AN ABSTRACT OF THE THESIS OF

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A study of heat transfer in beds of fluidized solids has been made in order to supply some of the basic information which was lacking when the wartime fluid catalytic cracking plants were built. Although a number of papers on fluidization of solids have been written in the past several years, with the exception of some preliminary experiments on indirect heat transfer by Parent, Yagol, and Steiner, nothing has been published on heat transfer in beds of fluidized solids.

In this thesis, equipment was designed and constructed for this purpose. A technique was developed which permitted a general study of the behavior of fluidized beds with respect to heat flow; and the magnitude of the coefficient of heat transfer between the upward flowing gas and the wall, with and without the presence of solid particles, was determined. In addition, an attempt was made to determine the coefficient of heat transfer between the gas and the solid particles.

The apparatus consisted essentially of two four inch tubes: one was sheet metal, ten feet high, insulated, and preceded by an air heater. A rotatable screen defined the bottom of the fluidized bed and seventeen thermocouples measured temperatures at various points in the gas stream, in the bed, in the metal tube, and in the insulation. The other tube was glass, ten feet high, and was used for visual observation of materials in order that their behavior in the metal tube might be known. Air velocities were measured by two thin-plate orifices.

Runs were made on 3A Catalyst, Ottawa Standard Silica Sand, wood charcoal, Utah hard coal, and with an empty tube. The particles ranged in size from -270 mesh to 8 mesh. In attempting to determine the coefficient of heat transfer between the air and the particles, hot air was introduced suddenly into the bed of fluidized solids and temperature measurements were made throughout the unsteady state portion of the run. It was hoped that the rate of heat flow to the solid could be calculated by difference between the input and the losses to the metal and the insulation but, partly because of unaccountable losses and partly because of the close approach to equilibrium between air and solid at the top of the bed, this method was unsuccessful. However, this did not affect the experimentation to determine the coefficient of heat transfer between the air and the metal wall.

The calculations of this latter value were based on combining the measured rate of heat flow through the insulation, after steady state had been reached,

with the resistance concept. The results are summarized below.

Coefficient of Heat Transfer between Air and Tube Wall

Material Fluidized	Approximate Avg. Particle Size	Mass Velocity lb per hr-sq ft	Coefficient of Heat Transfe Btu per hr-sq ft-OF
3A Catalyst	150 mesh	193	22
Sand	40 mesh	813	13
Coal	20 mesh	809	5.2
Coal	20 mesh	636	4.5
Coal	20 mesh	635	4.5
Charcoal	20 mesh	548	3.2
Empty Tube		637	0.89
Empty Tube		636	0.74

The following conclusions were drawn:

1. The film coefficient of heat transfer between the air and the tube wall is much greater when fluidized solids are present than when absent.

2. The film coefficient increases with increasing mass velocity. In fact,

it increases at about the same rate as the mass velocity.

3. The film coefficient increases with decreasing particle size. The increase is at least proportional to the particle size and may be much greater than that.

It is recommended that further extensive experimental work be carried out on the coefficient of heat transfer between the air stream and the tube wall and that an attempt be made to determine the coefficient of heat transfer between the air and the fluidized particles using continuous feed and discharge of the particles. HEAT TRANSFER IN BEDS
OF FLUIDIZED SOLIDS

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Table of Nomenclature

- a Subscript, indicating air.
- Ag Area of outside of metal wall, square feet.
- A; Area of insulation normal to the heat flow, square feet.
- As Total surface area of the particles in the bed, square feet.
- Am Cross-sectional area of the tube, square feet.
- Aw Inside surface area of the metal tube, square feet.
- A' Total mean area of the particles in the bed, square feet.
- C Specific heat, Btu per lb °F.
- G Mass velocity, 1b per (hour) (square foot).
- G' GAm, mass velocity lbs per hr.
- h_g Coefficient of heat transfer between the outside of the metal and the inside of the insulation, Btu per (hour) (square foot) (°F).
- hr Coefficient of heat transfer between the outside of the insulation and the room, Btu per (hour) (square foot) (°F).
- hs Coefficient of heat transfer between the air and the particles, Btu per (hour) (square foot) (OF).
- hw Coefficient of heat transfer between the air and the tube wall, Btu per (hour) (square foot) (OF).
- i Subscript, indicating insulation.
- K Thermal conductivity, Btu per (hour) (square foot) (OF per foot).
- 1 Distance, in the direction of the flow of heat, feet
- m Subscript, indicating metal.
- N Depth of bed, feet.
- P₁ Pressure, upstream of the orifice, inches of mercury.

- Δ P Pressure drop across the orifice, inches of water.
 - q $\frac{dQ}{d\Theta}$, rate of heat flow, Btu per hr
 - Q Quantity of heat, Btu.
 - r Radial distance from center of tube, feet.
 - r' Distance from center of tube to inside of insulation, feet.
 - s Subscript, indicating solid particles.
 - t Temperature, OF.
 - td Gas temperature at entrance of bed, OF.
 - te Gas temperature at exit from bed, OF.
 - t' Temperature at inside of insulation, OF.
 - t" Insulation temperature 1/4 inch out from metal wall, OF.
 - t₁, t₂, etc. Temperature at the thermocouples indicated by the subscript, ^oF.
 - to Room temperature, OF.
 - T average temperature of all the particles, OF.
- TC Abbreviation for thermocouple.
- Us Overall coefficient of heat transfer between the air and the particles, Btu per (hour) (square foot) (OF.).
- w Subscript, indicating tube wall.
- W Weight, pounds.
- x Distance, feet.
- \sim $\frac{K}{\text{Cp}}$, thermal diffusivity, square feet per hour.
- $\Delta_{\mathbf{S}}$ Temperature difference between the air and the center of a particle, ${}^{\mathsf{O}}\mathsf{F}$.
- Temperature difference between the air and the metal wall, F.
 - P Density, pounds per cubic foot.

- Pensity of the air in the orifice, pounds per cubic
 foot.
- 0 Time, hours.

HEAT TRANSFER IN BEDS OF FLUIDIZED SOLIDS

INTRODUCTION

This thesis was undertaken in order to supply information on the transmission of heat in beds of fluidized solids. Specifically, quantitative data are presented which show rates of heat transfer between beds of solid and the retaining wall of the bed. Some qualitative information is also submitted on the rate of heat transfer between the fluidized solids and air passing upwardly through the bed.

Although the technique of fluidization of solids has been known for seventy years, it did not attain industrial importance until the early part of World War II. At that time, a number of plants were built for the catalytic cracking of petroleum, using a process developed by the Standard Oil Development Company. This process was at first called 'jiggling' and, later, 'fluidization of solids' - terms which were very descriptive of the action of the solids. The essential idea was that a bed of fine solids - in this case catalyst - was suspended in a rising gas stream, leading to intimate contact between gas and solid. Under the pressure of wartime, these plants were built with little basic information. This thesis was written in order to supply some of that information.

Literature Survey

During the past few years, a number of papers have been written on fluidization of solids. With few exceptions, they have been concerned with either catalytic cracking of petroleum in fluidized beds or the flow characteristics of fluidized solids. As far as is known, the only information published on heat transfer in beds of fluidized solids has appeared in the following articles:

- 1. Kalbach, in his three papers (1,2,3) reviewing fluidization and discussing the information necessary for the design of fluidized equipment, mentioned the rapid transfer of heat as one of the chief advantages.
- 2. Kite and Roberts (4) discussed non-catalytic fluidization. Among possible uses they listed "---heat transfer
 operations wherein sensible heat is transferred from solid
 to gas phase or the reverse." They have applied this to
 the preheating of limestone in a stepwise calciner employing fluidization. However, no data were given. They also
 mentioned the uniformity of temperature in fluidized beds,
 a point emphasized by Dr. Warren K. Lewis in an address when
 receiving the Priestly Medal in 1947.

2. Kalbach, J.C. Fluidization in Chemical Reactions, Chemical Engineering 54:105-8, Jan. 1947.

3. Kalbach, J.C. Handling Solids-Gas Reactions by Fluidization, Chemical Engineering 54:136-8, Feb. 1947.

4. Kite, R.P. and Roberts, E.J. Fluidization in Non-Catalytic Reactions, Chemical Engineering 54:112-5, Dec. 1947.

^{1.} Kalbach, J.C. Improving Solids-Gas Contact by Fluidization, Chemical Engineering 51:94-8, June 1944.

- 3. Thomas and Hoekstra (5), in the course of a study of the catalytic cracking of gas oil, passed air through a bed of hot, spent catalyst in a two-inch pipe. The pipe was located in a thermostatted block. Temperatures were measured at four points in the bed and one above the bed by means of thermocouples inserted in a well. Heat was supplied by the burning carbonaceous deposit. No heat balances were given and no heat transfer rates calculated.
- 4. Parent, Yagol, and Steiner (6) carried out some preliminary experiments on indirect heat transfer in a steamjacketed copper tube. They measured the inlet and outlet
 temperatures of nitrogen flowing at approximately 0.4 and
 0.5 feet per second through the empty tube and again with
 -70+325 mesh coke present. At the lower velocity their
 heat transfer coefficient was 3 per cent lower when fluidized solids were present than when absent, whereas at the
 higher velocity the coefficient was 5 per cent higher when
 fluidized solids were present than when absent.
- 5. Furnas (7) studied heat transfer from a gas stream to beds of broken solids and reported that at his highest velocities the whole bed was lifted. However, in answer to

5. Thomas, C.L. and Hoekstra Fluidized Fixed Beds, Ind. and Eng. Chem. 37:332-4, April 1945.

7. Furnas, J.J. Heat Transfer from a Gas Stream to Beds of Broken Solids, U. S. Bureau of Mines Bulletin 361, 1932.

^{6.} Parent, J.D., Yagol N. and Steiner, C.S. Some Basic Observations on the Fluidizing Process in Laboratory Scale Equipment, Chem. Eng. Progress 43:429-436, Aug. 1947.

a letter about these experiments (8), he stated that he had not studied heat transfer under these conditions.

Although these are the only papers relating to heat transfer in beds of fluidized solids, as far back as 1878 a patent was granted to Frederic Luckenbach (9) for an improvement in apparatus for drying cereal grains. His apparatus consisted of a vessel with a conical false bottom through which hot or cold air or steam could be introduced into a bed of grain. This equipment was quite similar to that used today for fluidization. He pointed out that:

"By this means, a powerful and rapid drying, heating and cooling action was produced, rendering the process very rapid..." Other patents (10, 11, 12) mentioned the uniformity of temperature but none gave any indication of experiments to measure heat transfer rate.

Papers in related fields were also investigated and some of them have been helpful in designing the equipment and in understanding the problem. Among these might be mentioned, in addition to that by Furnas, the following:

- 1. Schumann's (13) mathematical analysis of heat flow
- 8. Furnas, J.J. Letter to J.S. Walton, Dec. 3, 1947.
- 9. Luckenbach, Frederic U. S. Patent 210,793, Dec. 10, 1878.
- 10. Canon, Frank A. U. S. Patent 1,355,105, Oct. 5, 1920.
- 11. Chappell, Frank L. U. S. Patent 1,892,233, Dec. 27, 1932.
- 12. Odell, William W. U. S. Patent 1,984,380, Dec. 18, 1934.
- 13. Schumann, T.E.W. Heat Transfer from a Liquid Flowing through a Porous Prism, Journal Franklin Inst.208:405-16, 1929.

to a bed of stationary solids from a liquid flowing through the bed. His results were in the form of time-temperature-position curves. Furnas used these curves in his work to calculate heat transfer coefficients from experimental data. However, Furnas' results were given in terms of volume coefficients* rather than the usual film coefficients..

2. Saunders and Ford (14) carried out experiments similar to Furnas' but with certain improvements, chiefly, reduction of wall effects and reduction of heat capacity of insulation. They plotted the generalized temperature difference, $\frac{t_g^2-t_o}{t_g-t_o}$ against the dimensionless group $\frac{\text{Ve}\Theta}{\text{dc}}$ for glass, steel and lead balls of several sizes and arrived, finally, at a value of $\ll = 11.25$ Btu per hour-square foot-°F.

d = particle diameter

e = time

V = velocity in feet per second to = initial temperature of solid

tg = gas temperature in

tg = gas temperature out

c! = heat capacity of unit volume of gas

^{*} Coefficient of heat transfer based on a cubic foot of loose solids rather than a square foot of solid surface. 14. Saunders, O.A. and Ford, H. Heat Transfer in the Flow of Gas through a Bed of Solid Particles, Iron and Steel Inst. Journal 141:291-328, #1 1940.

c = Heat capacity of unit volume of loose solids
3. Johnstone, Pigford, and Chapin (15) studied heat
transfer to clouds of falling particles. They dropped
particles in the 30 - 80 mesh range through a tube
heated electrically and measured the heat absorbed by
catching the particles in a calorimeter. They obtained
coefficients of heat transfer by convection, from gas
to solid, of 30 to 50 Btu per hour - square foot - OF.
4. Frey (16) studied pressure drop through beds of
fluidized solids in a glass tube and concluded, as did
Parent et al, (17) that the 'net' pressure drop was almost exactly equivalent to the weight of solid. By
'net' pressure drop is meant the measured drop minus
that due to friction loss of the gas passing through
the tube.

Scope

The problem, then, of the transfer of heat in a bed of fluidized solids has not been investigated, except as indicated above. In this thesis, equipment was designed and constructed for this purpose. A technique was developed

15. Johnstone, H.F., Pigford, R.L. and Chapin, J.H. Heat Transfer to Clouds of Falling Particles, Illinois Univ. Engineering Experiment Station Bulletin 330, 1941, 55 pages.

16. Frey, John S. The Flow Characteristics of Flowing Vapors through an Agitated Fluid Catalyst Bed, unpublished M.S. Thesis, Chemical Engineering Department, Oregon State College, 1947.

17. Parent, loc. cit.

which permitted a general study of the behavior of fluidized beds with respect to heat flow; and the magnitude of the film coefficient of heat transfer from the upward flowing gas to the wall, with and without the presence of solid particles, was determined. In addition, the film coefficient of heat transfer between gas and solid was determined qualitatively.

PROCEDURE

Method of Attack

The Study of heat transfer in fluidized beds appeared to be a fairly simple problem - the most obvious procedure being to pass a gas into a bed of fluidized solids and measure the inlet and outlet temperatures. However, there were a number of complications, chief among which were correct measurement of temperature, heat losses, and unsteady state operation.

Two methods of attack suggested themselves:

- 1. Continuous feed into and discharge from a bed of solids heated and fluidized by hot air. This was rejected because of the difficulties of maintaining constant feed and discharge rates; a difficulty which does not preclude its use at a more advanced stage of knowledge of the behavior of such systems.
- 2. A batch process in which hot air would be introduced suddenly into a fluidized bed. This was selected as a more convenient method for laboratory study.

Equipment

The scale of operations was limited, on the small side, by the seriousness of wall effects on the fluidization and, on the large side, by the volume of air available. As a result, a four inch tube was selected as the fluidization

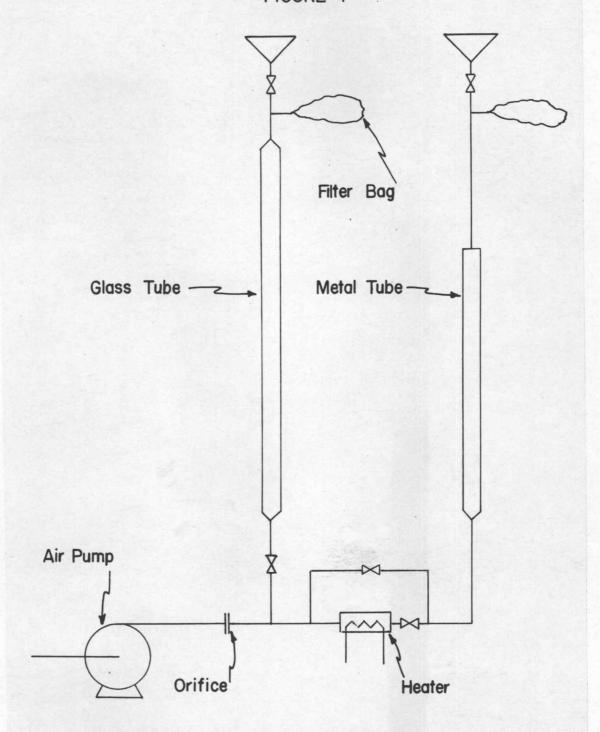
chamber.

The compressed air supply was of insufficient capacity and so a rotary vane air pump was used. This blower discharged a maximum of about 22 cubic feet per minute at atmospheric pressure, a value which corresponded to a linear velocity of 42 feet per second in the four inch tube. Air was used as the fluidizing gas throughout the experimental work. It was heated by an electric space heater and provision was made to by-pass the heater. The initial four inch tube was steel, 3/32 inches thick and six feet high. In addition, a four inch glass tube, ten feet high, in parallel with the steel tube, was used for visual observation of materials in order that their behavior in the metal tube might be known. The air leaving the top of the tubes passed, in each case, through a cloth filter bag which trapped practically all solid carryover. See figures 1 and 2.

The metal tube was insulated with a one inch layer of 85 per cent magnesia pipe lagging. This was done in preference to either a steam jacket or electric resistance heating for reasons of simplicity both in construction and operation, although greater heat losses were suffered thereby.

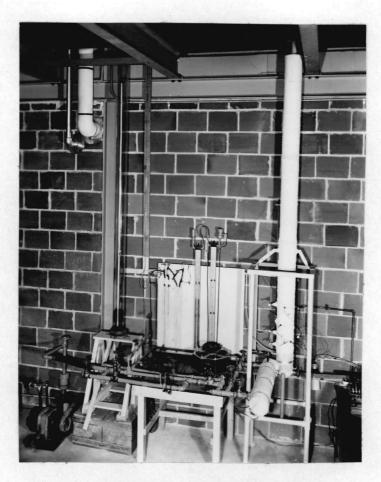
Measurements included velocity, pressure drop and temperature. Two calibrated thin-plate orifices were used to measure the air velocity. Pressure drop across part or

FIGURE I



PROCESS FLOW DIAGRAM

FIGURE 2



PHOTOGRAPH OF APPARATUS

all of the bed could be measured. Temperatures were measured at a number of points in the system by means of iron-constantan thermocouples. Initially, three types of thermocouples were used:

- 1. Thermocouples brazed onto the outside of the tube to read the metal temperature.
- 2. Bare thermocouples to read the air temperature in the high velocity approach section between the heater and the fluidization tube. This same type of thermocouple gave incorrect readings of air temperature in the tube itself with no solid present and a temperature, probably, somewhere between that of the solid and that of the air, when solid was present.
- 3. Suction thermocouples were used to determine the gas temperature in the fluidized bed. These were regular thermocouples inserted in wells through which the hot air could be drawn rapidly past the hot junction of the thermocouple. The fine solids were prevented from being sucked into the well by means of a fine screen placed over the end of the well.

Preliminary Operations

A number of preliminary runs were made to determine the operability of the equipment and the behavior of several fine solids at various air velocities. Runs A-1, A-6,

*Due to the low air velocity past the hot junction. †See appendix.

A-8, and A-13 were made in the glass tube using the following materials: Up to five pounds of 3A Catalyst, (a synthetic aluminum silicate, 60 per cent of which was - 200 mesh), a mixture of the largest 3A particles (10 per cent 50-100 mesh, 78 per cent 100-200 mesh, 12 per cent-200 mesh), and Ottawa Standard Silica Sand, (twice), respectively. This last material is a pure, white sand over 95 per cent of which is between 30 and 100 mesh in size. The individual grains are nearly spherical. These runs confirmed the conclusions of Parent et al (18) and Frey (19) that the pressure drop across a bed of fluidized solids is equal to the head of solids plus the friction loss through the empty tube at the same superficial velocity.*

Runs A-2, A-7, A-9, and A-14 were duplicates of the above runs except that they were made in the metal tube. It was found that bed depth and density could be estimated, approximately, by the difference in pressure between manometer taps. However, it was believed that the bed depth in the metal tube could be more accurately determined by comparison with the depth in the glass tube at the same mass velocity.

Runs A-3, A-5, A-10, A-11, A-12, and A-15 were tests of the heater, the thermocouples and the insulation. They were

^{18.} Parent, loc. cit.

^{19.} Frey, loc. cit.

^{*} See Appendix

made at several voltages and various air velocities. These runs are summarized in Table I.

TABLE I Preliminary Operations

Run	Material	Weight	Tube	Mass Velocity lb per hr-sq ft	Superficial velocity fps	Purpose	Remarks
A-1	Ottawa Std. Sand	1	Glass	250-800	1.0-3.4	Behavior of solid	Good fluidiza- tion between 350 & 500 pounds/ hr-sq ft
A-6	Ottawa Std. Sand*	5	Glass	350-1100	1.5-4.5	Behavior of solid	Slugging ¹ at all velocities
A-8	3A Catalyst	5	Glass	100-300	0.3-1.3	Behavior of solid	Good fluidiza- tion +
A-13	3A Catalyst (screened)*	5	Glass	90-230	0.25-0.9	Behavior of solid	Channels up to 150 pounds/hr-sq ft better fluidization above
A-2	Ottawa Std. Sand*	1&3	Metal	250-800	1.0-3.4	Behavior of solid	

^{*} Screen Analysis of Ottawa Standard Sand -16 + 30 mesh 2 ± 2% -30 + 50 " 70 ± 5%

^{26 ± 2%} -50 +100 "

[†] These terms will be explained later ‡ Screen Analysis of Large Size 3A Catalyst -50 +100 mesh 10%

^{-100 +150 &}quot;

^{-150 +200 &}quot;

^{-200 +270 &}quot;

⁻²⁷⁰

TABLE I CONTINUED

Preliminary Operations

Run	Material	Weight pounds	Tube	Mass Velocity lb per hr-sq ft	Superficial velocity fps	Purpose	Remarks
A-7	Ottawa Std. Sand	5	Metal	500-800	2.0-3.4	Behavior of solid	
A-9	3A Catalyst	5	Metal	110-180	0.36-0.66	Behavior of solid	
A-14	3A Catalyst (screened)	4-7/8	Metal	95-220	0.28-0.81	Behavior of solid	
A-3	too aso		Metal	660	3.0	Heater & TC operation	No preheat
A-5			Metal	700	3.1	11 11	100v across heater
A-10			Metal	175	0.63	ii ii	Heater burnt out
A-11		gas and	Metal	180	0.63	îi îi	110v across
A-12		-	Metal	660	3.0	îi îi	110v across
A-15			Metal	660	3.0	îi îi	100v across

First Heating Runs

The first four unsteady state heating runs were preliminary runs to determine the general behavior of the equipment and the solids. These runs are summarized in Table II.

TABLE II
Heating Runs

Run	Material		Mass Velocity lb per hr-sq ft	Remarks
1	3A Catalyst	5	180	Large % carry-
2	3A Catalyst (screened)	4-7/8	175	Carryover still serious
3	Ottawa Std. Sand	5	230	Poor fluidiza- tion
4	Ottawa Std. Sand	5	810	Slugging

The first run was made using five pounds of 3A Catalyst at an intermediate air velocity (for this material).
The chief difficulty was that, at the end of one hour, only
about 50 per cent of the original five pounds was still in
the tube. The other two and a half pounds had gradually
been carried into the filter bag by the rising gas stream.
Although the decrease in solid could be followed, roughly,
by means of the pressure drop across the bed, it did introduce another variable which would be rather difficult to
allow for in the calculations.

The second run was made on a fraction of 3A Catalyst, 78 per cent of which was between 100 and 200 mesh in size.

The carryover was reduced to about 9 per cent per hour which was still large but not excessive unless a run lasted for several hours. Throughout the heating period of this run, two bare thermocouples in the bed of solids (one near the bottom and one near the top) and two corresponding suction thermocouples in the bed all read the same temperature at any given time, within the accuracy of the readings. The two wall thermocouples at corresponding positions opposite the bed read two degrees lower. This indicated that the mixing in the bed was excellent and that the solid and air were very close to the same temperature (i.e. equilibrium) very shortly after the air entered the bed. This reasoning followed from the fact that the bare thermocouples, which gave a temperature between that of the gas and that of the solid, read the same as the suction thermocouples, which gave the true gas temperature.

Run 3 was made with five pounds of Ottawa Sand at a mass velocity of 210 pounds per hour per square foot. This velocity was so low that most of the solid was in the approach tube and in the cone connecting the one inch approach tube with the four inch main tube. Consequently, fluidization was poor.

Run 4 was also made with five pounds of Ottawa Sand but at a higher mass velocity namely, 810 pounds per hour per square foot. As in Run 2, the bare thermocouples and the suction thermocouples read the same temperature at any given time. However, the two wall thermocouples each read about

ten degrees lower. Losses due to carryover were less than $l^{\frac{1}{2}}$ per cent per hour of operation.

These runs set limitations on the equipment and helped to determine the direction of later experimentation. reasons against the use of very fine particles appeared. Firstly, the drifting of appreciable quantities of solid out of the fluidization tube introduced an undesirable variable. Secondly, if the air and solid particles were close to equilibrium throughout most of the bed, it would be impossible to calculate a heat transfer coefficient between the air and the solid. Increasing the particle size, would decrease the area available for heat transfer and tend toward non-equilibrium in the bed. However, the heat transfer coefficient might be so large that it would be impossible to obtain a measurable temperature difference between air and solid within the limitations of the equipment. This did not apply to the heat transfer coefficient between the air and the wall of the tube for, as was mentioned above, the wall temperature was 2°F below the air temperature when material as fine as 100 to 200 mesh was fluidized.

The poor fluidization of the sand indicated a limit to the maximum size and/or density of particles which could be used. In order to decrease the heat trasnfer area and still fluidize, it was necessary to use a material of lower density.

Runs 5 to 8, inclusive, were heating runs on wood

charcoal (Douglas fir) of 8 to 28 mesh. The charcoal had an apparent density of 9 pounds per cubic foot as compared to 87 pounds per cubic foot for the sand. Carryover of solid was at a high rate and there was also a loss of weight due to the moisture or other volatile matter in the charcoal. This amounted to 6 per cent of the initial weight in Run 7. Another difficulty in the use of charcoal was the attrition resulting from the impacts between particles and between the particles and the wall. In the case of a hard material like sand, the attrition over the operating period of a run was negligible. The table below shows the reduction of size in Run 8, not including the material in the filter bag.

TABLE III
Attrition in Run 8

Mesh	Per Cent at Start	Per Cent at End
-14 + 20	100.0	51.6
-20 + 28	0.0	36.0
-28 + 35	0.0	9.2
-35 ∓ 48	0.0	2.7
-48	0.0	0.5
	100.0	100.0

The four runs have been summarized below:

TABLE IV
Heating Runs on Charcoal

Run	Weight at Start in pounds	Mesh at Start	Mass Velocity lb per hr-sq ft	Remarks
5	1.0	-8 * 28	548	Operational difficulties
6	1.0	-8 + 20	535	Operational difficulties
7	1.0	-20 + 28	734	55% loss in weight
8	1.25	-14 + 20	523	36% loss in weight

Improved Apparatus

preliminary calculations on Run 8 brought out a serious disadvantage in the use of charcoal namely, that the rate of heat transfer to the metal wall was several times as great as the rate of heat transfer to the solid. This was a result of the combination of a comparatively large weight of metal with a high coefficient of heat transfer to the metal, as indicated by a very rapid rise in metal temperature. However, this difficulty did not affect experimentation to determine the coefficient of heat transfer between air and tube wall but only obscured the heat transfer between air and solid.

Two solutions to this problem presented themselves:

1. The use of a solid of much greater heat capacity
(i.e. weight) per unit volume. However, we have already seen
that, in order to decrease the heat transfer area of the
solid, it was necessary to go to a material of smaller

weight per unit volume. A compromise between these opposing demands was found in coal.

2. A decrease in the weight of metal. This was done by replacing the 3/32 (0.094) inches thick tube by a sheet metal tube of the same diameter but 0.016 inches thick.

Another important change in the equipment was made at the same time. In the original apparatus, the bottom of the bed of solids was located somewhere in the cone or the approach tube, depending on the solid, the size of the particles and the air velocity. In order to fix the bottom of the bed and thus eliminate one unknown factor, a fine screen was placed near the bottom of the new sheet metal tube and in a similar position in the glass tube.

Three groups of runs were made on the improved apparatus. Runs A-25 to A-28 were steady state runs with an empty tube. The original purpose of these runs was to determine the conductivity of the insulation. However, it was discovered that, although the temperature difference between two radial points, for example, the inside of the insulation and one fourth of an inch out from the metal wall, could be determined easily (by movable thermocouples) and was appreciable, the drop in temperature of the air as it flowed up the tube was very small, amounting to about 4°F in the first twelve inches above the screen. This reduced the accuracy of the heat balance and resulted in

considerable variation in the calculated values of conductivity on either side of the value given in the literature (20). Consequently, it was decided to use the published conductivity of 85 per cent magnesia, namely, 0.041 Btu per hour per square foot per degree Fahrenheit per foot. Two other interesting points were brought out by these runs.

- 1. There was a very sharp temperature drop between the metal and the inside of the insulation. Pulling the insulation more tightly about the tube did not decrease this temperature difference. This corresponded to a film coefficient of heat transfer of between 0.4 and 0.7 Btu per hour per square foot per degree Fahrenheit.
- 2. The film coefficient of heat transfer between the air and the tube corresponded with the published value (21) and was a function of the mass velocity to the 0.8 power. This last relationship has also appeared in the literature (22).

Runs A-31 and A-32 were unsteady state runs with an empty tube. The purpose of these runs was to determine whether the calculations led to a heat balance. Run A-31 was a double run. When steady state had been reached, the air was by-passed around the heater and an unsteady state

^{20.} McAdams, William H. Heat Transmission, Second Edition, New York, McCraw-Hill, 1942, p387.

^{21.} Ibid. pl74.

^{22.} Ibid. pl66

cooling run made.

Runs 9 through 12 were unsteady state runs using 14 to 28 mesh anthracite coal as the solid. Run 9 was both a heating and a cooling run. A summary of these runs appears below and a discussion follows in the next section.

TABLE V
Runs on Improved Apparatus

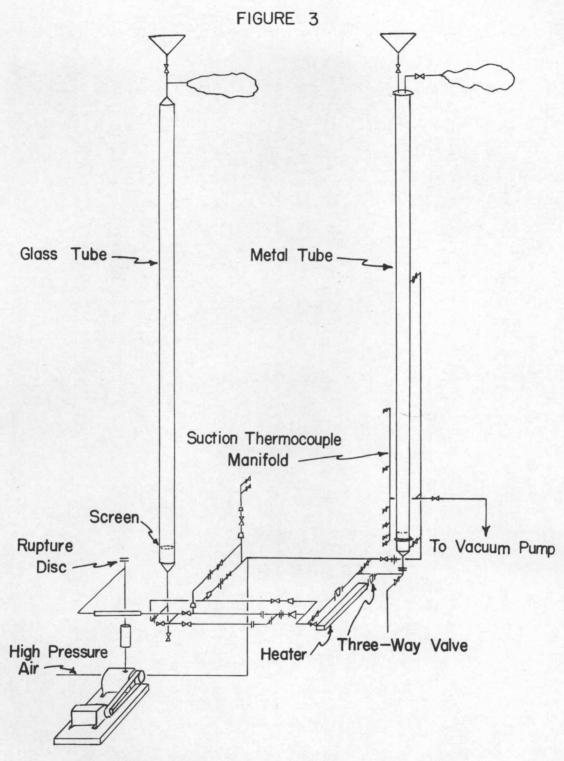
Run	Solid	Weight pounds	Mass Velocity lb per hr-sq ft	Type of Run
A-25	-	_	481	Steady State
A-26	-		485	Steady State
A-27		-	482	Steady State
A-28	-		481	Steady State
A-31	-	-	637	Unsteady State, Heating and Cooling
A-32	-	-	636	Unsteady State, Heating
9	Coal	2	630-636	Unsteady State, Heating and Cooling
10	Coal	2	638	Unsteady State, Heating
11	Coal	1	809	Unsteady State, Heating
12	Coal	12	635	Unsteady State, Heating

APPARATUS

Up to this point, the equipment and the changes made in it have been discussed only in outline. Inasmuch as some of the details were important to the operation and in the calculations, a more complete description of the final apparatus follows:

The air leaving the blower, which was driven by a one horsepower, variable-speed motor, passed through an oil mist filter consisting of a four inch length of three inch pipe filled with steel wool. A rupture disc made by clamping four sheets of waxed paper between the two halves of a flange union was located at an elbow just beyond this point. This part of the system was connected to the rest of the piping by a length of rubber hose in order to reduce vibration. Just above the orifice section was an air bleed which, in combination with the variable-speed motor, made possible control of the air flow at any rate between 80 and 1200 pounds per hour per square foot. Figure 3 is an isometric sketch of the piping and the layout.

To measure the velocity, two thin-plate orifices were used. The smaller was 9/32 inches in diameter in a one inch iron pipe and was used up to about 350 pounds per hour per square foot (in the four inch tube). The larger was 0.500 inches in diameter in a one inch brass pipe. The former was



ISOMETRIC SKETCH OF PIPING AND LAYOUT

used by a previous experimenter (23) on fluidization. The latter was designed according to the rule of thumb that pressure drop in inches of water across the orifice should not exceed the upstream pressure in pounds per square inch absolute (24). Recommendations of the ASME Fluid Meter Committee (25, 26) were followed as closely as possible without elaborate measurements or construction. * Both orifices were calibrated by means of a dry gas meter. The meter was located on the pump suction in some cases and below the orifice at the very lowest rates. This was done because the pump was operating at its minimum limit and a bleed was necessary to decrease the volume of gas passing through the orifice. It was necessary to measure the gas temperature because the heat of compression was appreciable and was not all dissipated in the lines between the pump and the orifices. In operation, the temperature was measured at two points, at the air bleed and about a foot below the orifice, and an arithmetic average taken for the true

23. Frey, loc. cit.

24. Rhodes, Thomas J. Industrial Instruments for Measurement and Control, New York, McGraw Hill, 1941, p197.

26. Fluid Meters - Part 1, 4th ed. American Society of Mechanical Engineers, 1937, 139pl.
 Since the recommendations were not followed exactly,

* Since the recommendations were not followed exactly, the published values of the coefficients were not used. Instead, the coefficients were determined experimentally.

^{25.} American Society of Mechanical Engineers Power Test Codes Part 5, Chap. 4, Flow Measurement by Means of Standardized Nozzles and Orifice Plates, American Society of Mechanical Engineers, 1940, 64p.

temperature at the orifice. The orifice calculations are shown in the appendix. A check valve was placed downstream of the orifices to prevent possible backflow of solids.

The air had three possible paths after it left the check valve:

- (1) It could flow through a 3/4 inch pipe to the bottom of the four inch diameter (4.055 inches average inside diameter) glass tube. From the top of this tube, the air flowed, via a 3/4 inch pipe, to a cotton filter bag and then out into the room.
- (2) It could flow through a one inch line to the electric heater and thence into the bottom of the four inch diameter (3.98 inch average inside diameter) sheet metal tube. From the top of the tube it flowed through a one inch pipe to a filter bag and then out to the room.
- (3) The third path for the air was through a one inch pipe to the base of the tube, by-passing the heater. A two-way, quick-acting plug valve was located in this line.

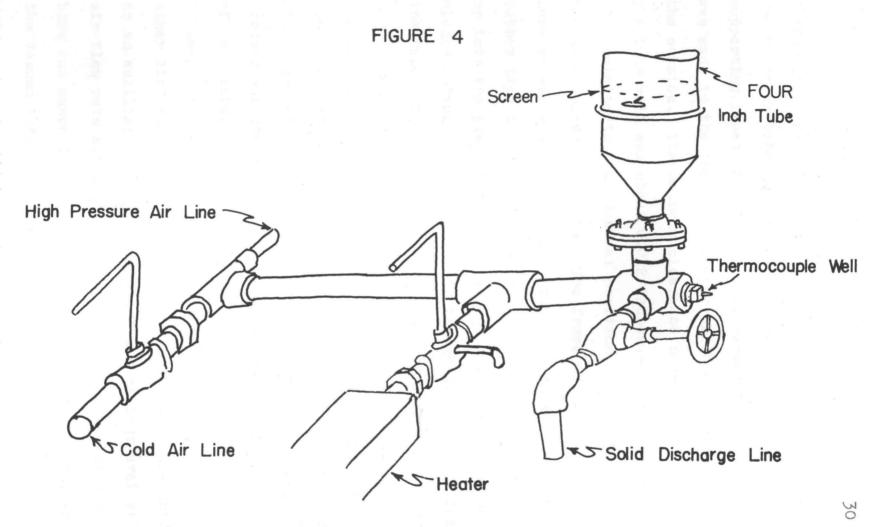
The heater was a 1400 watt, finned space heater, and was enclosed in a 2 inch by 3 inch by $18\frac{1}{2}$ inch sheet iron box with three baffles to force the air to flow parallel to the fins. Electrical contact was made by means of two spark plugs. The spark plugs were brazed to the box and the central terminals were connected to the heater leads. The current to the heater was controlled by a 7.5 'Powerstat', variable transformer. A voltmeter and an

ammeter measured the power input.

Between the heater and the four inch tube, there was a one inch three-way valve made from a 'stop and waste' plug valve. The side opening was a bleed to allow preheat of the air without heating the tube. Because the valve was a plug type, it was quick-acting and eliminated lag in switching from bleed to tube. The connection between the one inch entrance pipe and the four inch tube was a sheet metal cone 2-3/8 inches high. When the 3/32 inches thick tube was replaced by the sheet metal tube, the former was cut off about 2½ inches above the top of the cone and the connection to the sheet metal tube made by flanges. The two flanges were thermally insulated from one another by means of an asbestos paper gasket. The whole system from the heater entrance to the top of the four inch tube was insulated with one inch of 85 per cent magnesia. Figure 4 shows the piping at the base of the tube.

The solid was fed to the sheet metal tube from a hopper above the top of the tube and connected to it by a one inch pipe which extended about one foot into the tube. This was independent of the air exit line and so a small flow of air could be maintained during feeding to keep the bed from packing. In the case of the glass tube, the same pipe was used for feed and exit air and so this could not be done.

Before the screens were placed in the two tubes,



ISOMETRIC SKETCH OF PIPING AT BASE OF TUBE

discharge of particles, in each case, was through a tee in the approach tube a few inches below the tube itself. The supporting frame for the 100 mesh screen in the metal tube was made in the form of a damper which could be rotated from the outside. The maximum clearance between the frame and the tube wall was about 0.012 inches. After the screen had been rotated to the vertical position, discharge was made in the same manner as before. The frame of the 60 mesh glass tube screen was a brass ring around which was wrapped enough rubber tape to give a tight seal when the screen was forced up into the glass tube. The bed of solids was removed by disconnecting the glass tube at the bottom flange, lifting the tube onto a special frame, and then removing the screen.

Because a bed of fine solids packs quite tightly when air is not passing through it, air of several pounds pressure must be available to break it loose. Therefore, high pressure air from an eight cubic feet per minute compressor was connected into the line leading to the bottom of each tube. In addition, air from this source was used to keep the bed aerated during preheating operations. Another line was connected into the system above the orifices as an auxiliary supply, increasing the maximum available air flow rate slightly. High pressure air was also used to blow out manometer connections and to clear the screens on the vacuum thermocouples.

A pressure tap about $2\frac{1}{2}$ inches below the screen in the metal tube and another about five feet above the screen gave the combined pressure drop across the screen and the bed of solids. By subtracting the pressure drop at the same mass velocity with the tube empty, the 'net' pressure drop due to the bed was obtained. The pressure taps for the glass tube were located in the approach pipe and the air exit pipe. A water manometer was used to read the pressure drop and a second one gave the pressure drop across either orifice. A mercury manometer read the pressure at the upstream orifice tap.

Temperatures were measured by seventeen iron-constantan thermocouples. The thermocouple wires were number 20 gage, glass insulated. Lead wires were copper and extended from the cold junction to two multiple selector switches. The electromotive force was measured by a Type 8662 Leeds and Northup potentiometer. During calibration, the cold junction was located in an ice and water slush in a thermos bottle. During operation, the slush was replaced by water at 70°F.* All the thermocouples were calibrated against the freezing point of Bureau of Standards lead. Since the maximum deviation of any of the thermocouples was less than 3°F

^{*}The water in the thermos was adjusted to 70°F before each run and would remain at that temperature for several hours.

over a temperature difference (between hot and cold junctions) more than twice as great as that occurring during operation, a straight line deviation was assumed.

The location and type of each thermocouple are listed in Table VI and shown schematically in Figure 5.

Before Run 12, the well of No. 4 was bent into an 'S' curve so that the end was only about $\frac{1}{4}$ inch below the screen. No. 9 was moved to a position about $\frac{1}{4}$ inch below the screen following Run 12.

Nos. 15 and 16 were used to measure insulation temperatures. During operation, they were forced through the insulation until they hit the metal wall. Since there was no actual contact between the wall and the thermocouples, the temperature read on the thermocouples was that of the inside of the insulation. This was confirmed in the steady state Runs A-25 to A-28. In these runs, insulation temperatures were determined every \(\frac{1}{4}\) inch radially. When the temperatures were plotted against radial distance, the points fell on a smooth curve in every case.

Details of the several types of thermowells have been sketched in Figure 6. Following Run 12, the wells of thermocouples Nos. 5 and No. 7 were cut at a point just outside the tube wall and a piece of 'Micarta', a thermosetting plastic, inserted for thermal insulation. This plastic insert replaced the brass fitting used to screw

FIGURE 5
LOCATION OF INSTRUMENTS

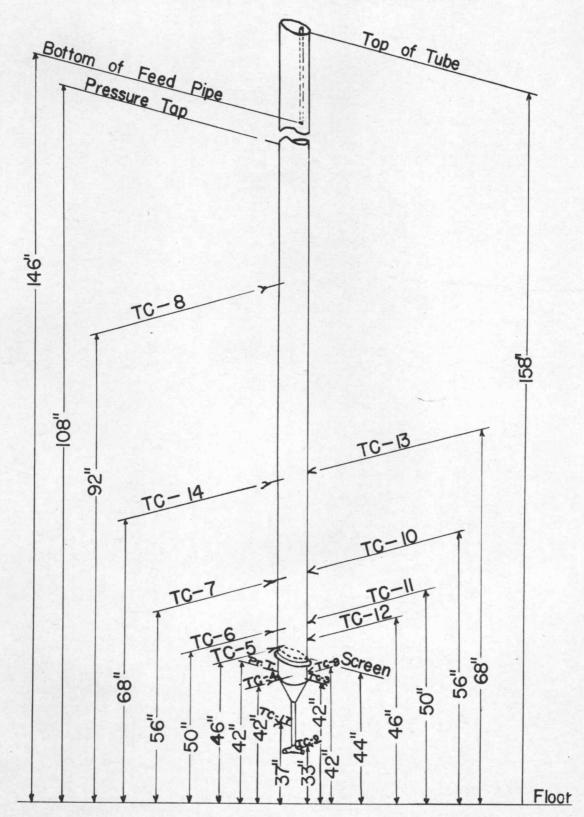


FIGURE 6
THERMOCOUPLE WELLS

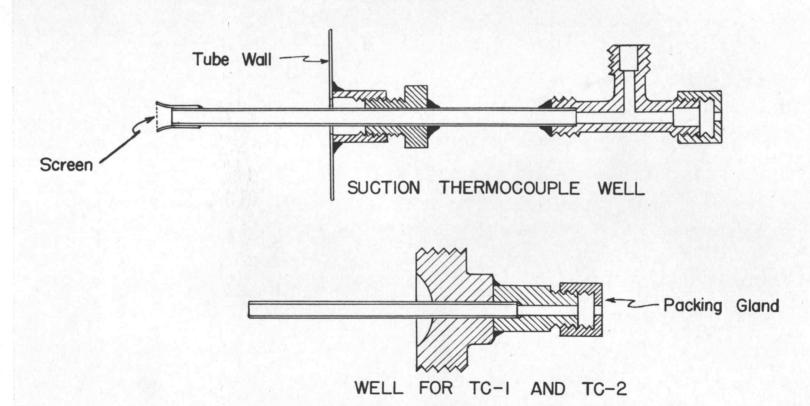


TABLE VI Thermocouples

Thermocouple	Type	Location	Well*	Location of Hot Junction in Relation to End of Well
7	Bare	Below orifices	Open	beyond
1 2	Bare	Approach tube downstream		a beyond a beyond
۵	Dare	of heater	opon	2
3	Bare	21 below screen	Open	1 beyond
	Suction	22" below screen	Open .	1/8" before
4 5		2" above screen	Screen	1/8" before
0	Suction	6" above screen	Screen	1/8" before
6	Suction	a (*) : (*) [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12] : [12]	Screen	1/8" before
7	Suction	12" above screen 48" above screen	Open	1/8" before
8	Suction	2½" below screen	open	
	Wall*			
10	Wall	12" above screen		
11	Wall	6" above screen		
12	Wall	2" above screen		
13	Wall	24" above screen	Comoon	1/8" before
14	Suction	24" above screen	Screen	1/0 perole
15	Insulation	above screen		
16	Insulation	Varying distances	-	
17	Bare	Approach tube downstream of heater	Open	± beyond

^{*} Wells were made of 3/16 inch outside diameter steel tubing.

† Screens were 80 mesh, brazed to flared opening of the well. See Figure 6.

‡ Brazed onto the outside of the metal tube.

the well into the wall fitting.

RESULTS AND DISCUSSION

Fluidization

Since the heat transfer studied in this thesis took place in fluidized beds, it will be desirable to discuss fluidization briefly.

When a stream of gas is passed upward through a bed of fine solids at a sufficient rate, the particles will be set in continuous motion of a rather complex pattern and the mass of solids will have the appearance of a boiling liquid. Carrying the analogy further, there is always some spraying of droplets above the surface of a rapidly boiling liquid. The same thing occurs in fluidization. The spray of fine solids is called the lean phase and the body of the bed, the dense phase. Entrainment of liquids corresponds to the carryover of solids which was mentioned earlier.

Three types of fluidization behavior have been recognized - channeling, 'slugging', and smooth fluidization.

At very low gas velocities, particularly with very fine particles, channels are cut through the bed and relatively poor contact results. If the particles are too coarse, too closely sized, or of too high density, 'slugging' takes place, that is, a large number of the particles bridge across the whole tube and rise as a single body or 'slug' for quite a distance before breaking up. During

smooth fluidization, the particles move along twisting, ever-changing paths and excellent mixing occurs.

Each of these types of fluidization could be recognized in the metal tube by observing the action of the manometer connected across the bed. When channeling was occurring, the liquid levels in the manometer were stationary and the net pressure drop across the bed was smaller than the head of solids. When the bed was slugging, the manometer levels fluctuated widely. During smooth fluidization, the liquid levels had small rapid fluctuations and the net pressure drop was approximately equal to the head of solids.

Theory

It will be recalled that the method of attack was to introduce hot air, suddenly, into a bed of fluidized solids. The air was consequently reduced in temperature and the solid became warmer. If there were no heat losses, all the heat given up by the air would have gone to heat the solid. Expressed mathematically, the heat gained* by the air in time do would be

(1)
$$dQ_a = GA_TCp_a(t_e - t_d)d\theta$$

and the heat gained by the solid would be

^{*}Heat given up will be considered negative

(2)
$$dQ_s = W_s Cp_s dT$$

Rearranging equation (1)

(3)
$$q_a = \frac{dQ}{d\theta}a = GA_TCp_a(t_e - t_d)$$

and dividing both sides of equation (2) by $d\theta$

(4)
$$q_s = \frac{dQ}{d\theta}s = W_s Cp_{sd\theta} \frac{dT}{d\theta}$$

If there were no losses, the equation would read

(5)
$$q_a + q_s = 0$$

However, this was impossible of attainment and so other terms were added to equation (5). The two main heat losses were to the metal wall and to the insulation. This made the heat balance

(6)
$$q_a + q_s + q_m + q_i = 0$$

The 'net' rate of heat flow to the metal can be expressed by an equation similar to that for the solid. By 'net' rate is meant rate at which heat is retained by the metal, i.e., not passed on through to the insulation.

$$q_{m} = W_{m} cp_{m} \frac{dt_{m}}{d\theta}$$

The rate of heat flow into the insulation can be determined from the Fourier conduction equation.

(8)
$$q_{i} = -k_{i}A_{i}\frac{dt_{i}}{dl}$$

The total rate of heat flow to the metal is, then, the sum of q_i and q_m . When this sum is equated to the expression for rate of heat flow through a gas film, the following expression is obtained:

$$(9) q_1 + q_m = h_W A_W \Delta_W$$

In this equation, 'hw' is the film coefficient of heat transfer between the air and the metal wall.

Under steady state conditions, the net rate of heat transfer to the metal would be zero resulting in

$$(10) q_i = h_{WAW}^{A}_{W}$$

Similarly, when steady state prevails, equation (6) reduces to

(11)
$$q_a + q_i = 0$$

The rate of heat transfer to the solid (4) can be equated to an expression similar to equation (9).

(12)
$$q_s = U_s A_s \Delta_s$$

where

(13)
$$\frac{1}{U_S} = \frac{1}{h_S} + \frac{1_S A_S}{k_S A^T}$$

It will be proved, later, that, even for the largest

particles used in these experiments, the second term on the right hand side of equation (13) is negligible, making

(14)
$$q_s = h_s A_s \Delta_s$$

where h_s is the film coefficient of heat transfer between the air and the particles.

The above equations formed the theoretical background for the experiments. In order to determine the value of some of the terms, it was necessary to modify some of them and to make some assumptions.

All the factors on the right hand side of equation (3) were either measured or were available in the literature. In equation (4), however, the solid temperature could not be determined directly and so q_s was left as an unknown to be obtained by difference from equation (6). The metal temperature, on the other hand, was measured at several points and was known as a function of time. In the use of equation (7), the assumption was made that the measured metal temperature was indicative of the true temperature through the metal wall and at all points around the wall at that height. A calculation of the temperature drop through the wall under conditions of maximum heat flow gave about 0.01°F temperature drop. In view of the cylindrical shape, the latter part of the assumption also appeared to be quite reasonable.

In order to determine the rate of heat flow to the insulation, equation (8) was modified by changing the derivative, $\frac{dt_i}{dl}$, to the finite increment form $\frac{\Delta^t}{\Delta l}$. Substituting in the correct area for cylindrical heat flow, i.e., the logarithmic mean area, the expression

(15)
$$q_{i} = -k_{i} 2\pi \left[\frac{r-r^{i}}{\ln \frac{r}{r^{i}}}\right] N \left[\frac{t-t^{i}}{r-r^{i}}\right]$$

was obtained. After changing to common logarithms and cancelling out the equality, the following equation resulted

(16)
$$q_i = \frac{2\pi N k_i(t'-t)}{2.30 \log \frac{r}{r}}$$

where t' was the temperature at the inner surface of the insulation, that is, at a distance r' out from the center and t was the temperature which corresponded to another point, a small finite distance further out from the center of the tube. Two methods were available for the determination of t. It could be measured directly as was done in the steady state runs or could be calculated graphically. The latter was chosen in order to avoid taking additional sets of temperature readings. This allowed readings which could be determined only by the actual measurement to be taken more frequently.

E. Schmidt developed a graphical method for the approximate solution of complex problems in unsteady state

heat conduction. It was based on the use of finite increments in place of differentials in the basic conduction equation for an infinite slab

$$(17) \qquad \frac{\partial t}{\partial \theta} = \sqrt{\frac{2}{x^2}}$$

where

(18)
$$\propto = \frac{k}{Cp}$$

Sherwood and Reed (27) discuss the method in some detail and Perry and Berggren (28) extend it to hollow cylinders. By the proper choice of the time increment, $\Delta\theta$, and the distance increment, Δx , the temperature at any point at any time becomes the arithmetic mean of the two temperatures x away on either side at a time $\Delta\theta$ previously. The relationship between Δx and $\Delta\theta$ which brings this about is

When the shape under consideration is a hollow cylinder, the abscissa scale is made logarithmic. An example of the method has been given in the Appendix.

27. Sherwood, Thomas K. and Reed, Charles E., Applied Mathematics in Chemical Engineering, New York, McGraw-Hill, 1939, pp241-255.

28. Perry, R.L. and Berggren, W.P., Transient Heat Conduction in Hollow Cylinders after Sudden Change of Inner-Surface Temperature, Univ. of Calif. Publications in Engineering Vol. 5, No. 3, pp59-88, 1944.

Typical Calculations

Run 11 has been selected in order to demonstrate the calculations. All the data on this run are given below:

TABLE VII

Run 11 - Heating Run on Dry Utah Coal

Material fluidized: Utah hard coal which had been heated at 300°F for several hours to drive off volatile matter and then cooled and kept in a desiccator.

Weight at start: 1.00 pounds

Weight at end: 1.00 pounds

Tyler standard screen analyses:

	Start	End
-8 + 14 mesh	0	0.005
-14 + 28 mesh	1.00	0.979
-28 + 48 mesh	0	0.019
-48 mesh	0 1.00 pounds	0.005 1.008 pounds

Room temperature: 70°F.

Vacuum pump flow when connected to one vacuum thermocouple: 0.18 cfm.

Duration of run: 83 minutes.

Fluidization: Good.

Orifice: 0.500 inch diameter.

Orifice and heater data:

Time minut		Air Bleed,	o _F	TC - 1	L - °F		inche		inche water		eater oltage
0 8 16 57 71		114		10	04 07 -		- 2.3 2.3 -		18.2 18.2 18.3		98 97 95
Tempe	rature	S:	181								
TC	-4	TC-	- 5	TC-	-7	TC-	-12	TC-	-10	TC.	-15 *
Time	Temp	Time	Temp	Time	Temp	Time	Temp	Time	Temp	Time	Temp
0:20 4:20 10:00 13:00 17:20 22:40 26:50 33:10 38:00 45:10 48:50 53:10 59:00 66:50 73:00	253 259 262 263 265 266 266 265 268	1:45 6:00 10:50 14:00 19:10 24:20 28:30 34:10 40:40 47:00 50:40 55:00 60:20 63:30 74:10	166 235 248 252 255 259 260 260 262 262 262 262 262 262	3:15 7:30 9:00 11:50 15:10 21:00 25:50 29:50 36:20 42:30 51:50 64:50 65:00 76:00	207 238 241 246 250 254 257 257 258 258 260 260 260 260 261	1:30 5:00 10:30 13:30 19:50 23:30 27:50 33:40 39:20 50:10 53:50 59:30 67:40 73:50	150 217 236 242 246 248 249 251 250 254 254 254 254	3:40 9:20 15:50 21:40 29:05 43:10 52:30 70:30	149 197 216 223 229 233 237 238	2:45 6:40 11:20 14:30 20:30 25:00 31:15 35:30 41:40 46:15 55:40 62:45 69:10 74:50	95 119 140 147 153 156 161 163 164 165 166 168

For the 0.500 inch orifice

where

(21)
$$\rho_0 = \frac{1.33(29.9 + P_1 - \frac{\Delta P}{13.6})}{t + 460}$$

*TC-15 located at inside of insulation 4 inches above the screen.
†Time in minutes and seconds after the start.

Using the arithmetic average of the air bleed temperature and TC-1 at 0 minutes

(22)
$$C_0 = \frac{1.33(29.9 + 2.3 - \frac{18.2}{13.6})}{109 + 460} = 0.0721 \text{ lb per cu ft}$$

and

(23)
$$G = 705\sqrt{0.0721(18.2)} = 806$$
 lb per hr-sq ft

The data for other times are summarized in Table VIII.

TABLE VIII

Mass Velocity

Time min	Air Bleed	TC-1	Average oF	Co lb per cu ft	G lb per hr- sq ft
0	114	104	109	0.0721	806
16	111	107	109	0.0721	809
71	113	110	111	0.0718	810

The variation in the orifice temperature was due to the fact that the pipes had not been fully heated by the heat of compression of the pump. Actually, there was a variation in mass velocity from about 796 to 820 pounds per square foot because of variation in the motor speed.

Using 809 pounds per hour per square foot as the average over the run, the flow of air in pounds per hour was calculated.

(24)
$$G^{1} = GA_{T} = \frac{809(3.97)^{2}\pi}{4(144)} = 69.6 \text{ lb per hr}$$

From a run in the glass tube (Run A-30), the depth of the one pound bed at the above mass velocity was determined. See Figures 7 and 8. Because the cross-sectional area of the glass tube was 1.04 times as large as that of the metal tube, the depth equivalent to 1.04 pounds was used. This gave five inches for the depth of the dense phase and approximately six inches more for the lean phase.

To date, it has not been possible to separate the calculations of the lean phase from those of the dense phase and so the calculations below have been based on the total depth of dense phase and lean phase combined.

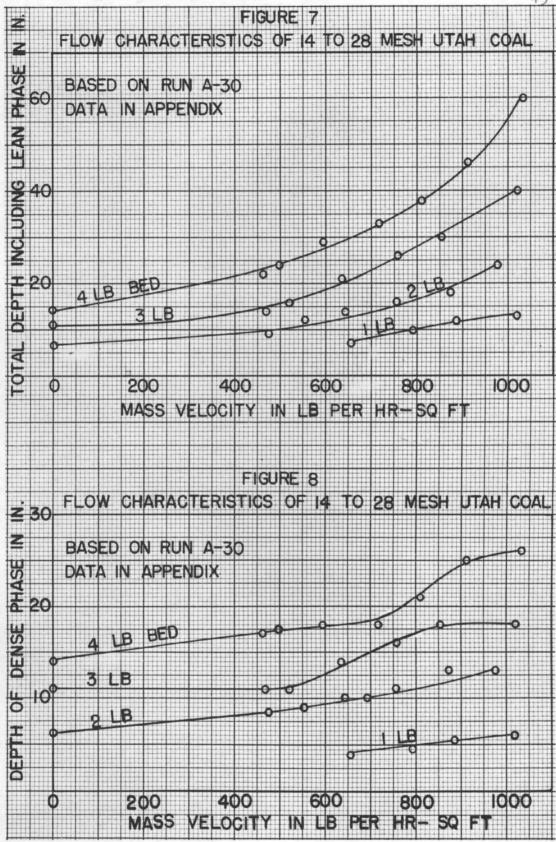
The weight of metal in contact with the bed was as follows:

- (25) Weight of tube wall = 0.0682 (11) = 0.75 pounds
- (26) Weight of two vacuum wells = 0.044 (2) = 0.09 pounds
- (27) Weight of screen and frame = 0.25 pounds

 The heat flow rate equations, based on equations (3), (4),

 (7), and (16), respectively, were:
- (28) $q_8 = 69.6 (0.240)(t_7-t_4) = 17.4(t_4-t_7)$

The specific heat of air is a function of temperature but varies only slightly within the temperature limits employed (70° F to 320° F). Consequently, it has been considered a constant. The bed actually did not extend from TC-4 to TC-7. There was a space of about $2\frac{1}{2}$ inches between TC-4 and the screen which has since been eliminated



and the top of the lean phase was about 11 inches above the screen, an inch below TC-7. The temperature drop of the air in this one inch, as indicated by the runs with an empty tube, was negligible.

(29)
$$q_s = 1.00(0.300) \frac{dT}{d\theta} = 0.300 \frac{dT}{d\theta}$$

The specific heat of the coal (0.300) was determined experimentally (See Appendix). This is also a function of temperature but was assumed constant.

(30)
$$q_{m} = \left[0.84(0.12) + 0.25(0.094)\right] \frac{dt_{R}}{d\theta} = 0.125 \frac{dt_{R}}{d\theta}$$

The values of the specific heat of the metal were obtained from McAdams (29). The temperatures read on the thermocouple number 12 were assumed to be representative of the metal temperature at all points in contact with the bed of solids. This was justifiable because, in earlier runs, all wall thermocouples opposite the fluidized bed read the same or very close to the same temperatures at any given time. For example, as mentioned on page 18, in Runs 2, 3, and 4 two wall thermocouples about twelve inches apart read the same. In Run 9, the maximum difference between wall thermocouples ten inches apart was 5°F. In Run 10, the readings on the same two thermocouples were inseparable. Another assumption, implied in the weight of metal used in

29. Op. cit., p399.

equation (30), was that the longitudinal flow of heat in the metal was negligible compared to radial flow.* This same assumption was made regarding the insulation.

In applying the Schmidt method, room temperature was used as a base, inasmuch as the insulation was all at that temperature at the start of the run.

Inserting the constants into equation (16)

(31)
$$q_{1} = \frac{2\pi(11)(0.041)(t'-t)}{12(2.30)\log\frac{r(12)}{2.0}} = 0.1028 \frac{t'-t}{\log\frac{r(12)}{2.0}}$$

Before these equations could be used, time-temperature plots had to be drawn. See Figure 9.

The rate of heat flow to the air was obtained after determining the difference between \mathbf{t}_4 and \mathbf{t}_7 at various times. For example, five minutes after starting the run, the difference was $29^{\circ}\mathrm{F}$ and, therefore

(32) $q_a = -17.4(29) = -505$ Btu per hr

The values for other times are given in Table IX

Rate of Heat Flow to Air

TABLE IX

Time minutes	(t ₄ -t ₇)	-qa Btu per hr
1	115	2000
2	74	1290
3	48	835
4	37	644
5	29	505

^{*}See Appendix

TABLE IX CONTINUED

Rate of Heat Flow to Air

Time minutes	(t ₄ -t ₇)	-q _a Btu per hr
6	25	435
8	18	313
10	16	278
12	14.5	252
14	13.2	230
16	12.4	216
18	11.8	205
20	11.4	198
25	10.2	177
30	8.1	141
35	7.0	122
40	7.0	122
45	7.5	130
50	8.4	146
60	7.0	122
70	7.9	137

The rate of change of the metal temperature was determined by measuring the slope of the time-temperature curve of TC-12. The data are shown below:

TABLE X

Rate of Heating of Metal

Time minutes	$(\frac{1}{60})\frac{dt}{d\theta}$ oF/min.	q _m Btu per hr
1	60.0	450
2	31.0	232
3	16.2	122
4	10.6	80
5	6.8	51
6	4.7	35
8	3.3	25
10	2.6	20
12	1.83	14
14	1.03	8

TABLE X CONTINUED

Rate of Heating of Metal

Time minutes	$(\frac{1}{60})\frac{dt}{d\theta}$ °F/min.	q _m Btu per hr
16	0.73	5½
18	0.48	31
20	0.60	5 1 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2
25	0.38	3
30	0.21	12
35	0.06	1 1 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2
40	0.00	Õ
45	0.42	3
50	0.18	11/2
60	0.00	0
70	0.00	0

Schmidt Method

The first step in finding the rate at which heat entered the insulation was to determine the value of the thermal diffusivity of the insulation. The density of the insulation was determined by weighing and measuring the volume of a piece of 85% magnesia. The value obtained was 12.5 pounds per cubic foot. The specific heat was taken from the literature (30).

(33)
$$\propto = \frac{k_1}{\text{Cp}_1 \ell_1} = \frac{0.041}{0.234(12.5)} = 0.0140 \text{ sq ft per hr}$$

Recalling that the key equation in the Schmidt method was

30. Perry, John H., ed. Chemical Engineers' Handbook, Second Edition, New York, McGraw-Hill, 1941, p543.

or, in terms of radial measurement,

(34)
$$\propto \frac{\Delta \theta}{(\Delta r)^2} = \frac{1}{2}$$

the size of the radial increment was arbitrarily selected as $\frac{1}{4}$ inch, a value which was small enough to insure accuracy and yet not so small as to make the graphical construction tedious. The time increment was then calculated using equation (34).

(35)
$$\frac{0.0140(\Delta\theta)}{\left[\frac{1}{4(12)}\right]^2} = \frac{1}{2}$$

(36) $\Delta\theta = 0.0155$ hours ~ 0.93 minutes

The values of the abscissa corresponding to the $\frac{1}{4}$ inch increments were found.

TABLE XI

Values of Abscissa in Schmidt Method

Radial Distance - inches 2 $2\frac{1}{4}$ $2\frac{1}{2}$ 2-3/4 3 Log $\frac{\mathbf{r}}{\mathbf{r}_1}$ 0.000 0.0512 0.0969 0.1383 0.1761

To account for the finite resistance to the flow of heat from the outside of the insulation to the room an additional increment was added. This increment had the following value (31):

$$\frac{k_{i}}{2.30h_{r}r_{o}}$$

where r_0 was the outside radius of the insulation and h_r had a value of 1.8 Btu per hour - square foot - o_F , a figure

obtained from the steady state runs, A-25 through A-28 (see Appendix). The numerical value of this term is not critical so that some inaccuracy in hr is unimportant. The abscissa was, then

(38)
$$0.1761 + \frac{0.041(12)}{2.30(1.80)(3.0)} \equiv 0.216$$

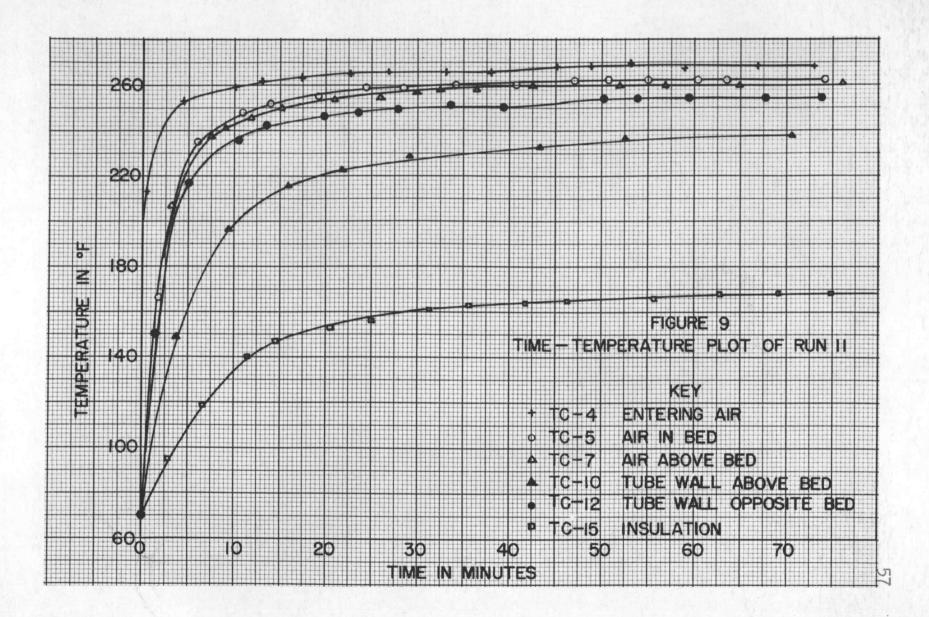
The temperature of the inside of the insulation which corresponded to the time increments (multiples of 0.93) was found next. The values listed in the third column of Table XII were taken from Figure 9. The graphical construction, the details of which are illustrated in the Appendix, was then carried out. Figure 10 shows the completed construction.

⁽³¹⁾ Perry and Berggren, loc. cit.

TABLE XII Flow of Heat into Insulation

Interval	Time minutes	t'*	t"† oF	t'-t"	qi Btu per hr
1 23 45 67 89 01 12 13 14 15 67 18 19 20 22 22 24 25 26 27 28 29 30 30 30 30 30 30 30 30 30 30 30 30 30	0.93 1.86 2.79 3.72 4.65 9.30 10.2 11.0 14.9 9.30 11.2 12.1 13.0 14.9 15.8 17.7 18.5 21.2 23.2 24.2 27.9	78 87.5 96 102 108 114 119 123 136 139 142 144 146 147 148 149 150 151 152 155.5 156.3 157.5 158.3 158.8	70 74 78.5 83.3 87.1 91.5 95.5 99.7 103.2 106.6 110.0 113.2 115.8 118.5 120.8 122.7 124.4 125.5 127.0 128.0 131.8 132.7 133.5 134.3 134.9 135.3 136.0	8 13.5 17.5 180.9 223.3 24.0 25.5 25.5 25.5 25.5 25.5 25.5 25.5 25	1665725679122210977666666665566
35 40 45 50 60 70 80	32.6 37.2 41.9 46.5 55.7 65.0 74.4	161 163 164.8 166 167 167.5 168	145.1	23.4	47

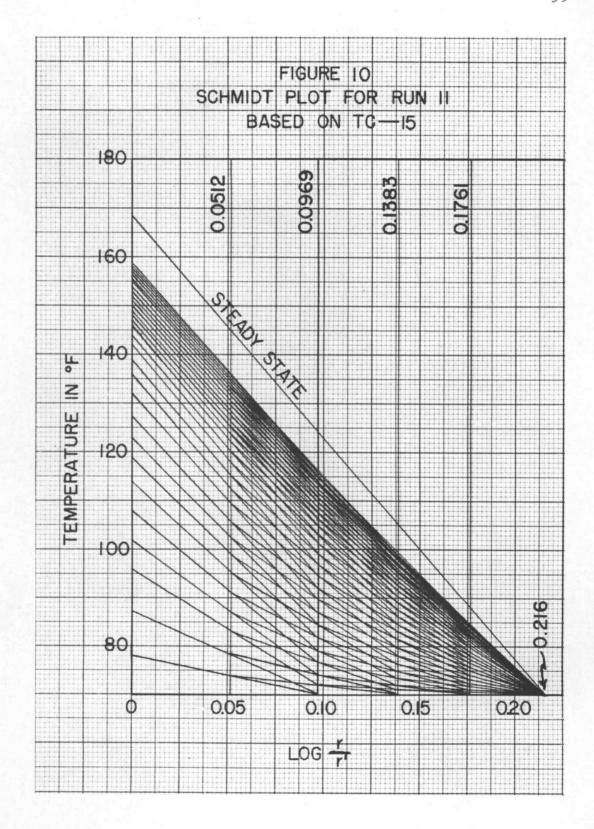
^{*}t'= t_{15} †t" was temperature $2\frac{1}{4}$ inches from center of tube.



The fourth column in Table XII was obtained from the point representing \(\frac{1}{4} \) inch on the temperature distribution curves discussed above. The differences between the first two points on the distribution curves were calculated from columns 3 and 4 in the same table. Assuming that the slope of the chord connecting these two points does not differ very much from the slope of the distribution curve at the inside of the insulation, * these differences were substituted into equation (31) and the rate of heat flow to the insulation was calculated. For example, at 4.65 minutes,

(39)
$$q_1 = \frac{0.1028(20.9)}{\log \frac{2.25}{2.00}} = 42$$
 Btu per hour

^{*}Perry and Berggren reported a maximum difference of 2% for the same number of increments.



The heat flow rates to the insulation are listed in Table XII and plotted against time in Figure 11. Note that after the sixth time interval, the rate of heat flow remained approximately constant. The extrapolation of the rate beyond the thirtieth time interval could be made, therefore, without difficulty.

Coefficient of Heat Transfer between Air and Tube Wall Referring back to equation (9),

(9) $q_1 + q_m = h_W A_W \Delta_W$

it is seen that both the terms on the left hand side are now known. The wall area was calculated on the basis of the eleven inch bed depth.

(40)
$$A_W = 3.97 \text{ II (11)} = 0.95 \text{ square feet}$$

The temperature difference between the air and the metal wall, Δ_W , was known at two points and was assumed to vary logarithmically. The value of Δ_W used in equation (9) was, then, the logarithmic mean of the two known differences.

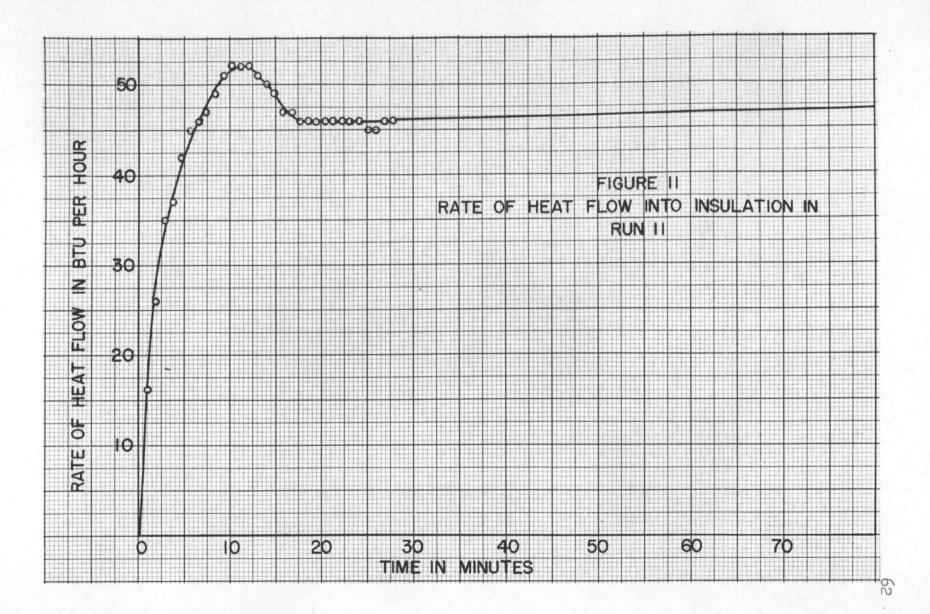
(41)
$$\Delta_W = \frac{(t_4-t_{12}) - (t_7-t_{12})}{t_4-t_{12}}$$

$$\frac{t_4-t_{12}}{t_7-t_{12}}$$

For example, at eight minutes, (t_4-t_{12}) was $27.7^{\circ}F$ and (t_7-t_{12}) was $9.5^{\circ}F$.

$$(42)\Delta_{W} = \frac{27.7 - 9.5}{\ln \frac{27.7}{9.5}} = 17.0 \, ^{\circ}F$$

Substituting all the known values into equation (9)



(43) $50 + 25 = h_w (0.95)(17.0)$

the coefficient of heat transfer from the air to the tube wall was found to be

(44) $h_{W} = 4.6 \text{ Btu per hr-sq ft - }^{\circ}F$

The values for other times are given in Table XIII

TABLE XIII

Run ll - Coefficient of Heat Transfer between Air and Tube Wall

Time minutes	q _i + q _m Btu_per hr	(t ₄ -t ₁₂)	(t7-t12)	o _F w	hw Btu per hr- sq ft - ° F
2 4 6 8 10 12 14 16 18 20	259 119 81 75 72 66 58 53 49 29 50	77.5 45.5 33.0 27.7 24.0 21.6 20.0 19.3 19.0 18.3	4.0 9.0 9.0 9.5 8.0 7.0 6.6 6.8 7.0	24.8 22.5 18.5 17.0 14.6 13.0 11.2 11.5 12.0	11.8 5.6 4.6 4.6 5.3 5.3 5.9 4.3 4.5
25 30 35 40 45 50 60 70	49 47 46 46 46 49 48 46 47	17.0 15.2 14.5 16.0 15.5 14.5 13.0	7.0 7.2 7.5 9.0 8.0 6.2 6.0	11.3 10.7 10.6 12.2 11.4 9.8 9.1 9.5	4.6 4.7 4.6 4.0 4.6 5.2 5.4 5.2

Eliminating the value for two minutes which was subject to large error because of the very rapid change in air temperature in the first few minutes, the average coefficient of heat transfer between the air and the metal tube wall was 4.9 Btu per hour - square foot - OF.

The same calculations were made on Runs 9, 10, and 12. The basic information on these runs is given in Table XIV and the complete data and calculations are presented in the Appendix.

TABLE XIV
Heating Runs on Utah Hard Coal

Run	Weight pounds	Mesh Size	Mass Velocity lb per hr-sq ft
9	2.00	-14+ 28	635
10	2.00	-14+ 28	638
11	1.00	-14+ 28	809
12	0.50	-14+ 28	636

Steady State Heat Flow

If the sole purpose of the above runs had been to find the coefficient of heat transfer between the air and the tube wall, the calculations could have been simplified somewhat by the use of equation (10),

(10) $q_i = h_W A_W \Delta_W$

which holds when steady conditions prevail. Using this equation, the rate of heat flow to the insulation can be calculated from the Schmidt plot. Taking the Schmidt plot of Run 11 (Figure 10) as an example, a straight line is drawn connecting the steady state value of inside insulation temperature, 168.5° F, with the maximum abscissa, 0.216, at 70° F. If $\log \frac{r(12)}{2.00}$ which represents distances along the abscissa, in equation (31)

(31)
$$q_i = 0.1028 \frac{(t'-t)}{\log \frac{r(12)}{2.00}}$$

is replaced by 0.216 and the corresponding values of temperature substituted in, q; can be found.

(45)
$$q_i = 0.1028 \frac{(168.5-70)}{0.216} = 47 \text{ Btu per hour}$$

The coefficient of heat transfer between air and tube wall is then found as before.

Carrying out the same calculation on Runs 9 and 12 (Run 10 was not continued to steady state), the following results were obtained:

TABLE XV
Steady State Values of Coefficient of Heat
Transfer between Air and Wall

Run	qi	Δ_{W}	h _a
	Btu per hr	\circ_{F}	Btu per hr-sq ft - °F
9	70	12.9	4.5
11 12	47 19글	9.5	5.2 4.5

Runs A-31 and A-32 were made with an empty tube at mass velocities of 637 and 636 pounds per hour per square foot, respectively. The only difference in the calculations from those made with solid present, was that the metal temperature varied, becoming lower higher up the tube. This caused a corresponding decrease in insulation temperatures. Consequently, averages had to be used for

the metal and insulation temperatures.

Using the steady state technique just discussed, calculations were made based on the following data:

TABLE XVI

Run A-31 - Steady State Temperatures

Thermocouple 5 12 7 10 15* 16*

Temperature °F 310 275 308 234 174 153

The rate of heat flow through the insulation was

(46)
$$q_i = \frac{2\pi(10)(0.041)(163-71)}{2.30(12)(0.216)} = 40$$
 Btu per hour

and, substituting in equation (10),

(47)
$$40 = h_W \left[\frac{3.97 \pi (10)}{144} \right] \frac{(308-234) - (310-275)}{\ln \frac{(308-234)}{(310-275)}}$$

from which

(48) hw = 0.89 Btu per hour-square foot - oF

TABLE XVII

	Run	A-32 -	Steady	State	Temperati	ares	
Thermocoup Temperatur		5 306	12 268	7 303	10 227	15 189	16 150
The rate of	f heat	flow	through	the in	nsulation	was	

^{*}Located at inside of insulation 4 inches above the screen.
†Located at inside of insulation 9 inches above the screen.
‡Located at inside of insulation opposite screen.

(49)
$$q_i = \frac{2\pi(10)(0.041)(150-76)}{2.30(12)(0.216)} = 36 \text{ Btu per hour}$$

The figure 159° F was obtained by assuming proportionality between t_{15} and t_{16} and calculating an inside insulation temperature halfway between TC-5 and TC-7.

(50) 36 =
$$h_W \left[\frac{3.97 \pi (10)}{144} \right] \left[\frac{(303-227)-(308-268)}{\ln \frac{303-227}{308-268}} \right]$$

(51) $h_{\rm W} = 0.74$ Btu per hour-square foot - ${}^{\circ}{\rm F}$

Although insulation temperatures were not measured in the early runs, it was possible, by means of the resistance concept, to calculate roughly the heat flow rate under steady state conditions and from that a value of $h_{\mathbf{w}}$. The resistance to the flow of heat from the air to the room was made up of the following parts:

- 1. The air film on the inside of the tube wall, $\frac{1}{h_{_{
 m W}}}$.
- 2. The metal wall, a resistance which was negligible compared to the others.
- 3. The air film between the outside of the tube and the insulation, a resistance which can be called $\frac{1}{h_g}$. As mentioned earlier, no matter how tightly the insulation fitted the tube, this resistance still existed.
- 4. The insulation itself.
- 5. The air film on the outside of the insulation, $\frac{1}{h_r}$. The method used in the last few pages, in effect, was to determine the heat flow rate from the known resistances,

(4) and (5), and the measured temperature difference between the inside of the insulation and room temperature. Knowing the heat flow rate and the temperature difference between the air and the tube wall, the coefficient of heat transfer between the air and the wall could be calculated.

Since the insulation temperatures were not measured in the early runs, it was necessary to combine resistances (3), (4), and (5) in order to determine the heat flow rate. However, before this could be done, the value of the coefficient of heat transfer across the gap between the metal and the insulation had to be determined.

The following equation expressed the relationship mathematically.

(52)
$$q_i = h_g A_g \Delta_g$$

where $\Delta_{\rm g}$ was the temperature difference between the metal and the inside of the insulation. The heat flow rates and the required temperatures either have already been given or appear in the appendix for Runs 9, 10, 11, and 12. Substituting these values into equation (52), we have, for Run 9

(53)
$$70.5 = h_g 4.00 \pi (14)(302-197)$$

(54) hg = 0.55 Btu per hour-square foot - oF

For Run 10

(55) 63 =
$$h_g \frac{4.00 \pi (14)}{144}$$
 (285-188)

For Run 11

(57)
$$47 = h_g \frac{4.00 \pi (11)}{144} (254-168.5)$$

For Run 12

(59)
$$19.5 = h_g \frac{4.00 \pi (4)}{144} (288-186)$$

The average of these four values is 0.55 Btu per hour-square foot - °F.

Using the same technique for this resistance as for that on the outside of the insulation, an increment which had the following value was added

(61)
$$\frac{k_1}{2.30 \text{ hgr}^2} = \frac{0.041(12)}{2.30(0.55)(2.00)} = 0.194$$

The three increments corresponding to resistances (3), (4), and (5) had, then, the following values: 0.194, 0.176, and 0.040, respectively. Their sum is 0.410. The equation for heat flow rate would then read

(62)
$$q_1 = \frac{2 \pi N(0.041)(t_m - t_0)}{2.30 (0.410)}$$

In Run 2, five pounds of a fraction of 3A catalyst averaging about 150 mesh in size* were fluidized. The air had a mass velocity of 193 pounds per hour per square foot. Although steady state was not reached, it was approached close enough so that temperatures could be extrapolated to this condition. These temperatures were:

TABLE XVIII

Run 2 - Steady State Temperatures

Thermocouple Vacuum Wall Room Temperature F 247 245 74

For the twelve inches between the two sets of thermocouples

(63)
$$q_1 = \frac{2\pi(12)(0.041)(245-74)}{2.30(12)(0.410)} = 46 \text{ Btu per hour}$$

and

(64) 46 = hw
$$\frac{4.00 \,\pi(12)}{144}$$
 (247-245)

(65) $h_W = 22$ Btu per hour-square foot - ${}^{\circ}F$

In Run 4, five pounds of Ottawa Standard Silica Sand were fluidized. This material had an average particle size of about 40 mesh.** The air had a mass velocity of 813 pounds per hour per square foot. The steady state temperatures were as follows:

^{*}See Table I and Appendix

TABLE XIX

Run 4 - Steady State Temperatures

Thermocouple Vacuum Wall Room Temperature OF 287 283 75

For the twelve inches between the two sets of thermocouples

(66)
$$q_i = \frac{2\pi(12)(0.041)(283-75)}{2.30(12)(0.410)} = 56 \text{ Btu per hour}$$

(67)
$$56 = hw \frac{4.00 \text{ Tr}(12)}{144} (287-283)$$

At the end of Run 5, there was approximately 0.8 of a pound of wood charcoal, which averaged about 20 mesh in size,* in the tube. The mass velocity of the air was 548 pounds per hour per square foot. The steady state temperatures were as follows:

TABLE XX

Run 5 - Steady State Temperatures

Thermocouple	Vacuum	Wall	Room
	(1) (2)	(1) (2)	
Temperature OF	256 247	235 241	81

For the twelve inches between the two sets of thermocouples

(69)
$$q_i = (0.269)(238-81) = 41$$
 Btu per hour

and

(70)
$$41 = h_W \frac{\pi}{3}$$

$$\frac{(256-235) - (247-241)}{\ln \frac{256-235}{247-241}}$$

^{*}See Appendix

(71) hw = 3.2 Btu per hour-square foot - OF

All the steady state values of the heat transfer coefficient between the fluidizing air and the tube wall are presented in Table XXI.

Material Fluidized	Approximate Avg. Particle Size	Mass Velocity lb per hr-sq ft	h _W Btu per hr- sq ft - °F
3A Catalyst	150 mesh	193	22
Sand	40 11	813	13
Coal	20 "	809	5.2
Coal	20 "	636	4.5
Coal	20 "	635	4.5
Charcoal	20 "	548	3.2
Empty Tube	하라요? 아이프랑 사람의 화면 없었다.	637	0.89
Empty Tube	-	636	0.74

The data are not extensive enough to derive equations relating the film coefficient to the mass velocity and the particle size. However, some qualitative conclusions can be drawn:

- 1. The film coefficient of heat transfer between the air and the tube wall is much greater when fluidized solids are present than when the tube is empty.
- 2. The film coefficient increases with increasing mass velocity. In fact, it increases at about the same rate as the mass velocity. This can be expressed mathematically as

(72)
$$h_{w} = f(G)^{m}$$

3. The film coefficient increases with decreasing particle size. The increase is at least inversely proportional to the particle size and may be much greater than that. In equation form

(73)
$$h_w = f'(D_p)^{-n}$$
 $n > 1$

Combining (72) and (73)

(74)
$$h_{W} = \phi (G)^{m} (D_{p})^{-n}$$

Coefficient of Heat Transfer between Air and Solid

Referring back to equation (6),

(6)
$$q_a + q_s + q_m + q_i = 0$$

it was believed that the rate of heat flow to the solid, q_s , could be calculated by difference and originally it was hoped that such rates could be calculated with sufficient accuracy to determine the coefficient of heat transfer between the air and the solid particles (Equation 14). However, as will be seen below, unaccountable losses made this method too inaccurate.

When the third column of Table IX, the rate of heat flow to the air, q_a , was combined with the third column of of Table X, the rate of heat flow to the metal, q_m , and the sixth column of Table XII, the rate of heat flow to the insulation, q_i , a series of values was obtained which should have, in the absence of other losses, given the rate of heat flow to the bed of fluidized solids. These data are given in Table XXII and plotted in Figure 12.

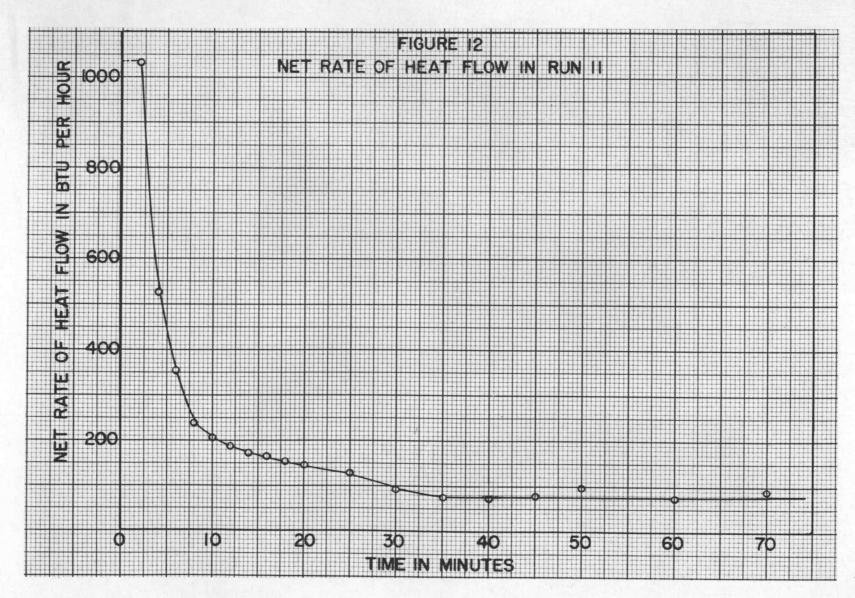


TABLE XXII

Run 11 - Net Rate of Heat Flow

Time Minutes	-q _a -(q _i +q _m) Btu per hr	q* Btu per hr
2 4 6 8 10 12 14 16 18 20	1031 525 354 238 206 186 172 163 155	559 195 90 54 34 24 20 17 13
25 30 35 40 45 50 60 70	128 93 75 75 80 98 75 90	7 4 2 4 2 4 0 0

The ordinate in Figure 12 is q, which equals $\frac{dQ}{d\theta}$, and the abscissa is time, θ . The area under the curve can be expressed as

(75)
$$\int_0^{\Theta} \frac{dQ}{d\Theta} d\Theta = Q$$

If there were no losses besides those to metal and insulation, Q would represent the heat absorbed by the solid and, from the known weight and specific heat, the average solid temperature could be calculated.

(76)
$$Q_{\rm S} = W_{\rm S} C_{\rm ps} (T-t_{\rm O})$$

^{*}Based on equilibrium at top of bed

For the case under consideration, at the end of 70 minutes the area under the curve amounted to 189 Btu, assuming a constant value of q from the start to two minutes. Setting the known values in (76)

(77) 189 = 1.00(0.300)(T-70) from which

$$(78)$$
 T = 700° F

a temperature which is obviously too high, inasmuch as the maximum entering gas temperature was only 268° F. Assuming that the solid was in equilibrium with the air leaving the bed at all times and that mixing was so rapid that the particles throughout the bed were all at the same temperature, an approximate rate of heat flow to the solid, $q_{\rm s}$, was calculated using equation (29)

(79)
$$q_s = 0.300 \frac{dt_7}{d\theta}$$

The values obtained are shown in the third column of Table XXII. It can be seen from the table that the approximate curve of q_s had the same shape as the curve plotted in Figure 12. The difference between the two curves, that is, the rate of heat flow to the unaccountable losses also had the same general shape, high in the first few minutes and decreasing rapidly with time.

In attempting to isolate the unaccountable losses, two unsteady state runs were made with an empty tube. These runs, A-31 and A-32, have already been discussed from the

standpoint of hw under steady state conditions. A resumé of the data and calculations on Run A-31 is given below. Run A-32 was similar to Run A-31.

TABLE XXIII

Run A-31 - Unsteady State Run with Empty Tube

Duration of Heating Cycle: 140 minutes Average mass velocity:

637 lb per hr-sq ft

TABLE XXIV

Run A-31 - Heat Flow Rate Equations

(80)
$$q_a = \frac{637(12.40)(0.240)}{144} (t_{14}-t_4) = -13.2(t_4-t_{14})$$

$$(81)^* \ q_m = \left[0.66(0.12) + 0.25(0.094)\right] \frac{dt_{12} + 0.82(0.12) + (0.084)}{d\theta} \frac{dt_{10}}{d\theta}$$

$$+ \left[0.41(0.12) + 0.25(0.094)\right] \frac{dt_{13}}{d\theta}$$

$$= 0.103 \frac{dt_{12}}{d\theta} + 0.110 \frac{dt_{10}}{d\theta} + 0.061 \frac{dt_{13}}{d\theta}$$

$$(82)^{\dagger} q_{1} = \frac{2\pi(13)(0.041)(t_{15}^{\dagger} - t_{15}^{"})}{2.30(12) \log \frac{2.25}{2.00}} + \frac{2\pi(11)(0.041)(t_{16}^{\dagger} - t_{16}^{"})}{2.30(12) \log \frac{2.25}{2.00}}$$

$$= 2.37(t_{15}^{\dagger} - t_{15}^{"}) + 2.00(t_{16}^{\dagger} - t_{16}^{"})$$

[&]quot;This equation given in three parts because of the variation in metal temperature.

[†] This equation given in two parts because of the variation in insulation temperature.

TABLE XXV

Run A-31 - Heat Flow Rates

Time Minutes	q _i Btu per hr	q _m Btu per hr	q _i + q _m Btu per hr	-q _a Btu per hr
2	22	270	292	340
5	52	148	200	240
10	68	85	153	160
15	70	26	96	115
20	71	19	90	105
30	72	11	83	80
40	74	8	82	80
60	76	3	79	80

The unbalance (q_i+q_m + -q_a) was small in the first few minutes and decreased to approximately zero after 30 minutes. The early unbalance can be traced to the weight of metal wall between TC-4 and the screen. As mentioned earlier, this difficulty has been corrected by locating the hot junction of TC-4 just below the screen. These calculations eliminated a number of possible sources of error such as incorrect measurement of mass flow rate, an air leak between the orifice and the tube, incorrect method of calculation, too large a volume of air flowing through the vacuum thermocouple wells, incorrect rate of heat flow to the metal, or incorrect rate of heat flow to the insulation.

From the shape of the curve of the rate of heat flow to the unaccountable losses, the fact that the rates balanced in the run with the empty tube, and the much more rapid rate of heat transfer to the metal when solid was present, it was believed that the greater part of these losses was due to the various metal fittings which extended through the insulation. As pointed out earlier (p33), plastic pieces were inserted to thermally insulate the thermocouple wells from the metal wall and from the outside. However, before trying out these changes, it was decided to make a calculation based on an assumed coefficient of heat transfer between the air and the particles of 5.2 Btu per hour-square foot - $^{\rm O}$ F, the value obtained in Run 11 for $h_{\rm W}$.

Assuming that the particles have a surface area 1.75***
times as large as spheres of the same average size, the
total surface area of one pound of 14 to 28 mesh coal
would be in the neighborhood of 50 square feet. Substituting into equation (14)

(14) $q_s = h_s A_s \Delta_s$

the value of qs for two minutes in Table XXII

(83) $559 = 5.2(50)\Delta_g$

the mean value of Δ_8 was found to be 2° F. Inasmuch as the temperature difference between air and solid--assuming the solid temperature was very close to the outgoing air temperature--was about 74° F at the entrance to the bed, the temperature difference would have to have been about 10^{-80} F to satisfy the logarithmic mean, i.e.,

^{*}See Appendix
**Personal communication from Prof. J. Schulein, Aug. 1948.

$$(84) \quad 2 = \frac{74 - \Delta_e}{\ln \frac{74}{\Delta_e}}$$

at the top of the fluidized bed. It can be seen from this that, if the film coefficient between air and particles is of the same order of magnitude as the coefficient between air and tube wall, the temperature difference between air and solid will be extremely small at the top of the bed. Therefore, even if all the heat could be accounted for, the method of determining the solid temperature by means of the area under the curve of $\mathbf{q}_{\mathbf{S}}$ versus time would not lead to a correct value of $\boldsymbol{\Delta}_{\mathbf{S}}$.

Despite the failure of the attempt to determine the coefficient of heat transfer between the air and the fluidized particles, it was felt that it was worth while to describe the calculations for unsteady state heat flow. Extension of the experiments to higher temperatures or larger particles might serve to break the equilibrium between the air leaving the bed and the solid particles.

Before closing this section it might be well to mention the relative resistances of the air film around the particle and the particle itself. The conductivity of coal is 0.111 Btu per hour (square foot) (°F) per foot (31) and half the diameter of the average particle in the 14 to 28 mesh range is about 0.0173 inches. If h_s were 5, the

(31) National Research Council, Chemistry of Coal Utilization, Volume I, John Wiley, 1945, pp320-6.

respective resistances would be

$$\frac{1}{K} = \frac{0.0173}{12(0.111)} = 0.013$$

(86)
$$\frac{1}{h_s} = \frac{1}{5} = 0.20$$

The resistance due to the particle itself is, therefore, only a small fraction of the total resistance.

CONCLUSIONS

The following values of the heat transfer coefficient between the air stream and the tube wall were obtained:

TABLE XXVI
Steady State Values of Coefficient of Heat
Transfer between Air and Wall

Material Fluidized	Approx. Avg. Particle Size	Mass Velocity lb per hr-sq ft	Heat Transfer Coefficient Btu per hr-sq ft - OF
3A Catalyst	150 mesh	193	22
Sand	40 11	813	13
Coal	20 11	809	5.2
Coal	20 "	636	4.5
Coal	20 "	635	4.5
Charcoal	20 "	548	3.2
Empty Tube		637	0.89
Empty Tube	-	636	0.74

From the above values, the following conclusions are drawn:

- 1. The film coefficient of heat transfer between an air stream and the containing tube wall is much larger when fluidized solids are present than when the tube is empty.
- 2. The film coefficient of heat transfer between the air and the tube is a function of the mass velocity to some power approximately equal to one.
- 3. The film coefficient of heat transfer between the air and the tube is an inverse function of the particle size to some power which is probably greater than one.
 - 4. The rate of heat transfer between a bed of

fluidized solids and the fluidizing gas is very rapid. One factor which brings this about is the large area available for heat transfer.

RECOMMENDATIONS

It is recommended that:

- 1. Further extensive experimental work be carried out on the coefficient of heat transfer between the air stream and the tube wall. The effect of the following factors should be investigated: mass velocity, particle size, type of fluidization, material fluidized, and particle concentration. These experiments can be done batchwise, with steady state heat flow.
- 2. An attempt to determine the coefficient of heat transfer between the air and the fluidized particles be made using continuous feed and discharge of solid particles. However, before this is attempted, a thorough study should be made of the various flow rates to determine whether the coefficient can be determined by this method.

APPENDIX

ADDITIONAL EQUIPMENT DATA

Length of sheet metal tube: 9 feet $7\frac{1}{2}$ inches

Weight of sheet metal tube: 7 pounds 14 ounces

Thickness of sheet metal tube: 26 gage

Average inside diameter of sheet metal tube: 3.97 inches

Average inside diameter of glass tube: 4.055 inches

Cross-section area of metal tube: $\frac{\pi}{4}$ (3.97)² = 12.90 sq in.

Cross-section area of glass tube: $\frac{\pi}{4}$ (4.055)²= 12.90 sq in.

Cross-section area of vacuum thermocouple well:

$$\frac{\pi}{4}$$
 (0.140)² = 0.0154 sq in.

Density of insulation

Volume of test piece: $\frac{\pi[8.5)^2 - (5.75)^2](36)}{4} = 0.321$ cuft

Weight of test piece: 4 lb 0 ounces

Density: $\frac{4}{0.321}$ = 12.5 lb per cu ft

ORIFICE CALIBRATION

TABLE XXVII

Nomenclature for Orifice Calibration

G = mass velocity; lb per hr-sq ft welocity in orifice, feet per second Um = superficial velocity in metal tube, feet per second = superficial velocity in glass tube, feet per second = gas temperature, OR To = gas temperature in orifice, OR TR = room temperature, OR = gas density, pound per cu ft = gas density in orifice, lb per cu ft CFM = cu ft of gas metered per min = upstream orifice pressure, inches mercury AP = pressure drop across orifice - inches water △P' = pressure drop across orifice @ 70° F and 30.2 in. mercury downstream pressure = orifice diameter - inches Do = orifice coefficient = pressure drop across orifice, feet of fluid flowing For metal tube 3.81 in. I.D.* (87) $G = \frac{(cfm)(60)(4)(144)(29)(492)}{\pi(3.81)^2(359)T_R} = \frac{30000(cfm)}{T_R}$ (88) $c = \frac{29(492)(29.9 + P_1 + \frac{\Delta P}{13.6})}{(359)T_0(29.9)} = \frac{1.33(29.9 + P_1 - \frac{\Delta P}{13.6})}{T_0}$ (89) $U_0 = \frac{c\sqrt{2gh}}{\sqrt{1 - (\frac{D_0}{1.049})^2}} = \frac{c\sqrt{\frac{2(32.2)(62.4)\Delta p}{12\rho_0}}}{\sqrt{1 - (\frac{D_0}{1.049})^2}}$ For 0.500 inch orifice (90) $U_0 = 20.80 \frac{\Delta P}{60}$

^{*}Orifices were calibrated with original 4 inch o.d.(3/32 in. thick tube. Conversion to sheet metal tube and glass tube is given on p93.

For 9/32 inch orifice

(91)
$$U_0 = 19.00\sqrt{\frac{\Delta P}{\rho_0}}$$

(92)
$$G = 3600 U_{0} (\frac{D_{0}}{3.81})^{2}$$

For 0.500 inch orifice

For 9/32 inch orifice

TABLE XXVIII

CALIBRATION OF 0.500 inch ORIFICE

Upstream Pressure Pl in. mercury	Pressure Drop AP in. water	Gas Flow cfm	Room Temp T _R OR	Orifice Temp OR	Density of Air - (o lb per cu ft	Corrected Press. Drop	Mass Velocity G - 1b per hr-sq ft	Remarks
0.3	0.7	2.7	536	537	0.0749	0.69	151)	Not used in
0.4	1.3	3.8	536	536	0.0750	1.29	212)	calculations
0.5	2.3	5.2	536	536	0.0751	2.28	291)	△P - too small
0.5	3.0	6.4	536	537	0.0749	2.96	358	
0.5	3.0	6.5	528	528	0.0760	3.01	369	
0.6	3.8	7.2	528	551	0.0730	3.66	409	
0.6	4.6	8.2	528	539	0.0745	4.52	465	
1.9	5.1	8.6	528	539	0.0775	5.22	488	
1.0	7.8	10.1	528	533	0.0757	7.79	574	
1.0	7.5	10.0	528	551	0.0737	7.28	568	
1.2	9.4	11.4	532	534	0.0760	9.42	643	
1.3	11.5	12.1	528	535	0.0756	11.48	687	
1.5	12.6	13.1	528	553	0.0735	12.20	744	
1.5	12.9	13.3	528	553	0.0735	12.50	755	
1.7	15.2	14.8	528	538	0.0755	15.15	840	
2.05	18.3	16.0	528	540	0.0755	18.23	909	
2.1	18.8	16.0	528	553	0.0737	18.30	909	
2.2	19.6	16.4	528	553	0.0740	19.13	931	
2.3	19.8	16.8	528	543	0.0753	19.7	955	
2.5	22.3	17.7	532	540	0.0758	22.3	1000	
2.55	22.8	17.8	532	537	0.0763	22.9	1004	
2.75	25.1	18.8	528	550	0.0745	24.7	1068	
3.2	29.2	19.8	528	530	0.0764	29.4	1123	
3.0	27.0	19.4	528	553	0.0745	26.5	llol	
5.6	32.5	21.2	528	543	0.0811	34.8	1204	

TABLE XXVIII CONTINUED

CALIBRATION OF 0.500 inch ORIFICE

Pressure Pa	Pressure Drop AP in. water	cfm	Temp	Temp OR	Density of Air - 6 1b per cu ft	Press. Drop	Mass Velocity G - lb per hr-sq ft	Remarks	
3.0 3.4 3.6	27.5 31.6 34.1	19.8 20.8 21.75	528	543	0.0745 0.0760 0.0750	27.0 31.7 33.8	1124 1181 1235		

TABLE XXIX

CALIBRATION OF 9/32 inch ORIFICE

Upstream Pressure P _l in•mercury	Pressure Drop AP in. water	Gas Flow cfm	Room Temp T _R OR	Temp OR	Density of Air - (o lb per cu ft		Mass Velocity G - 1b per hr-sq ft	Remarks
0.4	3.4	2.2	536	536	0.0746	3.35	123	Not used in calcutions
0.4	3.2	1.8	536	536	0.0746	3.15	101	
0.6	6.0	2.5	536	536	0.0746	5.90	140	
0.8	8.8	3.0	536	536	0.0746	8.67	168	
0.8	9.0	3.2	536	536	0.0746	8.86	179	
1.0	11.5	3.6	536	537	0.0748	11.35	201	
1.4	17.4	4.4	536	539	0.0744	17.10	246	
2.0	17.5	4.6	530	540	0.0755	17.45	260	
1.6	19.3	4.6	536	538	0.0747	19.03	257	
2.1	19.5	4.9	530	540	0.0755	19.45	277	
2.0	24.8	5.3	536	538	0.0746	24.4	296	
2.6	25.7	5.9	530	540	0.0755	25.6	334	
2.4	29.0	6.2	530	541	0.0744	28.4	351	

In order to put all the data on the same basis: Base temperature 70° F

Base downstream orifice pressure - 30.2 in. mercury

Co = 0.0758 under these conditions

(95)
$$C_0\Delta P = 0.0758 (\Delta P^1)$$

(96)
$$G = K_1 (\Delta P^1)^{\frac{1}{2}}$$

(97)
$$U_{\rm m} = \frac{G}{3600e}$$

Using method of residual summation

(98)
$$\log G = \log K_1 + \frac{1}{2} \log (\Delta P^1)$$

(99)
$$\sum \log G = N \log K_1 + \frac{1}{2} \sum \log (\Delta P^1)$$

From the data on the 0.500 inch orifice:

(100)
$$74.8505 = 26 \log K_1 + \frac{1}{2}(28.7658)$$

(101)
$$K_1 = 211.7$$

$$(103)$$
 C = 0.597

(104)
$$G = 1290 (0.597) \sqrt{\rho_0 \Delta P} = 770 \sqrt{\rho_0 \Delta P}$$

From the data on the 9/32 inch orifice:

(105) 28.1304 = 12 log
$$k_2 + \frac{1}{2}(13.3017)$$

$$(107)$$
 C = 0.604

(108) G = 372 (0.604)
$$\sqrt{\rho_0 \Delta P}$$
 = 224 $\sqrt{\rho_0 \Delta P}$

FIGURE 13 MASS VELOCITY IN 3.81 INCH DIAMETER METAL TUBE VS. PRESSURE DROP ACROSS ORIFICES F 80 1000 0.500 IN. ORIFICE LB PER HR-800 600 400 Z MASS VELOCITY 200 100 5 6 8 20 30 ΔP', PRESSURE DROP ACROSS ORIFICE AT 70°F IN IN. OF WATER

Conversion of mass velocities in original 3/32 inch tube to new sheet metal tube and glass tube:

Ratio of cross-section area of old tube to new tube: $\frac{11.36}{12.40}$

= 0.915

Ratio of cross-section area of old tube to glass tube:

TABLE XXX

Mass Velocity Equations

	Sheet Metal	Glass
0.500 inch orifice	705 √C0 AP	675 V COAP
9/32 inch orifice	206 (COAP	197 ₹ 60 AP

THERMOCOUPLE CALIBRATION

Calibration against Bureau of Standards Lead, the freezing point of which is 17.81 millivolts when the cold junction is at $32^{\circ}F$.

TABLE XXI
Thermocouple Calibration

Thermocouple	Readings in Millivolts	Average Deviation Millivolts
1	17.80 17.81	-0.005
2	17.79 17.75	-0.04
3	17.78 17.81	-0.01
4	17.82 17.81 17.81	0.00
5	17.82 17.76	-0.01
6	17.81 17.81 17.81	0.00
7	17.79 17.84	+ 0.04
8	17.84 17.79	0.00
9	17.83 17.81 17.81	0.00
10	17.76 17.74	-0.06
11	17.74 17.72	-0.08
12	17.74 17.79	-0.045
13	17.74 17.78	-0.05
14	17.75 17.78 17.74 17.74	-0.06

TABLE XXXI CONTINUED

Thermocouple Calibration

Thermocouple	Readings in Millivolts	Average Deviation Millivolts
15	17.83 17.83	+ 0.02
16	17.81 17.80	0.00
17	17.77 17.77	-0.04

TEMPERATURE VS MILLIVOLTS FOR IRON-CONSTANTAN THERMOCOUPLE WITH COLD JUNCTION AT 70°F

TABLE XXXII

	0	100	200	300	400
$\circ_{\mathbb{F}}$			Millivolts		
0 5 10 15 20 25 30 35 40 45 50 55 60 65 70 75 80 85 90 90 100	0 0.14 0.28 0.43 0.57 0.72 0.86	0.86 1.01 1.16 1.30 1.45 1.60 1.75 1.90 2.19 2.49 2.49 2.64 2.78 2.64 2.78 3.08 3.52 3.67 3.82	3.82 3.97 4.12 4.27 4.43 4.58 4.73 4.88 5.03 5.18 5.48 5.64 5.99 6.40 6.55 6.70 6.85	6.85 7.00 7.16 7.31 7.46 7.62 7.77 7.92 8.08 8.23 8.54 8.69 9.15 9.30 9.15 9.45 9.61 9.76 9.92	9.92 10.08 10.22 10.38 10.53 10.69 10.84 10.99 11.15 11.30 11.45 11.61 11.76 11.92 12.07 12.22 12.38 12.53 12.68 12.84 12.99

TABLE XXXIII

FLUIDIZATION CHARACTERISTICS OF 14 to 28 MESH UTAH COAL

The data below were taken in Run A-30, using the glass tube.

Weight pounds	Mass Velocity lb per hr-sq ft	Depth of Dense Phase - inches	Total Depth* inches	Type of Fluidization
1 1 1	656 792 886 1020	4 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	7 10 12 13	Smooth " " " "
2 2 2 2	475 552 642 754 871 978	8½ 9 10 11 13 13	9 12 14 16 18 24	Channeling Smooth " " " Slugging infrequently
3 3 3 3 3 3 3 3 3	469 520 635 758 854 1020	11 11 14 16 18 18	14 16 21 26 30 40	Channeling Slugging infrequently Slugging, more frequently Slugging, frequently Slugging, frequently Slugging
<u>λ</u> ₄ λ ₄	461 499	17 17½	22 21 ₄	Slugging

^{*}Including maximum depth of lean phase.

TABLE XXXIII CONTINUED

Weight pounds	Mass Velocity lb per hr-sq ft	Depth of Dense Phase - inches	Total Depth inches	Type of Fluidization
14	594	18	29	Slugging " " " " "
14	716	18	33	
14	810	21	38	
14	911	25	46	
14	1033	26	60	

NET PRESSURE DROP

It was stated earlier (pl3) that the 'net' pressure drop across a bed of fluidized solids is approximately equal to the head of solids. For example, in Run A-30, the following data were obtained:

Weight of bed: 3.0 pounds

Pressure drop across 0.500 inch orifice: 17.0 inches of

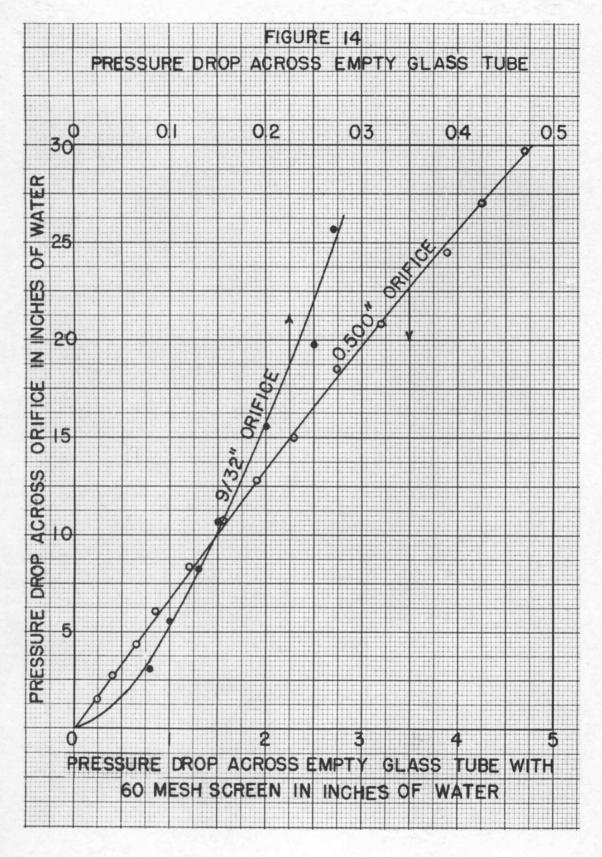
water

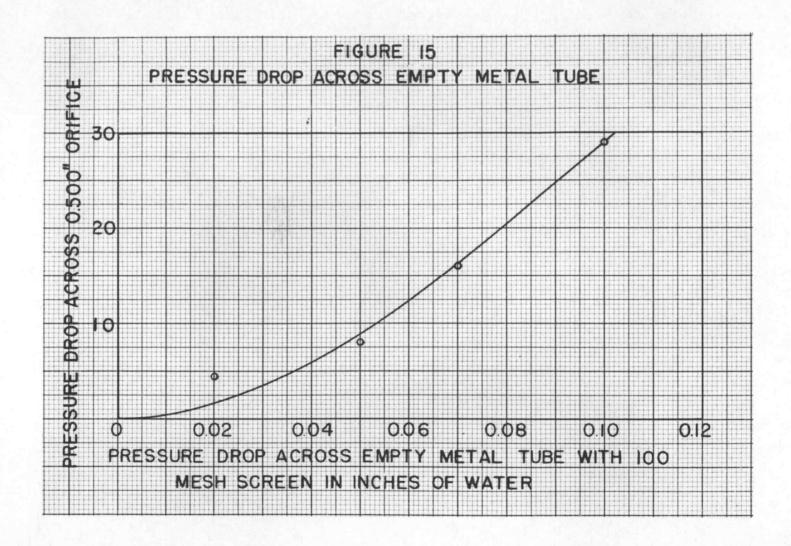
Pressure drop across bed: 9.0 inches of water
Mass Velocity: 758 lb per hr-sq ft

Referring to Figure 14, the pressure drop across the empty glass tube at this mass velocity was 2.6 inches of water. The net pressure drop was, therefore, 6.4 inches of water. One inch of water is equivalent to 0.0361 pounds per square inch of pressure or 0.0361 (12.90) = 0.465 pounds in the glass tube

The 6.4 inches of water was equivalent to

6.4 (0.465) = 3.0 pounds of solid in the tube





TYLER STANDARD SIEVES

TABLE XXXIV

Mesh	Opening inches
3	0.263
4	0.185
6	0.131
8	0.093
10	0.065
14	0.046
20	0.0328
28	0.0232
35	0.0164
48	0.0116
65	0.0082
100	0.0058
150	0.0041
200	0.0029

DENSITY OF PARTICLES

The density of the various solid particles was determined roughly by water displacement in a graduated cylinder. When the density was less than that of water, the particles were held beneath the surface by a piece of wire gauze.

TABLE XXXV

Material	Weight	Volume of Water before Particles Added - ml	Volume of Water after Particles added - ml	
Sand	74.0	43.0	71.7	2.58
Charcoal	8.1	45	67	0.37
Coal	30.0	51.2	76.8	1.17

APPARENT DENSITY OF PARTICLES

The apparent density of the various solid particles was determined by pouring them into a graduated cylinder slowly.

TABLE XXXVI

Material	Mesh	Volume ml	Weight	Apparent Density g per ml
3A Catalyst	-50+100 -100+150 -150+200 -200+270 -270	100 100 100 100	60.6 59.3 55.4 53.5 48.5	0.606 0.593 0.554 0.535 0.485
Sand	-20+28 -28+48 -48+65 -65+80 -80+100 -100+150 -150	4 100 100 15.5 17.7 5.8 1.3	5.6 153.8 147.0 22.0 24.9 7.9	1.40 1.54 1.47 1.42 1.41 1.36
Charcoal	-14+20	56	8.1	0.145
Coal	-4+8 -8+14 -14+28 -28+48	100 100 100 100	67.4 68.7 63.1 58.2	0.674 0.687 0.631 0.582

AVERAGE PARTICLE SIZE

The average particle size was determined by the following formula:

$$(109) \qquad x = \frac{100}{\sum \frac{f}{n}}$$

where

x = average particle size

f = volume per cent between two sieves

n = arithmetic average of adjacent sieve sizes

For example, Ottawa Standard Silica Sand had the following analysis

TABLE XXXVII

Average Size of Sand Particles

Mesh	Weight Grams	Volume*	f per cent	n	f/n
-20+28	5.6	4.0	0.7	0.0280	25
-28+48	611.9	398.5	70.6	0.0174	4060
-48+65	179.1	122.0	21.6	0.0099	2180
-65+80	22.0	15.5	2.8	0.0076	368
-80+100	24.9	17.7	3.1	0.0063	492
-100+150	7.9	5.8	1.0	0.0050	200
-150	1.3	1.3	0.2	0.0035	57
Totals	852.7	564.8	100.0		7380

The average particle size was

(110)
$$x = \frac{100}{7380} = 0.0135$$
 inches

which is equivalent to about 40 mesh.

^{*}Using apparent density values from Table XXXVI

TABLE XXXVIII

Average Size of Charcoal Particles at Start of Run 5

Mesh	Weight grams	Volume*	f per cent	n inches	f/n
-8+10	5	37	1.1	0.079	14
-10+14 -14+16	118	8 6 1 1198	26.7 37.1	0.056	480 860
-16+20	134	930	28.8	0.036	800
-20+28	31	194	6.0	0.028	210
-28	2	10	0.3	0.012	25
Totals	454	3230	100.0		2390

(111)
$$x = \frac{100}{2390} = 0.042$$
 inches

Average Size of Charcoal Particles at End of Run 5

TABLE XXXIX

Mesh	Weight grams	Volume*	f per cent	n inches	f/n
-8+10	0	0	0	0.079	0
-10+14	13.0	95	15.4	0.056	280
-14+16	25.8	189	30.9	0.043	720
-16+20	27.2	189	30.9	0.036	860
-20+28	21.5	135	22.0	0.028	790
-28+50	0.7	3.6	0.6	0.017	35
-50	0.3	1.4	0.2	0.009	22
Totals	88.5	613.0	100.0		2710
(112)	x = 100	_ = 0.037	inches		

The average at the end was equivalent to about 20 mesh.

^{*}Using apparent density values from Table XXXVI

SPECIFIC HEAT OF UTAH COAL

The specific heat of coal was determined roughly by measuring the change in temperature when a weighed quantity of coal was added to a weighed quantity of hot water in a thermos bottle. The data are given below

TABLE XL

Weight of Coal Grams	Temp of Coal oc	Weight of Water grams	Temp of water oc	Temp of Mixture oc	Specific Heat
100.0	33.4	277.9	52.8	50.9	0.302
150.0	32.8	196.4	66.65	60.4	0.297

STEADY STATE RUNS WITH EMPTY TUBE

TABLE XLI

Run A-26

Room temperature: 72° F

Orifice: 0.500 inch diameter

Pressure drop across orifice: 6.3 in. of water

Pressure upstream of orifice: 0.9 in. of mercury

Potentiometer readings:

Thermocouple Millivolts	0.12	9.11	4 8.41	5 8.29	6 8.13	8.09	8 7.44
Thermocouple Millivolts	9 7.26	10	11	12 7.04	13 4.99	14 7.87	17 8.63
Thermocouple Millivolts	16 [*] a	0 ¹¹ 3.39	2.15	1.52	3/4"	l" 0.64	
Thermocouple Millivolts	16b [†]	0" 2.34	1.79	1.22	3/4"	1"	

^{*}Measurements made six inches above the screen and at each ‡ inch through the insulation radially, from the metal wall (0") to the outside (1").

† Measurements made 24 inches above the screen.

TABLE XLII

Run A-27

Room temperature: 76° F

Orifice: 0.500 inch diameter

Pressure drop across orifice: 6.2 in. of water

Pressure upstream of orifice: 0.9 in. of mercury

TABLE XLII CONTINUED

Potentiometer readings:

Thermocouple Millivolts	0.18	9.71	4	5 8.89	6 8.87	7 8.86	8 7.84
Thermocouple Millivolts	9 7.79	10 5.72	11 6.45	12 7.18	13 5.15	14 8.61	17 8.94
Thermocouple Millivolts	16a*	0" 3.30	2.43	1.74	3/4"	1"	
Thermocouple Millivolts	16b	0 ¹¹ 3.04	2.30	1.57	3/4"	1"	
Thermocouple Millivolts	16°	0" 2.73	1.99	1.40	3/4" 0.86	1" 0.63	

^{*}Measurements made four inches above the screen. †Measurements made nine inches above the screen.

TABLE XLIII

Run A-28

Room temperature: 73° F

Orifice: 0.500 inch diameter

Pressure drop across orifice: 6.2 inches of water

Pressure upstream of orifice: 0.9 inches of mercury

^{*}Measurements made eighteen inches above the screen.

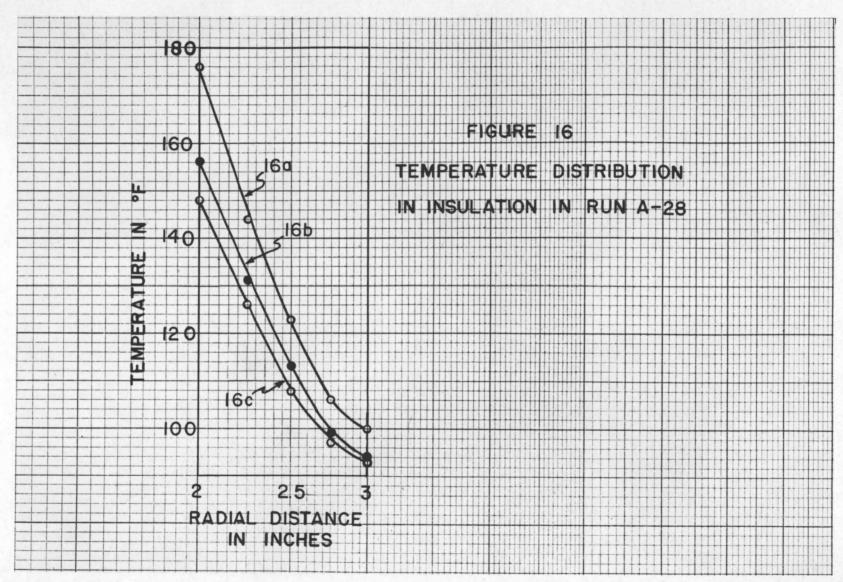
TABLE XLIII CONTINUED

Potentiometer readings:

Thermocouple Millivolts	0.19	4 7.78	5 7.72	6 7.64	7.64	8 6.79
Thermocouple Millivolts	9 6.60	10 4.89	11 5.54	12 6.26	13 4.57	14 7.41
Thermocouple Millivolts	16a*	0" 3.11	2.17	1.53	3/4"	1" 0.85
Thermocouple Millivolts	16 _b	0" 2.52	1.79	1.24	3/4"	l" 0.70
Thermocouple Millivolts	16c	0" 2.29	1.63	±n 1.10	3/4"	l" 0.66

Figure 16 is a plot of the insulation temperatures in Run A-28. The radial distances are plotted on a logarithmic scale.

^{* † \$} Same location as Run A-27



CALCULATION OF FILM COEFFICIENT BETWEEN THE OUTSIDE OF THE INSULATION AND THE ROOM

Under steady state conditions, the heat which flows into the insulation flows into the room. Expressed in equation form

$$(113) h_r A_r \Delta_r = \frac{K_i A_i \Delta_i}{1}$$

where $\Delta_{\mathbf{r}}$ is the temperature difference between the outside of the insulation and room temperature and $\Delta_{\mathbf{i}}$ is the temperature difference between the inside of the insulation and some point in the insulation, in this case the point. Using the data of 16c in Run A-28 as an example:

(114)
$$h_r = \frac{0.041(12)(2.5-2.0)(148-108)}{\frac{1}{2}(3)} l_n \frac{(2.5)}{(2.0)}(93-73)$$

= 1.47 Btu per hr-sq ft - OF

The results for the other two points of Run A-28 and for Runs A-26 and A-27 are presented below:

TABLE XLIV
Film Coefficient between Outside of Insulation and Room

Run	16a	16b	16c
A-26	2.35	1.51	-
A-27	1.56	1.84	2.21
A-28	1.45	1.51	1.47

The average was 1.8, the figure used in earlier calculations.

UNSTEADY STATE RUN WITH EMPTY TUBE

TABLE XLV

RUN A-31

Room temperature: 71° F

Duration of heating cycle: 140 minutes

Orifice: 0:500 inch diameter

Orifice and heater data:

Time Minutes	Air Bleed	T8-1 F	P _l in. mercury	ΔP in. water	Heater Voltage
0	107	93	_	2	99
6	_	-	-	11.4	-
20	107	101	1.4	11.4	99
50	108	103	_		-
90	108	104	-	-	-
146	109	105		- 1	-

Potentiometer readings:

TC	-4	TC-	5	TC-	7	TC-10)	TC-	11
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv
2:55	*7.51	1:30	7.26	4:20	7.34	4:55	2.28	35:40	5.14
8:50	7.29	7:45	7.27	9:50	7.21	10:20	3.30	47:20	5.28
14:15	7.20	12:55	7.20	15:15	7.08	16:10	3.80	74:30	5.31
30:00	7.14	25:20	7.09	28:00	7.07	32:00	4.34		
47:50	7.19	43:10	7.16	45:00	7.07	45:50	4.58		
70:50	7.19	69:00	7.14	58:20	7.22	53:30	4.68		
87:30	7.12	85:00	7.15	72:20	7.11	72:50	4.76		
109:20	7.19	108:10	7.15	89.00	7.01	88:30	4.80		
137:10	7.19	136:10	7.16	110.40	7.11	104:50	4.81		
				139.00	7.14	138:20	4.84		

^{*} Time in minutes and seconds after the start

TABLE XLV CONTINUED

TC-12	2	TC-13	3	TC-14	1	TC-la	5**	TC-16	₅ t
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv
2:07 8:30 13:45 29:00 43:10 52:10 70:00 84:30 103:30	5.29 5.54 5.76 5.95 6.01 6.00 6.04 5.99	0:59 7:25 12:20 26:30 39:30 51:20 68:30 82:40 101:30	2.64 3.08 3.63 3.86 4.04 4.19 4.25 4.23	0:23 6:45 11:40 23:30 24:40 40:30 66:30 83:30 107:10 135:10	6.87 6.89 6.80 6.85 6.96 6.93 6.96	3:30 9:25 14:50 30:40 44:30 52:40 71:30 87:00 104:10 133:00	1.14 1.53 2.00 2.20 2.26 2.37 2.41 2.44	5:20 10:50 16:30 32:30 46:20 54:00 73:20 89:50 105:40 133:40	0.95 1.25 1.61 1.80 1.89 1.99 2.06 2.07
132:30	0.07	TOTORO	TOOT	200.20	0.00	200,00	~ • 10		

^{*}Located at inside of insulation four inches above the screen.

[†]Located at inside of insulation nine inches above the screen.

UNSTEADY STATE RUNS WITH COAL

TABLE XLVI

Run 9 - Heating Run on Utah Coal

Material Fluidized: Utah hard coal

Weight at start: 2.00 pounds

Weight at end: 1.93 pounds (The loss was apparently by

volatilization).

Tyler Standard Screen Analyses:

	Start	End
-14 + 28 mesh	2.00	1.87
-28 + 48 mesh	0	0.05
-48 mesh	0	0.01
	2.00 lb	1.93 lb

Room temperature: 82° to 83° F

Duration of heating cycle: 90 minutes

Duration of cooling cycle: 47 minutes

Average weight of coal in bed during heating cycle: 1.98 lb

Approximate depth of dense phase: 10 inches

Approximate depth of lean phase: 4 inches

Fluidization: Infrequent slugging

Orifice and heater data:

Time minutes	${}^{ t Air_0}_{ t F}$	TC-1	P _l in.	AP in.	Heater Voltage
7		-	1.8	11.4	99
16	119	111	100	846	
25	119	111	-	11.4	100
47	122	113	1.8	11.4	99
81	126	115	-		-

Potentiometer readings:

TC-4		TC-5		TC-7		TC-10	0	TC-11	L
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv
1:30 9:50 19:30 28:40 37:45 48:25 60:20 68:20 72:40	6.77 7.34 7.39 7.49 7.25 7.42 7.53 7.57	3:20 11:20 20:40 30:30 39:25 49:30 61:45 71:50 85:10	3.71 5.84 6.58 6.91 6.87 7.13 7.17	5:55 12:40 21:45 31:45 40:45 50:50 62:50	4.79 5.98 6.57 6.88 6.83 6.98	9:20 18:45 27:50 37:15 47:50	0.48 5.10 6.08 6.47 6.57 6.65 6.85 6.89	54:15	6.62
82:00	7.58			T. G. 7			- 84	TC-16	.t
TC-12	3	TC-13	3	TC-14	1	TC-1	0	T.C-T	5
TC-12	mv	TC-13		Time		Time	mv	Time	mv

^{*}TC-15 located at inside of insulation four inches above the screen.

[†]TC-16 located at inside of insulation nine inches above the screen.

TABLE XLVII

Run 9 - Heat Flow Rate Equations

(115)
$$q_a = \frac{635(12.40)(0.240)(t_7-t_4)}{144} = -13.1(t_4-t_7)$$

(116)
$$q_s = 1.98(0.300) \frac{dT}{d\theta} = 0.594 \frac{dT}{d\theta}$$

(117)
$$q_m = \left[1.08(0.12) + 0.25(0.094)\right] \frac{dt_{10}}{d\theta} = 0.153 \frac{dt_{10}}{d\theta}$$

(118)
$$q_1 = \frac{2\pi(14)(0.041)(t!-t!!)}{12(2.30)\log \frac{2.25}{2.00}} = 2.55(t!-t!!)$$

(119)
$$q_1 + q_m = h_W \frac{3.97 \pi (14) \Delta_W}{144} = h_W (1.21) \Delta_W$$

TABLE XLVIII

Run 9 - Rate of Heating of Metal

Time minutes	(1) dt10 der min	qm Btu per hr
1 2 3 4 5 6 8 10 12 14 16 18 20	28.0 23.5 21.4 18.2 15.8 13.7 10.3 7.2 4.4 3.0 2.15 2.10 1.73	258 216 196 167 145 126 95 66 40 28 20 19
25 30	1.18 0.62	11 5½

TABLE XLVIII CONTINUED

Run 9 - Rate of Heating of Metal

Time minutes	(1)dt10 (60)de of per min	q _m Btu per hr
35 40 45 50 55 60 65 70 75 80 85 90	0.135 0.24 0.32 0.36 0.35 0.33 0.29 0.22 0.14 0.10 0.07 0.05	1 2 3 3 3 3 2 2 1 1 1 1 1 1 1 1 1 1 1 1

TABLE XLIX

Run 9 - Flow of Heat into Insulation

Interval	Time minutes	t'*	t" o _F	t'-t"	g _i Btu per hr
1 2 3 4 5 6 7 8 9 10 11 12 13 14 15 16 17 18	0.93 1.86 2.79 3.72 4.65 5.59 6.51 7.44 8.37 9.30 10.2 11.2 12.1 13.0 14.9 15.8 16.8 17.7	81 85 98 108 114 120 125 130 134 138 142 146 150 154 158 160 161 162 • 5 164	81 83 89.2 94.2 98.7 103.0 106.9 110.5 114.0 117.3 120.2 123.6 126.7 129.8 132.9 135.0 136.8 138.7	0 4 15 18.8 19.8 21.3 22.0 23.1 23.5 24.7 25.8 26.4 27.3 28.2 27.1 26.0 25.7 25.3	0 10 38 48 50 54 56 59 60 61 63 66 67 70 72 69 66 65 64

TABLE XLIX CONTINUED

Run 9 - Flow of Heat into Insulation

Interval	Time minutes	t'* oF	t" o _F	ti-ti	q _i Btu per hr
20 21 22 23 24 25 26 27 28 29 30	18.6 19.5 20.5 21.4 22.3 23.2 24.2 25.1 26.0 27.0 27.9	165.5 167 169 170 171 172.5 174 175 176 177	140.0 141.7 143.0 144.6 145.4 146.9 148.1 149.2 150.2 151.1 152.0	5.3 5.3 6.4 6.6 9.8 9.9 9.0 9.0 9.0 9.0 9.0 9.0 9.0	65 64 66 65 65 65 66 66 66 66
35 40 45 50 60 70 80 90	32.6 37.2 41.9 46.5 55.7 65.0 74.4 83.7	183 185 187 189 192 194 195 196 197	169.5	27.5	70

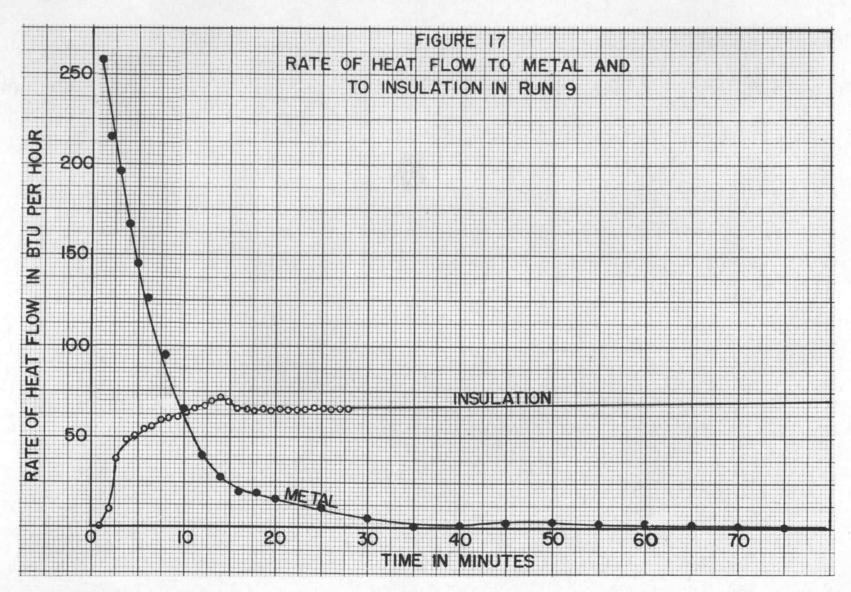


TABLE L

Run 9 - Coefficient of Heat Transfer between

Air and Tube Wall

Time minutes	qi+qm Btu per hr	t ₄ -t ₁₂ *	t7-t10	∽ _W ∘ _F	h _w Btu per hr- sq ft - °F
2 4 6 8 10 12 14 16 18 20	229 215 180 155 129 107 100 86 83 81	147 109 90 78 69 61 54 49 44.6 41.2	30 30 26 20 15 11 9 9	73.8 61.2 51.5 42.7 35.4 29.2 25.2 21.8 21.6	2.6 2.9 2.9 3.0 3.0 3.3 3.1 3.1
25 30 35 40 45 50 55 60 65 70 75 80 85 90	77 71 = 67 68 69 71 = 71 71 = 71 71 71 = 70 70 = 70	35.1 31.4 24.3 25.8 28.2 28.5 27.7 26.0 24.9 24.3 23.3 22.0 21.7	9.4 9.5 9.5 7.6 8.2 8.3 8.0 7.1 7.0 6.8 6.8	19.5 18.4 15.7 14.8 15.7 16.3 16.1 15.3 14.5 14.0 13.6 13.1 12.9	3.2 3.5 3.6 6.6 6.8 4.2 4.3 5.5 4.5 4.5 4.5

Although there was an upward trend with time, the values of the coefficient of heat transfer between the air and the metal wall were generally lower than those in Run 11.

^{*}Because of the small difference between t_{10} and t_{12} , the latter is used here.

TABLE LI
Run 10 - Heating Run on Utah Coal

Material fluidized: Utah hard coal

Weight at start: 2.00 pounds

Weight at end: 1.94 pounds (The loss was apparently due to volatilization)

Tyler standard screen analyses:

			Start	End
	-8 +14	mesh	0	0.02
	-14+28	mesh	2.00	1.87
	-28+48		0	0.04
	-48		0	0.01
om	Temperatu	re: 84°F	2.00 lb	1.94 lb

Duration of heating cycle: 45 minutes

Average weight of coal in bed: 1.97 pounds

Approximate depth of dense phase: 10 inches

Approximate depth of lean phase: 4 inches

Fluidization: Some slugging but not severe

Orifice and heater data:

Time minutes	Air Bleed	TC-1	Pl in. mercury	△P in. water	Heater Voltage
0 9 20 30 36	116 116 116	104	1.9	11.3 11.4 11.4 11.4	98 98 97 97

Potentiometer readings:

TC-4	1	TC-	ō	TC-	7	TC-10		TC-	11
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv
2:45 6:30 14:35 20:40 32:00 39:40	5.57 6.11 6.68 6.97 7.10 7.10	3:50 7:45 15:45 24:45 33:10 40:40	3.09 4.38 5.57 6.26 6.56 6.66	1:30 5:20 13:35 22:30 30:50 38:50	1.59 3.64 5.30 6.10 6.47 6.58	11:05 17:45 28:00 43:40	4.56 5.39 6.03 6.35	0:30 4:50 13:05 22:00 30:30 39:20	0.52 3.13 4.99 5.80 6.19 6.34
TC-	12	TC-	13	TC-	14	TC-	15*	TC-	16
TC-:	mv_	TC-	13 _mv_	TC-	14 	TC-	15**	TC-:	16 mv

^{*}Located at inside of insulation four inches above the screen.

TABLE LII

Run 10 - Heat Flow Rate Equations

(120)
$$q_a = \frac{638(12.40)(0.240)(t_7-t_4)}{144} = -13.2(t_4-t_7)$$

(121)
$$q_s = 1.97(0.300) \frac{dT}{d\Theta} = 0.591 \frac{dT}{d\Theta}$$

(122)
$$q_m = [1.08(0.12) + 0.25(0.094)] \frac{dt_{11}}{d\theta} = 0.153 \frac{dt_{11}}{d\theta}$$

(123)
$$q_i = \frac{2\pi 14(0.041)(t'-t'')}{12(2.30)} = 2.55(t'-t'')$$

(124)
$$q_1 + q_m = h_w \frac{3.97 \pi (14)}{144} \Delta_W = h_w (1.21) \Delta_W$$

TABLE LIII

Run 10 - Rate of Heating of Metal

Time minutes	(1)dt _{ll} 60 d 0 o _{F per min}	q _m Btu per hr
1 2 3 4 5 6 8 10 12 14 16 18 20	19.7 26.5 19.5 16.8 14.1 11.6 8.1 6.05 5.35 3.80 3.25 2.76 2.69	181 244 179 155 130 117 742 552 49 35 30 252 25

TABLE LIII CONTINUED

Run 10 - Rate of Heating of Metal

Time minutes	(1) dt ₁₁ 60 de F per min	q _m Btu per hr
25	1.71	16
30	1.01	9 =
35	0.53	5
40	0.37	3 2
45	0.10	1

TABLE LIV

Run 10 - Flow of Heat into Insulation

Interval	Time minutes	ti*	t" o _F	ti-ti	gi Btu per hr
1 2 3 4 5 6 7 8 9 10 11 12 13 14 15 16 17 18 19 20		o _F 86 91 95 102 108 115 119 124 128 133 136 139 142 145 145 155 157 159	84 85 87.3 89.5 93.2 96.8 101.0 104.0 107.5 110.6 114.4 117.0 120.0 122.3 124.9 127.5 129.8 131.6 133.5 135.3	2 6 7.7 12.5 14.8 18.2 18.0 20.0 20.5 22.4 21.6 22.0 22.7 24.1 23.5 23.4 23.5 23.7	5 15 20 32 38 46 46 51 52 57 55 56 56 58 62 60 59 60 60 60 60 60
21 22 23 24 25 26	19.5 20.5 21.4 22.3 23.2 24.2	161 163 165 167 168	137.3 139.0 140.5 142.2 144.0	23.7 24.0 24.5 24.8 24.0 24.9	60 61 62 63 61 63

TABLE LIV CONTINUED

Run 10 - Flow of Heat into Insulation

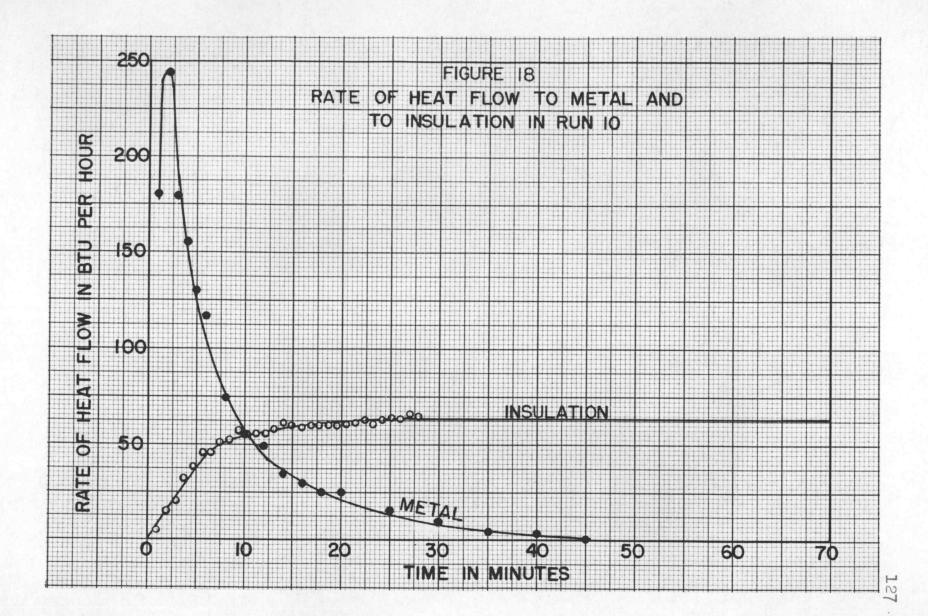
Interval	Time minutes	t'*	t" o _F	to-t"	g _i Btu per hr
27 28 29 30	25.1 26.0 27.0 27.9	172 173 175 176	147.0 148.0 149.1 150.4	25.0 25.0 25.9 25.6	64 64 66 65
35 40 45	32.6 37.2 41.9	180 183 185 188	151.7 155.5 158.5 163.4	24.6	63

TABLE LV

Run 10 - Coefficient of Heat Transfer between Air and Tube Wall

Time minutes	qi + qm Btu per hr	t4-t12 oF	t7-t10	o _F	h _w Btu per hr- sq ft - ^O F
2 4 6 8 10 12 14 16 18 20	261 188 162 126 111 106 = 94 90 87 = 87 = 87	127 101 87 69 62 59.5 54.1 48.4 45.5 42.2	11 11 14 8 8 11.0 11.1 10.5 10.8 10.4	47.5 40.6 39.7 29.3 26.4 28.8 27.2 24.9 24.1 22.8	4.5 3.8 3.4 3.6 3.1 2.9 3.0 3.2
25 30 35 40 45	80 74 = 70 68 = 66	36.1 31.7 28.5 26.0 24.1	11.1 10.4 10.0 9.6 8.8	21.2 19.1 17.7 16.5 15.2	3.1 3.2 3.3 3.4 3.6

Runs 9 and 10 were duplicates except for slight variations in heater voltage and air preheat time. Comparison



of the values of h_{W} for the two runs shows excellent agreement.

TABLE LVI

Run 12 - Heating Run on Utah Coal

Material fluidized: Utah hard coal which had been heated at 300° F for several hours and then cooled and kept in a desiccator.

Weight at start: 0.500 pounds

Weight at end: 0.498 pounds

Tyler standard screen analyses:

	Start	End
-8 + 14 mesh	0	0.006
-14+ 28 mesh -28+ 48 mesh	0.500	0.479
-48 mesh	0	0.002
	0.500 lb	0.498 lb

Room temperature: 74° F

Duration of heating cycle: 60 minutes

Approximate depth of dense phase: 2 inches

Approximate depth of lean phase: 2 inches

Fluidization: Good

Orifice and heater data:

Time minutes	AiroBleed	TC-1	P _l in. mercury	ΔP in. water	Heater Voltage
0 13	115 114	102 106	1.5	11.4	97 97 96
26 55	115	108		11.4	96

Potentiometer readings:

TC-4		TC-5		TC-6		TC-7		TC-12		TC-15	分
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv	Time	mv
0:30 4:20 9:10 15:10 21:00 26:50 32:50 42:45 50:00	3.15 4.92 5.86 6.39 6.66 6.83 6.93 7.04 7.05	1:30 5:30 10:50 16:20 22:10 27:30 34:00 44:00 51:30	2.49 4.55 5.55 6.06 6.37 6.54 6.68 6.80	2:25 6:50 12:10 18:40 23:50 28:40 36:10 46:00 53:20	2.82 4.54 5.56 6.12 6.23 6.44 6.60 6.70 6.69	3:20 7:50 13:20 19:30 24:45 30:30 37:20 41:20 48:00 59:50	3.39 4.88 5.64 6.10 6.32 6.48 6.59 6.66 6.73 6.70	1:10 5:10 10:30 15:50 21:40 27:10 33:30 44:50 51:00	1.62 3.91 5.01 5.59 5.94 6.12 6.29 6.44 6.44	1:50 6:20 11:20 17:50 22:40 28:00 35:40 45:30 52:30	0.57 1.53 2.17 2.65 2.86 3.05 3.23 3.36 3.44

^{*} Located at inside of insulation two inches above screen.

TABLE LVII

Run 12 - Heat Flow Rate Equations

(125)
$$q_a = \frac{636(12.40)(0.240)(t_6-t_4)}{144} = -13.1(t_4-t_6)$$

(126)
$$q_s = 0.50(0.300) \frac{dT}{d\theta} = 0.150 \frac{dT}{d\theta}$$

(127)
$$q_m = \left[0.32(0.12) + 0.25(0.094)\right] \frac{dt_{12}}{d\theta} = 0.0619 \frac{dt_{12}}{d\theta}$$

(128)
$$q_1 = \frac{2\pi(4)(0.041)(t'-t'')}{12(2.30) \log \frac{2.25}{2.00}} = 0.730(t'-t'')$$

(129)
$$q_1 = h_W = \frac{3.97 \pi (4)}{144} \Delta_W = h_W (0.346) \Delta_W$$

TABLE LVIII

Run 12 - Rate of Heating of Metal

Time	(<u>l</u>) ^{dt} 12	g
minutes	oF per min	Btu per hr
1 2 3 4 5 6 8 10 12 14 16 18	42.5 24.6 18.2 15.1 11.5 8.4 6.7 4.2 3.7 3.1 2.9 2.0	158 108 74 51 45 40 31 17 16 12 9
20	1.61	6

TABLE LVIII CONTINUED

Run 12 - Rate of Heating of Metal

Time minutes	$(\frac{1}{60})^{\text{dt}_{12}}$ oF per min	q _m Btu per hr
25 30 35 40 45 50	1.08 0.79 0.61 0.40 0.21 0.03	4 3 2 1 1 0

TABLE LIX
Run 12 - Flow of Heat into Insulation

Interval	Time minutes	t'* oF	t" o _F	ti-t"	q _i Btu per hr
1 2 3 4 5 6 7 8 9 10 11 21 13 14 14 15 16 17 18 19 20 21 22 23 24 25 25 26 26 26 27 27 28 28 28 28 28 28 28 28 28 28 28 28 28	0.93 1.86 2.79 3.72 4.65 5.59 6.51 7.44 8.37 9.30 10.2 11.2 12.1 13.0 14.9 15.8 16.8 17.7 18.5 20.5 21.4 22.3 23.2	82 90 98 105 112 118 124 128 136 140 143 146 149 152 154 156 158 160 162 163.5 165 168 169	74 78 81.8 86.3 90.6 95.2 99.5 104.0 107.3 110.8 114.0 117.3 120.0 122.7 125.1 127.8 129.6 131.6 133.4 135.2 137.0 138.3 141.2 142.3	8 12 16.7 21.8 21.8 24.5 24.7 22.5 26.7 26.8 26.8 26.8 26.7 26.7 26.8 26.8 26.7 26.7 26.7 26.7 26.7 26.7 26.7 26.7	6 9 2 3 5 6 8 7 8 8 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9
30 35 40 45 50 55	27.9 32.6 37.2 41.9 46.5 51.2	174 178 181 183 184.3 185	159.4	26.6	192

^{*}t'=t15

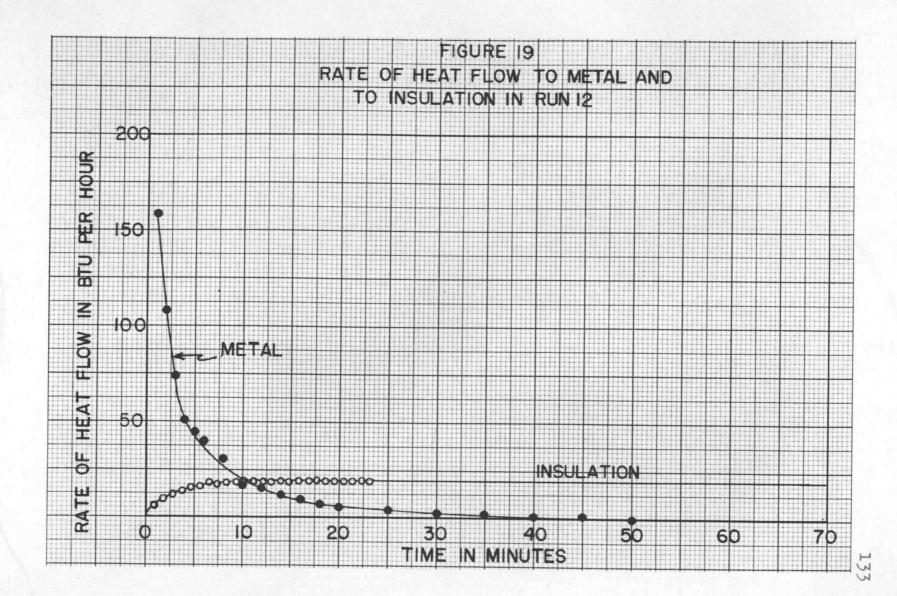


TABLE LX

Run 12 - Coefficient of Heat Transfer between

Air and Tube Wall

Time minutes	qi + qm Btu per hr	t4-t12	teoris	o _F w	hw Btu per hr- sq ft - OF
2 4 6 8 10 12 14 16 18 20	118 66 57 49 36 35 31 28 26 25 25	55 42 38 38.3 32.5 31.2 29.0 27.5 25.5 24.5	5 7 7 9.5 8.0 10.0 10.4 10.0 9.0 8.7	20.9 19.6 21.9 20.7 17.5 17.8 18.1 17.3 15.9	16.3 9.7 7.5 5.8 5.9 5.7 4.9 4.8 4.8
25 30 35 40 45 50	23-12-12-12-12-12-12-12-12-12-12-12-12-12-	23.2 21.8 21.1 20.3 19.5	8.5 8.3 8.0 7.8 7.5 7.5	14.7 14.6 13.4 13.1 12.6 12.6	4.6 4.4 4.6 4.5 4.7 4.5

The large values of $h_{\rm W}$ in the first few minutes were probably due to inaccuracies resulting from the small weight of solids fluidized. However, the steady state value agreed with that found in Run 9, a run made at approximately the same mass velocity.

UNSTEADY STATE COOLING RUNS

In attempting to find the coefficient of heat transfer between the air and the solid particles, a cooling run was made, that is, after steady state had been reached in Run 9, cool air was introduced into the fluidized bed. Additional data not appearing in previous tables are given below.

TABLE LXI

Run 9 - Cooling Run on Utah Coal

Average mass velocity during cooling: 632 lb per hr-sq ft
Average weight of coal in bed during cooling: 1.94 lb
Potentiometer readings:

TC-4	1	TC-	Ō	TC-7		TC-10		TC-11		
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv	
10:50 17:30 25:00 25:30	2.00 1.65 1.46 1.36	3:00 13:10 18:30 26:40 31:30 42:10	2.46 1.95 1.61 1.50	14:30 19:30 28:00 32:50	2.26 1.88 1.54 1.43	10:20 16:40 24:30 30:00	3.00 2.18 1.71 1.52			
TC-	12	TC-	13	TC-	14	TC-	15	TC-	16	
Time	mv	Time	mv	Time	mv	Time	mv	Time	mv	
	2.31 1.88 1.52 1.40	3:40 23:30		16:00 20:50 29:00 34:20	2.13 1.81 1.51 1.38	11:50	2.07 1.55 1.19 1.06	14:50 20:20 28:30 33:20	1.76 1.42 1.11 1.02	

The heat flow rate equations were the same as for the heating part of the run except for the slight change in mass velocity and in the average weight fluidized, assuming that the small loss in weight was spread out over the entire run.

TABLE LXII

Run 9 - Heat Flow Rate Equations during Cooling

(130)
$$q_a = \frac{632(12.40)(0.240)(t_7-t_4)}{144} = 13.1(t_7-t_4)$$

(131)
$$q_s = 1.94(0.300) \frac{dT}{d\theta} = 0.582 \frac{dT}{d\theta}$$

(132)
$$q_m = 0.153 \frac{dt_{12}}{d\theta}$$

(133)
$$q_i = 2.55(t'-t'')$$

The method of calculation was the same as in the heating part of the run except that the Schmidt plot of insulation temperatures had the line representing steady state heat flow as a base, rather than room temperature. Tables LXIII, LXIV, and LXV give the values of q_a , q_m , and q_i , respectively, at various times. The last is also plotted in Figure 21.

TABLE LXIII

Run 9 - Rate of Heat Flow to Air during Cooling

Time minutes	t75t4	q _a Btu per hr
1 2 3 4 6 8 10 12 14 16 18 20	83 79 72 63 48 36 28 22 17 14 11	1090 1035 930 825 630 470 370 290 220 185 145
25 30 35 40 45	7 6 4 2 0	90 80 50 25 0

Run 9 - Rate of Heat Flow to Metal during Cooling

Time minutes	- 1 dt ₁₂ 60 d 0 F per min	-q _m Btu per hr
1 2 3 4 6 8 10 12 14 16 18 20	20.2 17.8 16.0 14.2 11.3 7.8 5.8 4.7 4.0 3.2 2.6 2.0	185 163 147 130 104 71 53 43 37 29 24 18½
25 30 35 40 45	1.25 0.90 0.70 0.48 0.05	11 = 8 = 1 = 8 = 6 = 4 = 1 = 8 = 4 = 1 = 8

TABLE LXV

Run 9 - Rate of Heat Flow to Insulation during Cooling

		36			
Interval	Time minutes	t,*	t" °F	ti-ti	gi Btu per hr
1 2 3 4 5 6 7 8 9 0 1 1 2 3 1 4 5 6 7 8 9 0 1 2 2 2 2 2 2 2 2 2 2 3 3 3 3 3 3 3 3 3	93699259914470 10.554470 10.23.4.56.7.6554.3.2.2.100988776554.3.2.2.100988776554.3.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2.2.	186 183 180 175 167 163 159 155 151 147 144 140 137 133 130 128 126 122 120 119 118 117 116 114 113 111 110 109 107 106	161.4 160.0 158.8 155.8 154.2 151.9 148.6 145.8 140.0 137.3 135.0 129.6 124.1 122.0 120.3 118.6 116.7 115.0 113.8 111.6 110.5 109.2 107.1 106.2 105.5 104.6 103.8 103.8 103.7 102.0	24.66022814207004490743022458898542230 26.660211091998765655555554444444444444444444444444444	63 55 41 43 864 28 28 20 16 15 14 14 12 12 12 11 11 11 10

^{*}t' = t₁₆, located at inside of insulation 9 inches above screen.

TABLE LXV CONTINUED

Run 9 - Rate of Heat Flow to Insulation during Cooling

Interval	Time minutes	t' o _F	t" o _F	ti-tii	g _i Btu per hr
40 45 50	37.2 41.9 46.5	103** 101** 99*	99 97 95	4 4	10 10 10

^{*} Approximate values

On substituting into equation (6),

(6) $q_a + q_s + q_m + q_i = 0$

the following values of q were obtained:

TABLE LXVI

Run 9 - Rate of Heat Flow to Solid during Cooling

Time minutes	-q _s Btu per hr
1	970
2	920
3	825
4	735
6	660
8	425
10	340
12	265
14	200
16	170
18	135
20	115
25	90
30	85
35	55
40	30
45	10

Figure 22 is a plot of Table LXVI. By taking the area under this curve, values of $Q_{\rm S}$ were obtained for $v_{\rm A}$ rious lengths of time. Substituting these into equation (76) (134) $-Q_{\rm S} = 1.94(0.300)(T-310)$

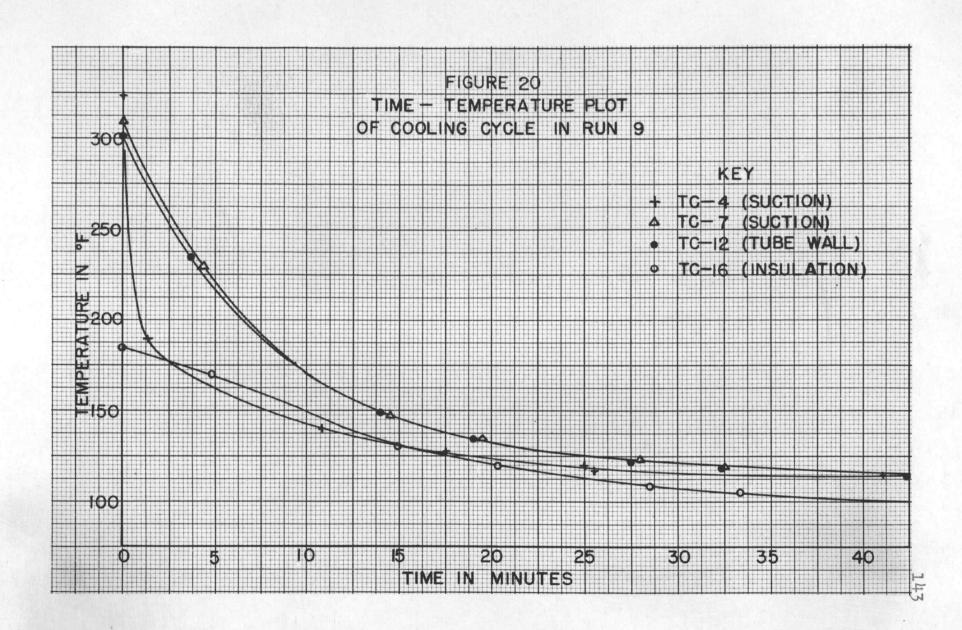
the calculated average solid temperature, T, was found. The results are given in Table LXVII.

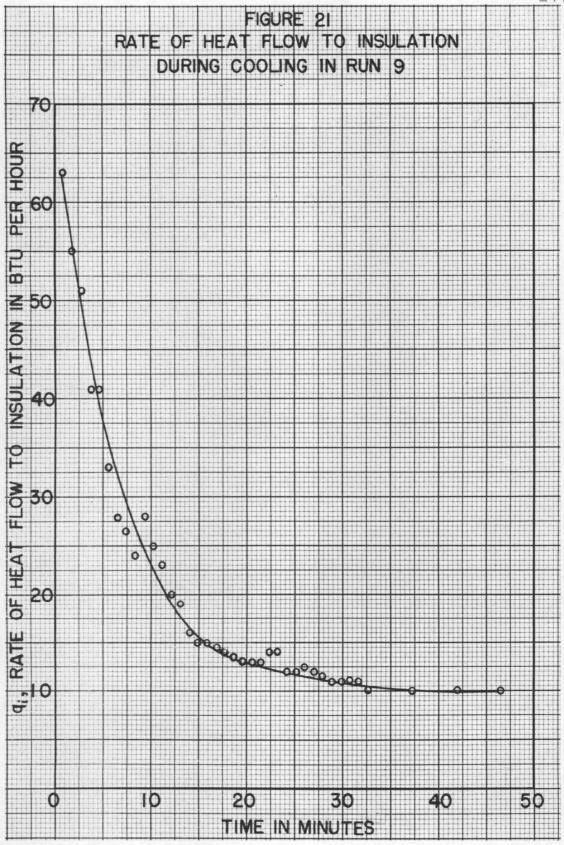
Table LXVII

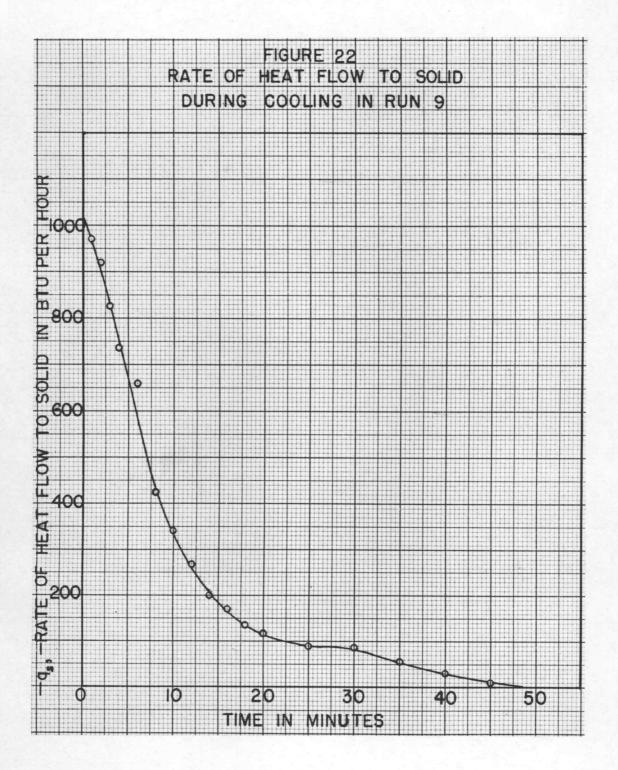
Run 9 - Solid Temperature during Cooling

Time minutes	$_{ m o_F}$
2 4 6 8 10 12 14 16 18 20 25 30 35 40	255 206 165 134 112 95 81 70 61 53 ¹ / ₂ 39 27 18 12
45 50	9

The temperatures are obviously too low although not as greatly in error as in the heating runs.







SUCTION THERMOCOUPLE OPERATION

The air was drawn past the hot junction of the thermocouple at a rate which was high enough to reduce radiation errors to a negligible amount. Actually, the air velocity was probably higher than necessary. A rough calculation of this velocity follows using the vacuum pump flow rate of Run 11 and the area of the well given on p85.

(135) Velocity =
$$\frac{0.18(144)}{(60)(0.0154)}$$
 = 28 feet per second

Although this quantity of air was small compared to the total mass velocity, it was cooled all the way to room temperature. As a result, the rate of heat loss was appreciable as shown below:

(136)
$$q = \frac{0.18(60)(0.075)(0.24)(300-70)}{3} = 15$$
 Btu per hr

This loss was calculated as if the suction were operating at all times whereas it was actually sucking air only about half the time. Note that three thermocouples are considered to be contributing to the loss.

TEMPERATURE DROP THROUGH THE METAL WALL

A calculation of the temperature drop, Δ , through the metal wall was made for a heat flow rate of 200 Btu per hour over a ten inch length of the tube.

(137)
$$q = \frac{-K 2 \pi r N \Delta}{1}$$

For iron, K = 26

The thickness of the sheet metal tube was 0.016 inches.

(138)
$$-200 = \frac{-26(2)\pi(2.0)(10)}{0.016(12)}\Delta$$

(139)
$$\Delta = 0.012^{\circ} \text{ F}$$

LONGITUDINAL FLOW OF HEAT

In Run 11, if it is assumed that all of the 16° temperature drop (steady state) between TC-10 and TC-12 occured in the approximately one inch between the top of the lean phase and TC-10, the following longitudinal heat flow rate is obtained:

(140)
$$q = \frac{K 2 \pi rl \Delta}{N}$$

(141)
$$q = \frac{+26(2)\pi(2.0)(0.016)(16)}{12(1)} = 7.0$$
 Btu per hr

FLOW OF HEAT THROUGH FITTINGS

The order of magnitude of the rate of heat loss through fittings can be determined if some approximations are made. Taking a vacuum well as an example:

Cross-section area:
$$\frac{\pi}{4} \left[(0.1875)^2 - (0.140)^2 \right] = 0.0122 \text{ sq in}$$

Length of section: ½ inch

Approximate outside temperature: 120° F

Approximate inside temperature: 250° F

(142)
$$q = \frac{26(0.0122)(12)(120-250)}{(144)(\frac{1}{2})} = 7 \text{ Btu per hr}$$

SCHMIDT METHOD

The Schmidt plot shown in Figure 10 was constructed as follows

- 1. At the end of the first time increment (0.93 minutes), the temperature of the inside of the insulation was 78° F. This point was connected with 70° F at the abscissa corresponding with a point half way through the insulation, c. (See Figure 23).
- 2. At 1.86 minutes, the inside temperature was $87\frac{1}{2}^{\circ}$ F. This was connected with point c and intersected the abscissa corresponding to a point one fourth of the way through the insulation at h. Point f was connected with d, which indicated room temperature (70° F) three-fourths of the distance through the insulation. The temperature for points one-fourth of an inch apart, at the end of the first time interval, was, as follows: 78° (a), 70° (b), 70° (c), 70° (d), 70° (e). At the end of the second interval, the corresponding values were 87.5° (e), 73.8° (f), 70° (c), 70° (d), 70° (e).
- 3. The line of intersected the halfway line at i. Continuing on, gi was drawn, intersecting the one-fourth line at K. The line hd was drawn next and intersected the halfway line at l. Points i and e were joined and intersected the three-fourths line at

m. The temperature distribution thus became ghide. Since d and e were at room temperature, the heat had not yet reached the outside of the insulation. It was stated earlier (p44) that the temperature at any point at any time is the mean of the two temperatures Δx ($\frac{1}{4}$ inch) away on either side at a time $\Delta \theta$ (0.93 minutes) previously. Thus, point h, located $\frac{1}{4}$ inch from the metal wall, is the mean of point e, which was the temperature of the inside of the insulation, one time interval earlier, and point c, which was located $\frac{1}{2}$ inch from the metal wall and also was the temperature one time interval before point h. This can be seen more readily in Tables LXVIII and LXIX.

minutes), the heat flow had just reached the outside of the insulation, i.e. the temperature distribution was jklme.

TABLE LXVIII

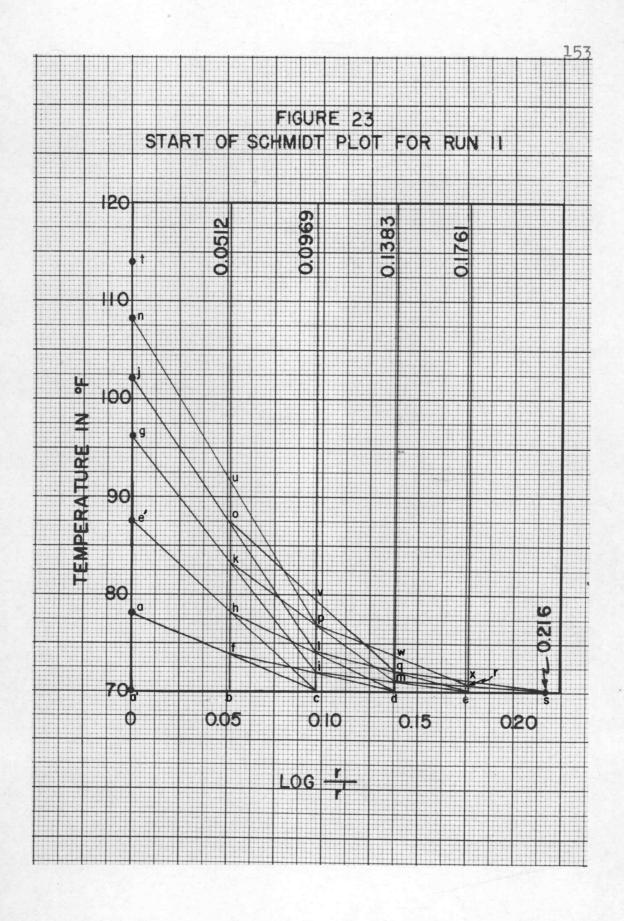
Temperature Distribution at Start of Schmidt Method Plot

Interval	2"(Inside)	2111	2111	2-3/4"	3"Outside)	
0	70	70	70	70	70	
1	78	70	70	70	70	
2	87.5	74	70	70	70	
3	96	78.5	71.7	70	70	
4	102	83.3	73.8	70.8	70	
5	108	87.1	76.7	71.9	70.5	
6	114	91.5	79.1	73.2	71.0	

TABLE LXIX

Location of Points in Temperature Distribution Curves

Interval	2"(Inside)	2111	2111	2-3/4"	3"(Outside)
0	a†	ъ	C	d	е
1	a	ъ	С	d	е
2	e1	f	C	d	0
3	g	h	i	. d	е
4	j	k	1	m	0
5	h	0	p	q	r
6	t	u	V	W	\mathbb{X}



5. If there were no resistance to the flow of heat from the outside of the insulation to the room (h_r = infinity), the temperature at the outside of the insulation would always remain 70° F and the graphical construction would pivot about this point. Because of the finite resistance, this pivot point was located at an abscissa of 0.216 (point s). The next temperature distribution was nopqrs. After the sixth time interval (5.59 minutes), the temperature distribution was tuvwxs. The construction was continued through the thirtieth time interval. By that time, as seen in Figure 10, the succeeding lines were very close together. Inasmuch as the rate of heat flow to the insulation was nearly constant by this time interval, extrapolation was made without difficulty.

ACCURACY OF RESULTS

An approximation can be made of the maximum error in the values of the coefficient of heat transfer between the air and the tube wall by substituting the worst possible combination into the flow rate equations.

(143)
$$q_{i} = \frac{2\pi N K_{i}(t^{i}-t_{0})}{2.30 \left[0.176 + \frac{K_{i}}{2.30 h_{r}r_{0}}\right]} = h_{w} 2\pi N r^{i} \frac{(t_{7}-t_{12}) - (t_{4}-t_{12})}{\ln(t_{7}-t_{12})}$$
 $K = -0.041 \pm 0.004$
 $h_{r} = 1.8 \pm 0.4$
 $t^{i} = \pm 10$
 $t_{7}, t_{12}, t_{4} = \pm 1$

Taking Run ll as an example, for

(144)
$$q_i = 47$$
 Btu per hr
$$\Delta_W = 9.5$$

$$K$$

$$2.30 h_r r_0 = 0.040,$$

$$h_W = 5.2$$

If the following extreme values are assumed:

(145)
$$\frac{K_i}{2.30h_r r_o} = \frac{0.037(12)}{2.30(1.4)(3.0)} = 0.046$$

(146) $\Delta_W = \frac{(251-253) - (270-253)}{\ln \frac{261-253}{270-253}} = 11.5^{\circ} \text{ F}$

and
$$h_{W} = 3.4$$

The maximum error is

(147) Error =
$$\left(\frac{5.2 - 3.4}{5.2}\right)^{100} = 35\%$$