness, that is, boiling has no effect as with temporary hardness.

Hardness can be removed from water in two general ways (in addition to boiling to remove temporary hardness, which has no extensive commercial application): by precipitation or by base exchange.

79. Precipitation Softening--Precipitation softeners may operate either by the cold process or by the hot process. Due to the fact that most of the slightly soluble calcium and magnesium salts exhibit a decreasing solubility with increasing temperature, the hot process of softening is more efficient than the cold process. There are cases, however, where the cold process must be used (such as softening of municipal water supplies). Following are detailed explan-

ations of the various processes.

The cold line and also process uses either quicklime or hydrated lime for the precipitation of temporary hardness and filter also for the production of a floc which aids in the coagulation of the precipitate and in the removal of bacteria and dirt. Figure XI-11 shows a typical softener operating with this process. The reactions are:

$$Ca(HCO_{3})_{2} + Ca(OH)_{3} \longrightarrow 2CaCO_{3}^{\dagger} + 2H_{2}O$$
 $Mg(HCO_{3})_{2} + 2Ca(OH)_{3} \longrightarrow Mg(OH)_{2}^{\dagger} + 2CaCO_{3}^{\dagger} + 2H_{2}O$
 $CO_{2} + Ca(OH)_{2} \longrightarrow CaCO_{3}^{\dagger} + H_{2}O$
 $Al_{2}(SO_{4})_{3} + 3Ca(OH)_{2} \longrightarrow 2Al(OH)_{3}^{\dagger} + 3CaSO_{4}^{\dagger}$

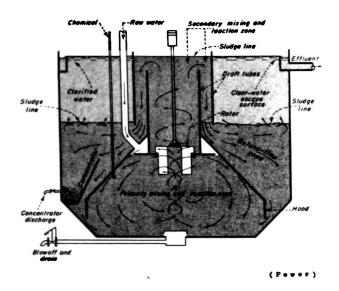


Fig. XI-11. Cold process softener.

The lowest hardness which can be produced by this process is about 70 parts per million, which is the solubility of calcium carbonate at ordinary temperatures.

The celd line-seds and alum process uses lime and alum, as in the previous process, plus soda ash to precipitate non-carbonate hardness. The apparatus employed is identical with that used for the lime and alum process. The additional reactions are:

In this process it is customary to use excess soda ash in order to depress the solubility of calcium carbonate by common ion effect. Figure XI-12 shows the hardness attainable by the use of excess soda ash.

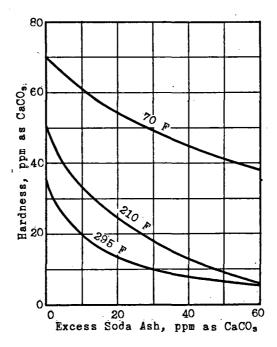


Fig. XI-12. Residual hardness resulting from use of excess sods ash.

The hot lime-soda process is similar to the cold lime-soda and alum process, except that the alum is not used because at elevated temperatures magnesium hydroxide forms a floc. The reactions are the same as previously given, and the residual hardness can be estimated from Figure XI-12. Figure XI-13 shows a typical hot process softener with its auxiliaries.

Example 32 shows the method of calculation of quantities of chemicals required for softening by the various

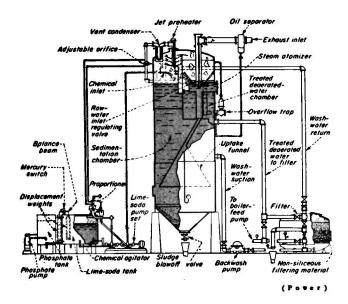


Fig. XI-13. Bot process softener.

precipitation methods. Useful chemical tables will be found in the Appendix.

Example 32-A.

A water of the following analysis is to be softened with lime and coagulated with 10 ppm of aluminum sulphate (filter alum): Ca = 62 ppm; Mg = 31 ppm; Na = 38 ppm; HCO₈ = 250 ppm; SO₄ = 139 ppm; Cl = 11 ppm; CO₂ = 16 ppm. Determine the weight of 90 o/o hydrated lime and 95 o/o crystalline filter alum required per thousand gallons of water.

Solution:

The first step in the solution is the conversion of the ionic analysis to the hypothetically combined form, as follows:

Ions	ppm	tepm	-epm	Combinations	epm	ppm
Ca	62	3.10		Ca(HCO.).	3.10	251.5
Ne Ne	31	2.55		Mg (HCO ₃)	1.00	73.2
Ne	38	1.65		Mg SO.	1.55	93.5

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Ions	ppm	tepm	-epm	Combinations	epm	ppm
HCO _s SO ₄ C1	250 139 11		4.10 2.89 0.31	Na ₂ SO ₄ NaCl	1.34 0.31	95.1 18.2
Sub.tot	531	7.30	7.3C		7.30	531.5
CO ₂	16					16
Total	547					547.5

It will be noted that the hypothetical combinations are made in a definite order, as follows:

Positive ions	Negative ions
Ca	HCO _s
Mg .	804
Na	C1

Next, the reactions are written with molecular weights and parts per million, as follows:

162.1 74.1
$$Ca(HCO_3)_2 + Ca(OH)_2 - 2 CaCO_3 + H_2O$$
251.5 $x_1 = 115 \text{ ppm}$

146.3 148.2
$$M_g(HCO_s)_2 + 2 Ca(OH)_2 \longrightarrow M_g(OH)_2 + 2 CaCO_s + 2 H_2O$$
73.2 $x_2 = 74.1 \text{ ppm}$

120.4 74.1
$$MgSO_4 + Ca(OH)_2 - Mg(OH)_2 + CaSO_4 + SO_5 + SO_5 + SO_5 + SO_6 + SO_6$$

44.0 74.1
$$CO_2 + C_2(OH)_2 - CaCO_3 + H_2O$$
16.0 $x_4 = 27.0$ ppm

666.4 222.3 Al₂(SO₄)₃ 18H₂O + 3 Ca(OH)₂ - 2 Al(OH)₃† + 3 CaSO₄† + 18 H₂O 10(0.95)
$$x_5 = 3.2 \text{ ppm}$$

Total 90 o/o hydrated lime required = 276.8/0.90 = 308 ppm Total 95 o/o filter alum required 10 ppm

The final water analysis would be: NaCl = 18 ppm; $Pa_2SC_4 = 95$ ppm; hardness about 70 ppm as $CaCO_3$.

One thousand gallons of water = 8.33×10^{-8} million lbs. Therefore, the lime required is 2.56 lbs per thousand gallons and the alum required is 0.083 lbs per thousand gallons.

Example 32-8.

The water of Example 32-A is to be softened by the lime soda and alum process. In order to depress the solubility of calcium carbonate, 25 ppm of excess soda ash is added. Petermine the weight of 95 o/o soda ash required in addition to the lime and alum.

Solution:

The reactions with molecular weights and parts per mill-ion are:

120.4

$$MgSO_4 + Ca(OH)_2 + Na_2CO_3 \longrightarrow Mg(OH)_2 + CaCO_3 + Na_2SO_4$$

92.5
 $y_1 = 82.3 \text{ ppm}$
142.0
 $y_2 = 82.3 \text{ ppm}$
142.0
 $y_1 = 82.3 \text{ ppm}$

The total 95 c/o soda ash required (including excess) is 111.8/0.95 = 118 ppm.

The final water analysis would be: NaCl = 18 ppm; $Na_2SO_4 = 211$ ppm; $Na_2CO_5 = 25$ ppm; hardness about 50 ppm as $CaCO_5$.

The soda ash required would be 0.98 lbs per thousand fallons.

If this water were to be softened by the hot lime soda process (atmospheric pressure), the alum would be omitted and the final water analysis would be: NaCl = 18 ppm; Na₂SO₄ = 205 ppm; Na₂CO₃ = 25 ppm; hardness about 22 ppm as $CaCO_3$. Slightly less lime would be used because of the omission of the alum.

80. Base Exchange Softening--Several natural minerals called sections (complex sodium aluminum silicates) have the property of base exchange, that is, they will exchange their sodium for calcium or magnesium, or vice versa, depending upon which is present in the greatest concentration in the solution passing over them. Synthetic zeolites of various types have been made which have greater exchange capacity

than the natural materials as well as the ability to operate with acids or bases.

The sodium cycle exchanges sodium for calcium or magnesium in the softening run, and vice versa in the regenerating run. Regeneration is accomplished with strong brine (ordinary salt) solution. Some of the reactions are:

Softening:

Regeneration:

Both magnesium and calcium chlorides are extremely soluble and are washed out, with the excess brine, with raw water. When the wash water becomes soft the washing is stopped and the zeolite is ready for another softening run. Figure XI-14 shows a typical zeolite softener which operates on the sodium cycle.

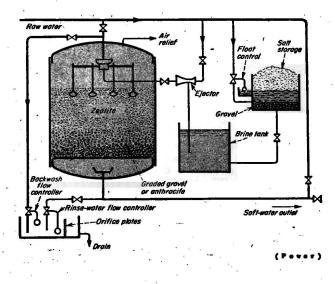


Fig. XI-14, Zeolite softener, sodium cycle.

The hydrogen cycle exchanges hydrogen for all metals, and the regeneration is accomplished with acid. Some of the reactions are:

Softening:

H₂Z + Ca(HCO₈)₂ - 2H₂CO₈ + CaZ

H₂Z + CaCl₂ - 2HCl + CaZ

Regeneration:

CaZ + H₂SO₄ - H₂Z + CaSO₄

The carbonic acid can be removed from the water by aeration. The other acids produced in this cycle are extremely corrosive, so the water is blended with effluent from a sodium cycle softener before aeration to produce essentially neutral water by the reaction of sodium bicarbonate with the acids. Figure XI-15 shows a typical zeolite softener which operates on the dual cycle.

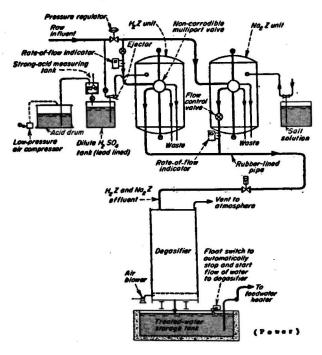


Fig. XI-15. Zeolite softener, duel cycle.

The acid adsorbing cycle employs a synthetic resin which has the property of being able to adsorb acids from solution and of being regenerated by alkali, as follows:

Adsorption:

R + HC1 -- R·HC1

Regeneration:

R·HCl + NaOH - R + NaCl + H2O

When the acid adsorbing cycle follows a hydrogen zeolite cycle and a deaerator water of almost as good quality as commercially distilled water is produced. The process is quite economical and requires much less investment than a distillation plant where pure water is required for industrial use. It does not appear, however, that the process has any advantages over the evaporator in a regenerative power cycle.

- 81. Boiler Water Treatment--Even in cycles which use evaporators for makeup water it is necessary to use some form of boiler water treatment because contamination of the boiler water results from even a small amount of condenser leakage or carryover from evaporators. It is even more necessary to treat boiler water in cycles where makeup is raw or softened water. Boiler water treatment has four objectives:
 - 1. To prevent corrosion of the boiler or power piping.
 - Z. To prevent embrittlement of boiler metal.
 - 3. To prevent formation of scale in the boiler.
 - 4. To prevent the formation of deposits in superheaters or turbines.

The methods of accomplishing these objectives are as follows:

Corrosion is prevented by maintaining the boiler water within a narrow alkaline range as indicated by the pH (which should be between 10 and 11) and by removing as much as possible of the dissolved gases in the condensate or boiler feed before it reaches the boiler.

The alkalinity is maintained by the use of caustic soda, soda ash or one or more of the polybasic phosphates. The phosphates have the advantage of helping to prevent scale; caustic soda has the disadvantage of inconvenience ir hand-

ling as well as the ability to cause boiler embrittlement when not closely controlled; and soda ash has the disadvantage of decomposing to caustic soda and releasing carbon dioxide which, when in solution, is corrosive to piping.

The removal of dissolved gases was discussed under feed-water heaters. The best deacrating feed-water heaters will not remove all of the dissolved oxygen from the boiler feed, so it is customary to maintain a residual of about 10 ppm of sodium sulphite in the boiler water to act as an oxygen scavenger, according to the reaction:

The sodium sulphate produced has the advantage of helping to inhibit boiler embrittlement.

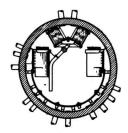
Bailer embrittlement, which is caused by caustic attack on intergranular cementite, is prevented primarily by avoiding riveted boiler drums, and inhibited in all cases by maintaining concentrations of certain materials in the boiler water. The best known inhibitor is sodium sulphate which is maintained at a ratio of about three to one with relation to the concentration of sodium carbonate in the boiler water. Trisodium phosphate has an embrittlement inhibiting property, and will replace an equal weight of sodium sulphate for this purpose. Organic inhibitors, such as sodium lignin sulphonate (from sulphite paper mill wastes) and quebracho extract (a tannin) are also used to some extent.

Scale formation is usually prevented or minimized by maintaining a residual of phosphate ion (about 50 ppm) in the boiler water. Calcium phosphate is less soluble than calcium carbonate or calcium sulphate, which are the principal constituents of boiler scale, but it precipitates as a sludge which is much less adherent than the other calcium precipitates. Sludge conditioners, such as sodium alginate (from certain seaweeds) and starches, are sometimes used to improve the fluidity of the sludge, thus making it easier to remove by boiler blowdown.

Deposits in superheaters and turbines are the result of carryover of solids from the boiler with the steam. These deposits can be prevented or minimized by careful attention to the maintenance of a practical minimum of dissolved solids in the boiler water by proper blowdown, and by the installation of efficient steam purifiers inside the boiler. Figure XI-16 shows several types of steam purifiers which are used in modern steam generators.



.mprove separation of steam from water and to clean steam involves baffles to change direction and "washing" of steam by feedwater contribugal separator in exit zone oi, flows to baffles, scrubbors, driers



Steam leaves baffle to enter

Steam enters phays water law

Fig. XI-16. Steam purifiers.

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CHAPTER XII

FUEL HANDLING EQUIPMENT

- 62. The simplest fuel handling equipment will be found in the plant which burns gas. The complexity of the system increases when fuel oil is used, and becomes a large part of the plant investment when coal is burned.
- 63. Ges Burning Equipment--The fuel handling equipment for a gas burning plant would consist of a pipe line bring-ing the gas to the plant, a gas holder (when the supply is subject to fluctuation), gas meters, control valves and burners. Figure XII-1 shows the elements of a gas fuel system and figure XII-2 shows a typical gas burner.

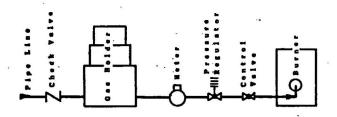


Fig. XIT-1; Gas fuel system.

84. Fuel Oil Burning Equipment -- The fuel handling equipment for a fuel oil burning plant would consist of unloading facilities (docks or railroad sidings), tanks (for about a 30 day supply), pumps, strainers, heaters, control valves and burners. Figure XII-3 shows the elements of a fuel oil burning system and Figure XII-4 shows a typical oil burner with mechanical atomizing and steam atomizing burner tir details. Figure XII-5 shows a so-called rotary cup oil burner which embodies a forced draft primary air fan, fuel oil pump and controls. This type of burner is well adapted to small



Fig. XII-2. Gas burner.

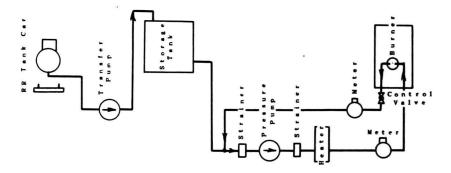


Fig. XII-3. Fuel oil system.

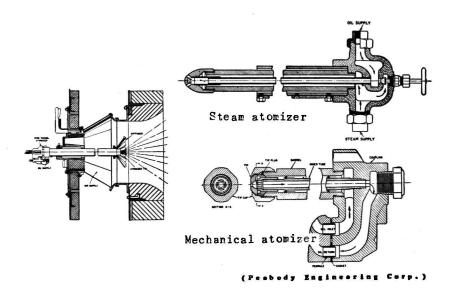
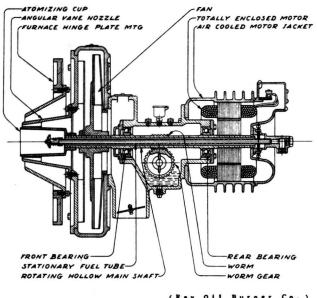


Fig. XII-4. Oil burner.

boilers, since burners of this type are available in sizes to handle up to about 200 boiler horsepower.

85. Coal Burning Fourpment-The variety of fuel handling equipment for a coal burning plant is quite large. Whether the plant burns lump coal or pulverized coal, a storage yard of sufficient size for about a 30 day supply of



(Ray Oil Burner Co.)

Fig. XII.5. Rotary cup oil burner.

coal is necessary; as well as conveying equipment for storing and reclaiming the coal, bringing the coal to sizing crushers and then to the plant bunkers. If the plant operates with lump coal it will be fed from the bunkers to the stokers through weighirg devices. Then, after the fuel is burned, the ash must be disposed of in some way (some companies sell it for road surfacing or for concrete block aggregate). Figure XII-6 shows a typical lump coal (stoker) burning system, and Figure XII-7 shows a pulverized coal system of the direct firing type ...

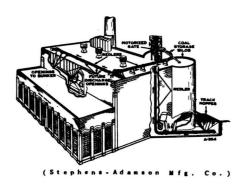


Fig. XII-6. Coal stoker system.

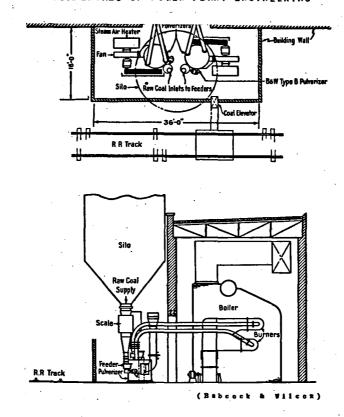


Fig. XII-7. Coal pulverizer system, direct firing.

Sometimes the pulverizers are operated to store pulverized coal in bunkers from which it is fed to the burners. The storage system requires smaller pulverizers than the direct firing system, but the storage of pulverized coal is hazardous and more plant space is taken up by the extra set of bunkers. Figure XII-8 shows a coal storage yard with the coal handling machinery.

86. Stokers--The equipment used for burning lump coal in a boiler furnace is known as a stoker, and it may be either underfeed or overfeed, depending upon whether the coal comes in under or over the air supply.

Overfeed stokers may be of the travelling grate type

(suitable for free burning coals), the chain grate type (suitable for caking coals), or the spreader type with dumping or travelling grates (suitable for a wide variety of coals). Figure XII-9 shows some of the various overfeed

stokers in use at the pres-

ent time.



Onderfeed stokers are of the retort type and are suitable for high volatile coals. There may be one or more retorts, depending upon the amount of fuel to be hurned. Figure XII-10 shows typical single and multiple retort stokers presently in use.

(Link Beit Co.)

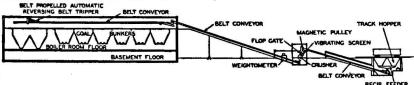


Fig. XII.S. Coul storage system.

Table XXI shows some approximate data for fuel burning rates per square foot of area for the various types of stokers.

TABLE XXI

FUEL BURNING RATES FOR STOKERS

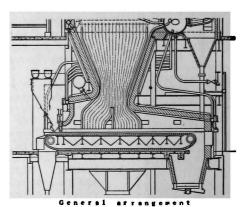
Pounds per hour per square foot

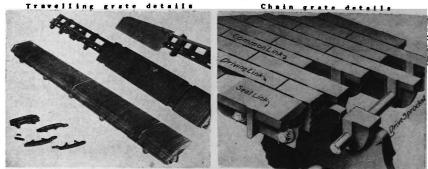
Cverfeed

Multirle retort

Spreader	 35	- 40
Travelling and chain grates		
Anthracite		35
Bituminous		6C
Underfeed		
Single retort	,	40

. 50 - 60





(a) Travelling grate or chain grate

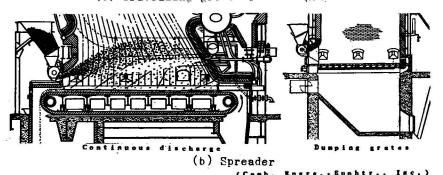
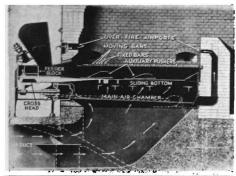
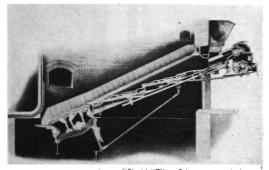


Fig. XII-9. Overfeed stokers





(comb. Engrg.-Supher., Inc.)
(a) Single retort

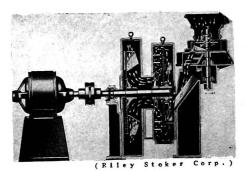


(Westinghouse Blec. Corp.)

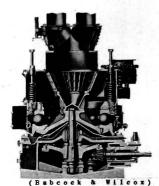
(b) Multiple retort

Fig. XII-18. Underfeed stokers.

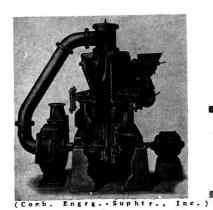
87. Pulverizers-The function of a pulverizer is to grind previously crushed coal to a fine powder (about 90 per cent through a 200 mesh screen) so that it will burn in a manner similar to gas or oil. The pulverizer may be of the types shown in Figure XII-11. Mill exhauster fans may be built into the pulverizer, as with the attrition mill, or they may be separate. Mill feeders may also be built in or separate. About twenty per cent of the total combustion air is used as primary air (always preheated) for conveying the coal from the pulverizers to the burners. Pulverizers require powerful motors, and Table XXII shows approximate power requirements for the various types of mill.

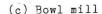


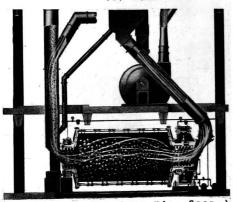
(a) Attrition mill



(b) Ball mill



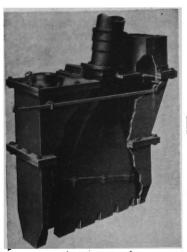




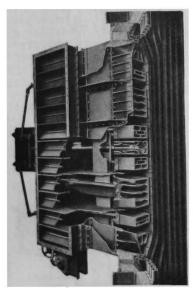
(Kennedy-Van Saun Mfg. Cor

(d) Tube mill

Fig. XII-11. Pulverizers.



(a) Vertical



(b) Tangential

TABLE XXII

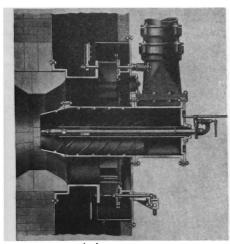
PULVERIZER FOWER REQUIREMENTS

KWH per ton

Ball and race mill	8 -	12
Bowl mill	11 -	16
Tube mill	11 -	17
Attrition mill	11 -	17

88. Pulverized Coal Burners--A pulverized coal burner is
designed to mix the coal and
primary air with secondary air
as it is blown into the furnace. A few of the many types
of pulverized coal burners are
shown in Figure XII-12.

89. Horizontal Cyclone Burner--A recent development



(c) Horizontal
(comb. Engrg.-Supher., Inc.)

Fig. XII-12. Pulverised coal burners.

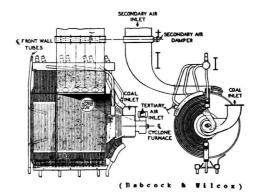
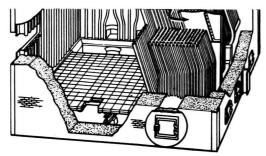
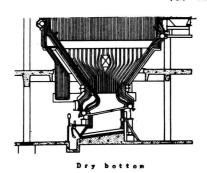


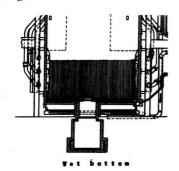
Fig. XII-13. Horisontal cyclone burner.

in coal burning equipment is the horizontal cyclone burner which is designed to burn low grades of coal at high rates without the necessity for pulverizing it. Figure XII-13 shows the burner which would be attached to the side of a furnace and connected into the waterwall circulation. The furnace is smaller than is usual with other



(a) Oil or gas





(b) Pulverized coal

Fig. XII-14. Furnace bottoms.

(Power)

types of fuel burning equipment because most of the combustion takes place in the cyclone. Coal, crushed to about four mesh size, is fed with primary air at high velocity into the rear end of the burner where it is met by a high velocity tangential stream of secondary air. The coal sticks into a layer of molten slag and burns as the air blows across it. The slag is continuously moving toward a drain hole at the bottom of the burner front from which it drips into the main furnace.

90. Furnace Bottoms--Furnace bottoms vary in design depending upon the type of fuel to be burned. Some of the various types of furnace bottoms are shown in Figure XII-14. Furnaces for gas or oil usually have sloping flat water cooled bottoms. Furnaces for stokers, naturally, have the bottom made up of grates or retorts. Furnaces for pulyerized

coal may be of the wet or dry bottom types. The wet bottom type has little or no cooling surface on the furnace bottom, so that the ash which falls on it remains in the molten condition in a pool, from which it is tapped, either continuously or intermittently. The dry bottom furnace has a completely cooled hopper bottom, so that the ash which falls to the bottom is cooled sufficiently to be removed in granular form.

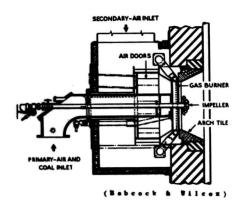


Fig. XII-15. Three-fuel burner

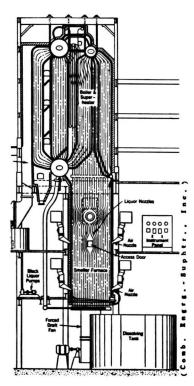


Fig. XII-16. Sulphate paper mill chamical recovery boiler.

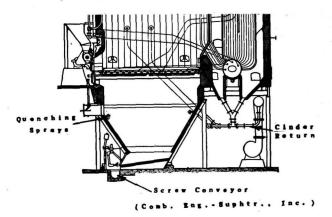


Fig. XII-17. Stoker ash disposal system.

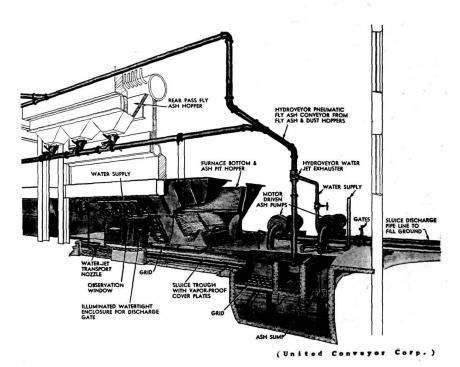


Fig. XII-18. Pulverised coal ash disposal system.

- 91. Multiple Fuel Burning Installations--In some localities more than one fuel may be available, and plants have
 been designed to burn two or three fuels separately or in
 combination. Figure XII-15 shows a three-fuel burner for
 pulverized coal, oil or gas. Other plants have been designed
 to burn industrial wastes in combination with some other
 fuel. Figure XII-16 shows a sulphate paper mill chemical
 recovery boiler which burns black liquor (containing lignin)
 in combination with fuel oil.
- 92. Ash Removal Systems--Coal burning plants must have some means of ash removal. Stoker firing produces a large cinder which can be dropped into a hopper car in an air lock under the boiler, or removed by a screw conveyor, as shown in Figure XII-17. Fulverized coal ash can be tapped from wet bottom furnaces into quench pits from which it can be transferred to any desired location by sludge pumps. Dry bottom furnaces may discharge into a conveyor or a quench pit as with the wet bottom furnace. One of these systems is shown in Figure XII-18.
- 93. Dust Collectors--Coal burning plants which are located close to population centers must be furnished with some sort of apparatus to remove dust and cinders from the flue gas in order to avoid the creation of a nuisance. Pulverized coal plants produce an extremely fine ash (called flyash), and at least half of the ash in the fuel goes along with the flue gas. Stoker fired plants do not produce significant amounts of flyash but some small cinders do travel through the steam generator. It is often thought that fuel oil produces no flyash, but some Florida cities have ordinances requiring dust collectors on fuel oil burning power plants within their municipal limits.

Cirder catchers operate on the principle of changing the direction of gas flow so that heavy particles will drop out of the gas stream. Figure XII-19 shows a cirder trap of this type.

Dust collectors may be of the centrifugal type, which operate in a manner somewhat similar to the cinder trap except that the gas velocity is higher; the electrostatic type which uses a high direct current potential to charge the

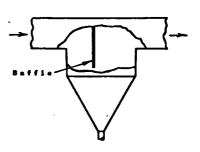
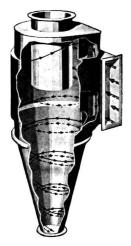
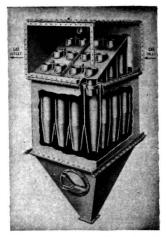


Fig. XII-19. Cinder trap.



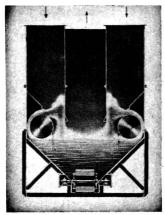
(Buell Bogrg. Co.)
(a) Single



(b) Multiple

Fig. XII-20. Cyclone dust collectors.

the dust particles so that they may be collected on electrodes; the bag type which employs fabric bags to filter the gas; and the wet scrubber which uses water for washing the gas. Figures XII-2C through XII-23 show these various types of dust collector.



(Schmieg Industries)

Fig. XII-21. Flue gas
scrubber.

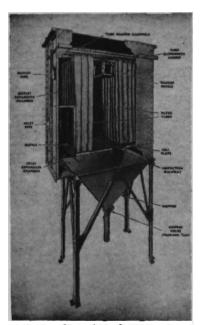
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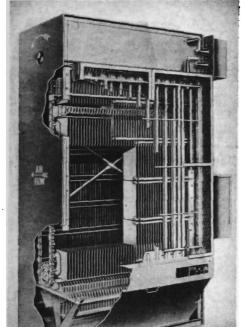
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Steam Generation," Transactions
A.S. N. E., Volume 50, 1927-28.



(Aser. Penadry Equip. Co.



(Aperican Air Filter Co.)

Fig. XII-22. Beg type dust collector.

dust collector.

CHAPTER XIII

DRAFT EQUIPMENT

94. Draft equipment consists of fans and stacks (or chimneys) whose purpose is to supply air for the combustion of fuel and to remove the products of combustion from the steam generator for disposal to the surrounding air.

95. Fans--Fans in use for the production of draft are almost always of the centrifugal type, although some small installations use axial flow fans and the new horizontal burners use axial flow blowers.

Axial flow fams follow the same laws of similarity as centrifugal pumps, but the skapes of the characteristic curves are considerably different. It will be noted from

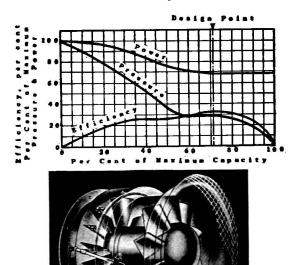
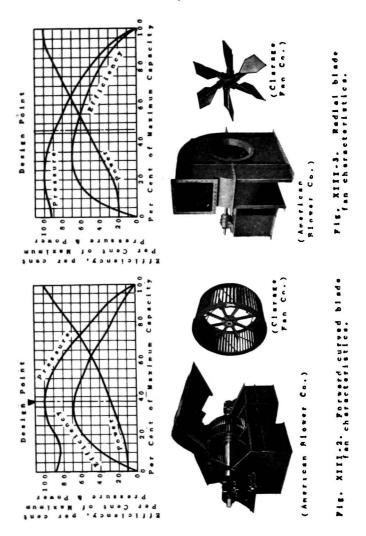


Fig. XIII-1. Axial flow

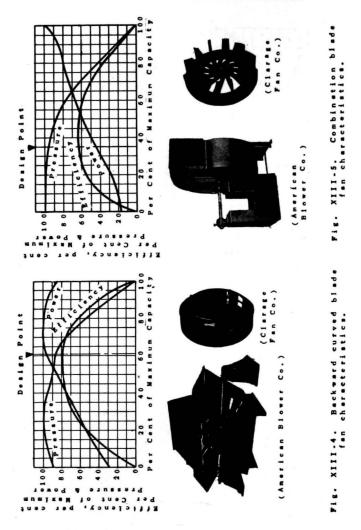
Figure XIII-1 that the pressure curve has an unstable region which makes it impossible to parallel two axial flow fans within the region of best operating efficiency.

Centrifugal jans follow the same laws of similarity as centrifugal pumps, but their characteristic curves will have somewhat different shapes due to the use of several different types of blade arrangement. Blade arrangements. as shown in Figures XIII-2 through XIII-5, may be either forward curved, backward curved, radial or combination.

(L. J. Wing Co.)



The forward curved blade far has a limited application in power plants because it is possible to overload the motor at zero head. In addition, it is impossible to operate these fans in parallel because of the existence of more than one point on the head-capacity curve which has the same ordinate. Such a characteristic leads to hunting (instability) in the system.



The radial blade fan is extensively used in pulverizer exhausters because it is difficult for the blades to become fouled. Otherwise, the power characteristic is unfavorable as with the forward curved blade fan.

The backward curved blade fan is the most popular type in use in power plants because, like the centrifugal pump,

it has a self-limiting power characteristic, and, due to the continually falling head-capacity characteristic, two or more fans can be operated in parallel without hunting.

The combination blade fam has characteristics intermediate between those of the radial and backward curved blade fams.

96. Heed and Work Relations in fans--As previously mentioned, centrifugal fans are similar to centrifugal pumps, the only difference being in the fluid pumped and the magnitude of the pressure rise. It can be seen that:

$$ahp = e_{f}(shp)$$

where: ahp and shp = air horsepower
and shaft horsepower
e, = fan efficiency

Air horsepower is the power required to move the air, or

$$ahp = \frac{M_a H_t}{60 \times 33000}$$

where: M_a = flow, lbs/hr
H_t = total head on fan
ft of fluid

Since pressures in the steam generator combustion air and flue gas circuits are stated in terms of inches of water and fan capacities are stated in cubic feet per minute, air horsepower is also

$$ahp = \frac{62.4 \ Q \ h_t}{12 \ X \ 33000}$$

where: Q = fan capacity, cu ft/min h_t = total pressure rise through fan, in. water

The total pressure rise through the fan is the sum of the friction losses in the ducts and apparatus; the increase in velocity head through the fan; the losses at the entrance and exit of the system; the difference in pressure between the inlet and discharge; and the static head (or chimney effect), or

$$h_t = h_f + \frac{\nabla_d^2 - \nabla_i^2}{2g} + h_{s_d} - h_{s_i} + h_{p_d} - h_{p_i} + h_{s_d} + h_{s_i}$$

Figure XIII-6 shows a steam generator with its fan and duct system and the location of the various positive and negative pressures.

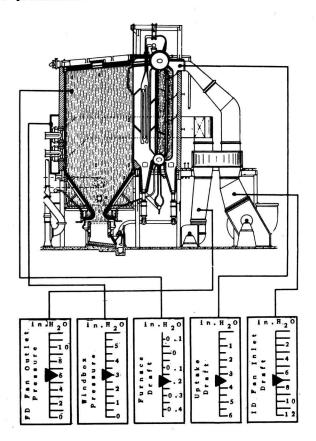


Fig. XIII-6. Steam generator fon and duct system.

97. Fan Speeds--The maximum efficiency of fans is approximately the same regardless of type or size. Larger fans will operate at slower speeds than smaller ones for the same head. Table XXIII shows the ranges of speeds and sizes to be expected for fans in use in power plants.

TABLE XXIII

ABBREVIATED PERFORMANCE—IMPROVED EXHAUSTERS—RIM OR L.S. WHEEL

Fan Size No	•.	11	13	16_	_ 17 _	19	21	_ 23 _	25	29	33	37	41
Wheel Dian	ĸ.	16"	20'	23′_	26"	30′	33′	36"	40'	46*	53 ″	80"	- 66"
2000 Q.V. R	PM PM HP	1172 1305 .66	1586 1040 .90	2170 905 1.23	2794 800 1.58	3526 694 2.00	4312 631 2.43	5176 679 2.92	5120 520 3.48	8248 452 4.58	10640 393 6.01	13396 346 7.56	16464 315 9.30
2500 O.V. R	PM HP	1465 1600 1.25	1982 1295 1.69	2712 1110 2.31	3492 984 2.97	4407 850 3.75	8390 776 4.58	6470 710 8.50	7650 640 6.50	10310 555 8.78	13300 483 11.30	16745 426 14.20	20580 388 17.50
3000 O.V. R	PM PM HP	1758 1865 2.02	2379 1490 2.73	3255 1295 3.74	4191 1140 4.80	5289 990 6.06	6468 905 7.40	7764 830 8.90	9180 745 10.50	12372 647 14.20	15960 562 18.30	20094 495 23.00	24696 450 28.40
3250 O.V. R	PM HP	1904 2065 2.70	2577 1650 3.66	3526 1430 8.00	4540 1266 6.45	5730 1100 8.14	7007 1000 9.91	8411 918 11.90	9945 825 14.10	13403 715 19.00	17290 623 24.50	21768 550 30.90	26754 500 38.00
3500 Q.V. R	PM HP	2051 2260 3.48	2775 1805 4.72	3797 1570 6.45	4889 1390 8.30	6170 1205 10.50	7546 1095 12.80	9058 1005 15.40	10710 900 18.20	14434 784 24.58	18620 683 31.70	23443 600 39.80	28812 548 49.00
3750 O.V. R	PM HP	2197 2440 4.34	2974 1950 5.86	4068 1695 8.08	5239 1495 10.35	6610 1300 13.05	8085 1180 15.95	9705 1080 19.15	11475 978 22.60	15465 845 30.50	19950 736 39.40	25117 650 49.50	30870 590 61.00
4000 Q.V. R	PM HPM HP	2344 2605 5.30	3172 2090 7.16	4340 1805 9.80	5588 1600 12.60	7052 1385 15.95	8624 1265 19.45	10352 1155 23.35	12240 1040 27.60	16496 903 37.20	21280 782 48 00	26792 692 60.50	32928 630 74.40
4800 O.V. R	PM IPM IHP	2637 2920 7.50	3568 2330 10.15	4882 2025 13.90	6286 1790 17.85	7933 1550 22.60	9702 1415 27.60	11646 1295 33.20	13770 1165 39.10	18558 1010 52.90	23940 880 68.20	30141 776 85 90	37044 705 105 50
5000 O.V. R	PM PM BHP	2930 3215 10.00	3965 2860 13.50	5425 2225 18.50	6985 1970 23.80	8815 1710 30.10	10780 1555 36.70	12940 1425 44.10	15300 1285 52.20	20620 1110 70.30	26600 968 90.75	33490 856 114.00	41160 776 140.50

ABBREVIATED PERFORMANCE-TYPE HV FANS, SINGLE WIDTH, SINGLE DILET

Pan Slee No.		ı.	11/4	11/2	1%	2	21/4	21/2	21/4	3	31/4	31/2	4	41/2	8
Wheel Dia.		13"	1814'	191/4"	2234"	26"	291/4"	321/4.	35%	39"	421/4"	451/2"	52"	581/2"	63"
1/200 O.V. 1200 O.V. 1000 V.P.	CFM RPM HP	1035 509 .12	407	327	3156 281 .29	4128 245 .38	217	197	7782 178 .75	9288 164 ,85	146	136	120	20880 106 2.00	25800 96 2.47
1500 O.V. ,106 V.P.	CFM RPM HP	1122 637 ,19		412	353			6981 247 1.00		10062 206 1.44		174	152	135	27950 122 3.87
34° 8.P. 1400 O.V. .123 V.P.	CFM RPM HP	1208 748 .28	598		3682 420 473	4816 367 .95	325	7516 294 1.48	264	245	222	206	180	160	30100 144 5.80
1° S.P. 1900 O.V. .140 V.P.	CFM RPM HP	1294 851 -39	2016 680 .61		478	417	369	8055 335 1.99	302		252	233	204	181	32250 163 7 .75
147 S.P. 1600 O.V. .160 V.P.	CFM RPM HP	1380 944 .52	754	618	530	463			336	309	280	26!	228	203	34400 182 • 10.2
14, 8.P. 1800 O.V. 202 V.P.	CFM RPM HP	1552 1039 .70	830	683	585	512	453	410	369	341	308	286	250		38700 200 13.8

MOTE for performance of double width, double intel NY from (evaluable in sizes No. 1% and larger) multiply (TM s 2 HPM s 1.015, MBP s 1.1 and OV s 1.11.

ABBREVIATED PERFORMANCE—TYPE W FANS, SINGLE WIDTH, SINGLE INLET

Pan Size No.		11/4	1%	11/4	1%	8	21/4	21/2	2%	*	31/4	31/2	3%	41/4	4%
Wheel Dia.		161%,	17%*	1017.	23%"	26"	291/4"	321/2"	35%"	39"	421/4"	451/2"	80%	551/4"	61%
V, B.P. 1000 O.V. .063 V.P.	CFM RPM HP	1344 809 -11	1627 735 .13	1935 620 .14	2630 532 .19	3440 464 -25	4350 412 -32	5370 372 .39	6485 338 .48	7740 310 57		266		218	196
14° S.P. 1200 O.V. .000 V.P.	CFM RPM HP	1613 1035 -24	1952 940 29	2322 792 29	3156 680 .40	4128 594 ,52	5220 528 .66	6444 476 .81	7782 432 ,98	396	366	340	208	280	250
1300 O.V. 100 V.P.	CFM RPM HP	1747 1188 .37	2115 1079 .45	2516 908 .43		4472 680 .77	\$655 604 .98	6981 544 1.20	8430 494 1.46	454	418	389	351	320	286
1° 8.P. 1400 O.V. .123 V.P.	CFM RPM HP	1882 1325 .51	2278 1203 62	2709 1012 .60	868	4816 760 1.08	676			506		434	392	358	
114" B.P. 1800 O.V. .140 V.P.	CFM RPM HP	2016 1449 .68	1317	2903 1108 ,80	948	5160 830 1.42	6525 736 1.79			11610 554 3.19	511	474	428	390	349
114" B.P. 1600 O.V. .160 V.P.	CFM RPM HP	2150 1567	1423	3096 1204 1.01	1032	\$504 902 1.80	6960 800 2.25	8592 722 2.81		602		516	465	424	31024 380 10.15

DAP 2 245 and OV 2 1.11.

- 98. Draft Losses -- Since a steam generator installation may include both forced and induced draft fans, the draft losses may be divided into two classifications, as follows:
 - 1. Forced draft (pressure) losses (air system).
 - a. Loss in duct between fan and air heater.
 - b. Loss in air heater.
 - c. Loss in duct between air heater and windbox.
 - d. Loss through fuel bed or burner.
 - 2. Induced draft losses (flue gas system).
 - a. Loss between furnace and boiler outlet.
 - b. Loss in duct between boiler and economizer.
 - c. Loss in economizer.
 - d. Loss in duct between aconomizer and air heater.
 - e. Loss in air heater.
 - f. Loss in duct between air heater and fan.
 - g. Loss in duct between fan and stack.

All of the draft losses, except those for the ducts, are usually taken from data supplied by the manufacturers, but the methods of estimating the losses are of interest.

Duct draft losses have been discussed in Chapter X.

Mir heater draft losses vary according to the arrangement of the surfaces. The draft loss on the gas side of tubular air heaters can be estimated by treating the air heater tubes as round ducts in parallel and using the average gas velocity; considering the inlet and outlet boxes as part of the main ducts; and adding 0.05 inches of water draft loss for each right angle turn. The draft loss on the air side of tubular air heaters may be estimated by using the empirical formula (adding 0.05 in. H₂0 for each right angle turn):

$$\Delta p = 0.0002307 \frac{f N (G/1000)^2}{p}$$

where: Ap = draft loss, in. water

f = a friction factor

= 0.06 for in-line tubes

= 0.09 for staggered tubes

N = number of restrictions

p = arithmetic mean gas

density, lbs/cu ft

G = mass flow of gas, lbs/hr

per sq ft of flow area

Fuel bed or burner draft losses can be estimated from Figures XIII-7 and XIII-8.

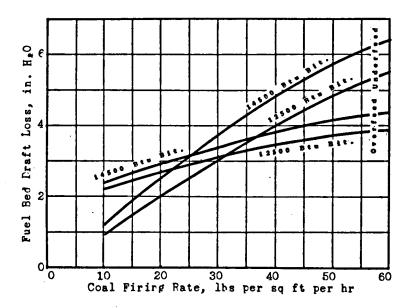
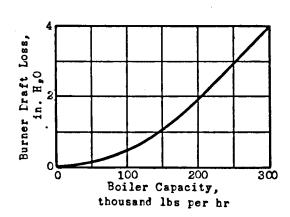


Fig. XIII-7. Fuel bed draft losses.



Pig. XIII-8. Burner draft leases

Bailer, superheater and economizer draft losses depend upon the arrangement of the surfaces and the mass flow of the gas. Draft losses for fire-tube boilers can be estimated in the same manner as for the gas side of tubular air heaters. Praft losses for water-tube boilers, superheaters and economizers can be estimated in the same way as for the air side of tubular air heaters.

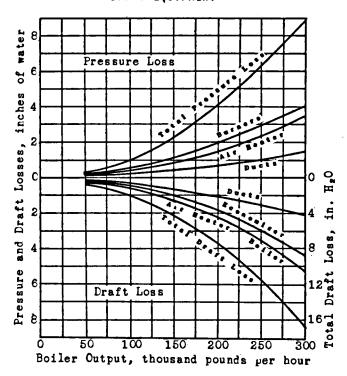
Due to the fact that ducts and boiler settings are never gas tight, an allowance of ten per cent on the gas volume and air volume should be made when specifying fan sizes. The normal practice is to control the furnace draft at between 0.1 and 0.2 inches of water to allow the opening of inspection doors without danger to the operators. Some modern steam generators, however, especially marine types and those utilizing the horizontal cyclone burner, dispense with the induced draft fan and operate entirely with forced draft. Figure XIII-9 shows some draft loss curves for a typical steam generator. It will be seen that, as in all devices involving fluid flow, the draft loss is proportional to the square of the quantity flowing.

Example 33 shows the method of making fan specifications.

Determine the fan specifications for an oil fired boiler producing 300,000 lbs per hour of steam at 925 psia and 900 F from feed-water at 300 F. The fuel is that of Example 15 with a heating value of 18500 Btu/lb, and it is burned with 15 o/o excess air to produce a boiler efficiency of 89 o/o. Air temperature is 100 F and flue gas temperature is 300 F. The boiler draft losses are shown in Figure XIII-9.

Solution:

Heat added to steam = $(1450.9 - 269.6)$	= 1180.3	Btu/1b
Oil fired = $\frac{300000(1180.3)}{0.89(18500)}$ =	21500	lbs/hr
Theoretical air	13.9	lbs/lb
Actual air = 13.9(1.15)	16.0	lbs/lb
Weight of air per hr.(incl. 10 o/o		
leakage)	378000	lbs/hr
Density of air at 100 F	0.0	7 lbs/ft ^s
Volume of air at 100 F	88800	cfm



Pig. XIII-9. Steam generator draft losses.

Wet gas produced 17.0 lbs/lb
Weight of gas per hr. (incl. 10 o/o
leakage) 402000 lbs/hr
Tensity of gas at 300 F 0.055 lbs/ft^a
Volume of gas at 300 F 122000 cfm

From Figure XIII-9, the forced draft loss is 8.8 in. of water, and the induced draft loss is 16.7 in. of water at a steam flow of 300,000 lbs/hr.

The fan specifications would be:

Forced draft fan: 89,000 cfm air at 100 F, 9 in. water.

Induced draft fan: 122,000 cfm gas at 300 F, 17 in. water.

99. Stacks--Small boilers may depend entirely upon stacks for the production of the necessary draft for operation, while large steam generators use stacks mainly for flue gas disposal, with draft production for emergency operation without the fans a secondary consideration. As a matter of fact, most large steam generators are completely incapable of being operated without the induced draft fan.

A stack produces draft because the hot gas inside is less dense than the air outside. The draft produced by a stack is proportional to the height of the stack and to the difference in density between the outside air and the flue gas, or (allowing for friction in the stack)

$$h = 0.254 \text{ B H } (\frac{1}{T_a} - \frac{1}{T_f}) - 0.77 \frac{p \text{ f H}^2}{D_4} (\frac{T_f}{T_a} - 1)$$

where: h = draft produced by stack, in. water

B = barometric pressure, in. Hg

H = height of stack above duct inlet, ft

m - - - A 11 - 2 - A - - - - - - D

Tα = outside air temp, deg R

 T_g = arithmetic mean gas temperature, deg R

p = density of gas, lbs/cu ft

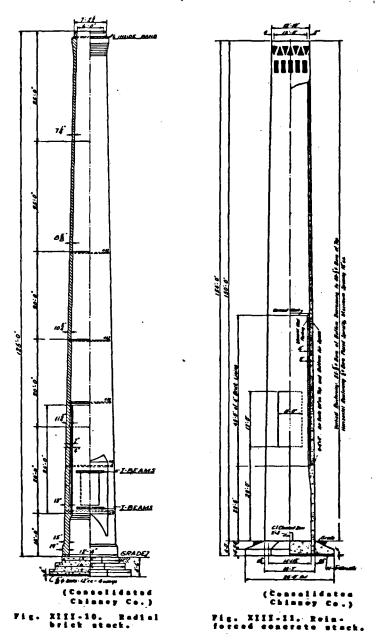
D = inside diameter of stack, ft

f = a friction factor

= 0.0013 for steel stacks

= 0.0018 for brick-lined stacks

It will be seen that, if given a certain draft to be produced, both stack height and diameter will be unknown, but if the first term on the right hand side of the equation is solved the minimum stack height will be obtained. Then the dimensions can be determined by trial with due consideration to stability of self-supporting stacks and to economy of construction. As a general rule, a guyed stack can be designed for maximum economy, a condition that is obtained when the product of the height and diameter is at the minimum. Self-supporting stacks must be designed to give maximum economy within the range of stability, a condition which will be met when the ratio of diameter to height is between C.O3 and O.O8 and the product of diameter and height is nearest to that for maximum economy. Figures XIII-10 and XIII-11 show two common types of power plant stacks, and Example 34 shows the method of calculating stack dimensions.



Example 34.

Determine the most economical dimensions for a brick stack to produce a draft of 0.6 inches of water for the boiler of Example 33. Barometric pressure is 30 inches of mercury and ambient temperature is 90 F.

Solution:

For a first approximation of the height,

0.6 = 0.254(30) H
$$(\frac{1}{550} - \frac{1}{760})$$

 $H_{min} = 156 \text{ ft}$

Then, by trial in the equation for stacks:

$$0.6 = 0.00384 \text{ H} - 0.0000226 \frac{\text{H}^2}{\text{D}_4}$$

H	160	170	180	190	200
D.	38.6	12.3	8.0	6.3	5.4
HD.	6180	2090	1448	1192	1070
D./H	0.241	0.072	0.045	0.033	C.027

It is apparent that the product HD; is still decreasing at a stack height of 200 ft. From the standpoints of economy and stability, the best choice would be the 6.3 ft diameter by 190 ft tall stack.

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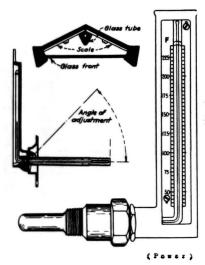
Combustion, August, 1938.

CHAPTER XIV

SUPERVISORY AND CONTROL EQUIPMENT

100. Every power plant contains a great variety of instruments and equipment for indicating, recording and controlling the performance of the plant in many details. The various instruments and controls may be divided into classifications as to the variable, such as temperature, pressure, speed, fluid level, combustion conditions, chemical characteristics and fluid flow.

101. Temperature--Temperature may be indicated by mercury thermometers, bourdon tube thermometers, thermocouples or electric resistance thermometers. In addition, all of these devices, except the mercury thermometer, may be made to control temperature by means of secondary devices such as valves or switches and may be made to record the temperature with relation to time. Figures XIV-1 through XIV-5 show some of the many temperature devices which are used in power plants.





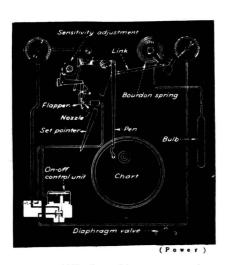


Fig. XIV-2. Air operated temperature controller.

102. Pressure--Fressures may be indicated by manometers using various fluids, bourdon gages or diaphragm gages, depending upon the pressure range to be covered. The ranges

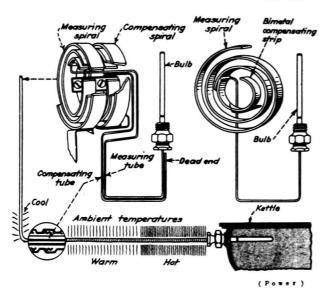


Fig. XIV-3. Bourdon

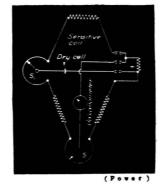


Fig. XIV-4. Electrical resistance tempera-

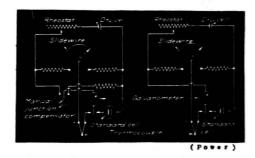
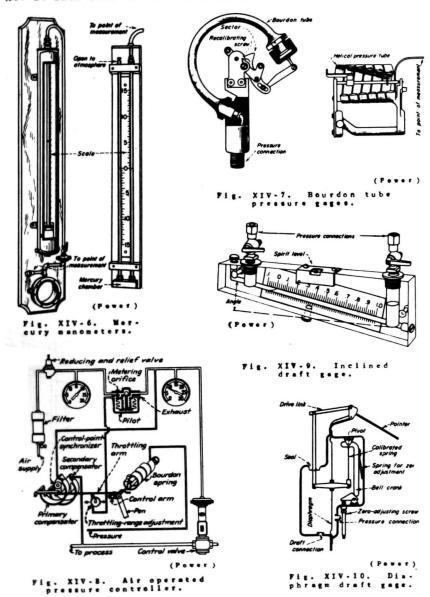


Fig. XIV-5. Thermocouple temperature potentioneters.

of usefulness of these various types overlap somewhat, but in general, manuscreters are used for indicating pressures in the range from 0.601 inch of water to 100 inches of mercury; diaphragm gages for indicating, recording or controlling pressures in the range from 0.01 inch of water to 10 pounds per square inch; and bourdon gages for indicating, recording or controlling pressures in the range from one pound per square inch to the maximum obtainable. Figures XIV-6 through XIV-10 show some of the instruments of this class.



103. Speed--The most common speed controlling device is the centrifugal flyball resisted by a spring, as shown in Figure XIV-11. The radial motion of the flyballs may be transferred to a throttle valve for controlling the speed of a turbine or engine within predetermined limits. A change

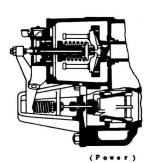
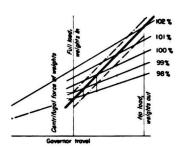


Fig. XIV-11. Flyball governor, direct acting.

in speed is always necessary to actuate the controlling device, therefore flyball governors have regulation characteristics as shown in Figure XIV-12. Since turbo-generators or engine-generators are constant speed apparatus, it is necessary to provide a device for readjusting the flyballs to a new load position. This device is arranged to adjust the governor spring tension or the throttle valve position, either locally or remotely by manual or automatic means.

The hydraulic governor is found on certain makes of steam turbines and consists essentially of an oil pump

consists essentially of an oil pump on the turbine shaft which produces a pressure which is proportional to speed in the same relation as with a flyball device. The pressure produced by the pump is resisted by a spring which has a load adjustment as in the flyball type of governor. A governor of this type is shown in Figure XIV-13.



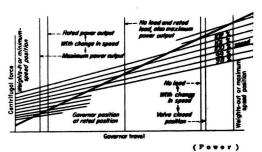
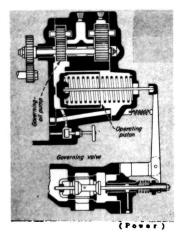


Fig. XIV-12. Governor characteristics.

The direct acting flyball or hydraulic governors are satisfactory for small turbines or engines of not more than a few hundred kilowatts caracity, but in larger sizes the governing forces are not sufficient to operate the heavy valve gear. To overcome this difficulty, hydraulic relays of various kinds are used to amplify the governor forces. Figures XIV-14 and XIV-15 show two of these relay systems.



Pig. KIV-13. Hydraulic governor, direct acting.

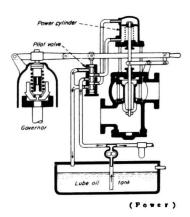


Fig. XIV-14. Flybell governor, relayed.

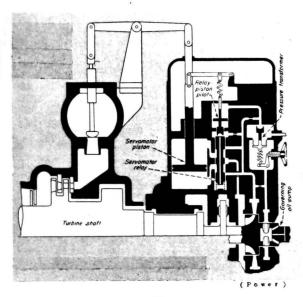
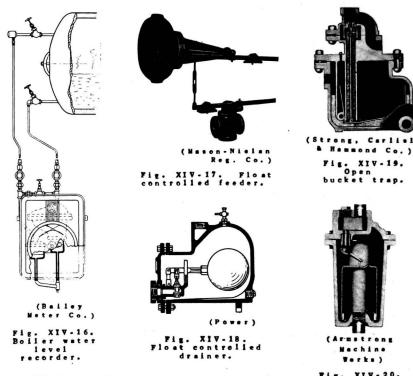


Fig. XIV-15. Hydraulic governor, relayed

104. Fluid Level--Fluid level instruments very frequently contain floats whose motion is transmitted to indicating, recording or controlling devices. The float type of instru-

ment is applicable to any pressure or level range for which a suitable float system can be devised. Figures XIV-16 through XIV-20 show several of the float type of fluid level instruments.



Fluid level in boilers may be control—

led by float types of instruments, but it
is much more usual to employ a thermostat—
ic device or a thermohydraulic regulator. The first, shown
in Figure XIV-21, utilizes a long inclined polished metal
tube connected at the bottom end to the water space and at
the top end to the steam space of the boiler. The tube is
arranged with a bell crank and lever so that any change in
length of the tube will be reflected as a change in water
control valve position. As the water level in the boiler
falls the tube lengthens, because the steam in the upper
part of the tube has a higher temperature than the water in
the lower part of the tube. This lengthening of the tube is
employed to open the water control valve. A rise in water

level causes a reverse action. These regulators have a falling water level characteristic with relation to boiler load, and the operating limits may be too wide for modern steam generators which are extremely sensitive to water level variations (with respect to steam quality). In these cases a second element which contains a

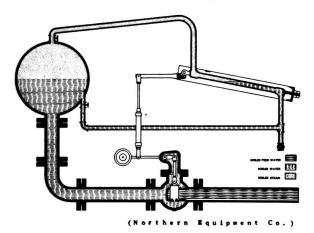


Fig. XIV-21. Thermostatic feed-water regulator.

spring loaded diaphragm is connected to the regulator in such a way that the pressure drop across the superheater is used to modify the thermostatic tube characteristics to maintain the water level within closer limits over the load range. This modification is shown in Figure XIV-22, and a hydraulic or air relay may be used to operate large valves.

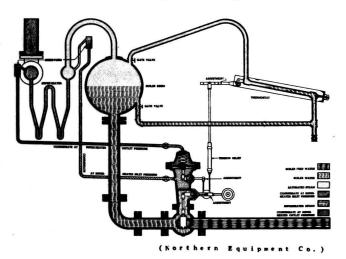
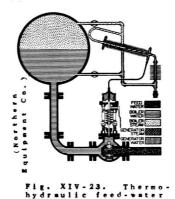


Fig. XIV-22. Two-element thermostatic feed-water regulator.



regulator.

The thermohydraulic boiler water level control employs an inclined tube similar to the thermostatic regulator except that it is surrounded by a jacket with heat radiating fins. The jacket is connected by tubing to a diaphragm type water control valve. The generator jacket, connecting tubing and diaphragm chamber are completely filled with water so that any decrease in water level in the boiler will result in an increase in pressure in the generator jacket and an opening action in the water control valve. This type of regulator is shown in Figure XIV-23. Its char-

acteristics are similar to those of the thermostatic type.

A third element, sensitive to water flow, may be added to the two-element regulator to coordinate water flow with steam flow and water level for smoother operation on rapid load changes, as shown in Figure XIV-24.

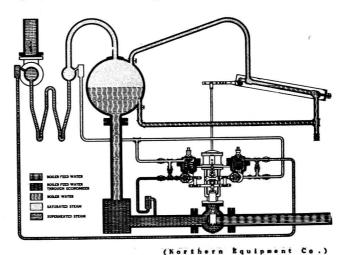


Fig. XIV-24. Three-element thermostatic feed-water

105. Combustion -- Some of the most valuable instruments in modern power plants are automatic combustion controls

which relieve the fireman from the necessity of manually controlling fuel and air flow, and which, incidentally, do a much better job than any human fireman could possibly do. Combustion controls are available for the smallest boiler as well as for the largest steam generator, and they will pay for themselves from fuel savings in a relatively short time.

All combustion controls for boilers or steam generators operate to maintain a constant pressure at the turbine throttle by varying the fuel and air supplied to the furnace. In addition, some controls have an element which maintains a constant draft in the furnace. Figures XIV-25 through XIV-27 show simple modulating or positioning electric controls for application to various fuels in small boilers. Large steam generators employ elaborate controls which meter fuel and air flows in exact relation to each other and to load. Some

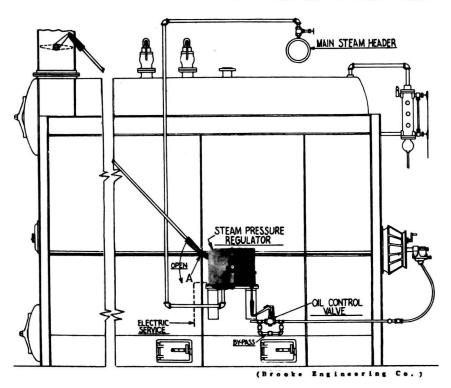


Fig. XIV-25. Modulating oil burner control.

of these control systems are shown in Figures XIV-28 through XIV-31.

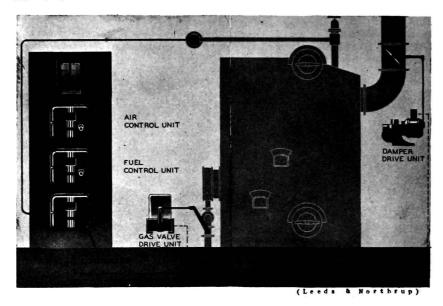


Fig. XIV-26. Modulating gas burner control.

106. Chemical Characteristics--Chemical characteristics of water in power plants are often controlled or supervised by electrical instruments. The chemical characteristics which are useful as control variables are pH and electrolytic conductivity. pH can be controlled in precipitation water softening because the pH of the water is affected by the softening chemicals. Electrolytic conductivity is most often used as a variable for operating alarm devices for signalling contamination of condensate by condenser leakage or faulty evaporator operation. Figures XIV-32 through XIV-34 show some of the instruments used for these purposes.

107. Fluid Flow--The most common fluid flow measuring devices utilized in power plants are nozzles and orifices with various types of secondary elements for measuring and recording the pressure differential produced by the primary element in terms of flow units. Figures XIV-35 through XIV-39 show some of the more common primary and secondary head type flow measuring devices.

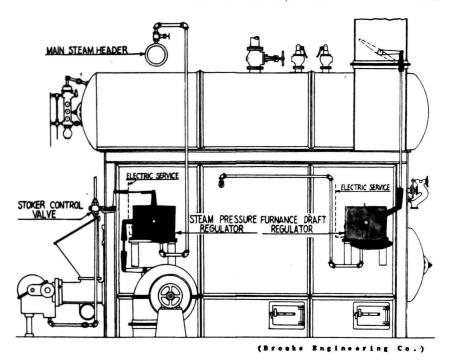


Fig. XIV-27. Modulating stoker control.

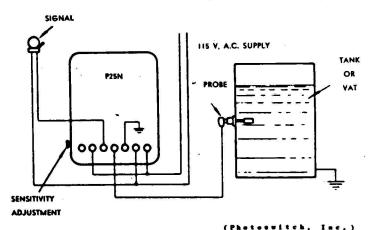
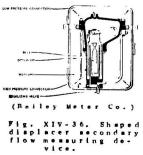


Fig. XIV-34. Conductivity siers.





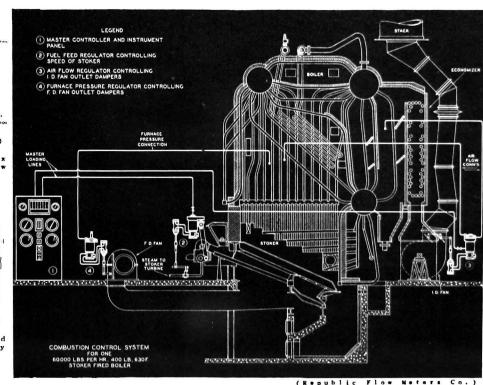
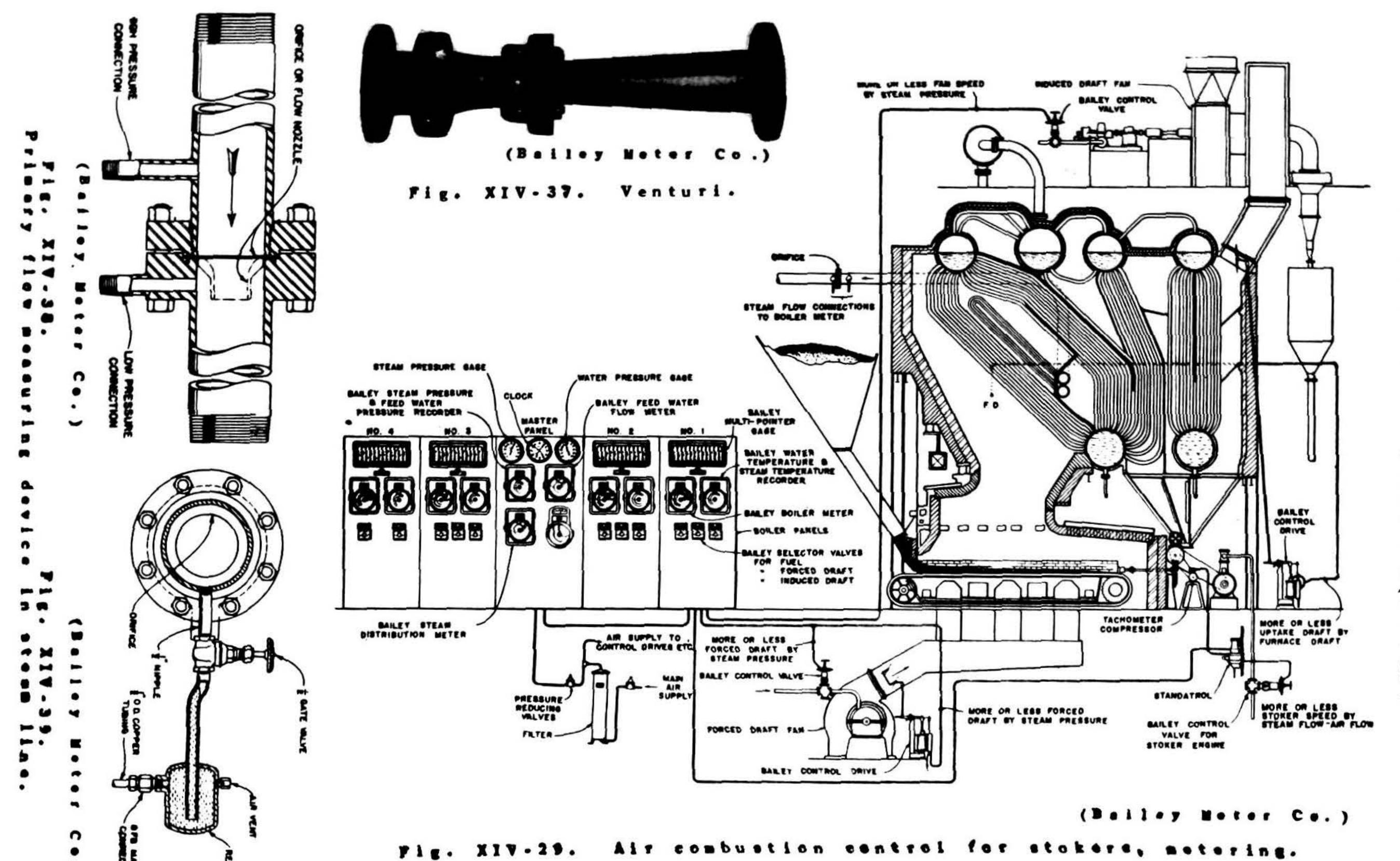


Fig. XIV-28. Air-hydraulic combustion control for stokers, metering.



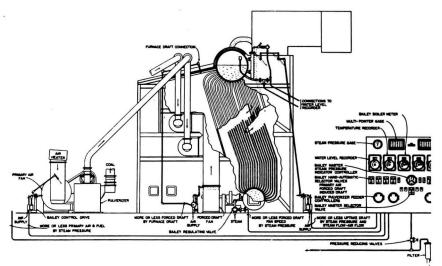


Fig. XIV-30.

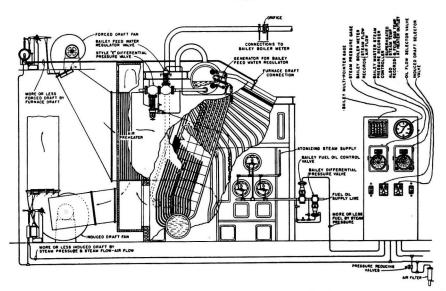


Fig. XIV-31. Air combustion control for

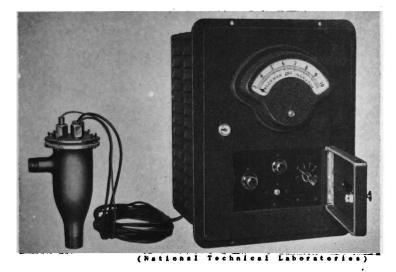
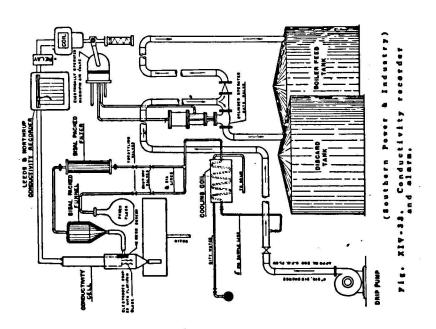
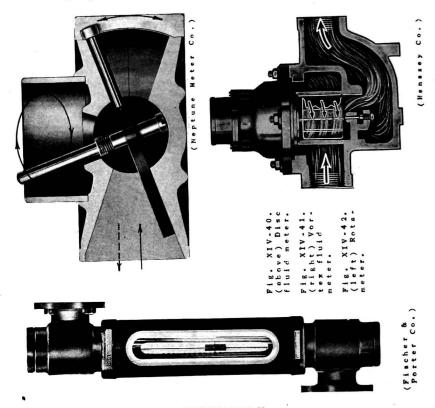


Fig. XIV-32, pH indicator and controller



Some special flow measuring applications make use of fluid meters of the disc or vortex types, as shown in Figures XIV-40 and XIV-41. Rotameters, one of which is shown in Figure XIV-42, have a float which rises inside a tapered tube in proportion to the flow.



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CHAPTER XV

THE COMPLETE POWER PLANT

108. The design of a power plant, it has been seen, involves the integration of many pieces of equipment into a whole system which must meet exacting economic requirements as well as requirements for reliability and ease of operation. The steps in the design of a power plant are essentially as follows:

- 1. A study of present and future load requirements.
- 2. A study to determine the best possible location for a new plant or for an extension to an existing plant in a system.
- 3. A study of cycle types and arrangements to reveal the most economical solution for the operating conditions.
- 4. Selection of the major equipment to fit the cycle decided upon.
- 5. Making a preliminary layout of the plant and analyzing it from the standpoint of ease of operation and economy of floor space and building volume.
- Making a final layout of the plant and writing specifications for the minor equipment.

Some of these steps may be transposed, or several may be carried on simultaneously. It is always true that subsequent discoveries may invalidate some of the conclusions drawn in earlier steps, thus requiring modifications. The final design should be the result of exhaustive analysis and detailed study from all standpoints.

After the consulting engineers have completed the design of the plant their function is limited to aiding in the analysis of bids for equipment and supervising the construction and initial operation of the plant. In some cases, especially with municipal projects, they will aid in securing the financing of the project.

109. Details of Design-The actual details of the design of a power plant are so voluminous that it is impossible to deal with even a small part of them in a text on fundamentals. Frobably the best way for a student to gain some

insight into the process of power plant design is to examine very carefully the design of actual plants, both in the field and from drawings, trying to determine the reason for each detail in the light of his knowledge of the fundamentals.

In order to show some of the multitude of design details, a portion of a set of power plant drawings is given in Figures XV-1 through XV-6.

It will be seen, from an examination of the complete power plant, that the mechanical equipment falls into several major groups, corresponding somewhat to the chapter headings in this book, as follows:

- 1. Prime movers.
- 2. Condensers, feed-water heaters and evaporators.
- 3. Steam generators and auxiliaries.
- 4. Fiping and duct-work.
- 5. Water handling equipment.
- 6. Fuel handling equipment.
- 7. Draft equipment..
- 8. Supervisory and control equipment.

Another method of grouping the equipment would be as follows:

- 1. Prime movers.
 - E. Lubricating oil system.
 - b. Generator cooling system.
 - c. Shaft gland sealing system.
- 2. Condensers.
 - a. Condensate pumps.
 - b. Circulating water pumps.
 - .c. Vacuum pumps.
- 3. Feed-water heaters and evaporators
 - a. Drip pumps.
 - b. Evaporator feed pumps.
- 4. Boiler feed system.
 - a. Boiler feed pumps.

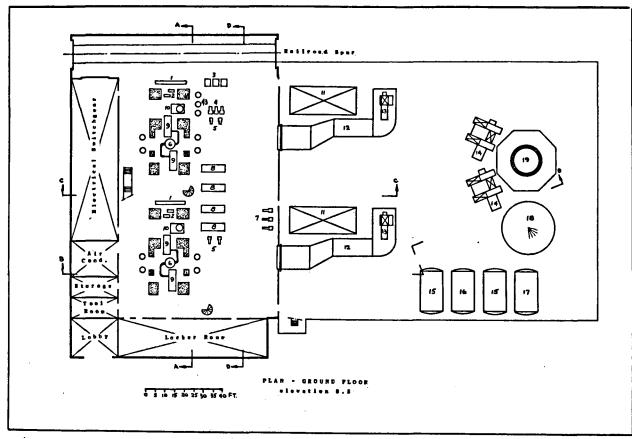


Fig. XV-1. City of Jacksonville, Florida, Power Plant No. 2.

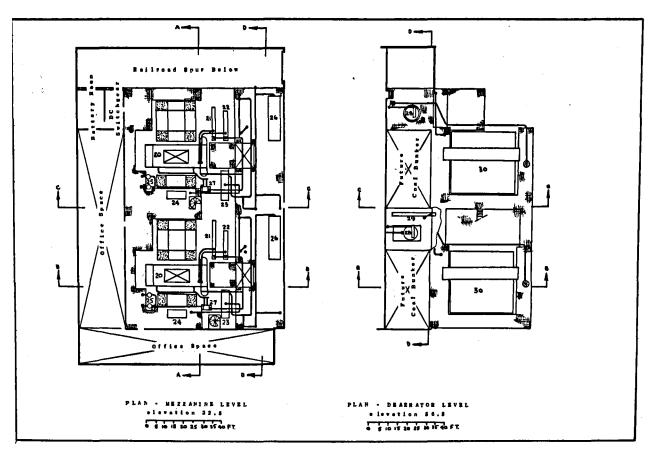


Fig. XV-2. City of Jacksonville, Florida, Power Plant No. 2.

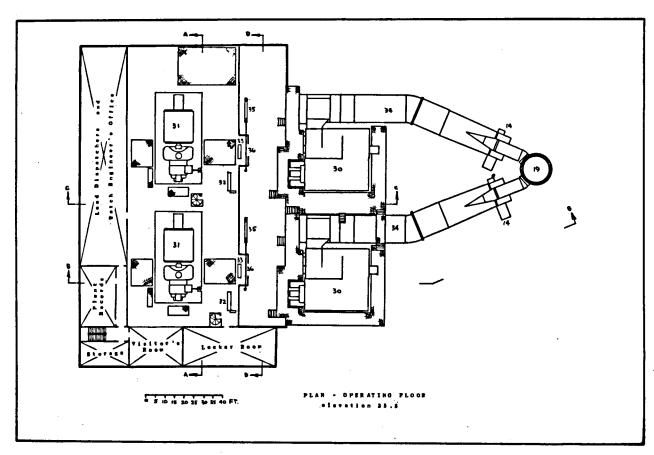
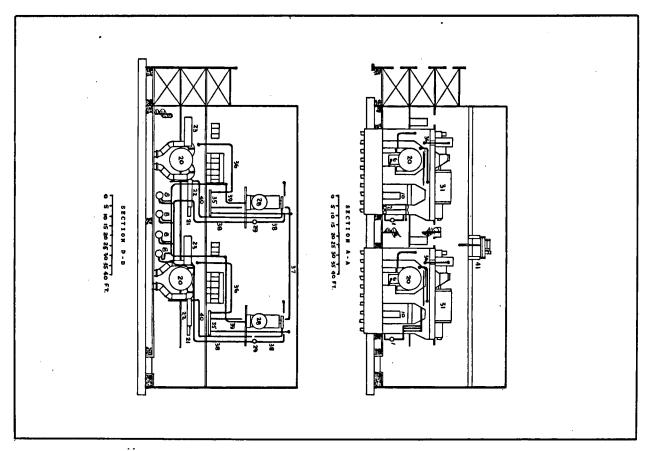


Fig. XV-8. City of Jacksonville, Florida, Power Plant No. 2.



.. Fig. XV-4. City of Jacksonville, Florida, Power Plant No. 2.

Fig. Ry-5. City of Jacksonville, Florida, Power Plant No. 2.

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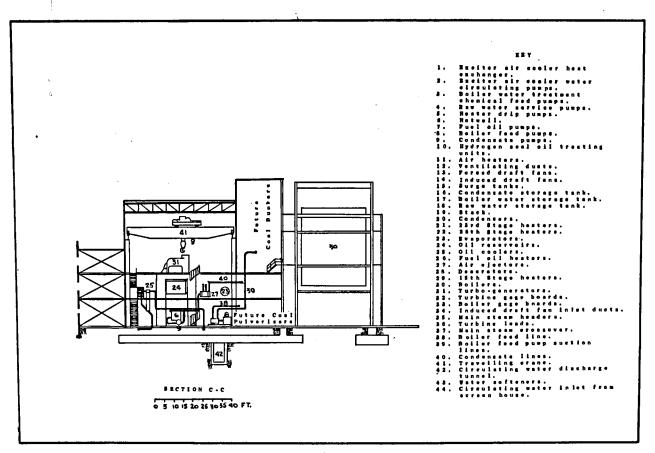


Fig. XV.6. City of Jacksonville, Florida, Power Pient No. 2.

- Eurge tanks and pumps.
- c. Feed-water regulators.
- Steam generator.
 - a. Boiler and furnace.
 - Superheater.
 - c. Economizer.
 - d. Air heater.
 - e. Fans and stack.
- Fuel system.
 - a. Fuel oil pumps or coal conveyors.
 - b. Fuel oil heaters or coal pulverizers.
 - c. Burners.
 - d. Ash disposal.
- Piping and duct-work.
 - a. Circulating water system.
 - b. Condensate system.
 - c. Boiler feed system.

 - d. Steam system.e. Service and fire protection water system.
 - f. Lubricating oil system.
 - g. Fuel oil, fuel gas or coal system.
 - h. Air ducts.
 - i. Gas ducts.
 - j. Compressed air system.
- 8. Water supply.
 - a. Condenser cooling system.
 - Boiler feed makeup.
 - c. Service and fire protection water.
- Fuel supply.
 - a. Unloading facilities.
 - b. Storage facilities.

110. Power Plant Buildings -- In past years the aim of power plant architecture was to give the impression of power and stability. This aim resulted in buildings that were much too expensive for their primary purpose of sheltering the power generating equipment. The last ten years has proven the ascendency of functional power plant architecture in the acceptance of the semi-outdoor and outdoor plant.

In the semi-outdoor plant more or less of the steam generating equipment is left exposed to the weather (with weatherproof casings) and the turbine equipment and boiler operating stations are completely sheltered. The outdoor power plant is just what the name implies, except that a

shelter is provided for the boiler and turbine operating stations.

At present, with few exceptions, in plants which are designed for complete shelter, as well as for semi-outdoor installations, the buildings are made as inexpensively as possible in order to reduce the capital outlay to the minimum. Frobably the cheapest type of power plant building construction (but not unattractive with proper treatment) makes use of corrugated asbestos-cement sheeting fastened to a structural steel frame. This construction is fire resistant and weatherproof, and has the additional advantage of being easily modified for expansion in any direction.

Figures XV-7 through XV-15 show some views of power plants of the various types of construction.

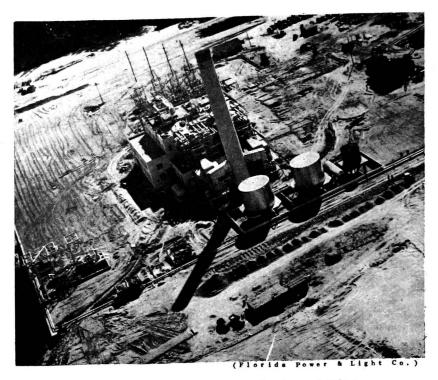
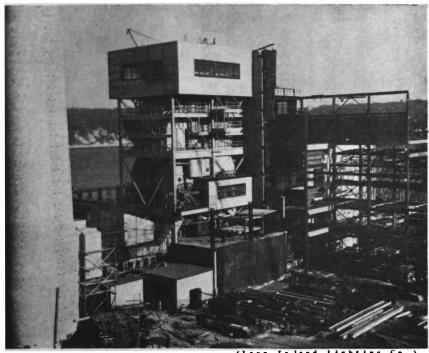
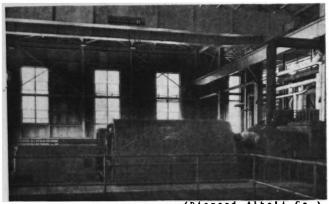


Fig. XV-7. Semi-outdoor plant, exterior.



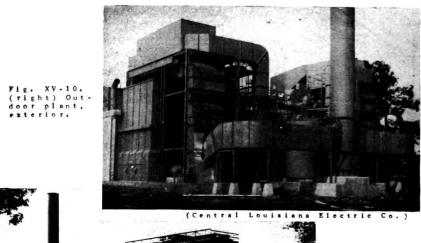
(Long Island Lighting Co.)

Fig. XV-8. Semi-outdoor plant, exterior.



(Dismond Alkali Co.)

Fig. My. 9. Semi outdoor plant, interior.



(Central Louisiana Electric Co.)

(left) Out-door plant, exterior.



Fig. XV-12. Outdoor plant, control shelter.



Fig. XV-13. Enclosed plant, exterior.

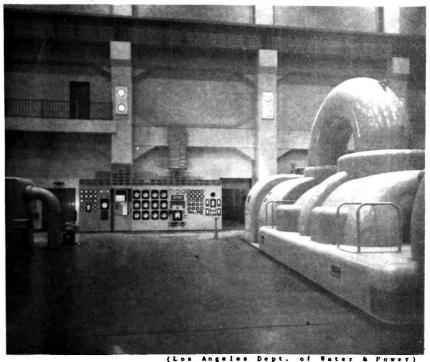
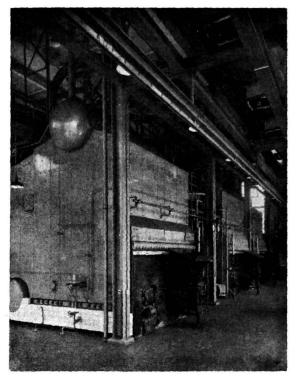


Fig. XV-14. Enclosed plant, interior.



(Fairmont Coal Bureau)
Fig. XV-15. Enclosed plant, interior.

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PROJECT IV

- 1. Specify the head, capacity and motor size for the following pumps (assume ordinary entrances and exits):
- (a) Circulating water: No static lift.
- (b) Condensate: Static lift 40 feet.
- (c) Boiler feed: Static lift 20 feet.
- 2. The analysis of the raw water for makeup is: Calcium bicarbonate, 150 ppm; magnesium bicarbonate, 85 ppm; calcium sulphate, 50 ppm; magnesium chloride, 30 ppm; sodium chloride, 60 ppm. How much 90 o/o hydrated line, 95 o/o soda ash and 95 o/o filter alum (using 8 ppm) would be required per thousand gallons of water to soften the water by the cold process, leaving a residual hardness of 50 ppm as calcium carbonate?
- 3. Estimate the draft and pressure losses in the air and flue gas circuits of the steam generator, and specify the capacity, pressure and motor size for the following fans:
- (a) Forced draft: Backward curved blades.
- (b) Induced draft: Forward curved blades.
- 4. Calculate the dimensions of the most economical brick stack to produce a draft of C.5 inches of water. Ambient temperature 90 F and barometer 29.5 inches of mercury.
 - 5. Make a preliminary plant layout drawing.

Note: The preceeding calculations apply to the power plant of Projects I, II and III.

CHAPTER XVI

ECONOMICS OF LOAD DIVISION

system the efforts of all operating and supervisory personnel are bent toward the production of power at the least possible cost. To a large extent the economy of operation is affected by the manner of division of load between units in a plant and plants in a system. It is the purpose of this chapter to discuss briefly the theory of the incremental rate method of load division and to show some of its applications.

MIL. The incremental Rate Theory-The theory of incremental rates postulates that each succeeding block of load should be thrown on the plant or machine which exhibits the lowest incremental rate (slope of the input-output curve), regardless of the absolute efficiency of the machine or plant. This theory contradicts the widely held opinion that the most efficient plant or machine should be loaded fully and the less efficient plants or machines used to supply the balance of the demand. According to the incremental rate theory, this opinion is true only in the special case in which the incremental rate of the most efficient plant or machine is always less than that of the less efficient plant or machines. The theory can be substantiated by a very simple demonstration, as follows:

Figure XVI- (a) represents the input-output curves of two machines, A and B. For simplicity these curves are made parabolic so that equations for them can be differentiated algebraically to obtain the equations of the incremental rate curves. It is not necessary to the theory, however, that the curves be capable of being expressed in the form of simple equations because any curve can be differentiated by graphical or tabular methods. The input-output curves A and B can be expressed as:

$$I_a = 17 + 0_a + 0.1 \ 0_a^2$$

$$I_b = 6 + 0_b + 0.15 \ 0_b^2$$

where: I and O = input and output in any desired units

The incremental rates for A and B are plotted in Figure XVI-1 (b), and can be expressed as:

$$\frac{dI_a}{dO_a} = R_a = 1 + 0.2 O_a$$

$$\frac{\mathrm{d}I_b}{\mathrm{d}O_b} = R_b = 1 + 0.3 O_b$$

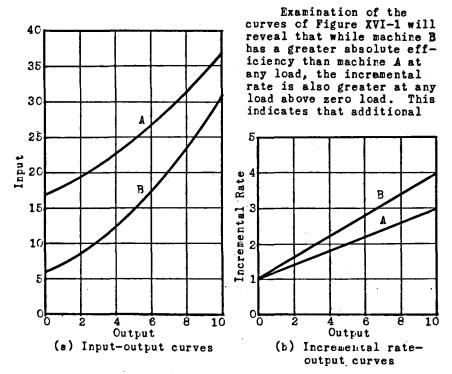


Fig. XVI-1. Machine operating characteristics.

blocks of load, according to the theory, will be handled at greater efficiency by machine A than by machine B.

113. Division of Load--A correlary to the requirement that succeeding blocks of load be supplied by plants or machines with the lowest incremental rate is that, for maximum overall efficiency, all plants or machines must be loaded to

the point at which the incremental rates are equal. This correlary can be substantiated by a simple demonstration, using the data for the machines of Figure XVI-1, as follows:

Suppose it is desired to produce a total output of 14 with machine A and machine B running together. Without regard to efficiency, it would be possible to obtain this output with an infinite number of combinations between 1C on one machine and 4 on the other. However, examination of the following table will show that one combination alone will produce an output of 14 with a minimum input. By interpolation the minimum total input is found to be at an output of

Ou 1	tput		nput	
A	. В	A	В	Total
10	4	37.0	12.4	49.4
8	5	34.1	14.8	48.9
8	6	31.4	17.4	48.8
7	7	28.9	20.4	49.3
6	8	26.6	23.6	50.2
5	9	24.5	27.1	51.6
4	10	22.6	31.0	53.6

5.8 from machine B and 8.2 from machine A. Examination of Figure XVI-1 (b) will show that the incremental rates for both machines are 2.64.

III. Loading Schedules--It is now possible to demonstrate the derivation of loading curves or schedules for maximum economy. The trend of loading of machines A and B of the previous discussion is such as to bring machine A to its maximum load first, and the load on machine B at which this occurs can be shown graphically by combination of the incremental rate curves of Figure XVI-1 (b), as shown in Figure XVI-2. Beyond this point machine B will supply the additional load to full plant capacity.

II5. Applications of the Incremental Rate Method--The term "machine" has been used rather loosely in the previous discussions to indicate that the incremental rate theory is applicable to any piece of apparatus or collection of apparatus for which input-output curves can be drawn.

Input-output curves for turbines may have shapes similar to those shown in the previous discussions, or they may be of a variety of other shapes and slopes depending upon the valve arrangement, variation of efficiency with load and

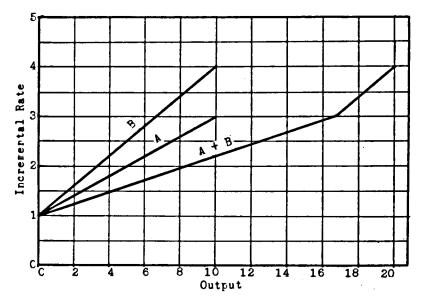


Fig. XVI-2. Plant leading curve,

the auxiliary power requirements. However, no matter what the shape, the basic theory is applicable to determination of the best load division.

The preceding remarks apply equally well to steam generators. Input-output curves can be derived from steam generator efficiency data and the incremental rate curves derived in turn from the input-cutput curves.

For the purposes of the system load dispatcher, the plant loading curves can be integrated to give plant inputcutput curves. These curves will be of use in deciding upon the system loading schedule. The plant loading curve of Figure XVI-2 was based upon the operation of two machines at all times, however, due to the better absolute efficiency of machine B, best economy would dictate the use of machine B alone up to a plant load of 1C. At this load machine A would be brought on the line and the load divided between the two machines. Figure XVI-3 shows the plant loading curve repletted on this basis. This curve can be integrated in three parts: from zero to 10; from 1C to 16.7; and from 16.7 to 20. The equations are:

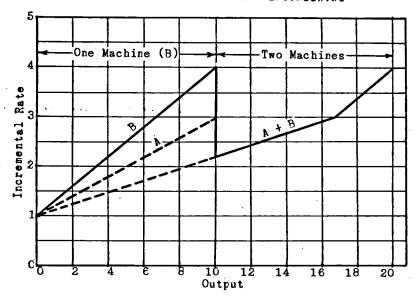


Fig. XVI-3. Plant loading curve.

$$I_{0}^{0} = {}_{0}^{20} = f (1 + 0.3 \text{ 0})d0 = 0 + 0.15 \text{ 0}^{2} + C_{1}$$

$$I_{0}^{0} = {}_{10}^{26} {}^{7} = f (1 + {}_{10}^{2} - {}_{10}^{2})d0 = 0 + 0.06 \text{ 0}^{2} + C_{2}$$

$$I_{0}^{0} = {}_{10}^{20} {}_{7} = f (-2 + 0.3 \text{ 0})d0 = -2 \text{ 0} + 0.15 \text{ 0}^{2} + C_{3}$$

When the constants are evaluated, the equations become:

$$I_{0}^{0} = {}_{0}^{10} = 6 + 0 + 0.15 0^{2}$$

$$I_{0}^{0} = {}_{10}^{16} {}_{7}^{-} = 23 + 0 + 0.06 0^{2}$$

$$I_{0}^{0} = {}_{16}^{20} {}_{7}^{-} = 48 - 2 0 + 0.15 0^{2}$$

The plant input-cutput curve is then plotted as shown in Figure XVI-4.

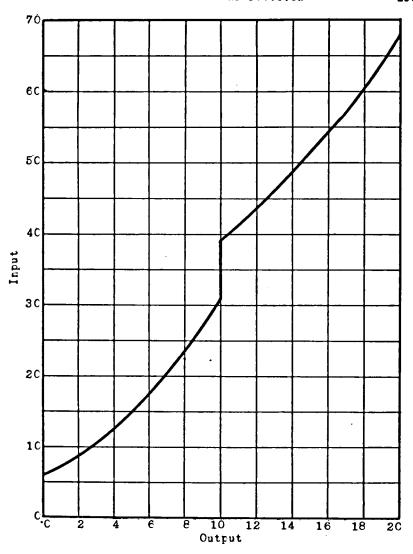


Fig. XVI-4. Plant input-output curve.

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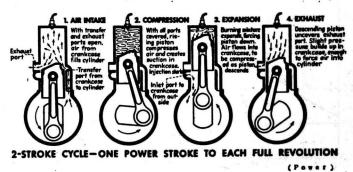
CHAPTER XVII

DIESEL-ELECTRIC POWER PLANTS

efficient machine (with the exception of the mercury-steam binary vapor cycle) available for the conversion of the energy in fuel to a useful form. It is most unfortunate that the engines are limited in size to about 7000 horsepower. As mentioned in Charter I, power plants with capacities up to about 7500 kw capacity may well contain Diesel engines, but above this capacity the steam turbine is supreme.

The Diesel engine takes its name from its inventor, Rudolph Diesel, who patented an engine in 1892 which was supposed to operate on the Carnot cycle. However, practical difficulties and development over the years has led to the modern Piesel engine which may be made to operate on the two-stroke or the four-stroke system. By far the greater number of modern Piesel engines operate on the two-stroke system because of the greater simplicity of the engine.

li7. Two-Stroke Diesel Cycle-The two-stroke Diesel cycle, as the name implies, requires two strokes of the piston to complete a cycle. Referring to Figure XVII-1, it can be seen that with the piston at the bottom of the cylinder



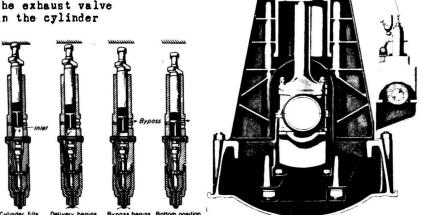
Pig. XVII-1. Two-stroke Diesel cycle.

scavenging air is blown in at one side of the cylinder and out at the other side, purging the cylinder of exhaust gases and charging it with fresh air. As the piston moves upward the scavenging and exhaust ports close and the air is compressed. When the piston reaches the top of its travel the

air temperature has risen to about 800 F, which is sufficient to cause ignition of the charge of fuel which is sprayed into the cylinder through an injector. The combustion of the fuel causes a further increase in temperature and the piston moves downward doing work. Sometimes the mechanical details are changed to employ a poppet valve in the cylinder head for the admission of scavenging air or for exhaust. Figure XVII-2 shows a fuel injector and Figures XVII-3 through XVII-5 show several two-stroke engines.

118. Four-Stroke Diesel Cycle--In this cycle, referring to Figure XVII-6, the first downward stroke of the piston is used for drawing in a charge of air through a poppet valve

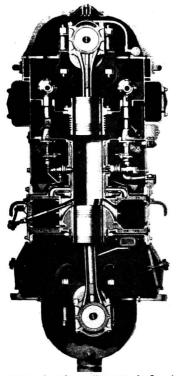
in the cylinder head. In the upward stroke the air charge is compressed. At the end of the compression stroke the fuel is sprayed into the cylinder through an injector and the piston moves downward on the power stroke. At the end of this stroke the exhaust valve in the cylinder



(Power)

Fig. XVII-2. Fuel injector.

Fig. XVII-3. Large low speed two-stroke engine.



(Fairbanks, Morse & Co.)
Fig. XVII-4. High speed opposed piston two-stroke engine.

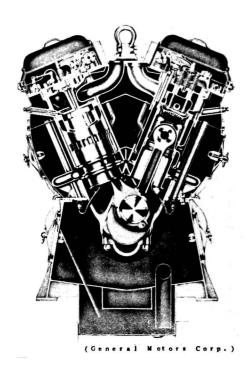
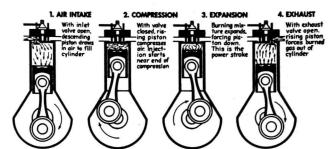


Fig. XVII-5. High speed V-type two-stroke engine.

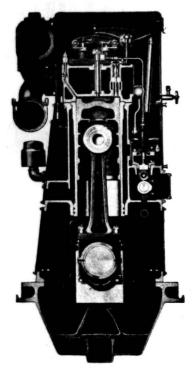


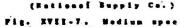
4-STROKE CYCLE—ONE POWER STROKE TO TWO FULL REVOLUTIONS

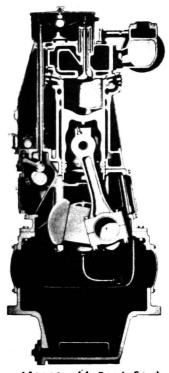
(Power.)

Fig. XVII-6. Four-stroke Diesel cycle.

head opens and on the upward stroke the exhaust gases are discharged from the cylinder. It would appear that the two-stroke engine would produce twice as much power for the same engine dimensions and speed as the four-stroke engine, but this is not true because the four-stroke engine has a much greater volumetric efficiency as an air compressor. Actually, the power produced per cylinder is almost the same, but the two-stroke cycle engine is preferred because of its greater simplicity. Figures XVII-7 through XVII-9 show several four-stroke cycle engines.







(Ingereall-Rand Co.)

Fig. XVII-8. Medium speed

four-stroke engine.

119. The ideal Diesel Cycle--The ideal Piesel cycle, shown in Figure XVII-1C, consists of a constant pressure addition of heat from A to B; an adiabatic expansion from B to C; a constant volume rejection of heat from C to D; and an adiabatic compression from Γ to A. The efficiency of the

ideal cycle can be expressed as follows:

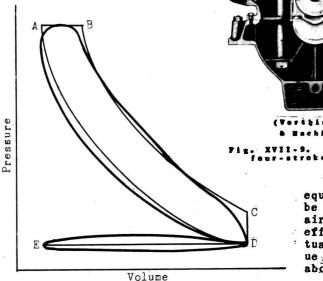
$$e_{t_c} = 1 - \frac{1}{r^n - 1} \frac{R^n - 1}{(n(R - 1))}$$

where: r = compression ratio,

 ∇_d/∇_a (15 to 18) R = fuel cutoff ratio,

n = polytropic exponent

If n is given a value of 1.4, which would represent the adiabatic value for air, the efficiency obtained from the



Ideal Diesel cycle. Fig. XVII-10.

equation would . be called the air standard. efficiency. Acttually, the value of n averages about 1.35.

The twostroke engine will follow the

shape of the air standard quite closely except for the diference in the value of n. The four-stroke engine, however, . will exhibit a negative work loop DED due to losses in discharging the products of combustion and drawing in a fresh

air charge. The departure of n from the adiabatic value is due to the necessity of providing the cylinders with water jackets to prevent overheating.

- 120. Supercharging.—The power output of Diesel engires can be increased by providing means for forcing more air into the cylinders, which in turn makes it possible to burn more fuel. The process involves increasing the initial air pressure, and the term applied is supercharging. Several systems of supercharging are used, as follows:
 - 1. Positive displacement blowers or compressors.
 - a. Direct connected.
 - b. Motor driven.
 - 2. Centrifugal or axial compressors.
 - a. Exhaust gas turbine driven.
 - b. Direct connected.
 - c. Motor driven.

The advantage of the supercharger lies in its ability to cause an increase in power of the engine which is greater than the power required to drive the supercharger. As a matter of fact, supercharging is necessary for any Diesel engine when it is operated at considerable distances above sea level because of the decrease in atmospheric pressure which results in lower peak cylinder pressures. The decrease in power of an unsupercharged engine with increasing altitude is such as to cause the engine to have only 70 per cent of its sea level power at 8000 feet and 50 per cent at 14000 feet. Figures XVII-11 through XVII-13 show several types of supercharger and scavenging blower.

121. Diesel Engine Efficiencies--The thermal efficiency of most Diesel engines will be about 30 per cent, and the heat balance will be about as follows:

Shaft output 30 o/o
Cooling water 30 o/o
Exhaust gas 30 o/o
Friction, radiation
compressor 10 o/o

A not inconsiderable item in the operation of a Diesel engine is the lubricating oil economy. Lubri-

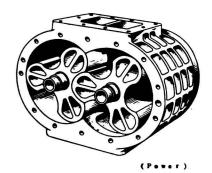
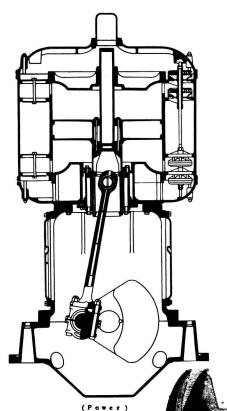


Fig. XVII-11. Positive



cating oil will be consumed because at each combustion the lubricating oil film burns from the cylinder wall. Figure XVII-14 shows some data on fuel and lubricating oil economy for Piesel-electric plants.

Power output is directly proportional to speed, so that high speed engines will be more economical of space for a given output than will low speed engines. However, high speed engines (600 to 1200 rpm) have shorter lives and require higher grade fuel oils (number 2 or 3), so that a careful economic analysis with respect to expected life and cost must be made before deciding upon the particular type of engine for any given installation.

Fig. XVII-12. Reciprocating scavenging pump.

I22. Diesel
Engine Auxiliaries--The auxiliaries for a Piesel
engine installation may be placed
in several classifications, as
follows:

Fig. XVII-13. Exhaust gas turbine driven centrifugal blower.

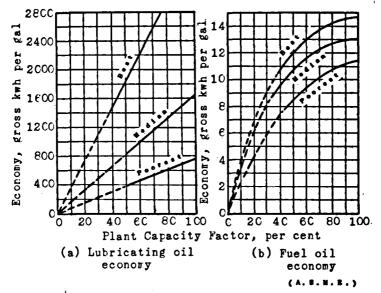


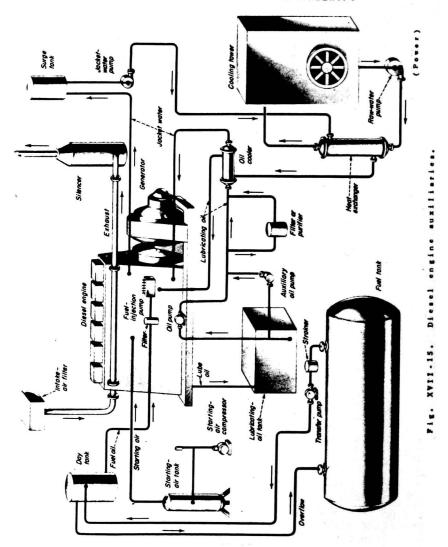
Fig. XVII-14. Diesel engine fuel and lubricating oil economy.

- 1. Fuel oil system.
- 2. Lubricating oil system.
- 3. Cooling water system.
- 4. Air intake system.
- 5. Exhaust system.
- Starting air system.

Figure XVII-15 shows the arrangement of the various systems in relation to an engine. The systems can be designed in accordance with the fundamentals of fluid flow presented elsewhere in this book, but especial attention will have to be given to the air intake and exhaust systems in order to avoid conditions which would lead to resonance in the ducts.

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CHAPTER XVIII

HYDRO-ELECTRIC POWER PLANTS

- 123. Hydro-electric power plants can be classified in two ways, as follows:
 - 1. Water impoundment volume. .
 - a. Storage plants.
 - b. Run-of-river plants.
 - 2. Head on turbines.
 - a. High head (over 500 feet).
 - b. Medium head (50 to 500 feet).
 - c. Low head (under 50 feet).

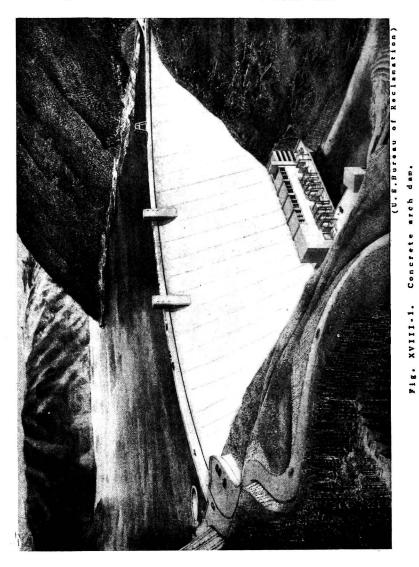
The impoundment volume will affect the amount of power available at any particular time, that is, the storage plant will be able to supply a swinging load with little or no water loss, while the run-of-river plant operates best on base load status (the base load, of course, varying with seasonal stream flow). The head available will affect the choice of type of prime mover. Bigh head plants characteristically use small water volumes and high rotational speeds with impulse types of turbines, while medium and low head plants use larger water volumes and lower rotational speeds with reaction type machines. Any of the types are available in sizes from 75C to 75,000 kilowatts capacity.

124. Impoundment Methods--Any hydro-electric plant will require a dam which may be high or low depending upon the topography of the dam site and the surrounding water-shed. Each case is unique and requires its own solution.

High daws (over 100 feet) may be of the arch type or of the gravity type. The arch dam depends upon the shape of the dam to transfer the water load to the enclosing rock walls which serve as abutments. Hoover Lam on the Colorado River is of this type. The gravity dam depends upon the mass of the dam itself to resist the water load and the enclosing rock walls merely serve to confire the water laterally. Grand Coulee Lam on the Columbia River is of the gravity type. The TVA system contains both types of dams.

Les dass (under 100 feet) are usually of the gravity type and may be of masonry construction, as with high dams, or they may be of earth fill with cores of rock, clay or concrete.

Figure XVIII-1 shows a concrete arch dam.



12b. Water Storage--The run-of-river plant requires little or no impoundment volume and the essential information for estimating the power capabilities of the site can

be obtaired from a river flow duration curve. The flow duration curve can be constructed from a hydrograph, which is similar to an electrical load curve. Figure XVIII-2 shows a river hydrograph, and Figure XVIII-3 shows the flow duration curve derived from it. The minimum power available and the yearly power cutput can be estimated from these curves.

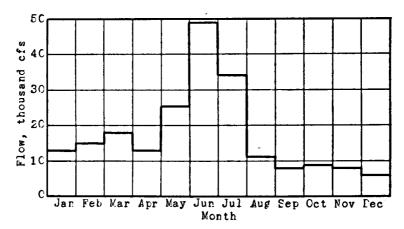


Fig. XVIII-2. River hydrograph.

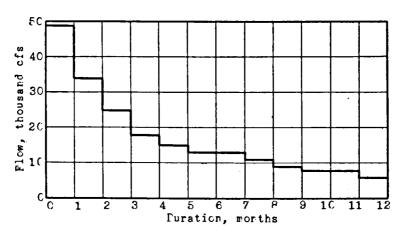


Fig. XVIII-3. River flow duration curve.

The storage plant makes use of a river mass flow curve in which curulative water-shed runoff is plotted versus time

during the year. Such a curve, derived from Figure XVIII-2, is shown in Figure XVIII-4. Using this curve it is possible to estimate the maximum possible sustained flow to base load turbines by calculating the slope of the line drawn tangent to two similar yearly peaks (A and B in Figure XVIII-4). Whenever the slope of the mass flow curve is less than that of the tangent, the reservoir level will be decreasing, and vice versa. The total impoundment volume should be sufficient to maintain a minimum head on the turbines when the reservoir level is at its lowest, which would be at point C.

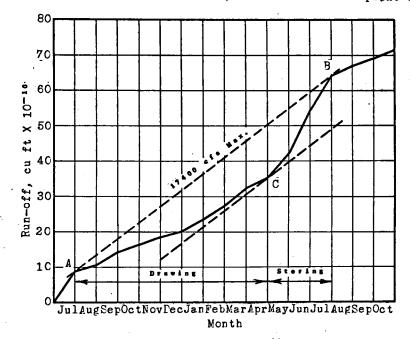


Fig. XVIII-4. River mass flow curve.

126. Hydraulic Turbines--The hydraulic impulse turbine is similar to the steam impulse turbine in that all of the pressure drop takes place in the nozzles. If it is assumed that the initial velocity is zero, the jet velocity is (assuming a perfect nozzle):

$$V = \sqrt{2gH} = \sqrt{2g(P_1 - P_2)v}$$

where: V = jet velocity, ft/sec H = head at jet inlet, ft of water F₁ and F₂ = jet inlet and outlet
 pressures, lbs/sq ft
v = specific volume of water, cu ft/lb

The power in the jet is:

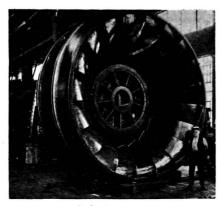
whp =
$$\frac{M \ H}{550} = \frac{M(P_1 - P_2)v}{550}$$

where: M = water flow, lbs/sec

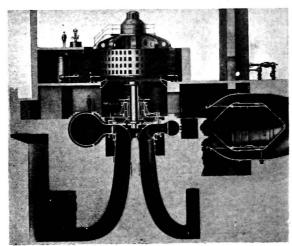
but due to the fact that it is not possible to slow the jet down to zero velocity, and due to mechanical friction, the shaft horsepower will only amount to about 85 to 90 per cent of the jet horsepower. The impulse turbine is for very high head developments (over 900 feet) and operates at speeds between 300 and 400 rpm. Figure XVIII-7 shows an impulse turbine runner of the Felton type with its needle nozzles. Two of the runners are overhung on one generator and the nozzles are so arranged that one per runner is the power nozzle and identical nozzles are used for surge relief. That is, as the power nozzle closes on a load decrease the relief nozzle opens an equal amount and then automatically closes slowly to prevent surges in the penstock (which is the name given to the pipe which delivers water to the nozzles from the reservoir. As an extra safety measure, the penstocks are equipped with surge tanks near the turbines.

For heads between 65 and 900 feet the Francis type of reaction turbine is used. The wheel, or runner, has curved blades which receive water through control gates from a volute chamber and discharge the water vertically downward. The action is very similar to that in a reaction steam turbine, some pressure drop taking place in the control gates giving some impulse effect, while the remainder of the pressure drop takes place in the runner. The water horsepower is the same as the jet horsepower for the impulse turbine, but the efficiency is higher, usually between 90 and 95 per cent. Figure XVIII-5 shows a Francis type turbine with an umbrella generator. These machines operate at speeds between 100 and 300 rpm.

For heads less than 65 feet the axial flow propeller type of reaction turbine is used. The runner may be of the Nagler type with fixed pitch blades or of the Kaplan type with adjustable pitch blades. The Kaplan runner is very well suited to run-cf-river plants because the power output and efficiency can be maintained under reduced head condi-

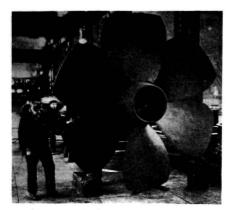


(a) Runner

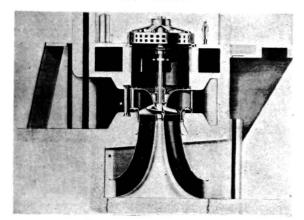


(b) General arrangement

Pig. XVIII-5. Francis turbine.

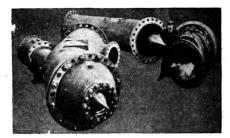


(a) Runner

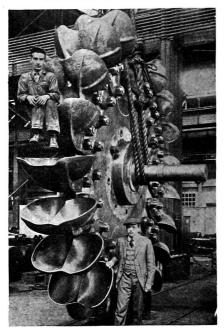


(b) General Arrangement

Fig. XVIII-6. Nagler turbine.



(a) Needle nozzles



(b) Runner
(Allis-Chalmers Mfg. Co.)

Fig. XVIII-7. Pelton turbine.

tions. Water is supplied to the propeller through control gates in a manner similar to that used with the Francis runner. Efficiencies vary between 85 and 90 per cent and speeds between 60 and 100 rpm. Figure XVIII-6 shows a typical Nagler turbine connected to an umbrella type generator.

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CHAPTER XIX

GAS TURBINE POWER PLANTS

427. In the last war the development of the gas turbine was given great impetus and at the present time several manufacturers are offering 5000 kilowatt units for use with heavy fuel oil or natural gas. Research is being conducted on pulverized coal burning units but difficulty is being experienced with flyash erosion in the turbines.

128. Ges Turbine Cycles--The simplest gas turbine cycle corresponds to the open Rankire vapor cycle, as shown in Figure XIX-1. Referring to the figure, air at atmospheric pressure and temperature is drawn in by the compressor, compressed and led to the combustion chamber where fuel is burned at constant pressure. The resulting high temperature combustion gases then expand to atmospheric pressure through a turbine and the unavailable energy is rejected to the at-

mosphere. Part of the turbine work is used to drive the compressor and the remainder is available for doing useful work. Due to the fact that present turbine blading materials will withstand temperatures of only about 1200 F for long periods. secondary air in sufficient quantity to cool the' combustion gases

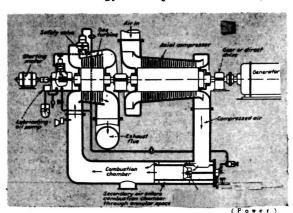


Fig. XIX-1. Simple open gas turbine cycle.

to a safe temperature must be supplied. The total air supplied amounts to between 100 and 150 pounds per pound of fuel oil. This large amount of excess air results in lowered cycle efficiency due to the increase in compressor power, but overall thermal efficiencies of 18 to 20 per cent are obtainable. An enthalpy-entropy diagram for the simple open cycle is shown in Figure XIX-2.

The simple open cycle may have a regenerator added to improve efficiency where the air temperature leaving the

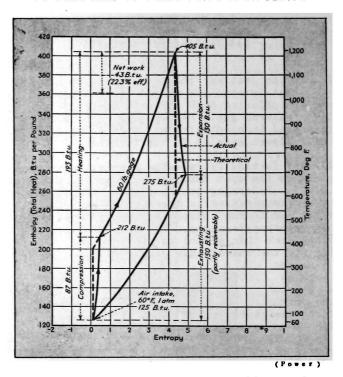


Fig. XIX-2. Simple open gas turbine cycle.

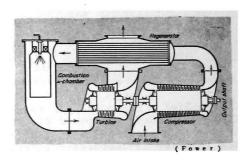
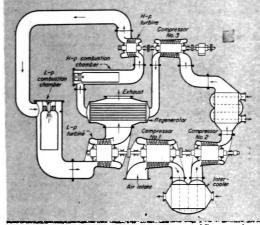


Fig. XIX-3. Open gas turbine cycle with regenerator.

compressor is less than the turbine exhaust temperature. Such an arrangement is shown in Figure XIX-3.

Other combinations with regeneration, multiple stage compression with intercooling and reheating will have higher efficiencies. Two of the various types of these more complex cycles are shown in Figures XIX-4 and XIX-5.

A closed gas turbine cycle has been proposed which makes use of a heat exchanger in the combustion chamber to make it possible to separate the combustion gases from the working fluid which can be made to operate at higher pressures



(Sechanical Engineering)
Pig. ZIX-4. Open gas turbine
cycle with rebent

where the density is greater, thus allowing smaller machines for the same output. Figure XIX-6 shows a diagram of the cycle. It will be noted that the

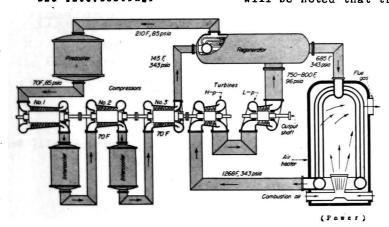


Fig. Il X-6. Cloud gas turbine cycle

cycle contains a precooler in which the mravailable energy is rejected. The efficiency of the closed cycle is slightly less than that of an equivalent open cycle, but the smaller size for the same output may overbalance the smaller efficiency.

129. Gas Turbine Plants--Up to the present time the principal application of the gas turbine in industry has been in the Houdry process of gasoline production where large quantities of compressed air are required. In these machines the excess turbine power is used to drive an oversized compressor and the extra air which is not needed for combustion goes to the process.

In the last few years several commercial gas turbo-generator installations with capacities up to 5000 kilowatts have been made. One of them is shown in Figure XIX-7. This particular machine is similar, except for the generator, to the units developed for use in locomotives.

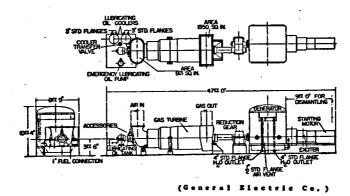


Fig. XIX-7. Locomotive type gas turbo-generator.

BIBLIOGRAPHY

Sawyer, R. T., The Modern Gos Turbine, Prentice-Hall, New York, 1945.

300-Pound Malleable Iron Fittings

WORKING PRESSURES

300 pounds steam or oil, 550° F.

14 to 1-inch — 2000 pounds cold water, oil, or gas, non-shock 114 to 2-inch — 1500 pounds cold water, oil, or gas, non-shock 21/2 and 3-inch — 1000 pounds cold water, oil, or gas, non-shock

Dimensions, in Inches

307

N Height

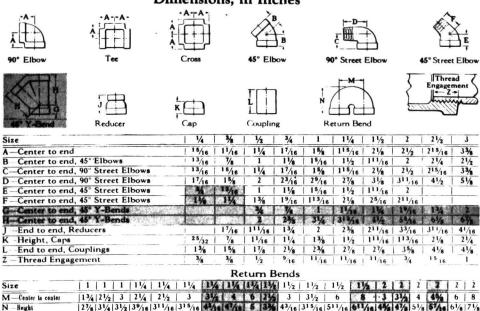


TABLE XVII continued

Standard Cast Iron Fittings

WORKING PRESSURES

125 pounds steam
175 pounds cold water, oil, or gas, non-shock

Dimensions, in Inches







Tee









90° Elbow

45° Elbow

Cross



45° Y-Bend



Return Bend



Size	A	В	C	D	E	F	G	J	K	Н	Z		-		Retu	rn Be	ends			
1/4	13/16	3/4		T			Ī	1			3/8	Size	M	N	Size	M	N	Size	M	1
3/8	15/16	13/16		1			İ				3/8	Clo	se Pat	tern	Ope	n Pat	ttern	Wide	e Pa	tter
1/2	11/8	7/8		13/10	3/4	21/4	11/0				1/2	1/2	11/4	123/32	3/4	17/8	27/32	1	3	1 3
3/4	15/16	1	13/4	15/10	1//	3	1 21/4	11/2			9/16	3/4	11/2	21/32	1	21/2	211/16	1	4	31
1	11/2	11/8	137	1	1	31/2	1.23/4	1*17/8			11/16	1	13/4	23/8	11/4	3	39/32	1	5	
11/4	13/4	15/16	173	11/16	10.4	41/4	31/4	21/8	21/8		11/16	11/4	21/4	229/32	11/2	31/2	33/4		6	
11/2	115/16	17/16		15/16	11/4	47/8	313/16	21/4	21/4		11/16	11/2	21/2	31/4	2	41/2	419/32	1		183
2	21/4	111/16	100	13/4	1//	53/4	41/2	27/16	27/16		3/4	2	31/4	331/32	1 21/2	51/2	57/16	11/4	4.	33
21/2	211/16	115/16	23/1	113/1	10/2	63/4	53/16	25/8	211/16		15/16	23/2	33/4	49/10	3	61/2	65/16	11/4	6	43
3	31/8	23/16	2.17	113/4	13/4	77/8	61/8	27/8	215/16		1	3	41/4	53/4		71/2	79/10	13/4		W.
31/2	137/1E	23/8		1		87/4	67/6	344	31/8		11/16	4	- 6	63.3/4				134	CVA	10
4	33/4	15/8	3	236	24/6	93/4	75/8	33/8	33/8		11/8		Ī	T	1	T		11/2	6	5
5	41/2	31/16	KW.	2.0	23//	11%	9 /4	37/8	37/8	23/8	11/4				i			11/4	8	
6	51/8	37/16	63%	27/8	21/4	137/16	103/4	43/8	43/8	25/8	15/16			Ī	I			2	6/4	MG.
8	69/16	1 41/4				1615/4	13%	51/4	51/4		17/16			1	1	1		2	6	55

Malleable Iron Unions and Union Fittings

WORKING PRESSURES

150 pounds steam

300 pounds cold water, oil, gas or gasoline, non-shock













Union Female

Union

nele Union on Run Male Union on Run

male	Male and I chale	Female Union		JINON	
		Dimens	ions,	in Inc	he

				ринены	O119, 111 I	IICIIES				
Size	1 1/4	%	1/2	3/4	1	11/4	11/2	2	21/2	3
A	1%	11/2	111/16	17/6	21/4	25/16	23/6	2%	31/4	30/10
В	23/16	2 1/8	25%	3	35/16	3%	318/16	41/4	41 5/16	5 1/2
C	13/16	15/16	11/8	15/16	11/2	13/4	115/16	21/4		
D	111/16	17/4	21/8	27/16	21/8	32/16	33%	31/8		
B	27/16	27/8	3	3%	315/10	47/14	47/6	51/2		

300 pounds steam, 550° F. 600 pounds cold water, oil, gas, or gasoline, non-shock















	- 6
-	90° Elbow
_	
	emale Union

90° Elbow Male Union

45° Elbow

45° Elbow Female Union, Male Union Female Union on Run Dimensions, in Inches

Male Union on Run

Tee Female Union on Outlet

Tee Male Union on Outlet

					DHILL	119101191	*** ****	, i ica					
Size	1/4	3/6	1/2	3/4	1	11/4	11/2	2	21/2	3	1 x %	1 x 1/2	1 x 3/4
A	16/16	11/16	11/4	17/16	1%	116/16	21/8	21/2	215/16	33/6	15/16	17/16	11/2
В	16/16	11/16	11/4	17/16	15%	115/16	21/8	21/2	216/16	3%	17/16	11/2	19/16
C	12/16	1/6	1	11/6	15/16	11/2	111/10	2	21/4	[
<u>b</u>	2	21/4	27/16	21/6	33/16	311/16	31 5/16	41/2	57/16	516/16		21/8	3
E	29/16	21/8	33/16	33/4	41/16	4%	41/6	53/6	67/14	73/16	311/18	313/10	316/10
<u> </u>	1 5/6	13/4	115/16	23/16	21/2	27/8		311/16					
G	23/18	21/16	211/16	31/16	37/16	313/16	41/6	49/16	53/16				

125-Pound Cast Iron Flanged Fittings

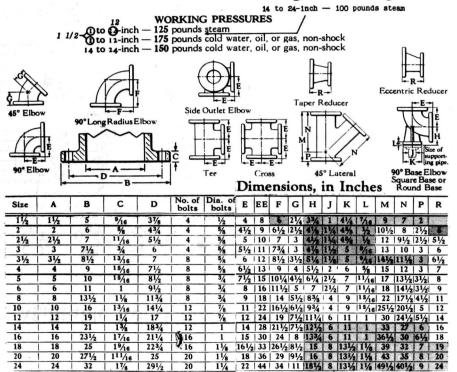


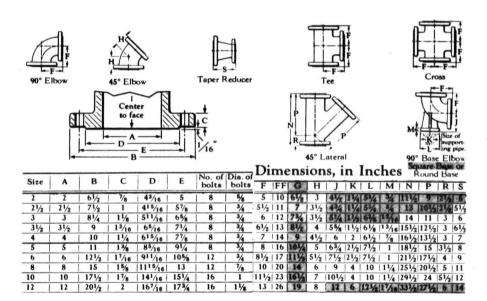
TABLE XVII continued

250-Pound Cast Iron Flanged Fittings

WORKING PRESSURES

250 pounds steam

400 pounds cold water, oil, or gas, non-shock



Cast Iron Flanges 125-Pound

250-Pound

WORKING PRESSURES

250 pounds steam 400 pounds cold water, oil, or gas, non-shock

WORKING PRESSURES

1 to 12-inch — 125 pounds steam 14 to 24-inch — 100 pounds steam

to 12-inch — 175 pounds cold water, oil, or gas, non-shock to 24-inch — 150 pounds cold water, oil, or gas, non-shock



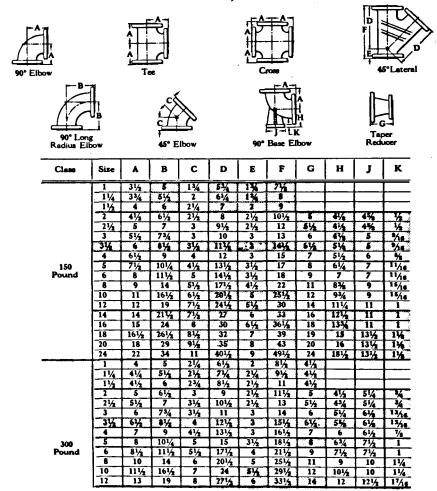
+				-
D				10.
• 4	1/1/1			140
		—в—	F .	1 1.

Size	Dimensions, in Inches									
of pipe	A	В	С	D	No. of Bolts	Dia. of Bolts				
Inches										
1	41/4	7/16	11/16	31/8	4	1/2				
11/4	45/8	1/2	13/16	31/2	4	1/2				
11/2	5	9/16	7/8	37/8	4	1/2				
2	6	5/8	1	43/4	4	5/8				
21/2	7	11/16	13/16	51/2	4	5/8				
3	71/2	3/4	11/4	6	4	5/8				
31/2	81/2	13/16	11/4	7	8	5/8				
4	9	15/16	15/16	71/2	8	5/8				
5	10	15/16	17/16	81/2	8	3/4				
6	11	1	19/16	91/2	8	3/4				
8	131/2	11/8	13/4	113/4	8	3/4				
10	16	13/16	115/16	141/4	12	1 7/8				
12	19	11/4	23/16	17	12	1 7/8				
14 OD	21	13/8	21/4	183/4	12	1				
16 OD	231/2	17/16	21/2	211/4	16	1				
18 OD	25	19/16	211/16	223/4	16	11/8				
20 OD	271/2	111/16	27/8	25	20	11/8				
24 OD	32	17/8	31/4	291/2	20	11/4				

		Dime	nsions, in	Inches		
A	В	ВС		E	No. of Bolts	Dia. of Bolts
41/6	211/14	11/10	7/6	31/4	4	%
51/4	31/16	3/4	1	31/8	4	%
61/8	39/16	13/16	11/8	41/2	1 -4	3/4
61/2	43/16	7/8	11/4	5	8	5/8
71/2	415/16	1	17/16	57/8	8	3/4
81/4	511/16	11/8	19/16	65/8	8	3/4
9	65/16	13/16	15/8	71/4	8	3/4
10	615/16	11/4	13/4	77/8	8	3/4
11	85/16	13/8	17/8	91/4	8	3/4
121/2	911/16	17/16	2	105/8	12	3/4
15	1115/16	15/8	23/16	13	12	7/8
171/2	141/16	17/8	23/8	151/4	16	1
201/2	167/16	2	29/16	173/4	16	11/8
23	1815/14	23/4	211/10	201/4	28	156
251/2	211/16	21/4	27/8	221/2	20	11/4
28	235/10	23/6	31/6	24%	24	11/4
301/2	25%	21/2	35/10	27	24	14
36	305/44	23/4	311/44	-,32	24	11/4

TABLE XVII continued

Cast Steel Flanged Fittings Dimensions, in Inches



Cast Steel Flanged Fittings

Dimensions, in Inches (Cont.)

	Size	A	С	D	E	F	G
-	1/2	31/4	2	53/4	13/4	71/2	5
	3/4	33/4	21/2	63/4	2	8%	5
	1	41/4	21/2	71/4	23/4	91/2	5
	11/4	41/2	23/4	8	21/2	101/4	5
	11/2	43/4	3	9	23/4	113/4	5
	2	53/4	41/4	101/4	31/2	133/4	6
	21/2	61/2	41/2	111/2	31/2	15	6%
	3	7	5	123/4	4	163/4	71/4
	31/2	71/2	51/2	14	41/2	183/4	73/4
600	4	81/2	6	161/2	41/2	21	83/4
Pound	5	10	7	191/2	6	251/2	101/
	6	11	71/2	21	61/2	271/2	111/
	8	13	81/2	241/2	7	311/2	131/
	10	151/2	91/2	291/2	8	371/2	154
	12	161/2	10	311/2	81/2	40	163/
	3	71/2	51/4	141/2	41/2	19	72%
	31/2	81/2	6	161/2	41/4	21	834
900	4	9	61/4	171/2	51/2	23	91/4
Pound	5	11	71/2	21	61/2	271/2	
	6	12	8	221/2	61/2	29	121/
	8	141/2	9	271/6	71/2	35	145/

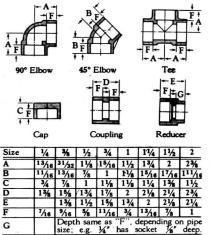
FORGER STEEL SOCKET WELFING FITTINGS

WORKING PRESSURES

Class	Size	A	С	D	E	F	G
	11/2	6	41/4	11	31/2	141/4	63/4
	2	71/4	43/4	131/4	4	171/4	
	21/2	81/4	51/4	151/4	41/2	1934	81/4
	3	91/4	54	171/4	5	223/4	91/4
	31/2	94	61/4	181/4	51/4	233/4	94
1500	4	103/4	71/4	191/4	6	251/4	
Pound	5	131/4	8%	231/4	71/2	303/4	139/
	6	137/8	9%	241/6	81/8	33	141/
	8	163/8	10%	297/8	91/4	39	17
	1/2	53/16			7 4 4		
	3/4	53/8	1	1,2	1 - 4		
	1	61/16	4				
	11/4	61/8	41/4				
	11/2	79/16	43/4				
2500	2	87/8	53/4	151/4	51/4	201/2	91/2
Pound	21/2	10	61/4	171/4	53/4	23	101/2
	3	113/8	71/4	193/4	63/4	261/2	113/
	4	131/4	81/2	23	73/4	303/4	131/
	5	155/8	10	271/4	91/4	361/2	153/
	6	18	111/2	311/4	101/2	413/4	18
	8	201/8	123/4	351/4	113/4	47	201/2
	10	25	16	431/4	143/4	58	251/2
	12	28	173/4	491/4	161/4	651/2	29

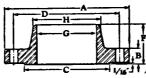
DIMENSIONS, IN INCHES
(For Standard and Extra Strong Fittings)

Pounds per Square Inch - Non-Shock Temp. No. 4 Carbon-Crane Molybdenum Steel Fittings Carbon Steel Fittings (Made to order) Deg. F. Stand- Extra Stand- Extra Strong ard ard Strong 1900 1800 2700 2550 1800 200 2250 2100 2250 2100 1400 1300 1400 1300 450 500 1650 1575 1500 1560 1440 1320 650 700 750 880 800 850 800 850 900 670 600 530 1200 960 810 900 540 1000

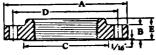


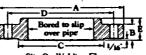
Forged Steel Flanges

Dimensions, in Inches



Welding Neck Flange .150 and 300-Pound





Screwed Flange 150 and 300-Pound

Slip-On Welding Flange 150 and 300-Pound

150 and 300-Pound						130 and 300-Pound					
Class	Pipe Size	A	В	С	D	Bo No.	lts Dia.	B	F	G	н
	1	41/4	1/10	2	31/2	4	1/2	3/4	23/10	1.05	1.32
	11/4	4%	1/4	21/2	31/2	4	1/2	13/18	21/4	1.38	1.66
	11/2	5	11/16	21/8	31/2	4	1/2	1/2	27/16	1.61	1.90
	2	6	1/4	35%	43/4	4	%	1	21/2	2.07	2.38
	21/2	7	1/2	41/6	51/2	4	%	11/6	23/4	2.47	2.88
	3	71/2	15/16	5	6	4	%	13/16	23/4	3.07	3.50
	31/1	81/4	//-	5/4	*			114	213/4	3.55	4.00
150	4	9	15/16	63/16	71/2	8	1%	15/16	3	4.03	4.50
Pound	5	10	15/16	75/1e	81/2	æ	3/4	17/16	31/2	5.05	5.56
	6	11	1	81/2	91/2	8	3/4	19/16	31/2	6.07	6.63
	8	131/2	11/6	10%	113/4	8	3/4	13/4	4	7.98	8.63
	10	16	13/16	123/4	141/4	12	1/8	115/16		10.02	10.75
	12	19	11/4	15	17	12	1/6	23/16	41/2	12.00	12.75
	14 OD	21	1%	161/4	183/4	12	1	21/4	5		14.00
	16 OD	231/2	17/16	181/2	211/4	16	1	21/2	_ 5	Bore	16.00
	18 OD	25	1%6	21	223/4	16	11/8	211/16	51/2	85 Of-	18.00
	20 OD	271/2	111/16	23	25	20	11/6	21/2	511/16	dered	20.00
	24 OD	32	11/4	271/4	291/2	20	11/4	31/4	6		24.00
	1	4%	11/16	2	31/2	4	%	11/16	27/10	1.05	1.32
	11/4	51/4	34	21/2	31/8	4	1%	11/18	29/16	1.38	1.66
	11/2	61/8	13/16	21/8	4 1/2	4	3/4	13/10	211/16	1.61	1.90
	2	61/2	1/4	31/4	5	8	*	15/16	23/4	2.07	2.38
	21/2	71/2	1	41/6	51/8	8	3/4	11/2	3	2.47	2.88
	3	81/4	11/6	5	6%	8	1/4	111/16	31/6	3.07	3.5u
	31/2	•	13/4	51/2	7%		1 %	TX.	34/4	3.55	
300	4	10	11/4	63/16	71/8	8	1.34	11/8	37/8		4.50
Pound	5	11	1%	75/16	91/4	8	X	2	31/8	5.05	
		121/2	17/16	81/2	10%	12	3/4	21/16	31/8	6.07	6.63
		15	15%	10%	13	12	1/6	27/16	4%	7.98	8.63
	10	171/2	11%	123/4	151/4	16	1	2%	4%		10.75
	12	201/2] 2	15	17%	16	11%	21/6	51/2	12.00	12.75

Forged Steel Flanges

Dimensions, in Inches (Cont.)

Class	Pipe Size	A	В	c	D		lts Dia.	E	F	G	н	C+1/6
	1 1/2	33/4	%16	1%	25%	4	1/2	7/6	25/16	21/16	0.84	
	3/4	4%	5%	111/16	31/4	4	1 %	1	21/2	21/4	1.05	111677 - 6771 8
	1	4 1/8	11/16	2	31/2	4	. %		211/16	27/16	1.32	
	11/4	51/4	13/16	21/2	31/8	4	-5%	11/6	21/8	2%	1.66	
	11/2	61/8	7/8	21/8	41/2	4	3/4	11/4	_ 3	23/4	1.90	⋠
	2	61/2	1	35%	5	8	%	17/16	31/6	21/8	2.38	C+7/6"
	21/2	71/2	11/8	41/8	57/8	8	3/4	1%	3%	31/8	2.88	TIIIIIII EZ
	3	81/4	11/4	5	65%	8	34	113/16	31/2	31/4	3.50	
	31/4	9	176	51/2	71/4	8	1/4	115/16	3%	3 1/8	4.00	8
600 Pound	5	103/4	13/2	63/16 75/16	81/2	8	7/8	21/4	41/4	41/	4.50	1 2 3
	6	1 14	1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	81/2	101/2	8	1	2 % 1 2 %	43/4	41/2	5.56	
	8	161/2	23/16	10%	133/4	12	11/4	3	4 1/6 5 1/2	51/4	8.63	
	10	20	21/2	123/4	17	16	111/4	33%	61/4	6	10.75	
	12	22	25%	15	191/4	20	11/4	35%	63%	61/8	12.75	
	3	91/2	11/2	5	71/2	8	7/8	21/8	41/4	4	3.50	
	31/4	103/4	1%	51/4	81/4	8	1	25/14	41/4	41/4	4.00	
900	4	111/2	13/4	63/16	91/4	8	11/8	23/4	43/4	41/2	4.50	J 700 5
Pound	5	133/4	2	75/16	11	8	11/4	31/8	51/4	5	5.56	5. ((((1) (((())))
	6	15	23/16	81/2	121/2	12	11/8	33%	53/4	51/2	6.63	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2
	8	181/2	21/2	10%	151/2	12	13%	4	6%	63/8	8.63	
	11/2	7	11/4	27/8	47/8	4	1	13/4	31/2	31/4	1.90	Borred to march 1. D M M M M M M M.
	2	81/2	11/2	35%	61/2	8	₹/8	21/4	41/4	4	2.38	[[]\$\f\[]\]
	21/2	9%	1%	41/8	71/2	8	1	21/2	4%	41/8	2.88	"# 4.5
1500	3	101/2	1%	5	8	8	11%	2 1/8	47/8	4%	3.50	(elding
Pound	31/2	11	2	51/2	81/4	8	11/4	31/14	5	4 1/4	4.80	#####################################
	4	121/4	21/8	63/16	91/2	8	11/4	39/16	51/8	41/8	4.50	å.
	5	143/4	21/8	75/16	111/2	8	11/2	41/8	63/8	61/8	5.56	
	6	151/2	31/4	81/2	121/2	12	1%	411/16	7	63/4	6.63	· -
	8	19	35/8_	10%	151/2	12	1%	5%	85/8	81/8	8.63	- B
	1/2	51/4	13/16	1%	31/2	4	3/4_	1%	31/8	21/8	0.84	1 222 × 222
	3/4	51/2	11/4	111/16	33/4 41/4	4	¾ ¾	111/16	33/4	31/2	1.05	Bored to alip C C C C C C C C C C C C C C C C C C C
	11/4	61/4	11%	21/2	51/8	4	1 /8	21/16	4	33/4	1.66	
	11/2	8	13/4	27/8	53/4	4	11/4	23/8	45%	43/4	1.90	
	2	91/4	2	35%	63/4	8	1	23/4	51/4	5	2.38	
2500	21/2	101/2	21/4	41/6	73/4	8	11/4	31/4	51/8	5%	2.88	eldin X
Pound	3	12	25%	5	9	8	11/4	35/8	67/8	65/8		Veldir
	4	14	3	63/16	103/4	8	11/2	41/4	73/4	71/2	4.50	Bored to slip over pipe CC+1/6, The Hall of the Hall o
	5	161/2	35%	75/16	123/4	8	13/4	51/8	91/4	9	5.56	80
	6	19	41/4	81/2	141/2	8	2	6	11	103/4	6.63	_
	8	213/4	5	10%	171/4	12	2	7	123/4	121/2	8.63	
	10	261/2	61/2	123/4	211/4	12	21/2	9	163/4		10.75	
	12	30	71/4	15	24%	12	23/4	10	181/2	181/4	12.75	1 200

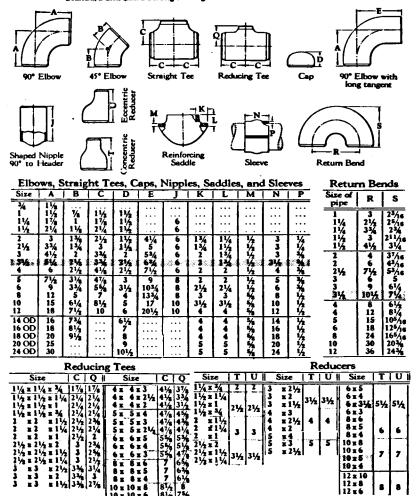
7 7

TABLE XVII

Steel Butt-Welding Fittings

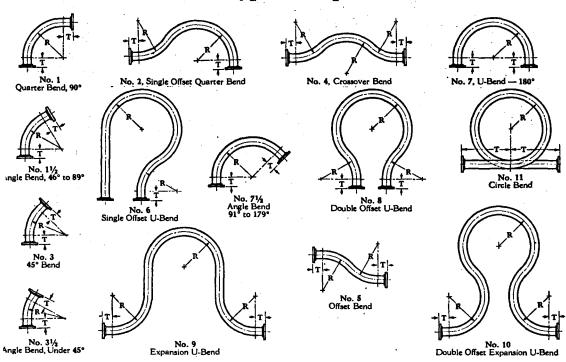
Dimensions, in Inches

Standard and Extra Strong Fittings have the same outside dimension



10 x 10 x 8 10 x 10 x 6 12 x 12 x 10 12 x 12 x 8 12 x 12 x 6

Standard Types of Pipe Bends



Standard Types of Pipe Bends

	Column "A"		Column "B"							Colum	ın "C"
	R Minimum Recommended		Shor	T Minimum Length of Tangent							
Size	Redius		Steel Pi	pe Only		Genu	ine Wro	ght Iro	n Pipe	Steel or	Genuine
Pipe	of Steel or Genuine Wrought Iron Pipe	Threaded Ends, Screwed, or Welded Flanges		Crar	Cranelap Joints		Threaded Ends, Screwed, or Welded Flanges		nelap nts	Wrought Iron Pipe Standard or Extra Strong Weight	
	Standard or Extra Strong Weight	Stand- ard Weight Pipe	Extra Strong Weight Pipe		Strong	Stand- ard Weight Pipe	Extra Strong Weight Pipe	Stand- ard Weight Pipe	Extra Strong Weight Pipe	Threaded Ends or Screwed Flanges	Cranelap Joints
1/4	11/4	1	-%			11/4	1			1	
4	1 1/8	11/4	3/4			17/8	11/2			11/4	
1/2	21/2	11/2	1	2	2	21/2	2	21/2	2	11/2	2
-1/4	33/4	13/4	11/4	3	3	3	21/2	3	3	13/4	2
_1	5	2	11/2	4	4	4	3	4	4	2	2
11/4	61/4	21/4	13/4	4	4	5_	4	5	4	2	21/2
11/2	71/2	21/2	21/2	5	5	6	5	6 8	- 5	21/2	3
21/2	121/2	5	41/4	10	10	10	8	10	10	3	5
3	15	8	6	12	12	12	10	12	12	l -	6
31/2	171/2	10	8	12	12	14	12	14	12		6
4	20	12	10	14	14	16	12	16	14	1-5-	6
- 5	25	18	14	18	14	20	15	20	15	6	Ť
6	30	22	15	22	15	26	18	26	18	, , , , , , , , , , , , , , , , , , , 	7
8	40	30	23	30	23	30	28	30	28	9	8
10	50	36	30	36	30	36	32	36	32	12	10
12	60	46	36	46	36	46	42	46	42	14	10
*14 OD	70	-60	48	60	48	60	54	60	54	16	14
•16 OD	96	80	60	80	60	80	70	80	70	18	16
*18 OD	108	90	66	90	66	90	80	90	80	18	18
*20 OD	120	100	72	100	72	100	90	100	90	18	18
*24 OD	144	144	108	144	108	144	122	144	122	18	20

^{*}For sizes 14-inch O.D. and larger, the radii shown are based upon pipe with a wall thickness of Ja-inch or lighter under the "Standard Weight Pipe" column, and a wall thickness of Ja-inch or heavier under the "Extra Strong Weight Pipe" column.

TABLE XXIV

FORMULAS AND MOLECULAR AND EQUIVALENT WEIGHTS OF SUBSTANCES FREQUENTLY APPEARING IN THE CHEMISTRY OF WATER CONDITIONING

Substance	Formula	Weight of Molecule or Radical	E - 87/4
			Eq. Wt.
Aluminum	Al	27.0	9.00
Aluminum chloride (crystal)	AICI.	133.0	44.33
Aluminum chloride (anhydrous)	AlCl _a ·6H ₂ O	241.0	80.33
Aluminum sulphate (crystal)	Al ₂ (SO ₄) ₄ · 18H ₂ O	666.4	111.06
Aluminum sulphate (anhydrous)	Al ₂ (SO ₄) _a	342.1	57.01
Ammonia	NH.	17.0	17.00
Ammonium alum	Al ₂ (SO ₄) ₂ (NH ₄) ₂ SO ₄ · 24H ₂ O	906.6	151.10
Potassium alum	$Al_2(SO_4)_2K_2SO_4 \cdot 24H_2O$	948.8	158.13
· Aluminum hydrate	Al(OH) _a	78.0	26.00
Ammonium chloride	NHCI	53.5	53.50
Ammonium (ion)	NH4	18.0	18.00
Ammonia water	NH ₁ OH	35.1	35.10
*Alumina	Al ₂ O ₃	102.0	17.00
Sodium aluminate	Na ₂ Al ₂ O ₄	164.0	27.33
Ammonium sulphate	(NH ₄) ₂ SO ₄	132.0	66.00
Barium	Ba	137.4	68.70
Barium carbonate	BaCO _a	197.4	98.70
Barium chloride	BaCl ₂ · 2H ₂ O	244.3	122.15
Barium sulphate	BaSO ₄	233.4	116.70
Barium hydrate	Ba(OH) ₂	171.0	85.50
Barium oxide	BaO	153.0	76.50
Calcium	Ca	40.1	20.05
Calcium bicarbonate	Ca(HCO ₂) ₂	162.1	81.05
Calcium carbonate	CaCO ₃	100.1	50.05
Calcium chloride	CaCl ₂	111.0	55.50
Calcium hydrate (pure)	Ca(OH) ₂	74.1	37.05
Calcium hydrate (90%)	Ca(OH) ₂		41.10
Calcium oxide (pure)	CaO	56.1	28.05
Calcium oxide (90%)	CaO		31.17
Calcium sulphate (anhydrous)	CaSO ₄	136.2	68.08
Calcium sulphate (gypsum)	CaSO₄ · 2H₂O	172.2	861.0
Calcium nitrate	Ca(NO ₃) ₂	164.1	82.05
Calcium phosphate	$Ca_3(PO_4)_2$	310.3	51.71
Carbon	C	12.0	3.00
Chloride (gas)	Cl ₂	70.9	35.45

Courtesy of the Cochrane Corp.

TABLE XXIV continued

		Weight of Molecule or	
Substance	Formula	Radical	Eq. Wt.
Copper Copper sulphate Iron (ferric) Iron (ferrous) *Ferric oxide	Cu	63.6	31.80
	CuSO ₄	160.0	80.00
	Fe'''	55.8	18.60
	Fe''	55.8	27.90
	Fe ₂ O ₈	159.6	26.60
*Iron oxide and alumina	R ₂ O ₄	109.0	18.20
Ferrous sulphate (copperas)	FeSO ₄ · 7H ₂ O	278.0	139.00
Ferric chloride	FeCl ₃	162.0	54.00
Ferric hydrate	Fe(OH) ₃	107.0	35.66
Ferric sulphate	Fe ₂ (SO ₄) ₃	400.0	66.66
Ferrous carbonate	FeCO ₃	116.0	58.00
Ferrous hydrate	Fe(OH) ₂	89.9	44.95
Ferrous oxide	FeO	71.8	35.90
Fluorine (gas)	F ₃	19.0	19.00
Hydrogen (gas)	H ₂	2.02	1.01
Hydrogen sulphide (gas) Lead Magnesium Magnesium oxide Magnesium bicarbonate	H _s S Pb Mg MgO Mg(HCO _s),	34.1 207.0 24.3 40.3 146.3	103.50 12.15 20.15 73.15
Magnesium carbonate	MgCO ₃	84.3	42.15
Magnesium chloride	MgCl ₂	95.2	47.60
Magnesium hydrate	Mg(OH) ₂	58.3	29.15
Magnesium nitrate	Mg(NO ₃) ₂	148.3	74.15
Magnesium phosphate	Mg ₂ (PO ₄) ₃	263.0	43.83
Magnesium sulphate Nitrogen (gas) Oxygen (gas) Potassium Potassium carbonate Potassium chloride	MgSO4 N2 O2 K K2CO2 KCI	120.4 28.0 32.0 39.1 138.0 74.6	60.20 8.00 39.10 69.00 74.60
Potassium hydrate	KOH	56.1	56.10
Potassium iodide	KI	166.0	166.00
Potassium nitrate	KNO₃	101.0	101.00
Potassium oxide	K₃O	94.2	47.10
Potassium permanganate	KMnO₄	158.0	158.00
Potassium sulphate	K _s SO ₄	174.0	87.00
Silicon	Si	28.1	7.03
Silver	Ag	108.0	108.00
Silver chloride	AgCl	143.3	143.30
Silver nitrate	AgNO ₈	169.9	169.90
Silica	SiO ₂	60.1	30.05
Sodium	Na	23.0	23.00
Sodium bicarbonate	NaHCO ₃	84.0	84.00

TABLE XXIV continued

	•	Weight of Molecule or	
Substance	Formula	Radical	Eq. Wt.
Sodium carbonate Sodium chloride Sodium hydrate	Na ₃ CO ₃	106.0	53.00
	NaCl	58.5	58.50
	NaOH	40.0	40.00
	NaNO ₃	85.0	85.00
Sodium nitrate Sodium oxide	Na ₂ O	62.0	31.00
Sodium sulphite Sulphur dioxide (gas) Trisodium phosphate Trisodium phosphate (anhydrous) Disodium phosphate	Na ₃ SO ₃	126.0	63.00
	SO ₂	64.1	32.05
	Na ₂ PO ₄ · 12H ₂ O (18.7% P ₂ O ₅)	380.2	126.73
	Na ₂ PO ₄ (43.3% P ₂ O ₅)	164.0	54.66
	Na ₂ HPO ₄ · 12H ₂ O (19.8% P ₂ O ₅)	358.2	119.40
Disodium phosphate (anhydrous)	Na ₂ HPO ₄ (50.0% P ₂ O ₅)	142.0	47.33
Monosodium phosphate	NaH ₃ PO ₄ · H ₂ O (51.4% P ₂ O ₅)	138.1	46.03
Monosodium phosphate (anhydrous)	NaH ₃ PO ₄ (59.1% P ₂ O ₅)	120.1	40.03
Metaphosphate (Hagan)	NaPO ₅ (69.5% P ₂ O ₅)	102.0	34.00
Sodium sulphate	Na ₅ SO ₄	142.0	71.00
Water	H ₂ O	18.0	9.00
Bicarbonate	ACID RADICALS HCO _a CO _a	61.0	61.00
Carbonate		60.0	30.00
Carbon dioxide	CO.	44.0	22.00
Chloride		35.46	35.46
Nitrate	NO ₃	62.0	62.00
Hydrate	OH	17.0	17.00
Phosphate	PO ₄	95.0	31.66
Phosphorous oxide	P ₂ O ₈	142.0	23.67
Sulphate	SO ₄	96.06	48.03
Sulphite	SO ₉	80.1	40.05
	ACIDS		
Hydrogen	H,	2.0	1.00
Acetic acid	HC,H,O,	60.0	60.00
Carbonic acid	H,CO,	62.0	31.00
Hydrochloric acid	HCI	36.46	36.46
Phosphoric acid	H,PO,	98.0	32.67
Sulphuric acid	H ₂ SO ₄	98.1	49.05
Nitric acid	HNO ₅	63.0	63.00
Sulpurous acid	H ₂ SO ₅	82.1	41.05

^{*} Al_2O_2 and Fe_2O_3 are commonly reported together as R_2O_3 , which for purpose of computation is given a mol. wt. of 109 and eq. wt. of 18.2 on the basis of 88% Al_2O_3 and 12% Fe_2O_3 .

APPERDIX

TABLE XXV

STEAM TABLES

Courtesy of Combustion Engineering-Superheater, Inc.

Saturated Steam: Temperature Table

Tomp.	A secutiff Lb. per	PRESSURE	Srac Sat.	nuc Aori	Sal.	Bat.	CHTHALPI	Sat.	Set.	Entrope	Bat.	Temp.
Tabr.	6g. la.	32 F.	Liquid	Evap.	Vapor	Liquid	Evap.	Vapor	birgid	Evap.	Vapor	Fahr.
	•		₹7	∀(g	Ve	No.	heg	h _e	-		*	
81	0.0886	0.1806	0.01602	3305.7	8305.7	•	1075.1	1075.1		2.1865	2.1865	- 85
14	0.0961	0.1957	0.01602	2060.4	2060.4	2.01	1074.0	1076.0	0.0041	2.1755	2.1796	74
24	0.1041	0.2120	0.01602	2836.6	2836.6	4.03	1072.9	1076.9	0.0082	2.1645	2.1727	36
20	0.1126	0.2292	0.01602	2632.2	2632.2	6.04	1071.7	1077.7	0.0122	2.1533	2.1655	36
40	0.1217	0.2476	0.01602	2445.1	2445.1	8.05	1070.5	1078.6	0.0162	2.1423	2.1585	4
44	0.1217	0.2978	0.01002	2271 8	2440.1 2271 B	10.06	1069.3	1079.4	0.0203	2.1314	2.1517	4
44	0.1420	0.2891	0.01602	2112.2	2112.3	12.06	1068.2	1080.2	0.0242	2.1207	2.1449	4
44	0.1532	0.3119	0.01602	1965.5	1965.5	14.07	1067.1	1081.2	0.0282	2.1102	2.1384	46
44	0.1652	0.3364	0.01602	1829.9	1829.9	16.07	1065.9	1082.0	0.0322	2.0995	2.1317	48
80	0.1780	0.3624	0.01602	1704.9	1704.9	18.07	1064.8	1082.9	0 0361	2.0891	2.1252°	60
- 52	0.1760	0.3905	0.01603	1588.4	1588.4	20.07	1063.6	1063.7	0.0400	2.0786	2.1186	64
- 14	0.2063	0.4200	0.01603	1482.4	1482.4	22.07	1062.5	1084.6	0.0439	2.0684	2.1123	- 14
94	0.2219	0.4518	0.01603	1383.5	1383.5	24.07	1061.4	1085.5	0.0478	2.0582	2.1060	56
86	0.2384	0.4854	0.01603	1292.7	1292.7	26.07	1060.2	1066.3	9.0517	2.0479	2.0996	56
80	0.2561	0.5214	0.01603	1208.1	1206.1	28.07	1059.1	1087.2	0.0555	2.0379	2.0934	••
84	0.2361	0.5597	0.01604	1129.7	1129.7	20.06	1057.9	1088.0	0.0594	2.0278	2.0934	=
84	0.2949	0.6004	0.01604	1057.1	1057.1	32.06	1056.6	1088.9	0.0632	2.0180	2.0812	==
66	0.3162	0.6438	0.01604	989.6	989.6	34.06	1055.7	1089.8	6.0670	2.0082	2.0752	66
68	0.3388	0.6898	0.01605	927.0	927.0	36.05	1054.5	1090.6	0.0708	1.9983	2.0691	66
70	0.3628	0.7387	0.01605	868.9	868.9	28.05	1053.4	1091.6	0.0745	1.9887	2.0632	70
70	0.3683	0.7906	0.01606	814.9	814.9	40.04	1052.3	1092.3	0.0743	1.9792	2.0575	73
74	0.4153	0.8456	0.01606	764.7	764.7	42.04	1051.2	1093.2	0.0820	1.9697	2.0517	74
76	0.4440	0 9040	0.01607	718.0	718.0	44.03	1050.1	1094.1	0.0858	1.9603	2.0461	76
76	0.4744	0.9659	0.01607	674.4	674.4	46.03	1048.9	1094.9	0.0895	1.9503	2.0403	78
•	0.5067	1.032	0.01607	633.7	633.7	48.02	1047.8	1095.8	0.0932		2 0347	
===	0.5409	1.101	0.01608	633.7 595.8	633.7 695.8	48.02 50.02	1047.8	1095.8	0.0932	1.9415 1.9321	2.0290	20 85
94	0.5772	1.175	0.01606	\$60.4	560.4	50.02 52.01	1045.5	1097.5	0.0908	1.9230	2.0236	=======================================
86	0.6158	1.253	0.01609	527.6	\$27.6	54.01	1044.4	1098.4	0.1042	1.9139	2.0181	26
64	0.6555	1.235	0.01600	497.0	497.0	56.00	1043.2	1099.2	0.1079	1.9047	2.0126	88
•	A AGAA	1.421	0.01610	468.4	468.4	88.00	1042.1	1100.1	0.1115	1.8958	2.0073	•
98	0.7429	1.513	0.01611	441.7	441.7	56.00 56.90	1040.9	1100.1	0.1151	1.8867	2.0078	=======================================
94	0.7902	1.609	0.01611	416.7	416.7	61.98	1039.8	1101.8	0.1187	1.8779	1.9966	94
94	0.8403	1.711	0.01612	393.2	893.2	63.98	1038.7	1102.7	0.1223	1.8692	1.9915	96
96	0.8030	1.818	6.01613	371.3	871.8	65.98	1037.5	1103.5	0.1250	1.8604	1.9863	96
100	0.9487	1.932	0.01613	350.8	350.8	67.97	1036.4	1104.4	0.1295	1.8517	1.9612	100
108	1.0072	2.051	6.01614	331.5	231.5	69.96	1035.2	1105.2	0.1330	1.8430	1.9760	102
104	1.0689	2.176	0.01614	313.5	313.5	71.96	1034.1	1106.1	0.1366	1.8345	1.9711	104
100	1.1338	2.308	0.01615	296.5	296.5	73.95	1033.0	1 (07.0	0.1401	1.8261	1.9662	106
100	1.3020	2.447	0.01616	280.7	280.7	75.94	1032.0	1107.9	0.1436	1.8179	1.9615	100
110	1.274	2.594	6.01617	265.7	265.7	77.94	1030.9	1106.6	0.1471	1,8096	1.9567	110
112	1.350	2.749	0.01617	251.6	251.6	79.93	1029.7	1109.6	0.1506	1.8012	1.9518	118
114	1.429	2.900	0.01618	238.5	238.5	81.93	1028.6	1110.5	0.1541	1.7930	1.9471	114
116	1.512 1.600	3.078 3.258	0.01619	226.2	226.2	83.92	1027.5	1111.4	0.1576	1.7848	1.0424	116
***	1.400	8.206	9.01620	214.6	214.5	66.92	1026.4	1112.3	0.1610	1.7767	1.9377	118
130	1.692	3.445	0.01620	203.45	203.47	87.91	1025.3	1113.2	0.1645	1.7687	1.0332	120
122	1.768	3.640	●.0162 L	193.16	193.16	89.91	1024.1	1114.0	0.1679	1.7606	1.9285	122
134	988. I 400. I	3.846 4.062	0.01622	183.44	183.46	91.90	1023.0	1114.9	0.1714	1.7526	1.9240	194
136	2.106	4.062	0.01623 0.01624	174.26	174.28 165.72	93.90 95.90	1021.8 1020.7	1115.7 1116.6	6.1748 6.1782	1.7446	1.9194	126
				0.70	104.75	43.90	1929.7	1119.5	€.1762	1.7368	1.9150	126
130	2.221	4.522	0.01625	157.66	157.57	97.89	1019.5	1117.4	0.1816	1.7289	1.9105	120
186	3.343 2.470	4.770 5.020	0.01626	149.83	149.86	99.80	1018.8	1118.2	0.1849	1.7210	1.9058	180
134	3.470 2.603	5.306	0.01626	142.50 135.78	142.61 135.75	101.80 103.88	1017.3	1119.1	0.1863	1.7134	1.9017	134
136	3.742	5.583	0.01628	120.26	120.28	105.85	1016.9	1119.9 1120.8	0.1917 0.1950	1.7056	1.8973 1.8930	136 138
								_	u		.4	
149	2.887 3.039	8.878	0.01629	123.16	123.16	107.88	1013.7	1121.6	0.1984	1.0004	1.8888	140
144	3 039	6.167 6.511	0.01630 0.01631	117.37 111.86	117.39	109.88 111.88	1012.5	1122.4	9.2017 9.2050	1.6828	1.8845	146
146	3.363	6.847	0.01632	106.72	106.74	113.86	1011.3	1124.1	0.2050 0.2063	1.6752	1.8802 1.8761	144 146
148	3.536		0.01633	101.62	101.84	115.07	0.0001	1124.9	0.2116	1.6604	1.8720	146
100	3.716	7.506	0.01634	97.18	97.20	117.87	1007.8	1125.7	0.2149	1.6530	1.8679	100
100	3.904	7.948	0.01635	92.79	92.81	119.87	1006.7	1125.7	9.2149 9.2161	1.6456	1.8639	100
104	4.100	8.348	0.01636	86.63	84.64	121.87	1005.5	1127.4	0.2214	1.6384	1.8598	154
106	4.304	8.765	0.01637	84.66	64.66	123.67	1004.4	1128.3	0.2247	1.6313	1.8560	156
•	4.518	9.196	0.01638	80.98	80.92	125,67	1003.2	1129.t	Q_2279	1.6341	1.8520	1.66
100	4.739	9.649	0.01639	77.37	77.30	127.67	1002.0	1129 0	6.2311	1.6169.	1.8480	100
148	4.970	10.12	0.01640	74.00	74.02	129.88	1000.8	1130.7	0.2343	1.6036	1.8441	166
164	8.210	10.61	0.01642	70.79	70.61	131.66	999.7	1131.6	0.2376	1.6029	1.8405	164
Les	& 460 & 730	11.12 11.65	0.01643	67.76	67.78	133.68	998.5	1132.4	0.2406	1.5958	1.6366	106
	- 1 A	11.00	6.01644	6LB7	64.80	135.88	907.3	1133.3	0.3430	1.588\$	1.8327	148

TABLE XXV
continued
Saturated Steam: Temperature Table—(Continued)

emp.	Assolute Lb. per 8q. In.	PRESSURE In. He.	SpE4	eine Vor	JME Sal	Sat.	STHALPY	Sat.	Sat.	ENTROPE	Set	Temp
ahr.		32 F.	Liquid	Evap.	Vapor	Liquid	Evap.	Vapor	Liquid	Evap.	Vapor	Fabi
<u>. </u>	P		¥¢.	▼1 €	Ve .	pt	hie	h _e	4	er _g	*	ŧ
170	5.990	12.20	0.01645	62.12	62.14	137.89	996.1	1134.0	0.2471	1.5319	1.8290	170
172	6.272	12.77	0.01646	59.50	59.52	139.89	995.0	1134.9	0.2503	1.5751	1.8254	177
174	6.565	13.37	0.01647	57.01	57.03	141.89	993.8	1135.7	0.2535	1.5683	1.8218	170
176 178	6.869	13.99	0.01648	54.64	54.66	143.90	992.6	1136.5	0.2566	1.5615	1.8181	170
178	7.184	14.63	0.01650	52.39	52.41	145.90	991.4	1137.3	0.2598	1.5547	1.8145	178
160	7.510	15.29	0.01651	50.26	50.28	147.91	990.2	1138.1	0.2629	1.5479	1.8108	180
162	7.849	15.98	0.01652	48.22	48.24	149.92	989.0	1138.9	0.2661	1.5412	1.8073	161
164	8.201	16.70	0.01653	46.28	46.30	151.92	987.8	1139.7	0.2692	1.5346	1.8038	184
186 188	8.566	17.44 18.21	0.01654	44.43	44.45	153.93	986.6	1140.5	0.2723	1.5280 1.5213	1.8003 1.7967	180
TOO	8.944	10.21	0.01656	42.67	42.69	155.94	985.3	1141.3	0.2754	1,3213	1.7307	100
190	9.336	19.01	0.01657	40.99	41.01	157.95	984.1	1142.1	0.2785	1.5147	1.7932	190
192	9.744	19.84	0.01658	89.38	39.40	159.95	98248	1142.8	0.2816	1.5081	1.7897	195
194	10.168	20.70	0.01659	37.84	37.86	161.96	981.5	1143.5	0.2847	1.5015	1.7862	194
196 196	10.605 11.057	21.59 22.51	0.01661 0.01662	36.38 34.98	36.40 35.00	163.97 165.98	980.3 979.0	1144.3 1145.0	0.2877 0.2908	1.4951 1.4885	1.7828 1.7793	194
	11.037	22.31	0.01002	01.80	25.00	103.98	979.0	1143.0	0.2908	1.2003	1.7793	194
200	11.525	23.46	0.01663	33.65	33.67	167.99	977.8	1145.8	0.2938	1.4822	1.7760	200
202	12.010	24.45	0.01665	82.37	32.39	170.01	976.6	1146.6	0.2969	1.4759	1.7728	201
204	12.512	25.47	0.01666	31.15	81.17	172.02	975.3	1147.3	0.2999	1.4695	1.7694	204
106 108	13.031	26.53 27.62	0.01667	29.99	30.01	174.03 176.04	974.1 972.8	1148.1 1148.8	0.3029 0.3059	1.4633	1.7662 1.7629	20
-05	13.568	27.62	0.01669	28.88	28.90							
R10	14.123	28.75	0.01670	27.81	27.83	178.06	971.5	1149.6	0.3090	1.4507	1.7597	21
113	14.696	29.92	0.01672	26.81	26.83	180.07	970.3	1150.4	0.3120	1.4446	L.7566	91
315	15.591		0.01674	25.35	25.37	183.10	968.3	1151.4 1153.3	0.3165	1.4352	1.7517	210
120 125	17.188		0.01677 0.01681	23.14 21.15	23.16 21.17	188.1 4 193.18	965.2 961.9	1155.1	0.3239 0.3313	1.4201 1.4049	1.7440 1.7362	33: 22
-44	18.915		U.U1001	-								
130	20.78		0.01684	19.371	19.388	198.22	958.7	1156.9	0.3386	1.3900	1.7286	23
235	22.80		0.01688	17.761	17.778	203.28	955.3	1158.6	0.3459	1,3751	1.7210	231
140	24.97		0.01692	16.307	16.324	208.34 213.41	952.1 948.7	1160.4 1162.1	0.3531	1.3607	1.7138	24
45	27.31 29.82		0.01696 0.01700	15.010 13.824	15.027 13.841	213.41	948.7 945.3	1162.1	0.3675	1.3402	1.6995	74 25
-50	29.52		9.01700								•	
55	32.53		0.01704	12.735	12.752	223.56	942.0	1165.6	0.3747	1.3181	1.6928	25
60	35.43		0.01708	11.754	11.771	228.65	938.6	1167.3	0.3817	1.3042	1.6859	26
66	38.54		0.01713	10.861 10.053	10.878 10.070	233.74 238.84	935. 3 931. 8	1169.0 1170.6	0.3888 0.3958	1.2906 1.2770	1.6794 1.6728	26 27
70 75	41.85 45.40	•	0.01717	9.313	9.330	243.94	928.2	1172.1	0.4027	1.2634	1.6661	27
•••				•								٠.,
188	49.20		0.01726	8.634	8.651	249.06	924.6	1173.7	0.4096	1.2500	1.6596	280
185	53.25		0.01731	8.015	8.032	254.18	921.0	1175.2	0.4165	1.2368	1.6533	28
190	57.55		0.01735	7.448	7.465	259.31 264.45	917.4 913.7	1176.7 1178.2	0.4234 0.4302	1.2237 1,2107	1.6471 1.6409	294
195	62.13		0.01740 0.01745	6.931 6.454	6.948 6.471	269.60	910.1	1179.7	0.4370	1.1980	1.6350	80
100	67.01		0.01743	0.454	0.171	200.00				••		
105	72.18		0.01750	6.014	6.032	274.76	906.3	1181.1	0.4437	1.1852	1.6289	30
10	77.68		0.01755	5.610	5.628	279.92	902.6	1182.5	0.4505	1.1727	1.6232	310
16	83.50		0.01760	5.239	5.257	285.10	898.8	1183.9 1185.3	0.4571. 0.4637	1.1587	1.6158	310
120	89.65		0.01765	4.897	4.915 4.601	290.29 295.49	895.0 891.1	1186.6	0.4703	1.1356	1.6059	324
125	96.16		0.01771	4.000	2.001	200.15		1100.0	0.2100			
130	103.03		0.01776	4.292	4.310	300.69	887.1	1187.8	0.4769	1.1234	1.6003	330
135	110.31		0.01782	4.021	4.039	305.91	883.2	1189.1	0.4835	1.1114	1.5949	330
40	117.99		0.01788	3.771	3.789	311.14	879.2	1190.3	0.4900	1.0994	1.5894	346
46	126.10		0.01793	8.539 8.324	3.557 3.342	316.38 321.64	875.1 871.0	1191.5 1192.6	0.4966 0.5030	1.0875	1.5841 1.5787	340
50	134.62		0.01799	9.324	9.012	321.0¶	9. 1.0	4.44.V	4.9030		1.0101	
55	143.58		0.01805	3.126	8.144	326.91	866.8	1193.7	0.5094	1.0640	1.5734	356
60	153.01		0.01811	2.940	2.958	332.19	862.5	1194.7	0.5159	1.0522	1.5681	360
65	162.93		0.01817	2.768	2.786	337.48	858.2	1195.7	0.5223	1.0406	1.5629	361
70	173.33		0.01823	2.607	2.625	342.79 348.11	853.8 849.4	1196.6 1197.5	0.5286 0.5350	1.0291	1.5577	370 376
75	184.23 195.70		0.01830 0.01836	2.458 2.318	2.476 2.336	348.11 353,45	844.9	1197.5	0.5350	1.0062	1.5526	386
80 85	195.70 207.71		0.01830	2.189	2.207	358.80	840.4	1199.2	0.5476	0.9949	1.5425	384
								1199.9	0.5540	0.9835	1.5375	396
16 16 16	220.29 233.47		0.01850 0.01857	2.064 1.9512	2.083 1.9698	364.17 369.56	835.7 831.0	1200.6	0.55602	0.9533	1.5325	294
=	247.25		0.01864	1.8448	1.8632	374.97	826.2	1201.2	0.5664	0.9610	1.5274	404
•	261.67	, 0	0.01871	1.7445	1.7632	380.40	821.4	1201.8	0.5727	0.9499	1.5226	408
	276.72	-	0.01878	1.6508	1.6696	385.83	816.6	1202.4	0.5789	0.9390	1.5179	410
	292.44		0.01886	1.5630	1.5819	391.30	811.7	1203.0	0.5851	0.9280	1.5131 1.508 2	415
	208.82		0.01894	1.4806	1.4995	396.78	806.7	1203.5	0.5912	U.917U	1.5082	-34
15	325.91		0.01902	1.4031	1.4221	402.28	801.6	1203.9	0.5974	0.9061	1.5035	435
	343.71		0.01902	1.3303	1.3494	407.80	796.5	1204.3	0.6036	0.8953	1.4989	430
16	362.27		0.01918	1.2617	1.2809	413.35	791.2	1204.6	0.6097	0.8843	1.4940	435
	381.59		0.01926	1.1973	1,2166	418.91	785.9	1204.8	0.6159	0.8735	1.4894	448
18	401.70		0.01934	1.1367	1.1560	424.49	780.4	1204.9	0.6220	0.8626	1.4846	448
10	422.61		0.01943	1.0796	1.0990	430.11	774.9	1205.0	0.6281	0.8518	p.4/99	-54
-4	444 94		0.0195	1.0256	1.0451	435.74	769.3	1205.0	0.6342	0.8410	1.4752	466
14 60	444.35 466.97		0.0195 0.0196	0.9745	0.9941	441.42	763.6	1205.0	0.6403	0.8303	1.4706	440
.	400.97 490.43		0.0190	0.9262	0.9459	447.10	757.8	1204.9	0.6463	0.8195	1.4658	445
			0.0198	0.8808	0.9006	452.84	751.9	1204.7	0.6524	0.8088	1.4612	470
70	514.70		0.0198	0.8379	0.8578	458.59	745.9	1204.5	0.6585	0.7980	1.4565	475

TABLE XXV continued

Saturated Steam: Temperature Table—(Concluded)

		Врас	arac Vol		1	Entralp			ENTROP		_
'omp. Fabr.	Abs. From Lb./8q. in	Set. Liquid	Evap.	Set. Vapor	Bot. Liquid	Evep.	Set. Vapor	Bat. Liquid	Eves.	Sat. Vapor	Temp Fahr.
t		24			hr hr	bre.					Table.
•	<u> </u>		Tig	<u>"e</u>		er(¢	<u> </u>		Me.	<u> </u>	
lae	566.12	0.0200	0.7972	0.8172	464.37	739.8	1204.2	0.6646	0.7873	1.4519	430
106	593.28	0.0201	0.7585	0.7756	470.18	733.6	1203.6	0.6706	0.7766	1.4472	485
100	621.44	0.0202	0.7219	0.7421	476.01	727.8	1203.3	0.6767	0.7658	1.4425	490
195	650.59	0.0203	0.6872	0.7075	481.90	720.8	1202.7	0.6827	0.7550	1.4377	496
J 00	680.80	0.0204	0.6544	0.6748	487.80	714.2	1202.0	0.6888	0.7442	1.4330	500
106	712.10	0.0206	0.6230	0.6436	493.8	707.5	1201.3	0.6949	0.7334	1.4283	806
510	744.55	0.0207	0.5932	0.6139	499.8	700.6	1200.4	0.7009	0.7225	1.4234	810
B16	777.96	0.0203	0.5651	0.5859	505. B	693.6	1199.4	0.7070	0.7116	1.4186	816
120	812.68	0.0209	0.5382	0.5591	511.9	686.5	1198.4	0.7132	0.7007	1.4139	890
945	848.37	0.0210	0.5128	0.5338	518.0	679.2	1197.2	0.7192	0.6898	1,4090	825
530	885.20	0.0212	0.4885	0.5097	524.2	671.9	1196.1	0.7253	0.6780	1.4042	530
606	923.45	0.0218	0.4654	0.4867	530.4	664.4	1194.8	0.7314	0.6679	1.3993	835
640	962.80	0.0214	0.4433	0.4647	536.6	656.7	1193.3	0.7375	0.6569	1.3944	540
545	1003.6	0.0216	0.4222	0.4438	542.9	648.9	1191.8	0.7436	0.6450	1.3895	545
800	1045.6	0.0218	0.4021	0.4239	549.3	640.9	L190.2	0.7498	0.6347	1.3845	550
846	1088.6	0.0219	0.3830	0.4049	555.7	632.6	1188.3	0.7559	0.6234	1.3793	545
848	1138.4	0.0221	0.3648	0.3869	562.2	624.1	1166.3	0.7622	0.6120	1.3742	840
946	1179.3	0.0222	0.3472	0.3694	568.8	615.4	1184.2	0.7684	0.6006	1.3690	346
870	1226.7	0.0224	0.3304	0.3528	575.4	606.5	1181.9	0.7737	0.5890	1.3627	878
178	1278.7	0.0226	0.3143	0.3369	582.1	597.4	1179.5	0.7810	0.5774	1.3584	975
100	1326.1	0.0228	0.2989	0.3217	588.9	588.1	1177.0	0.7872	0.5656	1.3528	600
646	1378.1	0.0230	0.2840	0.3070	595.7	578.6	1174.8	0.7936	0.5538	1.3474	546
500	1431.6	0.0232	0.2699	0.2931	602.6	568.6	1171.4	0.8000	0.5419	1.3419	800
898	1486.5	0.0234	0.2563	0.2797	609.7	658.7	l 168.4	0.8065	0.5297	1,3362	805
800	1543.2	0.0236	0.2432	0.2668	616.8	548.4	1165.2	0.8130	0.5175	1.3305	000
906	1601.8 1661.6	0.0239	0.2306	0.2545	624.1	537.7	1161.8	0.8196	0.5050	1.3246	605
616		0.0241	0.2185	0.2426	631.5	526.6	1158.1	0.6263	0.4923	1.3186	610
616 680	1723.4 1787.0	0.0244 0.0247	0.2068 0.1955	0.2312 0.2202	638.9 646.5	515.8 503.7	1154.2 1150.2	0.6330 0.6396	0.4795 0.4665	1.3125	415
es .	1852.4	0.0250	0.1845	0.2095	654.3	491.5	1145.B	0.8467	0.4531	1.2996	424
400	1919.8	0.0253	0.1740	0.1993	662.2	478.8	1141.0	0.8537	0.4394	1.2931	630
636	1989.0	0.0256	0.1638	0.1894	670.4	465.5	1135.0	0.8609	0.4252	1.2861	444
***	2060.3	0.0260	0.1539	0.1799	678.7	452.0	1130.7	0.8681	0.4110	1.2791	•
•	2123.5	0.0264	0.1441	0.1706	687.8	437.6	1124.9	9.8756	0.3961	1.2717	-
600	2206.8	0.0268	0.1348	0.1616	696.0	422.7	1116.7	0.8832	0.3809	1.2641	
686	2286.4	0.0278	0.1256	0.1529	705.2	407.0	1112.2	0.6910	0.3651	1.2561	646
600	2366.2	0.0278	0.1167	0.1445	714.4	390.5	1104.9	0.8991	0.3488	1.3479	644
005	2448.0	0.0283	0.1079	0.1362	724.5	872.1	1096.6	0.9074	0.3308	1,2382	644
670	2632.4	0.0290	0.0991	0.1281	734.6	353.3	1087.9	0.9161	0.3127	1.2288	670
975	2619.2	0.0297	0.0904	0.1901	745.5	332.6	1078.8	0.9253	0.2933	1.2186	678
900	2708.4	0.0306	0.0810	0.1118	757.8	310.0	1067.2	0.9352	0.2730	1.2072	600
•••	2800.4	0.0316	0.0716	0.1032	770.1	284.5	1054.6	0.9459	0.2485	1.1944	605
900	2895.0 2902.7	0.032 8 0.0345	0.0617 0.0511	0.0945 0.0856	784.2 801.3	254.9 219.1	1039.1 1020.4	0.9579 0.9790	0.2317 0.1897	1.1796 1.1617	600 605
700	3094.1	0.0300	0.0380	0.0758	823.9	171.7	995.6				
706	\$190.L	0.0440	0.035	0.0758 0.0507	823.9 870.2	171.7 77.6	995.6 947.8	0.9904 1.0305	0.1481 0.0661	1.1385	700 706
_	4° 2206.2	0.0841	•	9.0541	910.3	•	910.3	1.0645			706.

Critical temperature

Saturated Steam: Pressure Table

Abs. Press.	Temp.	Sat.	PIC VOL	Sat	Sat.	ENTHALP	Sat.	Set.	Ентиорт	Sat.	Abs. Pres
Lb./8q. In.	Fahr.	Liquid'	Evap.	Vapor Ve	Liquid be	Evap.	Vapor b _e	Jaquid er	Evap.	Vapor	Lb./Sq. I
0.0886	32.00	0.01602	3305.7	3305.7					Ng.	- GE	P
8.125	40.69	0.01602	2383.7	2383.7	0 8.74	1075.1 1070.2	1075.1 1078.9	0 0.0176	2.1865 2.1388	2.1865 2.1564	0.0886 0 125
0.250 0.500	59.31 79.58	0.01603 0.01607	1235.8 641.71	1235.8 641.73	27.38 47.60	1059.5 1048.0	1086.9 1095.6	0.0542	2.0414 1.9434	2.0956 2.0358	6.250
1	101.76	0.01614	333.77	333.79	69.72	1035.5	1105.2	0.1326	1.8443	1.9769	9.500 1
10 10	162.25 193.21	0.01641 0.01659	73.584 38.445	73.600 38.462	130.13 161.17	1000.7 982.1	1130.8 1143.3	0.2347 0.2834	1.6090 1.5042	1.8437 1.7876	10
14.696 15	212.00 213.03	0.01672 0.01672	26.811 26.303	26.828 26.320	180.07	970.3 969.6	1150.4	0.3120	1.4446	1.7876 1.7566	14.696
					181.11		1150.7	0.3135	1.4413	1.7548	16
20 30	227.96 250.34	0.01683 0.01700	20.093 13.746	20.110 13.763	196.16 218.83	959.9 945.2	1156.1 1164.0	0.3356 0.3680	1.3959 1.3312	1.7315 1.6992	20 10
40 50	267.24 281.01	0.01715 0.01727	10.489	10.506	256.02 250.09	933.7	1169.7	0.3919	1.2844	1.6763	40
60	292.71	0.01738	8.505 7.162	8.522 7.179	262.10	923.9 915.4	1174.0 1177.5	0.4110 0.4271	1.2473 1.2166	1.6583 1.6437	50 60
70	302.92 312.03	0.01748 0.01757	6.193 5.458	6.210 5.476	272.61 282.02	907.9 901.1	1180.5 1183.1	0.4409 0.4532	1.1905 1.167 7	1.6314 1.6209	70 80
80 90	320.27	0.01766	4.880	4.898	290.57	894.8	1185.4	0.4641	1.1472	1.6113	90
100 -	327.83	0.01774	4.415	4.433	293.43	899.9	1187.3	0.4741	1.1287	1.6028	100
110 130	334.79 341.26	0.01782 0.01789	4.032 3.710	4.050 3.728	305.69 312.46	883.3 878.1	1189.0 . 1190.6	0.4832 0.4916	1.1118	1.5950	110
130	347.31	0.01796	3.437	3.455 3.220	318.81	873 2	1192.0	0.4995	1.0820	1.5815	120 130
140	353.03 358.43	0.01803 0.01809	3.202 2.998	3.220 3.016	324.83 330.53	869.5 863.9	1193. 3 1194. 4	0.5069 0.5138	1.0686 1.0560	1.5755 1.5698	140 150
160 160	363.55	0.01815	2.816	2.834	335.95	838.6	1195.5	0.5204	1.0442 1.0327	1.5646	160
170 180 190	368.42 373.08	0.01821 0.01827	2.656 2.514	2.674 2.532	341.11 346.07	855.2 851.1	1196. 3 1197. 2	0.5266 0.5325	1.0220	1.5593 1.5545	170 180
190	877.55	0.01833	2.386	2.404	350.83	847.2	1198.0	0.5382	1.0119	1.5501	190
200	381.82	0.01839	2.270	2.288	355.40	843.3	1198.7	0.5436	1.0021	1.5457	200
210 220	385.93 389.89	0.01844 0.01850	2.165 2.067	2.183 2.086	359.80 364.05	839. 6 835.8	1199. 4 1199. 9	0.5488 0.5538	0.9929 0.9838	1.541 7 1.537 6	210 220
230 240	393.70 397.40	0.01855	1.9803	1.9989	368.16 372.16	832.2 828.7	1200.4 1200.9	0.5585	0.9752 0.9669	1.5337	230
240	400.97	0.01866	1.8244	1.8431	376.04	825.4	1201.4	0.5632 0.5677	0.9590	1.5267	240 250
260 270	404.43 407.79	0.01870 0.01875	1.7555	1.7742 1.7101	379.78 383.43	822.0 818.8	1201.8 1202.2	0.5720 0.5761	0.9513 0.9439	1.523 3 1.520 0	260 270
250 290	411.06	0.01880	1.6316	1.6504	386.99	815.5	1202.5	0.5802	0.9365	1.5167	280
300	414.24 417.33	0.01885 0.01890	1.5758 1.5237	1.5947 1.5426	390.47 393.85	815.5 812.4 809.3	1202.9 1203.2	0.5841 0.5879	0.9296 0.9228	1.5137 1.5107	390 300
350	431.71	0.01912	1.3064	1.3255	409.70	794.7	1204.4	0.6057	0.8915	1.4972	850
400 450	144.58	.0193	1.1416	1.1609	424.02	780.9	1204.9 1205.0	0.6215	0.8635	1.4850	406
500	456.27 467.00	0.0195	1.0123 0.9077	1.0318 0.9274	437.18 449.40	767.8 755.5	1204.9	0.6357 0.6488	0.8382 0.8153	1.4739 1.4641	450 500
850 600	476.94 486.21	0.0199 0.0201	0.8217	0.8416 0.7695	460.83 471.59	743.6 732.0	1204.4 1203.6	0.6609 0.6721	0.7939 0.7739	1.4548 1.4460	550 600
650	494.90	0.0203	0.6979	0.7082	481.73	721.0	1202.7	0.6826	0.7553	1.4379	650
708 . 750	503.09 510.83	0.0205 0.0207	0.6347 0.5884	0.6552 0.6091	491.49 500.8	710.1 699.4	1201.6 1200.2	0.6925 0.7019	0.7376 0.7206	1.4301 1.4225	700 750
800 850	518.20 525.23	0.0209 0.0210	0.5476 0.5116	0.5685 0.5326	509.7 518.3	689.1 678.9	1199.8 1197.2	0.7103 0.7194	0.7047 0.6893	1.4155 1.4087	800 850
900	531.94	0.0212	0.4794	0.5006	526. 6	669.0	1195.6	0.7276	0.6746	1.4022	900 950
966	538.38	0.0214	0.4503	0.4717	534.6	659.2	1193.8	0.7355	0.6605	1.3960	950
1000 1060	544.56 550.52	0.0216 0.0218	0.4240 0.4001	0.4456 0.4219	542. 4 550.0	649.5 640.0	1191.9 1190.0	0.7431 0.7504	0.6468 0.6335	1.3899 1.3839	1000 1050
1100	556.26	0.0219	0.3783	0.4002	557.4	630.4	1187.8	0.7575	0.6205	1.3780	1100
1150 1200	561.81 567.19	0.0221 0.0223	0.3583 0.3397	0.3804 0.3620	564.6 571.7	621.0 611.5	1185.6 1183.2	0.7644 0.7712	0.6079 0.5955	1.3723 1.3667	1150 1200
1250	572.39	0.0225	0.3228	0.3453	578.6	602.2	1180.8	0.7777	0.5835 0.5717	1.3612	1250
1300 1350	577.43 582.32	0.0227 0.0229	0.3067 0.2918	0.3294 0.3147	585.4 592.1	592.9 583.7	1178.3 1175.8	0.7840 0.7902	0.5602	1.3557 1.3504	1300 1350
1400	587.07 591.70	0.0231 0.0233	0.2780 0.2652	0.3011 0.2885	593.6 603.0	574.6 565.5	1173.2 1170.5	0.7963 0.8022	0.5489 0.5379	1.3452 1.3401	1400 1456
1500	596.20	0.0235	0.2530	0.2765	611.4		1167.7	0.8081	0.5269	1 3350	1500
1550	600.59	0.0237	0.2416 0.2309	0.2653	617.7	556.3 547.1 538.0	1164.8	0.8138 0.8195	0.5160 0.5054	1.3298	155 9 160 0
1550 1600 1650	604.87 609.05	0.0239 0.0241	0 2207	0.2549 0.2449	623.9 630.0	528.8	1161.9 1158.8	0.8250	0.4948	1.3249 1.3198	1650
1700 1750	613.12 617.11	0.0243 0.0245	0.2111	0.2354 0.2265	636.1 642.1	519.6	1155.7 1152.5	0.8304 0.8359	0.4843	1.3147	1700 1750
1800	621.00	0.0247	0.1033	0.2180	649.0	510.4 501.3	1149.3	0.8412	0.4740 0.4639	1.3051	1800
1850	624.82 628.55	0.0249 0.0252	0.1850 0.1770	0.2099 0.2022	653.9 659.9	492.0 482.5	1145.9 1142.4	0.8465 0.8517	0.4537 0.4434	1.3002 1.2951	1850 1900
1950	632.20	0.0254	0.1695	0.1949	665.8	473.0	1138.8	0.8569	0.4332	1.2901	1160
9000	635.78	0.0257	0.1622	0.1879	671. 7	463.5	1135.2	0.8620	0.4231	1.2851	2000
2100 2206	642.73 649.42	0.0262 0.0267	0.1486 0.1359	0.1748 0.1626	693.4 695.0	444.2 424.4	1127.6 1119.4	0.8722 0.8823	0.4029 0.3826	1.2751 1.2649	2100 2200
2300 2400	655.87 662.09	0.0274 0.0280	0.1240 0.1130	0.1514 0.1410	706.7 718.5	404.3 392.9	1111.0 1101.4	0.8923 0.9025	0.3624 0.3413	1.2547 1.2438	2300 2400
2500	668.10	0 0287	0.1026	0.1313	730.7	360.3	1091.0	0.9127	0.3195	1.2322	3500
2600 . 2700	673.91 679.54	0.0295 0.0305	0.0924 0.0818	0.1219 0.1123	743.1 756.1	337.0 312.2	1090.1 1068.3	0.9232 0.9342	0.2973 0.2740	1.2205 1.2082	3600 3700
2800 2900	684.98 690.26	0.0316 0.0329	0.0716	0 1032 0.0941	770.0 785.2	284.6 252.9	1054. 6 1038.1	0.9458 0.9586	0.2486	1.1944	2500 2900
3000	695.37	0 0346	0.0503	0.0349 +	802.6	216.7	1019.3	0.9731	0.1876	1.1607	3000 3100
3100 8200	700.29 705.04	0.0372 0.0443	0.0390 0.0153	0 0752 0 0596	824.6 871.3	169.4 75.3	994.0 946.6	0.9916 1.0311		1.1376 1.0958	3100 3200
2204.2*	705.34	0.0541	0.0155	0.0541	910.3	6	210.3	1.0645		1.0645	3206.3°

Critical pressure

TABLE XXV continued

Superheated Steam

Abs. Pres Lb./8q. I Set. Tem	e. n.	Set.	Bat. Steam	TEMPE 200°	BATURD 250°	-Dean	EES PAI	ERENHEI 400°	T450*	500*	600*	700*	800*	909*	1000°	1100*	1300*
								298.24 511.7 1240.2 2.1694			498.24 630.9 1333.9 2.2673	598.24 690.6 1382.1 2.3107	698.24 750.2 1431.0 2.3512	798.24 809.8 1480.8 2.3892	898.24 869.4 1531.4 2.4251	998.24 929.1 1583.0 2.4592	1098.24 988.7 1635.4 2.4918
6 (162.25)	Sh b b	0.0164 130.13 0.2347	73.600 1130.8 1.8437	87.75 78.17 1148.8 1.6710	87.75 84.24 1171.1 1.9043	137.75 90.21 1193.6 1.9349	187.75 96.26 1216.6 1.9642	237.75 102.19 1239.8 1.9920	287.75 108.23 1263.0 2.0182	337.75 114.16 1286.1 2.0429	437.75 126.11 1333.5 2.0898	537.75 138.05 1381.8 2.1333	637.75 149.99 1430.8 2.1738	737.75 161.91 1480.6 2.2118	837.75 173.83 1531.3 2.2478	937.75 185.80 1582.9 2.2820	1037.75 197.72 1635.3 2.3146
10 (193.21)	Bb b b	0.0166 161.17 0.2834	38.462 1143.3 1.7876	6.79 38.88 1146.7 1.7928				206.79 51.01 1239.3 1.9154									
14.696 (212.00)	Sh b	0.0167 180.07 0.3120	26.828 1150.4 1.7566		88.00 28.44 1169.2 1.7838	1.8148	1.8440	188.00 34.65 1238.9 1.8727	1.8989	1.9238	1.9709	2.0145	2.0551	2.0932	2.1292	2.1034	3.1900
16 (213.03)	Bb V b	0.0167 181.11 0.3135	26.320 1150.7 1.7548		36.97 27.86 1169.2 1.7816	1.8126	1.8424	186.97 33.95 1238.9 1.8705	1.8967	1.9216	1.9687	2.0123	2.0529	2.0910	2.1270	2.1612	3.1436
20 (227.98)		0.0168 196.16 0.3356	20.110 1156.1 1.7315			22.36 1191.1 1.7799	23.91 1214.8 1.8101	172.04 25.43 1238.4 1.8384	26.95 1261.6 1.8646	28.45 1285.0 1.8896	31.46 1332.7 1.9368	34.46 1381.2 1.9805	37.44 1430.3 2.0211	40.43 1480.2 2.0592	43.42 1531.0 2.0952	46.43 1582.6 2.1294	49.41 1635.1 2.1620
(240.07)	b	208.41	16.321 1160.4 1.7137		9.93 16.58 1166.3 1.7221	1.7570	1.7875		1.8422	1.8673	1.9146	1.9584	1.9990	2.037,1	2.0732	2.1074	3.1400
80 (250.34)	Bh b	0.0170 218.83 0.3680	13.763 1164.0 1.6992			1.7000	1.7043	149.66 16.89 1237.4 1.7930	1.5192	1.0444	1.0918	1.8337	1.9703	2.0145	2.0500	2.0845	3.1174
28 (259.28)	6h	0 0171 227.92 0.3807	11.90 7 1167.0 1.6869			1.7156	1.7468	140.72 14.45 1236.9 1.7758	1.8020	1.8274	1.8750	1.9189	1.9596	1.9978	2.0339	2.0681	2.1007
48 (267.24)	8b	0.0172 236.02 0.8919	10.506 1169.7 1.6768					132.76 12.62 1236.4 1.7606	1.7868	1.8123	1.8600	1.9040	1.9447	1.9829	2.0191	2.0533	2.0860
45 (274.45)		0.0172 243.38 0.4019	9.408 1172.0 1.6668				10.50 1211.1 1.7175	125.55 11.20 1235.8 1.7471	1.7734	1.7990	1.8468	1.8908	1.9315	1.9697	2.0059	2.0401	2.0728
80 (281.01)		0.0173 250.09 0.4110	8.522 1174.0 1.6583					118 99 10.06 1235.2 1.7349									
68 (287.07)		0.0178 256.30 0.4193	7.792 1175.6 1.6506			1.0004	1.6938	112.93 9.130 1234.6 1.7240	1.7507	1.7764	1.8244	1.8685	1.9093	1.9475	1.9837	2.0179	2.0512
66 (2 92.71)		0 0174 262 10 0.4271					7.821 1208.5 1.6834	107.29 8.353 1234.0 1.7139	8.882 1257.7 1.7407	9,398 1281.8 1.7665	10,42 1330.4 1.8146	11.44 1379.5 1.8588	12.44 1428.9 1.8996	13.44 1479.0 1.9378	14.44 1530.2 1.9741	15.45 1581.8 2.0083	16.45 1634.4 2.0410
66 (297.97)			6.654 1179.1 1.6376				47.04	102.03 7.696 1233.4 1.7047	147.00	107.08	907.00	907.00	407 00	707 AB	***		007.08
(3 02.92)	•	0 0175 272.61 0.4409	6.210 1180.5 1.6316				6.671 1206.7 1.6647	7.132 7.132 1232.8 1.6960	7.592 1256.7 1.7230	8.036 1281.0 1.7490	8.920 1329.9 1.7974	9.791 1379.0 1.8416	10.65 1428.6 1.8826	11.51 1478.7 1.9208	12.37 1530.0 1.9572	13.24 1581.6 1.9914	14.10 1634.3 2.0241
78 (307.60)	-	0 0178 277 44 0.4472	5 820 1181.6 1.6260				6.210 1205.8 1.6563	92.40 6.644 1232.2 1.6879	7.076 1256.2 1.7150	7.492 1280.6 1.7411	8.319 1329.6 1.7896	9.133 1378.8 1.8339	9.938 1428.4 1.8749	10.74 1478.6 1.9132	11.54 1529.8 1.9495	12.36 1581.5 1.9837	13.16 1634.2 2.0164

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

Abs. Pre Lb./Sq. (Sat. Ten	ee. In. Sat. ap.) Water	Sat. Steam	TEMPE 340°	BATURE-	-Degri	ES FAS	RENHEI 420°	450°	500°	600°	700°	800°	900*	1000*	1100°	1200
80 (312.03)	Sh ▼ 0.0176 h 282.02 • 0.4532	5.476 1183.1 -1.6209	27.97 5.720 1200.0 1,6424	47.97 5.889 1211.0 1.656Q	67.97 6.055 1221.2 1.6683	6.217 1231.5	6.384 1240.3	6.623 1255.7	7.015 1280.2	287.97 7.793 1329.3 1.7825	8.558 1378.5	9.313 1428.2	10.07	10.92 1529.7	11.58 1591.4	1634
85 (316.25)	8h v 0.0176 h 286.40 s 0.4587	5.169 1184.3 1.6159	23.75 5.368 1198.5 1.6339	43.75 5.528 1210.0 1.6481	63.75 5.685 1220.5 1.6608	83.75 5.839 1230.7 1.6728	103.75 5.995 1239.7 1.6831	133.75 6.226 1255.1 1.7003	183.75 6.594 1279.7 1.7266	283.75 7.329 1329.0 4.7754	383.75 8.030 1378.3 1.8198	483.75 8.762 1429.0 1.8609	583.75 9.472 1478.2 1.8992	683.75 10.18 1529.6 1.9357	783.75 10+90 1581.3 1.9699	883.7 11.6 1634. 2.002
90 (320.27)	8h v 0.0177 b 290.57 e 0.4641	1)85.4 1.6113	1.6264	1.6408	1.6538	1.6658	1.6763	1.6935	1.7200		1.8134	1.8546	1.8929	1.9294	1.9636	1.996
95 (324.13)	8h v 0.0177 b 294.58 a 0.4692	4.653 1186.4 1.6070	15.87 4.773 1196.0 1.6191	35.87 4.921 1208.0 1.6339	55.87 5.063 1219.0 1.6472	75.87 5.205 1229.3 1.6593	95.87 5.346 123 l.6 1.6700	125.87 5.552 1254.0 1.6872	175.87 5.886 1278.9 1.7138	275.87 6.547 1323.4 1.762 j	375.87 7.195 1377.8 1.8073	475.87 7.834 1427.7 1.8485	575.87 8.481 1478.0 1.8869	675.87 9.117 1529.4 1.9234	775.87 9.751 1581.1 1.9576	875.8 10.3 1633. 1.990
100 (327.83)	Sh v 0.0177 h 298.43 s 0.4741	1.6028	1.6124	1.6273	1.6409	1.6528	1.6645	1.6814	1.7080	1.7568	1.8015	1.8428	1.8814	1.9177	1.9520	1.984
105 (331.38)	Sh v 0.0178 h 302.13 e 0.4787	4.232 1188.2 1.5988	8.62 4.292 1193.5 1.6055	28.62 4.429 1205.9 1.6208	48.62 4.562 1217.2 1.6344	68.62 4.691 1227.6 1.6466										
11 0 (334.79)	Sh v 0.0178 h 305.69 e 0.4832	1.89.0	1192.2	1204.9	45.21 4.345 1216.4 1.6286	65.21 4.469 1226.9 1.6410	85.21 4.592 1236.9 1.6525	115.21 4.773 1252.4 1.6698	165.21 5.069 1277.5 1.6966	265.21 5.643 1327.4 1.7460	365.21 6.208 1377.1 1.7908	465.21 6.765 1427.1 1.8321	565.21 7.319 1477.5 1.8706	665.21 7.869 1529.1 1.9072	765.21 8.417 1580.8 1.9414	865.2 8.96 1633. 1.974
115 (338.08)	Sh v 0.0179 b 309.13 s 0.4875	3.882 1189.8 1.6915	•	21.92 4.022 1203.8 1.6088	41.92 4.146 1215.6 1.6230	61.92 4.266 1226.2 1.6355	4.384 1236.3	4.558 1251.9	4.843 1277.1	261.92 5.393 1327.1 1.7410	5.935 1376.9	6.469 1427.0	6.999 1477.4	7.525 1528.9	8.049 1580.7	8.57 1633.
~ 130 (341.26)	Sh v 0.0179 h 312.46 e 0.4916	8.728 1190.6 1.5879		18,74 3.845 1202.1 1.6028	38 74 3.963 7 1214 7 3 1.6173	58 74 4.079 1225.4 1.629	78.74 9 4.194 1235.1 9 1 6417	108.74 4.36 7 1251.4 7 1.659	158.7 4 63 1276 2 1 686	258,74 5 5.165 7 1326.8 3 1.7359	358.74 5.685 1376.7 1.7809	458.74 6.197 1426.8 1.8223	558.74 6.705 1477.2 1.8608	658.74 7.210 1528.8 1.8974	758.74 7.713 1580.6 1.9317	858.7 2.21 1633. 1.964
125	Bh v 0.0179 h 315.69 e 0.4956	1191.3			3.796 1213.7 1.6119	3 908 1224.5 1.6246	4 019 5 1235 0 5 1.6367	4.181 1250.8 1.6544	1276 1.081	255.66 4.954 3 1326.5 7 1.7314	5.454 1376.4 1.7764	5.947 1426.6 1.8179	6.435 1477.1 1.8565	1528.7 1.8931	7.403 1580.5 1.9274	7.88 1633. 1.960
130 (347.31)	8h v 0.0180 h 318.81 e 0.4995	8.45 5 1192.0 1.5815		12,69 3,528 1200.4 1,5918	32.69 3.641 1212.7 1.6066	52.69 3.750 1223.6 1.6194	72 69 3 857 1234.3 1.6317	1.0490	1.0108	252.69 4.760 1326.1 1.7267	1.7710	1.0134	1.6020	1.0001	1,5200	2.000
135 (350.21)	Bh v 0.0180 b 321.96 s 0.5032	8.333 1192.7 1.5784	-		3,497 1211.7 1.6015	3.603 1222.7 1.6144	1233.6 1.6269	1.0419	1.0/24	249.79 4.580 1325.9 1,7223	1./0/4	1.6090	1.0110	1.0013	1.9100	1.501
140 (353.03)	Sh v 0.0180 b 324.83 s 0.5069	3.220 1193.3 1.6755		6.97 3.258 1198.0 1.5813	26.97 3.364 1210.6 1,5965	46.97 3.467 1221.8 1.6097	66.97 3.567 1232.9 1.6225	96.97 3.715 1249.1 1.6406	146.97 •3.954 1275.0 1.6683	246.97 4.413 1325.5 1.7183	346.97 4.862 1375.7 1.7635	446.97 5.303 1426.0 1.8051	546.97 5.741 1476.6 1.8437	646.97 6.175 1528.4 1.8804	746.97 6.607 1580.2 1.9147	846.9 7.03 1633. 1.947
145 (355.76)	Sh v 0.0181 h 327.71 n 0.5104	8.114 1193.9 1.5726		4.24 8.136 1196.7 1.6760	24.24 3.240 7209.5 1.5914	44.24 3.340 1220.9 1.6048	64.24 3.438 1232.2 1.6178	94.24 3.581 1248.5 1,6360	144.24 3.812 1274.5 1.6638	244.24 4.257 1325.1 1.7139	344.24 4.692 1375.4 1.7592	444.24 5.119 1425.8 1.8009	544.24 5.541 1476.5 1.8396	644.24 5.961 1528.3 1.8763	744.24 6.378 1580.1 1.9106	*844.2 6.79 1633. 1.943
1 50 (358.43)	8b v 0.0181 b 330.53 e 0.5138	3.016 1194.4 1.5698			21.57 3.124 1208.4 1,5865	41.57 3.221 1220.0 1.6002	7 61.57 3.317 1231.4 1.6133			241.57 4.112 1324.9 1.7101						
185 (361.02)	Sb ▼ 0.0181 b 333.27 ■ 0.6172	2.921 1195.0 1.5671			1.5818	3.110 1219.1 1.5958	3 203 1230.7 1.6091	3.340 1247.5 1.6279	3.558 1273.6 1.6558	238.98 3.976 1324.5 1.7062	1374.9 1.7516	1425.4 1.7933	1476.2 1.8321	1528.0 1.8688	1579.9 1.9032	1632. 1.936
160 (363.55)	8h v 0.0182 h 335.95 e 0.5204	2.834 1195.5 1.5646			16.45 2.913 1206.0 1.5772	36.45 3.006 1218.3 1.5917	56.45 3.097 1230.0 1.£052	£6.45 3.230 1246.9 -1.6241	136.45 3.443 1273.2 1.6522	236.45 3 849 1324.1 1.7026	336.45 4.245 1374.7 1.7482	436.45 4.633 1425.2 1.7899	536.45 5.018 1476.0 1.8287	636.45 5.398 1527.9 1.8655	736.45 5.777 1579.8 1.8999	836.4 6.15 1632. 1.932

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pre Lb./Sq. (Sat. Tem	lo.	Bat. Water	Sat. Steam	Temper	420°	-Dron	460°	ASO°	T ₅₀₀ *	650*	600°	700*	800*	900*	1000*	1100*	1800*
168 (366.01)	h	0.0182 338.55 0.5236	2.752 1195.9 1.6619	33.99 2.909 1217.4 1.5874		73.99 3.084 1241.1 1.6144	3.170 1251.8	113.99 3.251 1262.4 1.6376	3.334 1272 B	3.533	3.729	1374.5	4.491 1425.0	4.864 1475.9	5.234 1527.8	5.601 1579.7	5.967 1632.7
176 (368.42)	h	0.0182 341.11 0.5266	2.674 1196.3 1.5593	31.58 2.816 1216.5 1.5832	51.58 2.903 1228.4 1.5969	71.58 2.988 1240.5 1.6105	3.071 1251.3	111.58 3.151 1261.8 1.6337	3.232 1272.3	3.426 1298.2	3.617 1323.5	3.991 1374.2	4.357 1424.9	4.720 1475.7	5.079 1527.6	5 43A	5.791 1632.7
176 (370.77)	Sh h	0.0182 343.61 0.5296	2.601 1196.7 1.5569	29.23 2.730 1215.6 1.5793	49.23 2.814 1227.6 1.5931	69.23 2.897 1239.9 1.6069	89.23 2.979 1250.8 1.6189	109.23 3.057 1261.3 1.6302	3.136 1271.9	3.325 1297.8	3.510 1323.2	3.875 1374.0	4.231 1424.7	4.584 1475.6	629.23 4.932 1527.5 1.8553	5.279 1579.5	5.625 1632.6
100 (373.08)	8h V h	0.0183 346.07 0.5325	2.532 1197.2 1.5545	26.92 2.648 1214.6 1,5751	46.92 2.731 1226.8 1.5891	66.92 2.812 1239.2 1.6030	86.92 2.892 1250.2 1.6151	106.92 2.968 1260.8 1.6265	126.92 3.045 1271.5 1.6378	176.92 3.229 1297.4 1.6641	226.92 3.410 1322.8 1.6886	326.92 3.765 1373.7 1.7345	426.92 4.112 1424.5 1.7765	526.92 4.455 1475.5 1.8154	626.92 4.794 1527.4 1.8522	726.92 5.132 1579.4 1.8866	826.92 5.468 1632.5 1.9196
188 (3/5.34)	h	0.0183 348.47 0.5354	2.466 1197.6 1.5522	24.66 2.570 1213.7 1,5712	44.66 2.651 1226.0 1.5853	64.66 2.731 1238.4 1.5992	2.809 1249.6	104.66 2.884 1260.3 1.6230	2.958 1271.0	3.139 1297.4	3.816 1322.4	3.661 1373.4	3.999 1424.3	4.333	4.664	4.992 1579.3	5.319 1632.4
190 (377.55)	h	0.0183 350.83 0.5382	2.404 1198.0 1.5501	1212.7	42.45 2.576 1225.1 1.5817	62.45 2.654 1237.7 1.5959	82.45 2.731 1249.0 1.6083	2.804 1259.8	2.877 1270.5	3.053 1296.6	3.225 1322.1	3.563 1373.1	3.893 1424.1	4.218 1475.2	4.540 1527.1	4.860 1579.2	822.45 5.179 1632.3 1.9136
198 (379.70)	h	0 0184 353.13	2.344 1198.4 1.5479	1211.7	40.30 2.505 1224.2 1.5780	60.30 2.581 1237.0 1.5924	80.30 2.656 1248.3 1.6048	100.30 2.728 1259.3 1.6166	120.30 2.799 1270.0 1.6279	170.30 2.972 1296.2 1.6545	220.30 3.140 1321.8 1.6792	\$20.30 3.470 1372.9 1.7252	420.30 3.791 1423.9 1.7673	520.30 4.109 1475.0 1.8063	620.30 4.423 1527.0 1.8432	720.30 4.735 1579.1 1.8777	820.30 5.046 1632.2 1.9107
200 (381.82)	b	0.0184 355 40	2.288 1198.7 1.5457	1210.8	2.437 1223.7	58.18 2.512 1236.3 1.5869	1247.9	98.18 2.656 1258.7 1.6133	2.726 1269.4	2.595 1295.6	3.059 1321.4	3.381 1372.5	3.697 1423.9	4.005 1474.9	4.311 1526.6	4.616 1579.0	1632.1
205 (383.89)	8h	0.0184 357.61	2.235 1199.0 1.5436	16.11 2.297 1209.7 1,5562	2.372 1222.5	56.11 2.446 1235.4 1.5854	1247.1	2.587 1258.2	2.656 1269.0	2.821 1295.4	2.982 1321.0	3.297 1372.4	3.604 1423.5	3.906 1474.7	4.205 1526.8	4.502 1578.9	816.11 4.798 1632.1 1,9052
210 (385.93)	8h	0.0184 359.80	2.183 1199.4 1.5417	14.07 2.237 1208.8 1,5527	34.07 2.311 1221.8 1.5676	\$4.07 2.394 1234.7 1.5821	74.07 2.454 1246.5 1.5951	2.522 1257.7	2.589 1268.5	2.751 1295.0	2.909 1320.7	3.216 1372.1	3.516 1423.3	3.812 1474.6	4.104 1526.6	4.395 1578.8	814.07 4.683 1632.0 1.9025
916 (367.93)	b	0.0185 361.95	2.134 1199.6 1.5395	1207.8	32.07 2.252 1221.0 1.5643	52.07 2.324 1234.0 1.5789	72.07 2.393 1245.9 1.5920	2.460 1257.2	2.526 1268.0	2.685 1294.6	2.839 1320.4	3.140 1371.9	3.433	3.722 1474.4	4.008 1526.5	4.292 1578.7	812.07 4.574 1631.9 1.9000
236 (389.89)	8h	364.05	2.086 1199.9 1.5376	1206.8	30.11 2.196 1220.1 1.5610	50.11 2.267 1233.2 1.5757	70.11 2.335 1245.2 1.5889	90.11 2.400 1256.7 1.6013	110.11 2.465 1267.5 1.6127	160.11 2.621 1294.1 1.6397	210.11 2.772 1320.0 1.6647	310.11 3.067 1371.6 1.7112	410.11 3.354 1422.9 1.7536	510.11 3.637 1474.2 1.7928	610.11 3.916 1526.4 1.8298	710.11 4.193 1578.6 1.8644	810.11 4.469 1631.8 1.8974
225 (201.81)	8h	0.0185 366.11 0.5562	2.042 1200.2 1.5358	8.19 2.072 1205.8 1.5423	25.19 2.142 1219.2 1.5577	48.19 2.212 1232.3 1.5724	68.19 2.279 1244.5 1.5858	89.19 2.344 1256.2 1,5984	2.407 1267.1	2.560 1293.7	208.19 2.708 1319.6 1.6619	2.997 1371.4	3.278 1422.7	3.555 1474.1	3.828 1526.3	4.100 1578.5	808.19 4.369 1631.7 1.8948
220 (293.70)	ь	368.16	1.9989 1200.4 1.5337	6.30 2.021 1204.9 1.5330	26.30 2.091 1218.3 1.5514	46.30 2.160 1231.6 1.5693	66.30 2.226 1243.8 1.5827	86.30 2.289 1255.6 1.5954	2.352 1266.7	2.502 1293.3	2.647 1319.3	2.930 1371.1	3.205 1422.5	3.477	3.744 1526.2	4.010 1578.4	806.30 4.274 1631.6 1.8923
935 (395.56)		3 70.17	1.9573 1200.7 1.6320	1203.9	24.44 2.042 1217.5 1.5513	44.44 2.110 1230.8 1.5662	64.44 2.175 1243.2 1.5798	84.44 2.237 1255.0 1.5925	2.298	2.446 1292.9	2.589 1219 0	2.866 1370 9	3.136 1422 3	3.402 1473.8	2.664 1526.0	3.924 1578.3	804.44 4.182 1631.6 1.8899
948 (397.40)	8h	0.0186 372.16	1.9176 1200.9 1.6301		22.60 1.995 1216.6 1.5482	42.60 2.062 1230.0 1.5633	62.60 2.126 1242.5 1.5770	82.60 2.187 1254.4 1.5898	102.60 2.247 1265.7 1.6017	152,60 2,392 1292,5 1,6289	202.60 2.532 1218.6 1.6541	2.805 1370.5 1.7009	402.60 3.069 1422.1 1.7435	502.60 3.330 1473.6 1.7828	602.60 3.586 1525.9 1.8199	702.60 3.841 1578.2 1.8545	802.60 4.095 1631.5 1.8876
(369.30)	Sh.	0.0186 374.11	1.8797 1201.1 1.5283		20.80 1.950 1215.6 1.5450	40.80 2.015 1229.1 1.5602	60.80 2.078 1241.8 1.5742	2.139 1253.8	2.199 1265.2	2.341 1292.0	200.80 2.479 1318.3 1.6517	2.746 1370.8	3.006 1421.9	3.261 1473.5	3.513 1525.8	3.762 1578.1	4.011 1631.4

Sh = superheat, deg. F.
v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb.
a = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pre	in. Sat.	Sat. Steam	TEMPE 430°	BATURB 440°	-Dran	EES FAI	ERENGE 500°	520°	550*	600*	700*	800°	900°	1600°	1100*	1200*
250 (400.97)	Sh y 0.0187 h 376.04 a 0.5677	1.8431 1201.4	19.03 1.9065 1214.6	39.03 1.9711 1228.3	59.03 2.0334 1241.0	79.03 2.0932 1253.2	99.03 2.1515 1264.7	119.03 2.2085 1274.5	149.03 2.2920 1291.6 1.6238	199.03 2.4272 1317.9 1.6492	299.03 2.6897 1370.0 1.6961	399.03 2.9444 1421.7 1.7388	499.03 3.1949 1473.3 1.7782	599.03 3.4416 1525.6 1.8153	699.03 3.6867 1578.0 1.8500	799.03 3.9299 1631.3 1.8831
256 (402.71)	Sh v 0.0187 b 377.91 s 0.5698	1,5249	1.5388	1.5543	1.5684	1.5816	1.5938	1.6041	1.6212	1.6466	1.6937	1.7364	1.7759	1.8130	1.8477	1.8808
960 (404.43)	Sh ▼ 0.0187 h 379.78 s 0.5720	1.7742 1201.8 1.5233	15.57 1.8246 1212.8 1.5359	35.57 1.8876 1226.6 1.5514	55.57 1.9482 1239.5 1.5656	75.57 2.0063 1252.0 1.5790	95.57 2.0631 1263.6 1,5912	115.57 2.1185 1273.8 1.6017	145,57 2,1991 1290,8 1,6188	195.57 2.3299 1317.1 1.6442	295.57 2.5833 1369.5 1.6914	395.57 2.8289 1421.3 1.7342	495.57 3.0701 1473.0 1.7737	595.57 3.3077 1525.4 1,8109	695.57 3.5437 1577.8 1.8456	795.57 3.7778 1631.1 1.8787
265 (406.12)	Sh v 0.0187 h 381.62 a 0.5741	1.7416 1202.0 1.5217	13.88 1.7858 1211.9 1.5330	33.88 1,8481 1225.7 1,5485	53.88 1.9080 1239.7 1.5628	73.88 1.9654 1251.2 1.5762	93.88 2.0213 1263.0 1.5386	113.88 2.0759 1273.4 1.5993	143.88 2,1554 1290.4 1.6164	193.88 2.2840 1316.9 1.6419	293.88 2.5331 1369.3 1.6892	393.88 2.7744 1421.1 1.7320	493.88 3.0114 1472.9 1.7715	593.88 3.2446 1525.3 1.8087	693.88 3.4761 1577.7 1.8434	793.88 3.7061 1631.1 1.8765
270 (407.79)	8h v 0.0188 h 383.43 a 0.5761															
276 (409.44)	Sh v 0.0188 h 385.22 s 0.5782	1.6798 1202.3 1.5183	10.56 1.7127 1210.0 1.5271	30.56 1.7735 1224.1 1.5429	50.56 1.8318 1237.3 1.5574	70.56 1.8879 1250.0 1.5711	90.56 1.9422 1262.0 1.5837	110.56 1.9956 1272.6 1.5946	140.56 2.0725 1289.5 1.6116	190.56 2.1973 1316.1 1.6373	290.56 2.4382 1368.7 1.6847	390.56 2.6714 1420.7 1.7277	490.56 2.9002 1472.6 1.7673	590.56 3.1253 1525.0 1.8045	690.56 3.3486 1577.5 1.8393	790.56 3.5704 1630.9 1.8724
380 (411.06)	Sh ▼ 0.0188 h 386.99 6 0.5802	1.6504 1202.5 1.5167	8.94 1.6780 1209.0 1.5241	1.7381 1223.2	48.94 1.7957 1236.5 1.5547	1.8512 1249.4	88.94 1.9048 1261.5 1.5813	1.9575 1272.2	2.0334 1289.1	2.1562 1315.7	2.3932 1368.5	2.6226 1420.5	2.8475 1472.4	3.0688 1524.9	3.2883 1577.4	3.5062 1630.8
295 (412.66)	Sh ▼ 0.0188 h 388.74 s 0.5822			1.7040 1222.3 1.5375	1.7610 1235.6 1.5521		1.8687 1260.9 1.5790	1.9207 1271.8 1.5902	1.9955 1288.6 1.6071		2.3499 1368.2 1.6806	2.5756 1420.3 1.7237	2.7968 1472.2 1.7633	3.0143 1524.7 1.8005	3.2300 1577.3 1.8353	3.4443 1630.7 1.8684
290 (414.24)	6h v 0.0189 h 390.47 n 0.5841	1.5947 1202.9 1,5137	5.76 1.6122 1207.0 1.5184	25.76 1.6710 1221.4 1.5346	45.76 1.7273 1234.8 1.5493	65.76 1.7815 1248.0 1.5635	85.76 1.8338 1260.4 1.5766	105.76 1.8853 1271.4 1.5879	135.76 1.9590 1288.2 1,6048	185.76 2.0783 1315.0 1.6307	285.76 2.3080 1367.9 1.6784	385.76 2.5302 1420.1 1.7215	485.76 2.7478 1472.1 1.7612	585.76 2.9616 1524.6 1.7984	685.76 3.1738 1577.2 1.8332	785.76 3.3844 1630.6 1.8663
996 (415.80)	Sh v 0.0189 b 392.17 a 0.5861	1.5122	1.515/	1.5319	1.5467	1.5611	1.5742	1.5857	1.6026	1.6286	1.6763	1.7195	1.7593	1.7965	1.8313	1.8644
300 (417.33)	Sh ▼ 0.0189 h 393.85 • 0.5879	1.5426 1203.2 1.5107	2.67 1.5506 1205.2 1.5130													
\10 (420.35)	Sh v 0.0189 h 397.16 s 0.5917	1203.5		1217.8	1231.5	1245.3	79.65 1.7054 1258.0 1.5673	1269.6	1286.4	1313.5	1366.9	1419.3	1471.5	1524.1	1576.8	1630.3
320 (423.29)	Sh v 0.0190 h 400.40 n 0.5953	1.4479 1203.8 1.5052		1.5189	1.5342	1.5494	76.71 1.6472 1256.8 1.5629	1.6954 1268.6 1.5751	1.7637 1285.6 1.5922	1.6185	2.0844 1366.3 1.6667	2.2874 1418.9 1.7102	2.4857 1471.2 1.7501	2.6804 1523.8 1.7874	2.8735 1576.6 1.8224	3.0643 1630.1 1.8556
330 (426.16)	Sh v 0.0190 h 403.56 a 0.5988	1204.0					73.84 1.5925 1255.5 1.5582									
\$40 (428.96)	Sh v 0.0191 h 406.65 a 0.6023	1204.2		1.5086	1.5243	1.5399	71.04 1.5409 1254.2 1.5538	1.5666 1.5666	1.583.8	1.6106	1.6591	1.7027	1470.5 1.7428	1.7802	1.8152	1.8485
250 (431.71)	Sh v 0.0191 h 409.70 e 0.6057	1204.4		8.29 1.3472 1210.3 1.5038	28.29 1.3976 1224.8 1.5197	48.29 1.4460 1239.5 1.5355	68.29 1.4923 1252.9 1.5496	88.29 1.5377 1265.5 1.5626	118.29 1.6016 1282.9 1.5801	168.29 1.7041 1310.6 1.6069	268.29 1.8991 1364.7 1.6556	368.29 2.0863 1417.6 1.6993	468.29 2.2687 1470.2 1.7395	568.29 2.4475 1523.0 1.7769	663.29 2.6246 1576.0 1.8120	768.29 2.8000 1629.6 1.8453
340 (434.39)	8h v 0.0192 h 412.67 s 0.6090	1.2889 1204.5 1.4946		5.61 1.3035 1208.5 1,4991	25.61 1.3532 1223.1 1.6151	45.61 1.4008 1238.0 1.5311	65:61 1.4463 1251.5 1.5453	85.61 1.4909 1264.5 1.5587	115.61 1.5536 1282.0 1.5763	165.61 1.6538 1309.9 1.6033	265.61 1.8441 1364.1 1.6521	365.61 2.0266 1417.2 1.6960	465.61 2.2044 1469.9 1.7362	565.61 2.8784 1522.8 1.7737	665.61 2.5506 1575.8 1.8088	765.61 2,7213 1629.4 1,8421

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pro	la.	Sat. Water	Sat. Steam	TEMPE 460°	RATURE 480°	-Deca	226 FAI	TRENEE	T 560*	680*	600*	700*	800*	900°	1000°	1100°	1300°
\$70 (437.01)						62.99 1.4028 1250.2 1.5412	82.99 1.4466 1263.4 1.5548	102.99 1.4881 1275.2 1.5667	122.99 1.5286 1286.7 1.5781	142.99 1.5675 1298.3 1.5894	162.99 1.6063 1309.1 1.5997	262.99 1.7921 1363.6 1.6488	362.99 1.9703 1416.8 1.6928	462.99 2.1435 1469.6 1.7331	562.99 2.3131 1522.5 1.7706	662.99 2.4809 1575.6 1.8058	762.99 2.6471 1629.2 1.8391
250 (439.59)							1.4045 1262.3 1.6510	100.41 1.4452 1274.2 1.5630	1.4850 1286.0 1.5747	1.5232 1297.5 1.5859	1.5612 1308.4 1.5963	1.7428 1363.0 1.6455	1.9168 1416.4 1.6896	2.0859 1469.2 1.7299	2.2512 1522.2 1.7675	2.4148 1575.4 1.8027	2.5768 1629.1 1.8361
\$90 642.11)								97.89 1.4046 1273.2 1.5593									
400 (444.58)								95.42 1.3660 1272.4 1.6561									
410 (447.00)	•	0.6244	1.4828	1.4933	1.5101	1.5252	1.5399	93.00 1.3291 1271.2 1.5524	1.5040	1.5759	1.5865	1.6362	1.0800	1.7212	1.7589	1.7943	1.8277
420 (449.38)	•	0.6278	1.4805	1.4892	1.5000	1.0218	1.5361	90.62 1.2942 1270.2 1.5489	1.5612	1.5726	1.5832	1.6331	1.6776	1.7184	1.7561	1.7915	1.8249
430 (451.72)								88.28' 1.2607 1269.1 1.5455									
440 (454.01)	•	Ų.633U	1.4762	1.4812	1.4981	1.5138	1.5291	85.99 1.2288 1268.0 1.5422	1.5550	1.5664	1.5772	1.6275	1.6722	1.7130	1.7508	1.7862	1.8197
450 (456.27)	8h 4	0.0195 437.18 0.6357	1.0318 1205.0 1.4739	3.73 1.0401 1207.9 1.4771	23.78 1.0824 1223.7 1.4941	43.73 1.1230 1238.7 1.5099	63.73 1.1617 1253.8 1.5255	83.73 1.1982 1266.9 1.5387	103.78 1.2337 1280.0 1.5517	123.73 1.2681 1291.8 1.5632	143.73 1.3013 1303.1 1.5740	243.73 1.4593 1359.1 1.6245	343.73 1.6092 1413.4 1.6694	443.73 1.7539 1467.0 1.7103	543.73 1.8951 1520.3 1.7481	643.73 2.0345 1573.9 1.7836	743.78 2.1720 1627.8 1.8171
460 (458.48)	8h 4 6	0.0196 439.69 0.6384	1.0092 1205.0 1.4719		21.52 1.0545 1222.0 1.4902	41.52 1.0946 1237.2 1.5062	61.52 1.1329 1252.5 1.5220	81.52 1.1690 1265.8 1.5354	101.52 1.2039 1279.0 1.5485	121.52 1.2379 1291.0 1.5602	141.52 1.2706 1302.3 1.5710	241.52 1.4258 1358.6 1.6217	341.52 1.5729 1413.0 1.6667	441.52 1.7147 1466.6 1.7076	541.52 1.8530 1520.0 1.7455	641.52 1.9896 1573.7 1.7811	741.52 2.1243 1627.7 1.8146
470 (460.66)	8h 4 6	0.0196 442,17 0.6411	0.9875 1205.0 1.4699		1220.2	1235.7	1.518 5	79.34 1.1410 1264.7 1.5321	1.5453	1.6570	1301.5 1.5680	1.6189	1412.5 1.6639	1466.3 1.7050	1.7429	1573.5 1.7785	1.8120
480 (462.80)	ь	444.60	0.9668 1205.0 1.4679		17.20 1.0021 1218.6 1.4825	37.20 1.0416 1234.2 1.4989	57.20 1.0789 1249.9 1.5151	77.20 1.1141 1263.5 1.5288	97.20 1.1482 1277.0 1.5422	117.26 1.1813 1289.1 1.5539	137.20 1.2131 1300.8 1.5650	237.20 1.3630 1357.5 1.6161	837.20 •1.5049 1412.1 1.6612	437.20 1.6413 1466.0 1.7023	537.20 1.7742 1519.5 1.7402	637.20 1.9054 1573.3 1.7758	737.20 2.0347 1627.3 1.8093
490 (464.91)	8b	0.0197 447.00 0.6462	0.9466 1204.9 1.4659				1.0535 1248.4 1.5116	75.09 1.0884 1262.3 1.5256	1.1220 1276.0 1.5392	1.1548 1288.3 1.5511	1.1860 1300.0 1.5622		1.4729 1411.7 1.6588	1.5067 1465.6 1.6999	1.7371 1519.2 1.7379	1.8659 1573.1 1.7736	1.9927 1627.1 1.8071
800 (467.00)	84	0.0197 449.40 0.6488	0.9274 1204.9 1.464]		1.4/52	1.4922	1.5079	73.00 1.0636 1261.1 1.5225	1.5363	1.5482	1.5596	1.6110	1.6564	1.6975	1.7356	1.7714	1,8052
\$10 (469.05)	•	0.6518	0.9090 1204.8 1.4621		10.95 0.9310 1213.5 1.4714	30.95 0.9695 1229.6 1.4883	50.95 1.0056 1245.6 1.5048	70.95 1.0397 1259.9 1.5192	90.95 1.0727 1274.0 1.5332	110.95 1.1046 1286.6 1.5454	130.95 1.1350 1298.4 1.5566	230.95 1.2780 1355.7 1.6082	330.95 1.4127 1410.9 1.6538	430.95 1,5418 1465.0 1,6951	530.95 1.6675 1518.7 1.7332	630.95 1.7915 1572.6 1.7689	730.95 1.9135 1626.8 1.8026
050 (471.07)		0.0198 454.07 0.6537	0.8912 1204.7 1.4601			0.9472 1228.1 1.4849		68.93 1.0169 1258.6 1.5160	88.93 1.0494 1272.9 1.5302	108.93 1.0810 1285.6 1.5425	128.93 1.1110 1297.6 1.5539	228.93 1.2519 1355.1 1.6057	328.93 1.3844 1410.4 1.6514	428.93 1.5113 1464.6 1.6928	528.93 1.6347 1518.4 1.7310	628.93 1.7565 1572.4 1.7668	728.93 1.8763 1626.6 1.8005
(473.05)	•	0.0199 456.35	-0.8741 1204.6 1.4584		6.95 0.8879 1210.0 1.4642	26.95 0.9258 1226.5 1.4816	46.95 0.9612 1242.8 1.4984	66.95 0.9948 1257.3 1.5130	86.95 1.0269 1271.8 1.5274	106.95 1.0582 1284.8 1.5400	126.95 1.0878 1296.8 1.5514	226.95 1.2267 1354.6 1.6035	326.95 1.3571 1410.0 1.6493	426.95 1.4818 1464.3 1.6908	526.95 1.6031 1518.1 1.7290	626.95 1.7228 1572.2 1.7648	726.95 1.8402 1626.4 1.7985

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Superheated Steam—(Continued)

Abs. Pre	88.																
Lb./Sq. (Sat. Ten	ln.	Sat. Water	Sat. Steam	TEMPE: 500°	TATURE-	DEGR 540°	560°	580°	[*] 600°	650°	700°	750°	800°	900°	1000°	1100°	1200°
540 (475.02)	Sh v h s	0.0199 458.62 0.6585	0.8576 1204.5 1.4565	24.98 0,9051 1225.0 1,4781	44.98 0.9401 1241.4 1.4950	64,98 0.9736 1256.1 1.5098	1.0054 1270.7	104.98 1.0363 1283.8 1.5370	1.0655 1296.0	1.1356 1325.6	1.2025 1354.0	1.2671 1382.1	1.3309 1409.6	1.4535 1463.9	1.5727 1517.8	624.98 1.6903 1572.0 1.7625	724.98 1.8056 1626.2 1.7962
550 (476.94)	Sh v h s	0.0199 460.83 0.6609	0.8416 1204.4 1.4548	23.06 0.8851 1223.4 1.4748	43.06 0.9198 1240.0 1.4919	63.06 0.9530 1254.8 1,5068	83.06 0.9846 1269.6 1.5215	103.06 1.0151 1282.9 1.5344	123.06 1.0441 1295.2 1.5461	173.06 1.1132 1324.9 1.5735	223.06 1.1791 1353.5 1.5987	273.06 1.2428 1381.6 1.6224	323.06 1.3055 1409.2 1.6447	423.06 1.4262 1463.6 1.6862	523.06 1.5434 1517.5 1.7244	623.06 1.6590 1571.7 1.7603	723.06 1.7724 1626.0 1.7940
560 (478.85)	Sh v h s	0.0200 463.04 0.6632	0.8253 1204.3 1.4530	21.15 0.8658 1221.8 1.4714	41,15 0.9003 1238.5 1.4886	61.15 0.9332 1253.5 1.5038	81.15 0.9644 1269.5 1.5187	101.15 0.9947 1232.0 1 5318	121.15 1.0233 1294.4 1.5436	171.15 1.0917 1324.2 1.5711	221.15 1.1566 1352.9 1.5964	271.15 1.2193 1381.1 1.6292	321.15 1.2810 1408.7 1.6425	421,15 1,3998 1463,2 1,6841	521.15 1.5151 1517.2 1.7224	621,15 1.6289 1571.5 1.7584	721.15 1.7403 1625.8 1.7921
570 (480.73)	Sh V h	0.0200 465.22	0.8114 1204.1	19.27 0.8472 1220.2	39.27 0.8814 1236.9 1.4853	59.27 0.9141 1252.2	79.27 0.9450 1267.3	99.27 0.9749 1281.0	119.27 1.0033 1293.5	169.27 1.0708 1323.5	219.27 1.1348 1352.3	269.27 1.1966 1380.6	319.27 1.2575 1408.3	419.27 1.3744 1462.9	519.27 1.4879 1517.0	619.27 1.5998 1571.3	719.27 1.7093 1625.6
580 (482.58)	Sh V h s	0.0201 467.37 0.6677	0.7968 1204.0 1.4494	17.42 0.8291 1218.6 1,4648	37.42 0.8631 1235.5 1.4822	57.42 0.8956 1250.9 1.4978	77.42 0.9263 1266.1 1.5128	97.42 0.9558 1280.0 1.5264	117.42 0.9839 1292.6 1.5384	167.42 1.0506 1322.8 1.5662	217.42 1.1137 1351.6 1.5916	267.42 1.1747 1380.0 1.6156	317.42 1.2347 1407.8 1.6381	417.42 1.3498 1462.5 1.6799	517.42 1.4616 1516.7 1.7183	617.42 1.5714 1571.0 1.7543	717.42 1.6794 1625.4 1.7881
590 (484.41)	Sh v h s	0.0201 469.50 0.6699	0.7831 1203.8 1.4477	15.59 0.8116 1217.0 1.4616	35.59 0.8455 1234.0 1.4791	55.59 0.8778 1249.6 1.4949	75.59 0.9082 1265.0 1.5101	95.59 0.9373 1278.9 1.5236	115.59 0.9653 1291.8 1.5359	165.59 1.0310 1322.1 1.5638	215.59 1.0934 1351.1 1.5894	265.59 1.1535 1379.5 1.6134	315.59 1.2128 1407.4 1.6360	415.59 1.3262 1462.2 1.6778	515.59 1.4360 1516.4 1.7162	615.59 1.5442 1570.8 1.7522	715.59 1.6505 1625.3 1.7861
600 (486.21)	Sh v h	0.0201 471.59		13.79 0.7945 1215.6	33.79 0.8284 1232.5	53.79 0.8605 1248.3	73.79 0.8907 1263.7		113.79 0.9471 1290.9	163.79 1.0123 1321.4	213.79 1.0738 1350.5	263. 79 1.1332 1379.0	313.79 1.1915 1407.0	413.79 1.3032 1461.8	513.79 1.4115 1516.0	613.79 1.5179 1570.5	713.79 1.6224
610 (487.99)	Sh v h s	0.0202 473.67 0.6743	0.7565 1203.5 1.4444	12.01 0.7781 1213.8 1.4552	32.01 0.8120 1230.9 1.4728	52.01 0.8436 1246.9 1.4890	72.01 0.8736 1262.5 1.5044	92.01 0.9022 1276.8 1.5183	112.01 0.9296 1290.0 1.5309	162.01 0.9942 1320.6 1.5591	212.01 1.0548 1350.0 1.5850	262.01 1.1135 1378.5 1.6091	312.01 1.1708 1406.5 1.6318	412.01 1.2809 1461.5 1.6738	512.01 1.3878 1515.8 1.7123	612.01 1.4928 1570.3 1.7484	712.01 1.5964 1624.9 1.7823
620 (489.75)	Sh v h	0.0202 475.72 0.6764	0.7438 1203.3 1.4427	10.25 0.7622 1212.2 1.4520	30,25 0,7960 1229.5 1,4698	50.25 0.8275 1245.5 1.4860	70.25 0.8572 1261.3 1.5016	90,25 0.8856 1275.8 1.5157	110,25 0.9127 1289.1 1.5284	160.25 0.9765 1319.9 1.5568	210.25 1.0364 1349.3 1.5827	260.25 1.0943 1377.9 1.6068	310.25 1.1505 1406.1 1.6296	410.25 1.2596 1461.2 1.6717	510.25 1.3648 1515.5 1.7102	610.25 1.4677 1570.1 1.7464	710.25 1.5707 1624.7 1.7803
630 (4 91. 4 9)	Sh v h	0.0202 477.75 0.6785	0.7316 1203.1 1.4410	8.51 0.7466 1210.6 1.4488	28.51 0.7802 1227.8 1.4665	48.51 0.8117 1244.1 1.4830	68.51 0.8413 1260.1 1.4988	88.51 0.8694 1274.7 1.5130	108.51 0.8963 1288.3 1.5260	158.51 0.9595 1319.2 1.5545	208.51 1.0187 1348.7 1.5805	258.51 1.0757 1377.4 1.6047	308.51 1.1312 1405.7 1.6276	408.51 1.2387 1460.8 1.6697	508.51 1.3423 1515.2 1.7083	608.51 1.4445 1569.9 1.7445	708.51 1.5449 1624.5 1.7784
640 (4 93.21)					26.79 0.7651 1226.3 1,4636												
650 (494.90)	Sh v h	0.0203 481.73 0.6826	0.7082 1202.7 1.4379	5.10 0.7171 1207.3 1.4427	25.10 0.7504 1224.8 1.4607	45.10 0.7816 1241.3 1.4774	65.10 0.8107 1257.6 1,4935	85.10 0.8384 1272.5 1.5080	105.10 0.8648 1286.5 1.5213	155.10 0.9269 1317.8 1.5501	205.10 0.9846 1347.6 1.5764	255.10 1.0404 1376.3 1.6006	305.10 1.0944 1404.7 1.6236	405.10 1.1988 1460.1 1.6659	505.10 1.2999 1514.7 1.7046	605.10 1.3987 1569.4 1.7408	705.10 1.4958 1624.1 1.7748
560 (496.58)	Sh v h	0.0204 483.77 0.6847	0.6969 1202.5 1.4363	3.42 U.7031 1205.7 1.4396	23.42 0.7361 1223.2 1.4576	43,42 0,7672 1240.0 1,4746	63.42 0.7962 1256.4 1.4908	83.42 0.8237 1271.4 1.5054	103.42 0.8499 1285.5 1.5188	153.42 0.9113 1317.1 1.5479	203.42 0.9686 1347.0 1.5742	253.42 1.0234 1375.8 1.5985	303.42 1.0769 1404.3 1.6216	403.42 1.1803 1459.7 1.6639	503.42 1.2797 1514.4 1.7027	603.42 1.3774 1569.2 1.7390	703.42 1.4727 1624.0 1.7730
670 (498.23)	Sh	0.0204 485.61 0.6867	0.6861 1202.3 1.4349	1.77 0.6892 1204.0 1.4367	21.77 0.7224 1221.7 1.4549	41.77 0.7531 1238.7 1.4721	61.77 0.7820 1255.1 1.4883	81.77 0.8093 1270.2 1,5030	101.77 0.8354 1284.5 1.5166	151.77 0.8963 1316.3 1.5459	201.77 0.9527 1346.3 1.5723	251.77 1.0072 1375.3 1.5968	301.77 1.0599 1403.9 1.6200	401.77 1.1617 1459.4 1.6624	501.77 1.2600 1514.1 1.7012	601.77 1.3560 1569.0 1.7376	701.77 1.4503 1623.8 1.7716
680 (499.87)	Ŀ	0.0204 487.64	0.6757 1202.1 1.4332		20.13 0.7089 1220.2 1.4519	0.7397	0.7683 1253.9	80.13 0.7954 1269.1 1,5004	0.8212 1283.6	0.8814 1315.6	0.9375 1345.8	0.9912 1374.7	1.0432 1403.4	400.13 1.1440 1459.0 1.6603	1.2408 1513.8	1.3357 1568.7	1.4283 1623.6
690 (501.49)	Sh	0.0205 489.56	0.6652 1201.8 1,4316		18.51 0.6956 1218.5 1.4488	38.51 0.7263 1235.8 1.4663	58.51 0.7549 1252.5 1.4828	78.51 0.7818 1263.0 1.4978	98.51 0.8075 1282.7 1.5118	148.51 0.8675 1314.9 1.5415	198.51 0.9225 1345.1 1.5681	248.51 0.9758 1374.2 1.5927	298.51 1.0272 1402.8 1.6159	398.51 1.1267 1458.7 1.6586	498.51 1.2223 1513.6 1.6975	598.51 1.3162 1568.5 1.7339	698.51 1.4075 1623.4 1.7680
		٠			16,91 0.6830 1217,1 1.4461	0.7133	∩ 741Q	76.91 0.7687 1266.8 1.4953	0.7041	0.8534	. 0 0084	U 06U8	1 0117	396.91 1.1096 1458.2 1.6567	1 2043	1 2065	1.3870
				-													

heat, deg. F. ac volume, cu. ft. per lb.

31 324 3206.2*

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pres Lb./8q. Id (Sat. Tem)	p.) \	Bat. Water	Bat. Steam	820*		860°	580°	e00°	630"	650*	700*	750°	800°	\$00°		1100*	
726 (507.01)		0.0206 496.2 0.6973	0.6314 1200.9 1.4263	12.99 0.6524 1212.8 1.4385	32.99 0.6827 1230.9 1.4568	52.99 0.7109 1248.0 1.4737	72.99 0.7373 1264.0 1.4892	92.99 0.7624 1279.1 1.5036	112.99 0.7864 1293.8 1.5173	142.99 0.8203 1312.3 1.5342	192.99 0.8740 1343.0 1.5612	242.99 0.9250 1372.3 1.5859	292,99 0,9745 1401,3 1,6094	392.99 1.0697 1457 4 1.6522	492.99 1.1612 1512.5 1.6913	592.99 1 2511 1567.7 1.7279	692.99 1.3383 1622.7 1.7621
750 (510.83)	3b	9.0207 500.8 9.7019	0.6091 1200.2 1.4225	9.17 0.6237 1208.8 1,4313	29.17 0.6538 1227.4 1.4501	49.17 0.6818 1244.9 1.4674	69.17 0.7080 1261.0 1.4830	89.17 0.7326 1276.6 1.4979	109.1 7 0.7561 1291.4 1.5117	139.17 0.7896 1310.5 1.5291	189.17 0.8419 1341.5 1.5564	239.17 0.8918 1371.0 1.5813	289.17 0.9399 1400.2 1.6049	389.17 1.0326 1456.5 1.6479	489.17 1.1212 1511.8 1.6871	589.17 1.2078 1567.1 1.7237	689.17 1.2928 1622.3 1.7580
776 (614.57)	8b 4 (0.0208 505.3 0.7064	0.5882 1199.5 1.4189	5.43 0.5969 1204.7 1.4242	25.43 0.6268 1223.7 1.4434	45.43 0.6545 1241.5 1.4610	65.43 0.6803 1258.2 1.4772	85.43 0.7047 1271.9 1.4902	105.43 0.7278 1289.0 1.5062	135.43 0.7607 1308.6 1.5241	185.43 0.8119 1340.0 1.5518	235.43 0.8606 1369.7 1.5769	285.43 0.9073 1399.0 1.6006	385.43 0.9977 1455.6 1.6438	485.43 1.0838 1511.1 1.6832	\$85.43 1.1676 1566.6 1.7200	685.43 1.2505 1621.8 1.7543
900 (518.20)	Bh D	0.0209 509.7 0.7108	0.5685 1198.8 1.4155	1.80 0.5714 1200.3 1.4170	21.80 0.6013 1220.0 1.4369												
828 (521.75)	Bb •	0.0210 514.0 0.7152	0.5500 1198.0 1.4121		18.25 0.5774 1216.1 1.4304												
880 (826.23) ·	Bb •	0.0210 518.3 0.7194	0.5326 1197.2 1.4087			1.4429	1.4600	1,4758	1.4905	1.5101	1.5386	1.5643	1.5883	1.6321	1.8720	1.7090	1./435
978 (528.62)	85	0.0211 522.4 0.7236	0.5162 1196.4 1.4056														671.38 1.1045 1619.9 1.7403
(531.94)	•	0.7276	0.5006 1195.6 1.4022		1.4106	1.4306	1.4484	1.4649	1.4803	1.5004	1.5296	1.5559	1.5801	1.0245	1.6647	1.7017	668.05 1.0727, 1619.3 1.7362
925 (536.20)	8b 7 b	0.0213 530.6 0.7316	0.4858 1194.7 1.3991	•													664 80 1.0432 1618.9 1.7337
988 (538.38)	Bh T	0.0214 534.6 0.7355	0.4717 1193.8 1.3960	•	1.62 0.4741 1195.8 1.3980												661.62 1.0148 1618.5 1.7305
975 (541.50)	8h	0.0215 538.5 0.7393	0.4583 1192.9 1.3929)		1.4128	1.4322	1.4433	1.4039	1.65/3	1.31/6	1.5449	1.5695	1.0145	1.0221	J.6928	658.50 0.9880 1618.0 1.7274
1 000 (544.56)	83		0.4456 1191.6 1.3896														655.44 0.9626 1617.5 1.7244
1028 (547.57)	Bb	0.0217	0.4334 1191.0 1.387	} •													652.43 0.9384 1617.0 1.7216
1 050 (550.52)	Bb	8.0216 \$50.0 8.750	0.4211 1190.0 1.383	}		1.3942	1.4100	1.4341	1.4312	1.4/39	1.3038	1.5330	1.5590	1.6047	1.6458	1.6833	649.48 0.9354 1616.5 1.7184
1978 (553.42)		853.1 0.754	0.410: 7 1188: 9 1.38:			0.4198 1196.3 1.3883	0.4446 1218.7 1,4100	0.4672 1238.8 1.4292	0.4879 1257.2 1.4464	0.5169 1282.8 1.4698	0.5592 1319.4 1.5020	0,5980 1352.6 1,5300	0.6349 1384.1 1.5555	0.7042 1444.3 1.6015	0.7696 1502.5 1.6428	0.8322 1559.3 1.6804	646.58 0.8933 1615.9 1.7156
1100 (856.26)	61	0.021 657.6 0.767	0.400 1187. 1.378	2 8 0		3.76 0.4056 1192.3 1.3826											643.74 0.8724 1615.4 1.7128
1196 (650.07)		0,023 561 0.761	0.390 0 1186. 0 1.375	;			20.93 0.4167 1211.3 1.3991	40.93 0.4392 1232.3 1.4190	60.93 0.4598 1251.8 1.4370	90.93 0.4883 1277.7 1.4610	140.93 0.5301 1315.9 1.4946	190.93 0.5678 1349.6 1,5231	240.93 0.6036 1381.7 1.5491	340.93 0.6706 1442.6 1.5956	440.93 0.7333 1500.8 1.6369	540.93 0.7938 1558.2 1.6749	640.93 0.8524 1614.9 1.7101

Sh = superheat, deg. F.
v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb.
s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pr. Lb./Sq. (Sat. Ter	mp.)	Water								T00*	720*	750*	800°	900*	1000*	1100*	1200
11 50 (561.81)	Sh b	0.0221 564.6 0.7644	0.3804 1185.6 1.3723	18.19 0.4035 1207.4 1,3934	38.19 0.4259 1228.7 1.4137	58.19 0.4468 1248.6 1.4323	78.19 0.4659 1266.9 1.4491	98.19 0.4839 1284.4 1.4649	118.19 0.5005 1299.5 1.4783	138,19 0.5165 1314.1 1.4910	158.19 0.5317 1327.9 1.5028	188.19 0.5537 1348.1 1.5197	238.19 0.5889 1380.4 1.5458	338.19 0.6549 1441.7 1.5926	438.19 0.7166 1500.2 1.6341	538,19 0.7760 1557.8 1.6723	638.19 0.8333 1614.5 1.707.5
1178 (564.54)	Sh b	0.0222 568.2 0.7678	0.3710 1184.4 1.3694	15.46 0.3911 1203.6 1.3880	35.46 0.4133 1225.4 1.4088	55.46 0.4339 1245.4 1.4275	75.46 0.4530 1264.6 1.4451	95.46 0.4708 1281.5 1.4603	115.46 0.4874 1297.3 1.4743	135.46 0.5032 1312.1 1.4872	155.46 0.5183 1326.1 1.4991	185.46 0.5400 1346.4 1.5161	235.46 0.5747 1379.0 1.5425	335.46 0.6396 1440.6 1.5896	435.46 0.7005 1499.4 1.6313	535.46 0.7586 1557.0 1 6694	635.46 0.8149 1614.0 1 7048
1300 (567.19)	Sh b	0.0223 571.7 0.7712	0.3620 1183.2 1.3667	12.81 0.3793 1200.2 1.3831	32.81 0.4013 1222.1 1.4040	52.81 0.4219 1242.6 1.4232	72.81 0.4408 1261.5 1.4405	92.81 0.4585 1279.2 1.4565	112.81 0.4750 1295.3 1.4707	132.81 0.4907 1310.3 1.4838	152.81 0.5056 1324.4 1.4958	182.81 0.5271 1345.0 1.5131	232.81 0.5613 1377.7 1.5395	\$32.81 0.6251 1439.5 1.5867	432.81 0.6853 1499.0 1.6289	532.81 0.7423 1556.6 1.6671	632.81 0.7975 1613.6 1.7025
1 225 (569.82)	8b V b	0.0224 575.1 0.7745	0.3534 1182.0 1.3640	10.18 0.3669 1195.6 1.3771	30.18 0.3895 1218.6 1.3990	50.18 0.4102 1239.7 1.4188	70.18 0.4290 1258.8 1.4363	90.18 0.4466 1276.7 1.4524	110.18 0.4631 1293.3 1.4671	130,18 0.4786 1308.3 1.4802	150.18 0.4934 1322.7 1.4925	180.18 0.5146 1343.3 1.5097	230.18 0.5483 1376.4 1.5365	330.18 0.6113 1438.6 1.5840	430.18 0.6702 1498.0 1.6262	530.18 0.7264 1556.0 1.6646	630.18 0.7806 1613.0 1.7000
1250 (572.39)	Sh b	0.0225 578.6 0.7777	0.3453 1180.8 1.3612	7.61 0.3549 1190.9 1.3710	27.61 0.3782 1215.0 1.3939	47.61 0.3991 1236.7 1.4142	67.61 0.4177 1256.0 1.4319	87.61 0.4354 1274.5 1.4486	107.61 0.4517 1291.2 1.4634	127.61 0.4672 1306.6 1.4768	147.61 0.4817 1321.0 1.4891	177.61 0.5027 1341.9 1.5066	227.61 0.5360 1375.2 1.5335	\$27.61 0.5980 1437.7 1.5813	427.61 0.6558 1497.2 1.6235	527.61 0.7113 1555.4 1.6620	627.6 0.7644 1612.6 1 6975
1276 (574.93)	8b V b	0.0226 582.0 0.7809	0.3371 1179.5 1.3584	5.07 0.3437 1186.5 1.3651	25.67 0.3672 1211.2 1.3887	45.07 0.3881 1233.5 1.4095	65.07 0.4068 1253.3 1.4277	85.07 0.4244 1272.0 1.4445	105.07 0.4406 1289.0 1.4596	125.07 0.4560 1304.6 1.4732	145.07 0.4705 1319.4 1.4858	175.07 0.4912 1340.3 1.5033	225.07 0.5241 1373.9 1.5305	325.07 0.5852 1436.7 1.5785	425.07 0.6420 1496.3 1.6207	525.07 0.6968 1554.7 1.6594	625.07 0.7488 1612.0 1.6950
1300 (577.43)	8b 8 8	0.0227 585.4 0.7840	0.3294 1178.3 1.3557	2.57 0.3329 1182.0 1.3593	22.57 0.3567 1207.7 1.3837	42.57 0.3776 1230.3 1.4049	62.57 0.3965 1250.6 1.4235	82.57 0.4140 1269.6 1.4406	102.57 0.4301 1287.0 1.4560	122.57 0.4453 1302.8 1,4698	142.57 0.4598 1317.8 1.4826	172.57 0.4803 1338.9 1.5002	222.57 0.5127 1372.6 1.5275	322,57 0.5730 1435.8 1.5758	422.57 0.6290 1495.7 1.6183	522.57 0.6826 1554.2 1.6570	622.57 0.7340 1611.6 1.6926
1325 (579.89)	h	588.7	0.3220 1177.0 1.3530		1.0700	1,1005	*.4100	1.4000	4.4020	1.1000	140.11 0.4493 1315.9 1.4792	1.10/1	1.3240	1.0/31	1.0108	1.0040	1.6903
1350 (582.32)	Sh V h	0.0229 592.1 0.7902	0.3147 1175.8 1.3504		17.68 0.3363 1200.0 1.3734	37.68 0.3576 1223.8 1.3957	57.68 0.3766 1245.0 1.4151	77.68 0.3940 1264.6 1.4328	97.68 0.4105 1283.2 1.4492	117.68 0.4252 1299.1 1.4631	137.68 G.4393 1314.2 1.4760	167.68 0.4594 1335.8 1.4941	217.68 0.4911 1370.0 1.5218	317.68 0.5497 1433.8 1.5705	417.68 0.6042 1494.3 1.6134	\$17.68 0.6559 1552.8 1.6521	617.68 0.7057 1610.6 1.6681
1878 (584.71)	Sh V h	0.0230 595.3 0.7932	0.3078 1174.5 1.3477								135.29 0.4295 1312.3 1.4726						
1400 (587.07)	8h 4 4	0.0231 598.6 0.7963	0.3011 1173.2 1.3452								132.93 0.4203 1310.6 1.4696						
1425 (589.40)	Sb V b	0.0232 601.8 0.7992	0.2947 1171.8 1.3425								130.60 0.4112 1308.9 1.4665						
1450 (591.70)	h	605.0	0.2885 1170.5 1.3401								128.30 0.4025 1307.0 1.4634						
1475 (593.97)	8b 8 6	0.0234 608.2 0.8052	0.282 4 1169.1 1.3375		1.3400	1.3723	1.3939	1.4135	1.2308	1.1109	126.03 0.3939 1305.0 1.4601	1.9/09	1.9081	1.5582	1.0010	1.0414	1.0//4
1 500 (596.20)	Sh H	0.0235 611.4 0.8081	0.2765 1167.7 1.3350								123.80 0.3856 1303.0 1.4569						
1 525 (598.41)	Sh b s	0.0236 614.5 0.8109	0.2708 1166.2 1.3323								121.59 0.3777 1301.2 1.4538						
1550 (600.59)	h	0.0237 617.7 0.8138	0.2653 1164.8 1.3298	÷		19.41 0.2883 1194.5 1.3576	39.41 0.3084 1220.5 1.3814	59.41 0.3261 1243.3 1.4020	79.41 0.3420 1264.0 1.4203	99.41 0.3567 1282.7 1.4365	119.41 0.3702 ,1299.6 1.4510	149.41 0.3891 1323.0 1.4706	199.41 0.4185 1359.4 1.5001	299.41 0.4719 1426.2 1.5511	399.41 0.5208 1488.3 1.5951	499.41 0.5669 1548.2 1.6348	599.41 0.6113 1607.0 1.6713

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Abs. Pre		901		Trupe	BATTER	-Dear	EER FAI		iT.								
Abs. Pre Lb./8q. (Sat. Ten	1p.)	Water	Bteam	630°	640	660°	680°	700°	720*	740*	760*	790*	900°	900*	1000*		1200°
1878 (602.74)	8b	0.0238 620.8 0.8166	0.2599 1163.4 1.3273	17.26 0.2804 1190.3 1.3524	37.26 0.3008 1216 9 1,3768	57.26 0.3186 1240.3 1.3979	77.26 0.3345 1261.3 1.4165	97.26 0 3492 1280.5 1.4332	117.26 0.3627 1297.6 1.4478	137.26 0.3755 1313.5 1.4612	157.26 0.3877 1328.8 1.4738	177.26 0.2993 1343.5 1.4858	197.26 0.4105 1357.8 1.4972	297.26 0.4633 1424.9 1.5485	397.26 0.5117 1487.3 1.5927	497.26 0.5573 1547.4 1.6325	597.26 0.6010 1606.2 1.6691
1600 (604.87)	8b 6	0.0239 623.9 0.8195	0.254B 1161.9 1.3249	15.13 0.2730 1186.3 1.3477	35.13 0.2935 1213.7 1.3728	55.13 0.3114 1237.6 1.3943	75.13 0.3274 1258.9 1.4132	95.13 0.3421 1278.4 1.4302	115.13 9.3555 1295.7 1.4449	135.13 0.3682 1311.8 1.4585	155.13 0.3802 1327.3 1.4713	175.13 0.3919 1342.2 1.4834	195.13 0.4031 1356.7 1.4950	295.13 0.4554 1424.1 1.5465	395.13 0.5032 1486.8 1.5909	495.13 0.5482 1547.0 1.6308	595.13 0.5914 1605.8 1.6674
1 028 (606.97)	8b 7 6	0.0240 627.0 0.8222	0.2497 1160.4 1.3223	13.03 0.2656 1182.1 1.3425	33.03 0.2864 1210.2 1.3683	53.03 0.3044 1234.5 1.3902	73.03 0.3203 1256.2 1.4094	93.03 0.3348 1275.9 1.4266	113.03 0.3484 1293.5 1.4416	133.03 0.3610 1310.0 1.4555	153.03 0.3729 1325.5 1.4683	173.03 0.3845 1340.6 1.4806	193.03 0.3957 1355.2 1.4923	293.03 0.4474 1423.0 1.5440	393.03 0.4948 1486.0 1.5887	493.03 0.5391 1546.3 1.6287	593.03 0.5816 1605.2 1.6652
1656 (609.05)	8h Y	0.0241 630.0 0.8250	0.2448 1158.8 1.3198	10.95 0.2583 1177.6 1.3373	30.95 0.2794 1206.8 1,3641	50.95 0.2976 1231.7 1.3865	70.95 0.3136 1253.7 1.4060	90.95 0.3280 1273.6 1.4233	110.95 0.3417 1291.8 1.4389	130.95 0.3542 1308.4 1.4528	150.95 0.3661 1324.1 1.4658	170.95 0.3776 1339.3 1.4782	190.95 0.3887 1354.0 1.4899	290.95 0.4399 1422.1 1.5420	390.95 0.4867 1485.3 1.5867	490.95 0.5305 1545.7 1.6268	590.95 0.5724 1604.7 1.6634
1676 (611.10)					28.90 0.2726 1203.2 1.3597												
1700 (613.12)					26.88 0.2659 1199.4 1.3549												
1798 (615.13)					24.87 0.2593 1195.6 1.3505												
1750 (617.11)					22.89 0.2529 1191.7 1.3460												
1778 (819.07)	-		0.2222 1150.9 1.8076		90.09	40.02	80.09	90.09	100.09	100.00	140.00	100.00	100.00		380.93 0.4493 1481.3 1.5767	***	
1906 (621.00)	8b.	0.0247 648.0	0.2180 1149.3 1.3051		19.00 0.2405 1183.7	39.00 0.2598 1212.7	59.00 0.2764 1237.6	79.00 6.2912 1259.8	99.00 0.3045 1279.6	119.00 0.3168 1297.4	139.00 0.3283 1313.9	159.00 0.3393 1329 9	179.00 0.3499 1345.3	279.00 0.3986 1416.0	379.00 0.4425 1480.6 1.5748	479.00 0.4834 1542.0	579.00 0.5224
1828 (622.92)	- 1	650.9	0.2139 1147.7 1.3028		17.08 0.2345 1179.5 1.3319	0.2540	57.08 0.2708 1234.9 1.3815	0.2855	0.2991 1277.8	1295.6	1312.2	0.3335	0.3441 1343 9	0.3924	377.08 0.4357 1479.3 1.5728	0.4763	0.5147
1850 (624.82)	6h	0.0249 653.9 0.8465	0.2099 1145 9 1.3002		15.18 0.2285 1175.1 1.3269	35.18 0.2482 1205.8 1.3546	55.18 0.2651 1231.8 1.3776	75.18 0 2799 1254.8 1.3976	95.18 0.2936 1275.6 1.4154	115.18 0.3056 1293.5 1.4305	135.18 0.3170 1310.5 1.4445	155.18 0.3279 1326.7 1.4577	175.18 0.3384 1342.4 1.4705	275.18 0.3863 1413.9 1.5248	375.18 0.4293 1479.0 1.5710	475.18 0.4695 1540.8 1.6120	575.18 0.5075 1600.8 1.6492
1976 (626.69)	8h	0.0251 656.9 0.8491	0.2060 1144.2 1.2977		13.31 0.2225 1170.5 1.3218	33.31 0.2427 1202.3 1.3504	53.31 0.2597 1229.0 1.3741	73.31 0.2746 1252.3 1.3943	93.31 0.2882 1273.5 1.4124	113.31 0.3003 1291.8 1.4278	133.31 0.3115 1308 7 1.4418	153.31 0.3224 1325.2 1.4552	173.31 0.3328 1340.8 1.4677	273.31 0.3804 1412.9 1.5227	373.31 0.4229 1478.0 1.5689	473.31 0.4626 1540.2 1.6101	573.31 0.5003 1600.3 1.6475
1900 (628.55)	•	659.9	0.2022 1142.4 1.2951		11.45 0.2165 1165.6 1.3163	31.45 0.2371 1198.8 1.3462	51.45 0.2543 1225.9 1.3702	71.45 0.2694 1249.8 1.3910	91.45 0.2828 1271.0 1.4091	111.45 0.2950 1289.7 1.4249	131.45 0.3063 1307.0 1.4392	151.45 0.3171 1323.5 1.4526	171.45 0.3274 1339.4 1.4653	271.45 0.3747 1411.9 1.5207	371.45 0.4170 1477.5 1.5672	471.45 0.4562 1539.7 1.6084	571.45 0.4934 1599.8 1.6457
1925 (630 38)		662 8	0.1985 1140.6 1.2926		9.62 0.2107 1160.9 1.3111	29 62 0.2317 1195 0 1.3419	49.62 0.2491 1223.1 1.3667	69.62 0.2642 1247.3 1.3878	89.62 0 2776 1268.7 1.4061	109 62 0.2898 1287.7 1.4221	129.62 0.3010 1305.1 1.4364	149.62 0.3118 1321.8 1.4500	169.62 0.3221 1337.8 1.4628	269.62 0.3690 1410.7 1.5185	369.62 0.4109 1476.6 1.5652	469.62 0.4497 1538.9 1.6065	569.62 0.4864 1599.0 1.6438
1988 (632.20)	6h	0.0254 665.8 0.8569	0.1949 1138.8 1.29Q1		7.80 0.2049 1155.8 1.3056	27.80 0.2264 1191.3 1.3376	47.80 0.2440 1220 0 1.3630	67.80 0.2591 1244.8 1.3846	87.80 0.2726 1266.5 1.4031	107.80 0.2848 1285.8 1.4194	127.80 0.2960 1303.4 1.4339	147.80 0.3067 1320.2 1.4476	167.80 0.3170 1336.3 1.4604	267.80 0.3636 1409.7 1.5165	367.80 0.4052 1476.0 1.5635	467.80 0.4436 1538.4 1.6048	567.80 0.4800 1598.8 1.6424
1978 (634 00)	64	0.0256 668.7 0.8595	0 1913 1137.0 1 2877		6.00 0 1992 1150 5 1.3000	26.00 0.2212 1187.5 1.3334	46.00 0.2391 1217.0 1.3595	66 00 0.2544 1242 4 1.3816	86 00 0 2678 1264.4 1.4004	106.00 0.2798 1283.6 1.4165	126.00 0.2910 1301.5 1.4313	146.00 0.3016 1318.3 1.4450	166.00 0.3117 1334.4 1.4579	266.00 0.3581 1408.4 1.5144	366.00 0.3994 1475.1 1.5617	466.00 0.4376 1537.8 1.6032	566.00 0.4737 1598.4 1.6409

Sh = superheat, deg. F. v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

FUNDAMENTALS OF POWER PLANT ENGINEERING

TABLE XXV continued

Abs. Pr Lb./Sq. (Sat. Te	ess. In. mp.)	Sat. Water	Sat. Steam	Tenr	ERATURI 680°	— Dea	120°	T40°	760°	780*	800°	820*	850*	900°	1000°	1100°	1300*
2000 (635 78)	8b	0.0257 671 7 0.8620	0 1879 1135.2 1.2851	24.22 0 2162 1183 7 1.3289	0 2344 1214 3 1.3560	64.22 0 2498 1240 0 1.3783	0.2633 1262.4 1.3975	0.2752 1281.8 1.4138	0 2862 1299.6 1.4285	0 2966 1316.4 1.4422	164.22 0.3067 1332.7 1.4552	0.3165 1348.6 1.4677	0 3305 1371.1 1.4851	0 3528 1407.2 1.5122	0.3940 1473.5 1.5592	0.4319 1537.4 1.6015	0.4678 1598.6 1.6393
2028 (637.54)	8h	0.0258 674 7 0.8646	0.1845 1133.3 1.2826	22.46 0.2112 1179.9 1.3246	42.46 0.2296 1211.1 1.3523	62.46 0.2452 1237.5 1.3752	82.46 0.2587 1260.3 1.3947	102.46 0.2707 1280.1 1.4114	122.46 0.2816 1297.9 1.4261	142.46 0 2919 1314.8 1.4398	162 46 0.3020 1331.3 1.4530	182.46 0.3116 1347.1 1.4654	212.46 0.3257 1370.1 1.4832	262.46 0.3478 1406.2 1.5103	362.46 0.3887 1473.8 1.5582	462.46 0.4262 1537.0 1.6001	562.46 0.4617 1598.1
2050 (639,29)	8b	0.025 <mark>9</mark> 677.6 0.8671	0.1812 1131.4 1,2800	20.71 0.2062 1175.7 1.3199	40.71 0.2248 1207.7 1.3483	60.71 0.2405 1234.6 1.3717	80.71 0.2541 1257.9 1.3916	100.71 0.2661 1278.0 1.4085	120.71 0.2769 1295.9 1.4233	140.71 0.2872 1313.0 1.4372	160.71 0.2972 1329.5 1.4504	180.71 0.3067 1345.3 1.4628	210.71 0.3208 1368.6 1.4808	260.71 0.3428 1405.1 1.5082	360.71 0.3833 1472.8 1.5562	460.71 0.4205 1536.2 1.5982	560.71 0.4556 1597.4 1.6362
2078 (641,02)	8b 7 (0261 680.5 0.8697	0.1780 1129.5 1.2776	18.98 0.2013 1171.5 1.3154	38.98 0.2200 1204.3 1.3445	58.98 0.2358 1231.6 1.3682	78.98 0.2494 1255.2 1.3884	98.98 0.2615 1275.7 1.4056	118.98 0.2724 1294.0 1.4208	138 98 0.2828 1311.5 1.4350	158.98 0.2927 1328.1 1.4483	178.98 0.3022 1344.0 1,4608	208.98 0.3161 1367.3 1.4788	258.98 0.3374 1403.1 1.6056	358.98 0.3782 1472.1 1.5546	458.98 0.4151 1535.7 1.5967	558.98 0.4499 1597.0 1.6348
2100 (642.73)	Sh h	0.0262 683.4 0.8722	0.1748 1127.6 1.2751	17.27 0.1964 1167.3 1.3108	0.2152 1200.4	57.27 0.2310 1228.2 1.3643	0.2447 1252.2	0.2568 1273.0	0.2679 1292.0	0.2783 1309 6	157.27 0.2882 1326.4 1.4458	0.2977 1342.6	0.3114 1365 8	0.3319 1400 8	0.3730	0.4096 1534 R	0.4441
2128 (644.43)					0.2108 1197.3 1.3364		0.2406 1250.2 1.3820	0.2527 1271.3 1.3994	0.2637 1290.2 1.4151	0.2739 1307.8 1.4294	155.57 0.2838 1324.8 1.4430	0.2933 1341.2 1.4559	0.3070 1364.7 1.4740	0.3280 1400.8 1.5011	0.3682 1470.5 1.5505	0.4045 1534.5 1.5929	0.4386 1596.0 1.6311
2150 (646.11)											153.89 0.2793 1322.7 1.4408						
2178 (647.77)	8h 7.0	0.0266 692.0 0.8798	0.1655 1121.5 1.2676	12.23 0.1828 1154.3 1,2971	32.23 0.2018 1189.7 1,3284	52.23 0.2182 1219.8 1,3546	72.23 0.2321 1244.9 1.3760	92.23 0.2442 1266.5 1.3942	112.23 0.2552 1286.1 1,4104	132.23 0.2654 1304.0 1.4250	152.23 0.2752 1321.3 1.4388	172.23 0.2847 1338.0 1.4519	202.23 0.2983 1361.9 1.4704	252.23 0.3196 1399.4 1,4985	352.23 0.3587 1468.9 1.5478	452.23 0.3943 1533.1 1.5903	552.23 0.4279 1595.0 1,6288
2200 (649.42)	b	695.0	1119.4	1148.0	1185.8	1216.2	1241.8	1263.8	1283.6	1302.9	150.58 0.2710 1319.5 1.4362	1336.2	1360.3	1398 0	1467.9	1532.3	1594.3
2225 (651.06)	Sb C	0269 697.9 .8848	0.1597 1117.4 1.2625	8.94 0.1727 1142.7 1.2852	28.94 0.1931 1182.1 1,3201	48.94 0.2097 1213.3 1.3472	68.94 0.2237 1239.1 1.3693	88.94 0.2360 1261.7 1.3983	108.94 0.2471 1282.0 1.4051	128.94 0.2574 1300.5 1.4201	148.94 0.2671 1318.0 1.4341	168.94 0.2765 1335.2 1.4476	198.94 0.2900 1359.2 1.4661	248.94 0.3109 1397.0 1.4944	348.94 0.3495 1467.0 1.5441	448.94 0.3847 1531.9 1.5871	548.94 0.4175 1593.7 1.6255
2280 (652.67)	Sh v 0	.0270 700.8 .8873	0.1569 1115.3 1.2599	7.33 0.1680 1137.2 1.2795	27.33 0.1889 1178.0 1.3156	47.33 0.2055 1209.7 1.3432	67,33 0,2197 1236,5 1,3661	87.33 0.2320 1259.2 1.3852	107.33 0.2432 1279.8 1.4022	127.33 0.2535 1298.6 1.4175	147.33 0.2632 1316.3 1.4317	167.33 0.2726 1333.5 1.4452	197.33 0.2860 1357.8 1.4640	247.33 0.3066 1395.5 1.4922	347.33 0.3449 1465.8 1.5421	447.33 0.3800 1531.2 1.5854	547.33 0.4125 1593.1 1.6239
2278 (654.27)	Sh V 0	.0272 703.8 .8898	0.1542 1113.2 1.2573	5.73 0.1630 1130.8 1.2731	25.73 0.1848 1174.3 1.3116	45.73 0.2018 1207.1 1.3401	65,73 0.2159 1233.9 1.3630	85.73 0.2281 1256.7 1.3822	105.73 0.2393 1277.6 1.3995	125.73 0.2496 1296.8 1.4151	145.73 0.2594 1314.8 1.4295	165.73 0.2688 1332.1 1.4431	195.73 0.2821 1356.5 1.4619	245.73 0.3026 1394.4 1.4903	345.73 0.3407 1465.1 1.5405	445.73 0.3755 1530.7 1.5839	545.73 0.4077 1592.8 1.6225
2306 (655.87)	. 0	.8923	1.2547	1.2665	1.3070	1.3366	1.3596	1.3790	1.3964	1.4123	144.13 0.2556 1312.9 1.4270	1.4407	1.4097	1.4884	1.5388	1.5823	1.6210
2326 (657.74)											142.26 0.2520 1311.5 1.4249						
2350 (659.00)	8b v 0 h	.0277 712.6 .8974	0.1462 1106.5 1.2495	1.00 0.1479 1109.7 1.2524	21.00 0.1725 1161.2 1.2980	41.00 0.1901 1196.6 1.3287	61.00 0.2046 1225.2 1.3532	81.00 0.2172 1249.8 1,3739	101.00 0.2285 1271.7 1.3920	121.00 0.2389 1291.6 1.4082	141.00 0.2484 1309.8 1.4227	161.00 0.2576 1327.3 1.4365	191.00 0.2706 1352.1 1.4557	241.00 0.2910 1391.0 1.4848	341.00 0.3282 1462.3 1.5354	441.00 0.3621 1528.3 1.5791	541.00 0.3936 1590.9 1.6180
2376 (660.55)	Sh V 0	.0278 715.6 .9000	0.1436 1104.0 1.2467		19.45 0.1686 1156.9 1.2935	39.45 0.1863 1193.0 1.3249	59.45 0,2010 1222.4 1,3501	79 45 0.2137 1247.5 1.3712	0.2250 1269.5 1.3893	0.2353 1289.6 1.4057	139.45 0.2450 1308.3 1.4207	0.2541 1325.8 1.4344	0.2671 1350.8 1.4537	0.2873 1389.8 1.4830	0.3244 1461.7 1.5340	0.3580 1527.8 1.5777	0.3892 1590.4 1.6166
2400	Sh b b	.0280 718.5 .9025	0.1410 1101.4 1.2438		17.91 0.1646 1152.2 1.2887	37.91 0.1824 1189.1 1.3208	57.91 0.1974 1219.4 1.3467	77.91 0.2101 1244.7 1.3680	97.91 0.2214 1267.1 1.3865	117.91 0.2317 1287.3 1.4030	137.91 0.2415 1306.5 1.4183	157 91 0.2506 1324.2 1.4323	187.91 0.2636 1349.6 1.4519	237.91 0.2836 1388.5 1.4810	337.91 0.3205 1460.8 1.5323	137.91 0.3538 1526.9 1.5761	537.91 0.3848 1589.8 1.6152

Sh = superheat, deg. F.
v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb. s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued

Superheated Steam—(Concluded)

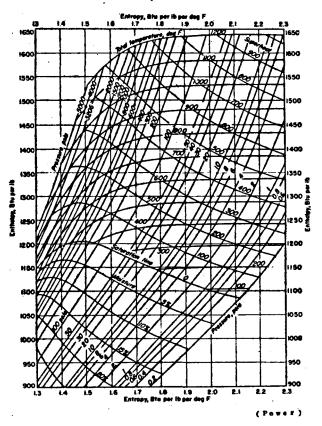
Abs. Pro Lb./8q. (Sat. Ton	ia. np.)	Set. Water	Bat. Steam	TEMPE	84TURB 700°	-Dear	EES PAI	160°	780*	800*	820*	850*	900*	950*	1000°	1100°	1900*
\$450 (665.12)	8h	0.0283 724.6 0.9076	0.1360 1096.3 1.2381	14.88 0.1567 1142.3 1.2787	34.88 0.1750 1181.3 1.3127	54.88 0.1902 1212.9 1.3397	74.88 0.2032 1239.5 1.3620	94.88 0.2147 1262.8 1.3813	114.88 0.2250 1283.5 1.3981	134.88 0.2347 1302.9 1.4136	154.88 0.2440 1321.4 1.4282	184.88 0.2569 1347.1 1.4481	234.88 0.2766 1386.3 1.4774	284.88 0.2954 1423.8 1.5045	334.88 0.3130 1459.0 1.5290	434.88 0.3458 1525.6 1.5732	534.88 0.3764 1588.9 1.6125
280 0 (668.10)								91.90 0.2083 1258.6 1.3761									
2560 (671.03)	8b	0.0291 736.7 0.9179	0.1264 1085.6 1.2265	8.97 0.1408 1120.0 1.2568	28.97 0.1606 1164.7 1.2957	· 48.97 0.1766 1199.8 1.3257	68.97 0.1902 1228.9 1.3502	88.97 0.2020 1254.0 1.3709	108.97 0.2125 1276.0 1.3888	128.97 0.2223 1296.5 1.4052	148.97 0.2315 1315.6 1.4202	178.97 0.2440 1341.5 1.4402	228.97 0.2634 1381.8 1.4704	278.97 0.2817 1419.8 1.4979	328.97 0.2989 1455.6 1.5228	428.97 0.3308 1522.9 1.5674	528.97 0.3604 1586.7 1.6070
2600 (673.91)	Bb b	0.0295 743.1 0.9232	0.1219 1080.1 1.2205	6.09 0.1323 1106.0 1.2433	26.09 0.1541 1157.0 1.2877	46.09 0.1706 1194.0 1.3193	66.09 0.1842 1223.9 1.3444	86.09 0.1961 1249.7 1.3657	106.09 0.2066 1272.2 1.3840	126.09 0.2164 1293.1 1.4008	146.09 0.2256 1312.6 1.4161	176.09 0.2380 1338.9 1.4364	226.09 0.2573 1379.7 1.4670	276.09 0.2754 1418.1 1.4947	326.09 0.2924 1454.2 1.5199	426.09 0.3237 1521.6 1.5645	526.09 0.3530 1585.8 1.6044
2650 (676.75)	8b	0.0300 749.5 0.9287	0.1173 1074.5 1.2147	3.25 0.1235 1090.1 1.2284				83.25 0.1903 1245.1 1.3606									
2706 (679.54)	8h h	0.0308 756.1 0.9342	0.1123 1068.3 1.2082					80.46 0.1846 1240.3 1.3550									
2760 (682.28)	Bh b	0.0310 763.0 0.9399	0.1077 1061.8 1.2016		17.72 0.1335 1127.1 1.2583	37.72 0.1520 1172.2 1.2969	57.72 9.1679 1207.4 1.3265	77.72 0.1794 1236.0 1.3501	97.72 0.1902 1260.8 1.3703	117.72 0.2000 1283.0 1.3881	137.72 0.2094 1304.2 1.4048	167.72 0.2215 1331.4 1.4258	217.72 0.2402 1373.3 1.4572	267.72 0.2576 1412.2 1.4853	317.72 0.2740 1449.2 1.5110	417.72 0.3043 1518.0 1.5566	517.72 0.3322 1582.8 1.5969
2200 (684.98)	Bh b	0.0318 770.0 0.9458	0.1032 1054.6 1.1944		15.02 0.1267 1115.9 1.2476	35.02 0.1461 1164.5 1.2892	55.02 0.1615 1201.5 1.3203	75.02 0.1741 1231.3 1.3449	95.02 0.1851 1257.0 1.3658	115.02 0.1950 -1279.8 1.3840	135.02 0.2045 1301.7 1.4013	165.02 0.2166 1329.3 1.4226	215.02 0.2351 1371.5 1.4542	265.02 0.2521 1410.4 1.4823	315.02 0.2685 1447.9 1.5084	415.02 0.2983 1517.0 1.5542	515.02 0.3258 1581.9 1.5945
2889 (687.65)	8h	0.0322 777.5 0.9521	0.0985 1046.6 1.1864					72.35 0.1690 1226.5 1.3397									
\$900 (690.26)	ь	785.2	0.0941 1038.1 1.1785		9.74 0.1126 1089.2 1.2228	29.74 0.1348 1148.3 1.2733	49.74 0.1511 1189.5 1.3080	69.74 0.1641 1221.6 1.3345	89.74 0.1754 1249.1 1.3569	109.74 0.1855 1273.3 1.3762	129.74 0.1949 1295.7 1.3939	159.74 0.2069 1324.2 1.4159	209.74 0.2250 1367.1 1.4480	259.74 0.2418 1406.9 1.4767	309.74 0.2578 1444.8 1.5032	409.74 0.2870 1514.8 1.5495	509.74 0.3138 1580.2 1.5902
\$980 (692.83)	84	0.0337 793.6 0.965\$	0.0895 1028.9 1.1697					67.17 0.1593 1216.6 1.3290									
9000 (695.37)	8h 5	0.0346 802.6 0.9731	0.0849 1019.3 1.1607		4.63 0.0972 1054.0 1.1907	24.63 0.1236 1130.3 1.2559	44.63 0.1410 1176.4 1.2947	64.63 0 1946 1211.4 1.3236	84.63 0.1661 1240.6 1.3474	104.63 0.1763 1266.0 1.3677	124.63 0.1856 1289.0 1.3858	154.63 0.1978 1319.0 1.4090	204.63 0.2155 1362.5 1.4416	254.63 0.2322 1403.4 1.4711	304.63 0.2478 1441.7 1.4978	404.63 0.2763 1512.4 1.5447	504.63 0.3027 1578.9 1.5860
8050 (697.84)	8b	0.0357 812.9 0.9818	0.0804 1007.7 1.1501		2.16 0.0868 1025.0 1:1650		1.2014	62.16 0.1499 1205.9 1.3179	1.5401	1.5155	1.5519	1.4030	1.7300	1.1001	1.4932	1.3424	1.0001
8100 (700.29)		0.037 2 824.6 0.9916	0.0752 994.0 1.1376			19.71 0.1128 1110.8 1.2374	39.71 0.1312 1162.4 1.2808	59.71 0.1456 1200.9 1.3126	79.71 0.1576 1232.5 1.3383	99.71 0.1680 1259.4 1.3599	119.71 0.1771 1282.7 1.3782	149.71 0.1891 1313.4 1.4019	199.71 0.2068 1358.2 1.4355	249.71 0.2231 1399.5 1.4653	299.71 0.2385 1438.7 1.4926	399.71 0.2663 1509.9 1.5398	499.71 0.2921 1577.0 1.5815
\$1 86 (702.69)							0.1266 1155.3 1.2738	0.1410 1194.7 1.3064	0.1532 1227.6 1.3331	0.1636 1254.9 1.3550		0.1850 1310.5 1.3983	0.2024 1355.5 1.4320	0.2187 1397.5 1.4623	0.2341 1437.2 1.4900	0.2615 1508.6 1.5373	0.2871 1576.1 1.5792
9900 (705.04)		0.044 3 871. 3 1.0311	0.0598 946.6 1.0958					54.96 0.1368 1189.3 1.3011									
\$906.2° (705.34)			0.0541 910.3 1.0645			14.66 0.1018 1088.5 1.2170	34.66 0.1211 1145.8 1.2652	\$4.66 0.1363 1188.5 1.3005	74.66 0.1488 1222.9 1.3284	94.66 0 1591 1250.6 1.3506	114.66 0.1682 1274 8 1.3696	144.66 0.1805 1307.3 1.3947	194.66 0.1980 1353.4 1.4293	244.66 0.2140 1395.3 1.4596	294.66 0.2290 1434.8 1.4872	394.66 0.2564 1507.3 1.5351	494.66 0.2816 1575.1 1.5773

Critical pressure

Sh = superheat, deg. F.
v = specific volume, cu. ft. per lb.

h = enthalpy, B.t.u. per lb.
s = entropy, B.t.u. per deg. F. per lb.

TABLE XXV continued Mollier Diagram for Steam



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