

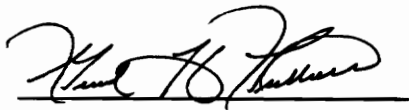
**AN EVALUATION OF SEPARATION METHODS FOR THE
SELECTIVE COAGULATION OF ULTRAFINE COAL**

by
Jason R. Pyecha

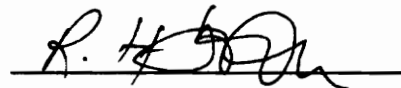
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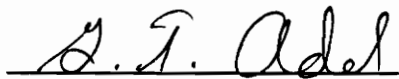
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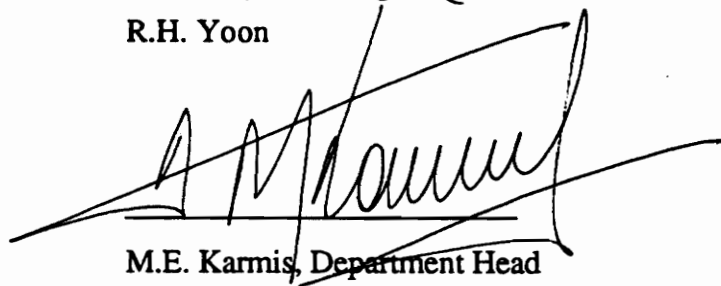
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G.H. Luttrell, Chairman

Mining and Minerals Engineering

(ABSTRACT)

A novel technique for selectively coagulating and separating coal from dispersed mineral matter has been developed at Virginia Tech. The process, which is known as *Selective Hydrophobic Coagulation* (SHC), differs from oil agglomeration, shear or polymer flocculation, and electrolytic coagulation processes in that it does not require reagents or additives to induce the formation of coagula. In most cases, simple pH control is all that is required to (i) induce the coagulation of coal particles and (ii) effectively disperse particles of mineral matter. If the coal is oxidized, a small dosage of reagents may be used to enhance the coagulation process.

During the SHC development, it was discovered that the hydrophobic coagula were very difficult to separate from dispersed mineral matter due to their very small size and their susceptibility to breakage. Using the SHC technique, an evaluation of new methods for coagula recovery was conducted. In this effort, several methods for improving the separation of the coal coagula from dispersed mineral matter were examined. These included lamella thickening, centrifugal sedimentation, vacuum filtration, drum screening, and froth flotation. Each separation method was optimized using statistically-designed test matrices to determine the best separation method based on overall process performance. The thickener was found to be the best method for separating hydrophobic coagula from dispersed mineral matter based on overall process performance (e.g., recovery and grade), unit capacity, and engineering feasibility. Further testing of the thickener separation unit was conducted in an attempt to improve the process performance and the unit throughput.

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Chapter 1 Introduction

1.1 Ultrafine Coal Processing

The need for suitable ultrafine coal cleaning methods has arisen from the need to meet environmental regulations and the feasibility of superclean (< 2% ash) coal as an alternative liquid-fuel source. The fine grinding of run-of-mine coal increases the liberation of impurities, increasing the potential for ash and sulphur rejection. While the liberation size of the impurities is a characteristic of the coal seam, most U.S. coals require grinding to an ultrafine (< 10 microns) particle size to achieve pyrite liberation (Kneller and Maxwell, 1985; Attia, 1985). If processed effectively, superclean and ultraclean (< 0.8% ash) coal could be formulated into high-quality fuels suitable as replacements for oil and gas applications. The effective processing of ultrafine coal would also reduce the need for scrubbers while improving power plant performance.

The difficulties in treating ultrafine particles are due mainly to their small mass, their increased specific surface area, and their high surface energy (Fuerstenau, 1980). Normal coal cleaning methods fail when treating ultrafine particles by producing low separation efficiencies and/or low throughput. When processed, the small mass of the ultrafine particles causes them to become entrained in the liquid medium of the process. For flotation methods, this causes low bubble-particle collision frequency (reduced recovery) and high hydraulic entrainment (reduced ash rejection). The increase in specific surface area of the ultrafine particles increases the required reagent consumption for processing. The high surface energies of the ultrafine particles causes the preferential adsorption of the reagents onto them, while the coarser particles may not be adequately reagentized. This causes a decrease in the recovery of coarser particles. The high surface energies associated with the ultrafine particles also cause difficulty in the selectivity between the coal and the mineral matter. As the particle size is reduced the differences between the coal and mineral matter surface properties decrease, reducing the selectivity of a surface-chemical separation process.

The ultrafine particle processing methods developed over the last few decades have focused on particle surface-chemical properties and their differences. The processes increase the apparent size of the valuable particles through agglomeration, flocculation, or coagulation. The processes also attempt to minimize mineral matter entrapment while creating sufficient aggregate strength to permit the use of separation techniques such as flotation, sedimentation, or screen separation. The surface chemistry of fine coal is advantageous for these aggregate-forming type processes, in that the coal macerals are typically hydrophobic while the mineral matter is hydrophilic.

a) Oil Agglomeration

Oil agglomeration preferentially wets hydrophobic coal particles in an aqueous suspension with a collecting liquid (typically an oil or a low molecular weight hydrocarbon), forming aggregates of coal particles. The collecting liquid is the bridging mechanism between the coal particles while the hydrophilic mineral matter remains dispersed in the aqueous suspension. The main factors behind the oil agglomeration process are i) the free energy relationships at the liquid-liquid-solid interface, ii) the amount of collecting liquid used in relation to the solids content, and iii) the type and intensity of mixing employed.

Higher rank coals are more hydrophobic than lower rank or oxidized coals, which have more hydrophilic surfaces. Surface conditioning agents such as long-chain surfactants may be used to improve the hydrophobicity of these coals (Venkatadri and Wheelock, 1988). The use of heavier oils as collecting liquid has also been shown to increase the wettability of these coals, increasing the effectiveness of the oil agglomeration process (Robbins et al., 1992).

The aggregate structures formed using the oil agglomeration process depend on the amount of collecting liquid used. The use of coal particles less than 0.5 mm and intense agitation produces the following aggregate structures. Low collecting liquid dosages of 1/2-5% by weight of the coal (low oil agglomeration) form unconsolidated

flocs with pendular bridges between the coal particles. These flocs are two-dimensional, loosely packed, less than 1 mm in diameter, and tend to trap hydrophilic mineral matter and water. With larger dosages of collecting liquid, 5-15% by weight of the coal, the number of collecting liquid junctions increases. Funicular to capillary bridges are formed, creating three-dimensional flocs from 1 to several mm in diameter. Finally, in the capillary wetting region, 15-20% collecting liquid by weight of the coal, compact spherical pellets are formed with the voids being filled with collecting liquid. The entrapment of mineral matter and water is minimal at the higher collecting liquid dosages, providing the best ash rejection and dewatering characteristics (Capes and Germain, 1982).

Intense initial mixing in the oil agglomeration process disperses the emulsified collecting liquid and the ultrafine particles. After dispersion, the mixing intensity is reduced to allow agglomeration. Slow speed mixing produces larger, weaker agglomerates while higher speed mixing allows the formation of smaller, stronger agglomerates (Rao, 1988). The agglomerate size depends on the separation method employed, be it flotation, sedimentation, or screen separation.

The oil agglomeration process is able to produce ultraclean coal using staged agglomeration. The main drawback of the process is the high collecting liquid consumption. To decrease the high cost of the collecting liquid several processes have been developed which recycle the collecting liquid, such as the Otisca T-process. While these recovery processes work well, there are still high capital and operational costs. Another method to reduce the collecting liquid consumption has been to focus on processing techniques in the low oil agglomeration range (< 5% collecting liquid by weight of coal), in which pyrite rejection is favored. Unfortunately, the low oil agglomeration method does not handle well on screens and additional drying is required to decrease the moisture levels of the aggregates (Capes et al., 1988).

b) Selective Flocculation

Selective flocculation bridges the desired particles together in an aqueous solution of two or more solid constituents using soluble macromolecules such as starches and polyacrylamides (flocculants). In coal slurries, the coal particles flocculate due to hydrophobic interactions between the polymeric chains and the coal surface while the hydrophilic mineral matter remains dispersed. The success of the selective flocculation process depends upon the appropriate application of surface-chemical principles. Selective xanthate-containing dispersants for coal have been developed which increase the ultrafine pyrite rejection of the flocculation process (Attia, 1985).

Flocculation produces flocs which are typically larger and contain a more loose structure than coagula, allowing hydrophilic mineral matter and water to become entrapped within the flocs. This entrapment increases both the ash and moisture contents of the flocs. While the flocs are larger than coagula, they are also significantly stronger than coagula produced using inorganic coagulants (Ray and Hogg, 1986). As the dosage of flocculant is increased, floc breakage decreases, creating stronger, more tightly packed flocs. Unfortunately these tightly packed flocs also suffer from the entrapment of mineral matter and water.

Flocculation of ultrafine particles (3-6 microns) requires increased flocculant dosages, forming flocs only 10-15 microns in size (Attia et al., 1990). This size difference between the flocs and the dispersed mineral matter makes the separation process difficult. Floc separation methods include flotation, sedimentation, and screen separation. Screening requires strong flocs, able to withstand breakup, which increases the entrapped mineral matter and water. Flotation requires an understanding of the interactions between the flocculant and the flotation reagent. While these problems must be overcome, the main drawbacks to the flocculation process are low selectivity due to the entrapment of hydrophilic mineral matter and high flocculant consumption when treating ultrafine particles. The incorporation of treating separate size fractions, pre-

cleaning steps, and multiple flocculation stages has produced results of less than 3% ash, showing that it is a feasible process for ultrafine coal cleaning (Attia et al., 1990).

c) Selective Coagulation

Selective coagulation agglomerates dispersed hydrophobic coal particles without an agglomerant, while hydrophilic mineral matter remains dispersed. There are no bridging mechanisms involved in coagulation, unlike oil agglomeration or flocculation. Selective coagulation occurs because of the differences in surface potentials between the coal particles and the mineral matter particles. When the attractive force (hydrophobic interaction force) between particles is greater than the repulsive force (electrostatic force), coagulation can occur without additional external energy. With the existence of a repulsive force larger than the attractive force, external energy (in the form of mixing/kinetic energy) may be introduced to overcome the repulsive force and induce coagulation.

For selective coagulation to occur, the chemistry of the suspension must be adjusted with appropriate pH modifiers and/or electrolytes so that the component particles have the same surface-charge-sign. The selectivity of coagulation is based on the differences in the surface potentials of the component particles. In addition to pH modifiers and/or electrolytes, control of the surface potentials can be done with dispersants and chelating agents.

Several laboratory studies conducted at Virginia Tech (Honaker, 1988; Honaker 1992) have shown that selective coagulation of coal produces superclean and ultraclean coal products with high recovery values. These studies have also shown the simplicity of the selective coagulation of ultrafine coal. They reveal that high rank, unoxidized coals usually have sufficient hydrophobic force to naturally induce coagulation while oxidized coals and low rank coals may require treatment with a small amount of hydrocarbon oil (well below oil agglomeration dosages) to enhance their hydrophobicity. The studies also

demonstrate the cost effectiveness of the selective coagulation process due to the reduced reagent requirements.

1.2 Research Objectives

One of the major advantages of selective coagulation is that it does not require the use of agglomerating agents such as hydrocarbon oils or flocculants. However, the coagula formed by selective coagulation are relatively weaker than those formed when agglomerating agents are used. As a result, attention must be given to the proper selection and design of systems to recover these coagula. In Chapter 2, five different continuous flow devices were evaluated to determine the most effective methodology for the separation of dispersed mineral matter from coagulated coal. These included (i) a lamella thickener, (ii) a centrifuge, (iii) a vacuum drum filter, (iv) a froth flotation column, and (v) a drum screen. Each unit was tested using bench-scale continuous units at throughputs up to 0.5-1.0 lb/hr. These separators were tested at the optimum reagent conditions established from batch tests carried out on (i) a high ash, low sulphur Elkhorn No. 3 seam coal sample and (ii) a high sulphur Pittsburgh No. 8 seam coal sample.

After determining the most effective method of separating the dispersed mineral matter from coagulated coal, the best possible separator was selected and tested in greater detail using the Elkhorn No. 3 seam coal sample (see Chapter 3). This selection was based on overall process performance (e.g., recovery and grade), unit capacity, and engineering feasibility. In making this selection, the results of the work conducted in the DOE project (DE-AC22-90PC90174) were also considered.

Chapter 2 Advanced Separation Methods

2.1 Introduction

One of the major advantages of the SHC process is that it does not require the use of agglomerating agents such as hydrocarbon oils or flocculants. However, the coagula formed by the SHC process are relatively weaker than those formed when agglomerating agents are used. As a result, attention must be given to the proper selection and design of systems to recover these coagula. In the present work, five different continuous flow devices are being evaluated to determine the most effective methodology for the separation of dispersed mineral matter from coagulated coal. These include (i) a froth flotation column, (ii) a lamella thickener, (iii) a vacuum drum filter, (iv) a centrifuge, and (v) a rotating drum screen. Each unit was tested using bench-scale continuous units at throughputs up to 0.5-1.0 lb/hr. These separators were tested at the optimum reagent conditions established from batch tests carried out on each of the two base coals (Elkhorn No. 3 and Pittsburgh No. 8 seam coals).

2.2 Sample Acquisition

The process development test work was conducted using run-of-mine coal samples from the Pittsburgh No. 8 and Elkhorn No. 3 coal seams. The Pittsburgh No. 8 sample was obtained from a Consol, Inc. property in northern West Virginia (Consol Sample Designation #LL-1022-45-1). The Elkhorn No. 3 sample was obtained from United Coal Company's Wellmore No. 8 preparation plant in Buchanan County, Virginia. Approximately 225 kgs (500 lbs) of each coal was collected at the mine site and shipped to Virginia Tech in sealed 55-gallon drums. In an attempt to minimize oxidation, each drum was flushed with nitrogen prior to shipment. After delivery, each of the two coal samples were pulverized to minus 6.5 mm (1/4 inch) using a pilot-scale roll crusher. The crushed samples were split into representative lots of approximately 1 kg each and transferred to air-tight plastic

containers. In order to minimize oxidation, each sample container was stored in a freezer below 20°C. Head assays for the two coal samples are provided in Table 2.1.

2.3 Sample Preparation

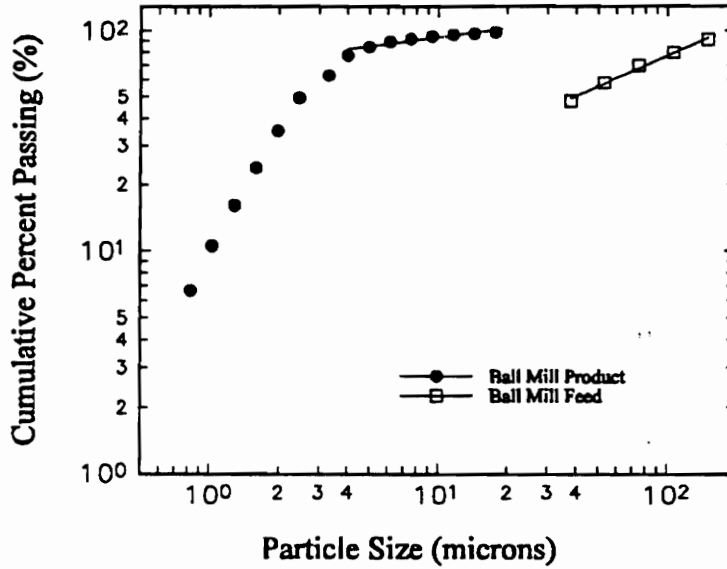
All of the coal samples used in the process development test work were prepared using the same procedure. In each series of experiments, a laboratory hammer mill was used to pulverize 1 kg samples of minus 6.5 mm (1/4 inch) coal to a mean size of approximately 75 microns. The sample was then split into lots of approximately 500 gm each and wet-ground for 30 minutes at 30% solids using a 13.33 cm (5-1/4 inch) diameter stirred ball mill loaded with 1.6 mm (1/16 inch) stainless steel grinding balls. This procedure resulted in a mean particle size of approximately 4 microns as measured using an Elzone 80-XY particle size analyzer. Typical size distributions of the micronized samples of the Elkhorn No. 3 and Pittsburgh No. 8 seam coals are shown in Figure 2.1 and tabulated in Appendix A.

After wet-grinding, the pulverized coal samples were diluted using cold tap water to the desired solids content. Hydrochloric acid (HCl) and sodium hydroxide (NaOH) were used to adjust the pH to the desired value. When used, ethylenediaminetetraacetic acid (EDTA) was added to the sample prior to pH adjustment to tie-up free ions in solution that may be detrimental to coagulation selectivity. The pH of the suspension was allowed to stabilize prior to testing.

Table 2.1 - Head sample assays for the Pittsburgh No. 8 and Elkhorn No. 3 samples used in the process development test work.

Sample	Type	Feed Ash %	Feed Sulfur %
Elkhorn No. 3	ROM	49.8	0.35
Pittsburgh No. 8	ROM	15.4	3.6

Elkhorn No. 3 (50% Ash)
(30 minute grind time)



Pittsburgh No. 8
(30 minute grind time)

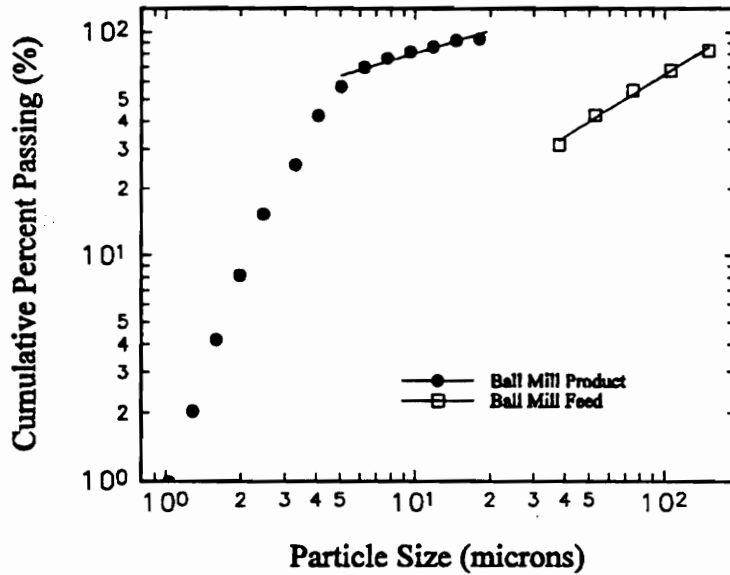


Figure 2.1 Typical size distributions of the micronized samples of Elkhorn No. 3 and Pittsburgh No. 8 coals.

2.4 Batch Testing

During many of the trial runs with the continuous test units, difficulties were experienced in obtaining good separation efficiencies. Therefore, a series of exploratory batch coagulation tests were performed to resolve these difficulties and to identify the optimal chemical conditions for treating each coal. This data would also provided a "yardstick" for evaluating the effectiveness of the continuous separators.

a) Experimental

The sedimentation tests utilized a 6-inch diameter 3-liter mixing cell with baffles. One liter of the minus 20 micron coal sample at 3% solids was added to the mixing cell. The slurry was agitated with a 3-inch diameter impeller. When used, the chelating agent (EDTA) was added to the slurry in the first cleaning stage and conditioned for 15 minutes. The pH was adjusted to the desired value using hydrochloric acid (HCl) or sodium hydroxide (NaOH). After the pH had stabilized for 15 minutes at the desired value, the impeller was halted to allow coagulation and sedimentation. After 5 minutes, the supernatant of dispersed mineral matter was siphoned from the mixing cell. The remaining coagula was considered the product. Figure 2.2 provides a schematic overview of the steps used in the batch sedimentation tests.

To run multiple cleaning stages, the next stage began with the addition of cold tap water to the product of the previous stage, diluting the slurry to one liter. The slurry was then agitated to break the coagula and remove entrapped mineral matter. There was no EDTA addition in preceding cleaning stages, since the intended use of the EDTA was to tie-up the free ions acquired from the fine grinding process. The pH was adjusted and allowed to stabilize for 15 minutes. After pH stabilization, the impeller was halted and 5 minutes were given for coagulation and sedimentation. After the 5 minutes, the supernatant was siphoned from the mixing cell and the remaining sedimented coagula was taken as clean coal product. These steps were repeated for the desired number of cleaning stages.

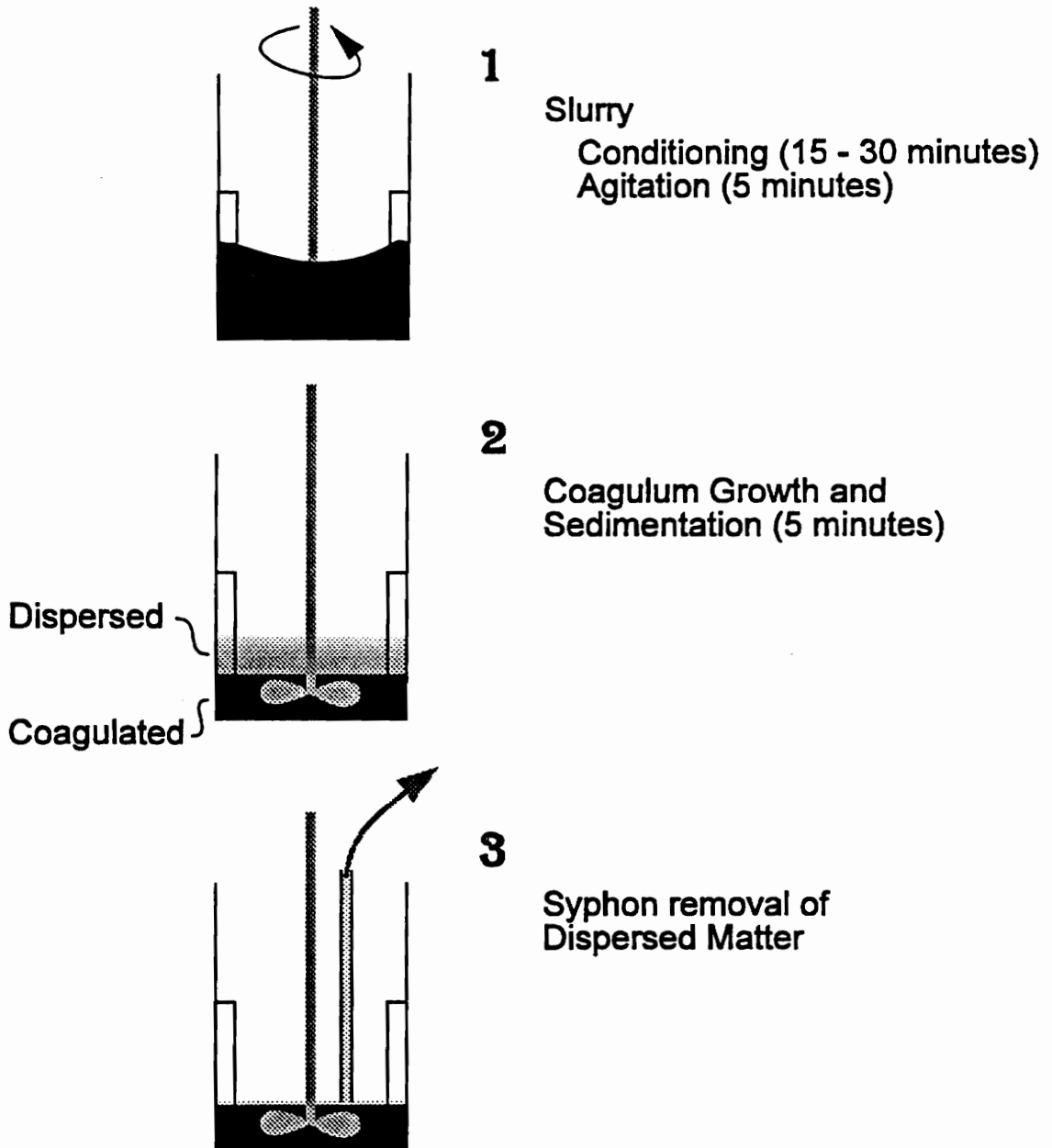


Figure 2.2 Schematic representation of the SHC batch testing procedure.

b) Results and Discussion

The test results and specific operating conditions for the batch SHC tests conducted using the Elkhorn No. 3 seam coal are provided in Appendix B. The test data are summarized in Figure 2.3. As shown, both single- and three-stage tests were conducted as a function of pH and EDTA (ethylenediaminetetraacetic acid) addition. In all tests, the maximum separation efficiency for this particular coal occurred at pH 8.8. The addition of EDTA slightly widened the pH window over which effective separations could be achieved. However, it is unlikely that the small improvement would justify the additional cost of adding EDTA. An increase in the number of cleaning stages also improved the separation efficiency.

The results of batch SHC tests conducted using the Pittsburgh No. 8 seam coal are also provided in Appendix B and are summarized in Figure 2.4. As shown, this coal responded less favorably to upgrading by the SHC process as indicated by the very low separation efficiencies. Even after adding EDTA and three stages of cleaning, a maximum separation efficiency of only 26% was achieved. Recent work has shown that this coal is difficult to treat by the SHC process due to the specific adsorption of multivalent cations from solution. Thus, the testing of additional chelating agents should be considered to improve the cleanability of this particular coal.

Test results obtained with the Elkhorn No. 3 and Pittsburgh No. 8 coals indicated that multiple cleaning stages improved the performance of the batch SHC tests. In order to investigate this capability in greater detail, each coal was tested under optimal reagent conditions as a function of cleaning stage number. As shown in Figure 2.5, the Elkhorn No. 3 coal has a final percent ash of 16.5% after 5 stages of cleaning. The increase in separation efficiency with multiple cleaning stages decreased to less than 5% after 3 stages of cleaning with only a slight decrease (< 1%) in recovery. Minimal improvements in cleaning performance were also noted after three stages of cleaning with the Pittsburgh No. 8 coal (Figure 2.5).

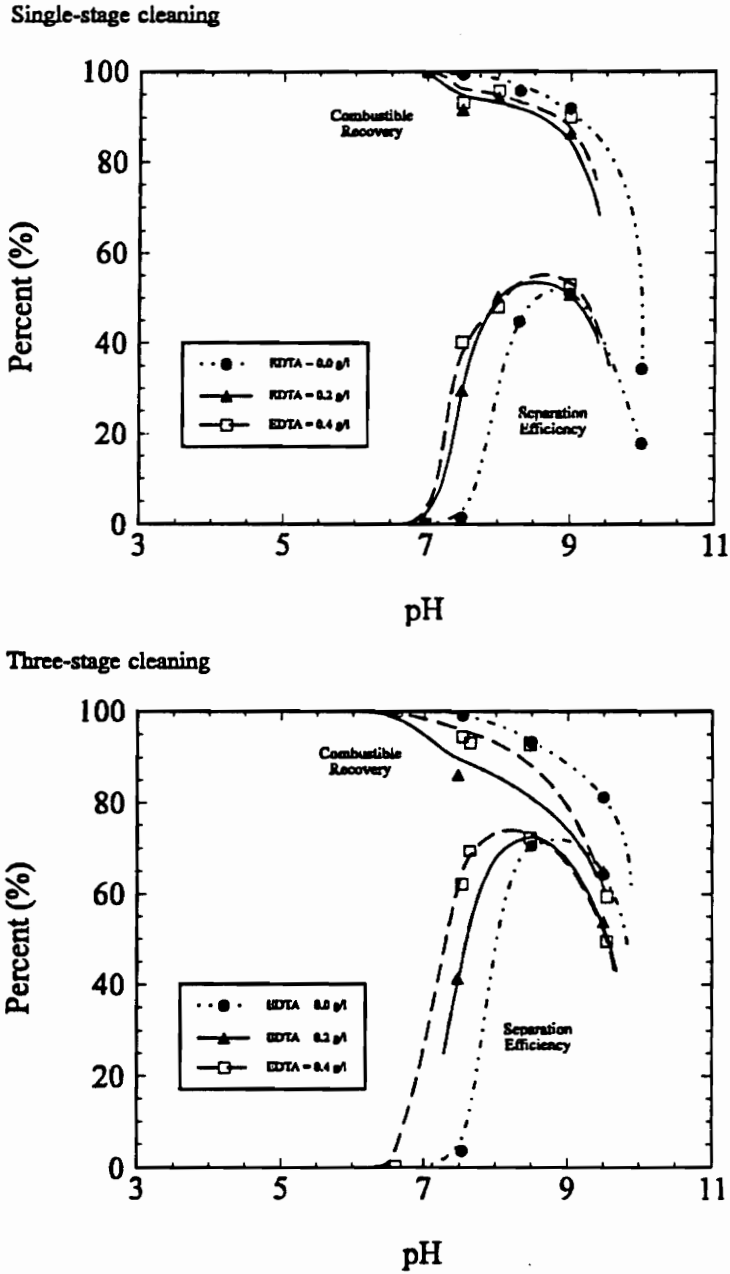


Figure 2.3 Responses versus pH for single- and three-stage batch SHC tests conducted using the Elkhorn No. 3 coal.

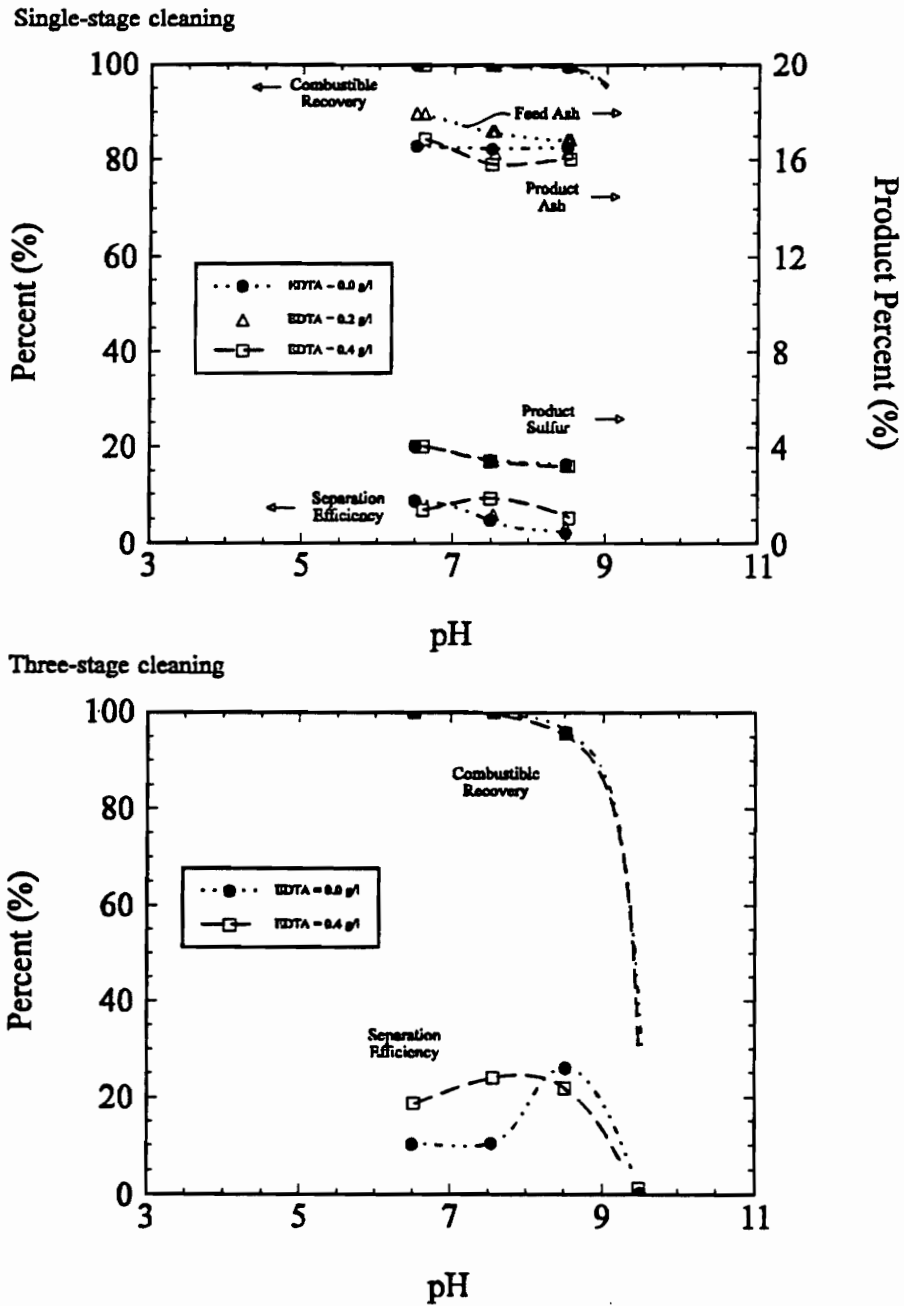


Figure 2.4 Responses versus pH for single- and three-stage batch SHC tests conducted using the Pittsburgh No. 8 coal.

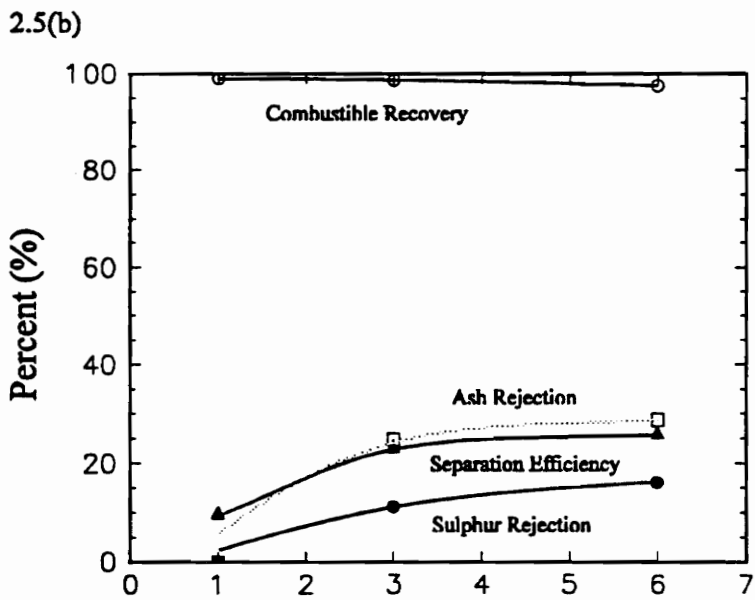
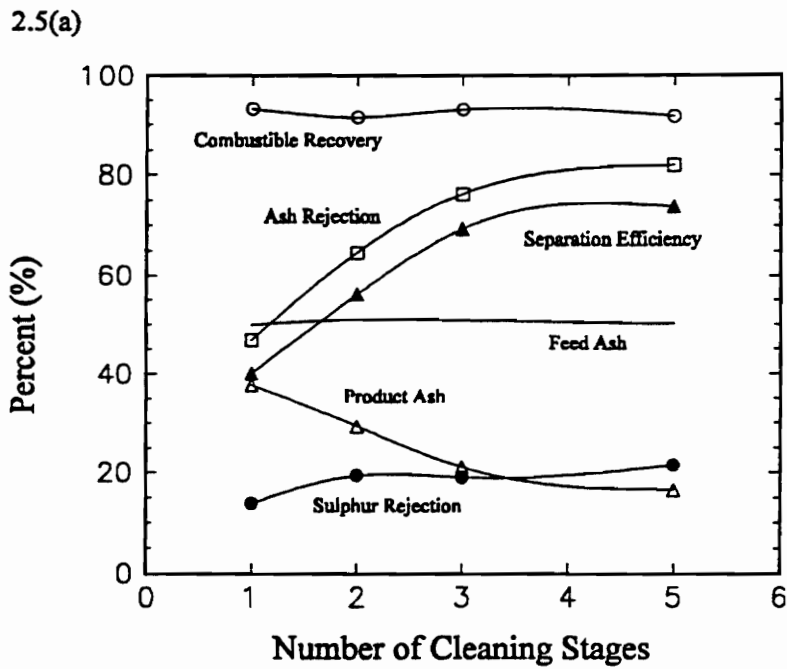


Figure 2.5 Responses versus cleaning stages for (a) Elkhorn No. 3 and (b) Pittsburgh No. 8 coals at pH 7.5 with EDTA dosages of 0.4 grams/liter of slurry.

2.5 Lamella Thickener

A series of tests were conducted to determine the effectiveness of a lamella-type thickener in separating clean coal coagula from dispersed mineral matter. In principle, this technique is similar to the sedimentation/elutriation technique employed in the DOE Contract No. DE-AC22-90PC90174 project (Yoon, R.H. and Luttrell, G.H., 1992). However, the lamella design should offer significant improvements in throughput compared to conventional open-tank settling vessels.

a) Experimental

A prototype lamella thickener constructed of Plexiglas was used for this task (Figure 2.6). Design information for this unit is given in Appendix C. The unit included provisions for introducing countercurrent wash water to the lower section of the separator as a means of minimizing the entrainment of fine mineral matter. In each test, the unit was filled with de-ionized water and adjusted to pH 7.5 with HCl and NaOH. Cold tap water in a sump was adjusted to pH 7.5 and pumped with a peristaltic pump to the wash water tube. The conditioned feed slurry was pumped from a mixing sump with a peristaltic pump through ten feet of tubing to a four-way splitter. The feed slurry then entered the middle of each plate in the thickener. The mudline was allowed to build to a fixed level, then the underflow peristaltic pump was started, removing the product slurry. The underflow pump was adjusted as needed to maintain the fixed mudline level. After the underflow rate became constant, the test was run for a minimum of one retention time to ensure test reliability. Table 2.2 shows the standard operating parameters used for the lamella thickener experiments.

Both coals were evaluated in 8-run factorial designs which examined the effects of wash water, feed rate, and feed solids content. The test levels examined for each parameter are given in Table 2.3. The evaluation tests for both the Elkhorn No. 3 and Pittsburgh No. 8 seam coals were performed in random order to minimize experimental bias. In addition, several replicate tests were conducted throughout the test program

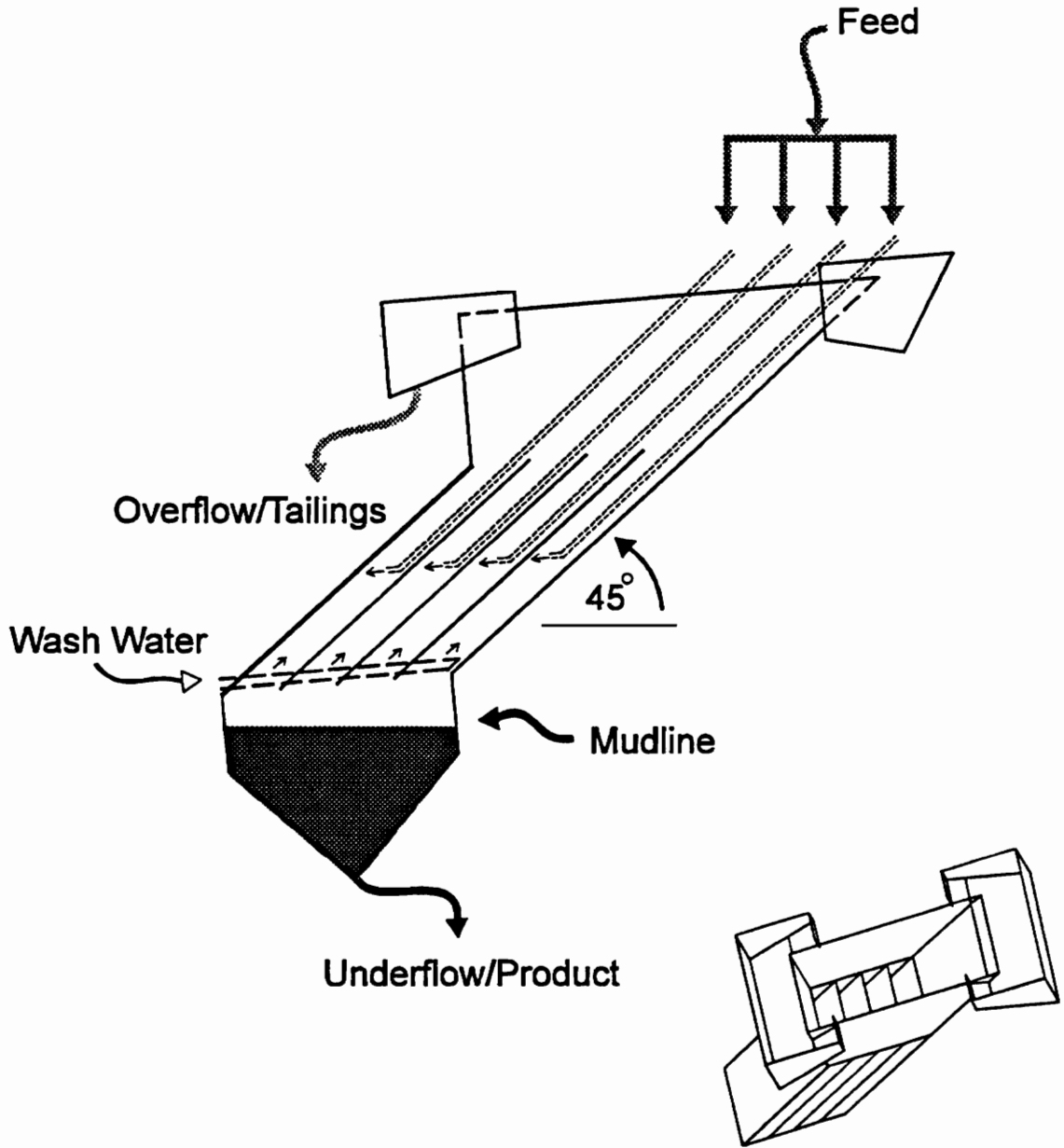


Figure 2.6 Schematic of the lamella thickener separation unit used in coagula recovery test work.

Table 2.2 - Standard operating conditions for the lamella thickener tests.

Parameter	Value
pH	7.5
pH Condition Time	15 minutes
Chelating Agent	EDTA
Chelate Dosage	0.4 gm/liter
Chelate Condition Time	15 minutes
Feed Particle Size	-20 microns
Mudline	constant
Lamella Plate Angle	45 degrees
Cell Actual Area	42 inch ²
Cell Effective Area	170 inch ²
Cell Volume	11 liters

Table 2.3 - Experimental levels for the lamella thickener parametric tests.

Coal Seam	Parameter	Units	Parameter Range		
			Low (-1)	Normal (0)	High (1)
Elkhorn No. 3 (50% Ash)	Feed Solids	(%)	2	4	6
	Wash Water	(ml/min)	0	20	40
	Feed Rate	(ml/min)	50	100	150
Pittsburgh No. 8	Feed Solids	(%)	2	4	6
	Wash Water	(ml/min)	20	60	100
	Feed Rate	(ml/min)	50	75	100

under "normal" operating conditions in order to provide a check on the reproducibility of test procedures and assay determinations. The experimental data were adjusted using a material balance program to ensure that a consistent data set was obtained in each case.

Statistical analysis of the experimental test results was carried out using the Design-Expert software. The program was used to generate the necessary test matrix and to complete all required regression analyses. The data analysis procedure was an iterative process since, in most cases, several analyses were required in order to obtain a suitable relationship between the operating variables and lamella thickener performance. After selecting an appropriate model, a variety of statistical routines were executed to identify the goodness-of-fit. For all test runs, the values of regression coefficients, standard errors, variances, t-tests, etc., were determined as a means of quantifying the statistical significance of each of the coefficients in the model. After developing a reasonable mathematical model of the desired response, contour plots were created to demonstrate the effects of the various operating parameters on the process performance.

b) Results and Discussion

Prior to conducting the parametric tests, the lamella thickener was sampled as a function of time to ensure that all data points were taken under steady-state conditions. Figure 2.7 and Appendix C show the variation between samples taken during a single continuous test run. For all practical purposes, the lamella thickener had reached steady-state operation prior to taking the first sample point (i.e., after 20 minutes of operation), although a slight downward drift in ash rejection was observed until after 90 minutes of continuous operation. Therefore, all test runs were allowed to proceed for 90 minutes prior to taking samples during the parametric test program.

The parametric test data obtained using the Elkhorn No. 3 seam coal is summarized in Appendix C. The best separation efficiency (i.e., 51.0%) was obtained at a combustible recovery of 92.4%. At this level of recovery, it was possible to achieve ash and sulfur rejections of 58.6% and 15.3%, respectively. However, the maximum

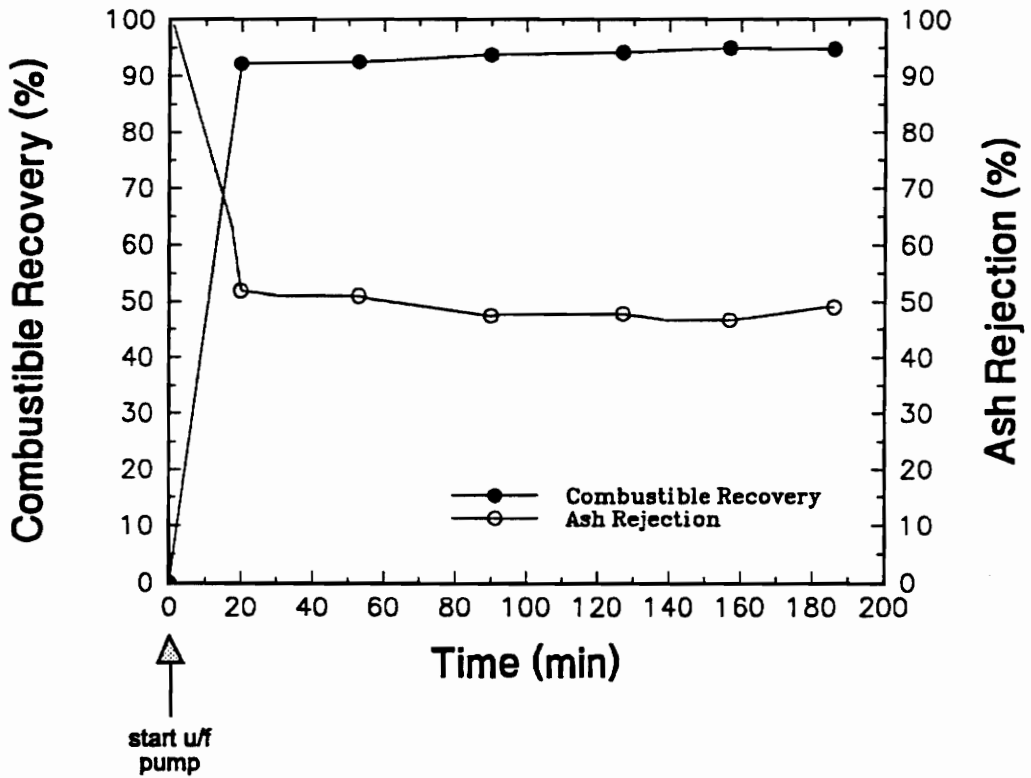


Figure 2.7 Combustible recovery and ash rejection versus retention time in the lamella thickener unit using Elkhorn No. 3 coal.

throughput of the lamella thickener was only 0.11 lb/hr for this combination of operating conditions. A significantly higher throughput of 0.33 lb/hr was obtained at a slightly lower separation efficiency of 47.5%. The penalty for operating under this set of conditions was a loss of approximately 2 percentage points in combustible recovery.

The results of the statistical computations performed using the Elkhorn No. 3 seam coal parametric test data are provided in Appendix C. In general, very good correlations were obtained between the model predictions and experimental results. The resultant response surface plots are given in Figure 2.8, while Table 2.4 shows the effect of increasing operating parameters on lamella thickener responses. According to the response surface plots, the separation efficiency was increased with decreasing feed solids content. The decrease in performance can be attributed to the increased entrapment of ash-forming minerals within the hydrophobic coagula at higher solids concentrations. The plots also indicate that an increase in wash water rate improved the separation efficiency. Higher wash water flows tended to expand the settled coagula bed and improve the removal of fine mineral matter. Finally, higher feed rates resulted in a loss of clean coal coagula to the thickener overflow, while low feed rates allowed mineral matter to settle and be recovered with the clean coal product. In summary, these analyses show that the highest overall separation efficiencies were obtained at a low feed solids content, moderate feed rate and high wash water rate.

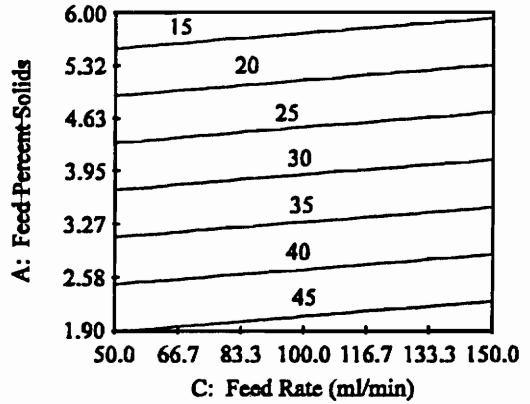
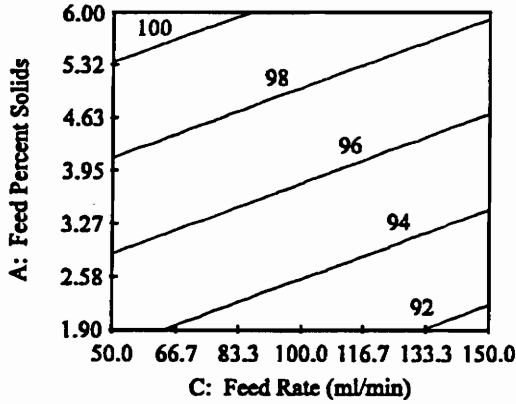
The parametric test data for the Pittsburgh No. 8 seam coal is summarized in Appendix C. Because of the difficult nature of this coal, the best separation efficiency was only 30.3%. This corresponds to a combustible recovery, ash rejection and total sulfur rejection of 93.3%, 36.9% and 23.07%, respectively. The maximum throughput of the lamella thickener for this set of operating conditions was approximately 0.23 lb/hr.

The statistical computations of the Pittsburgh No. 8 parametric test data are provided in Appendix C. The model predictions were found to be in good agreement with the experimental results. The resultant response surface plots are given in Figure 2.9. Table 2.5 shows the effect of increasing operating parameters on lamella thickener

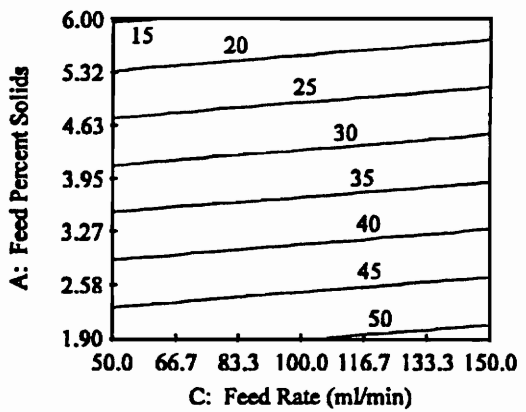
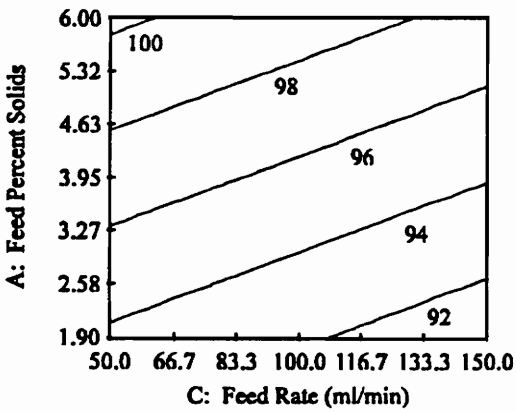
Response:
Combustible Recovery (%)

Response:
Separation Efficiency (%)

Actual Constants: Wash Water = 0 ml/min



Actual Constants: Wash Water = 20 ml/min



Actual Constants: Wash Water = 40 ml/min

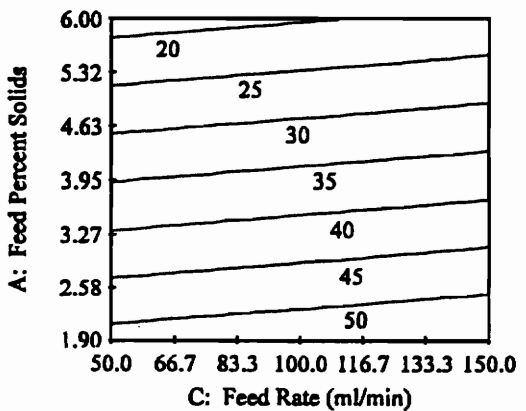
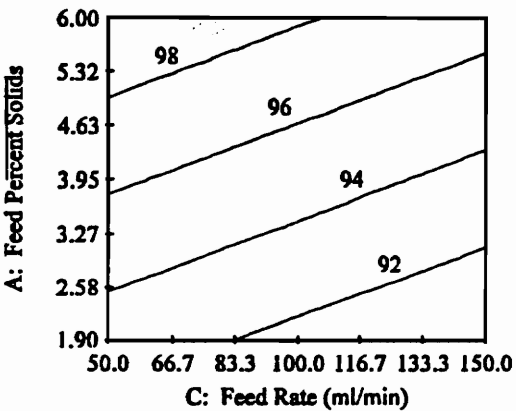


Figure 2.8 Combustible recovery and separation efficiency response surface plots obtained from the parametric testing of the Elkhorn No. 3 coal.

Table 2.4 - Correlation matrix for the lamella thickener (Elkhorn No. 3 coal).

Process Response	Effect of Increasing:		
	Feed Solids	Feed Rate	Water Rate
Clean Coal Yield	L↑	S↓	M↓
Combustible Recovery	M↑	S↓	S↓
Ash Rejection	L↓	M↑	M↑
Sulfur Rejection	M↓	S↑	S↑
Separation Efficiency	L↓	S↑	M↑

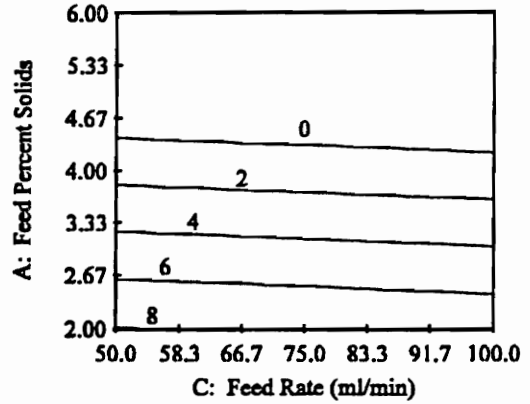
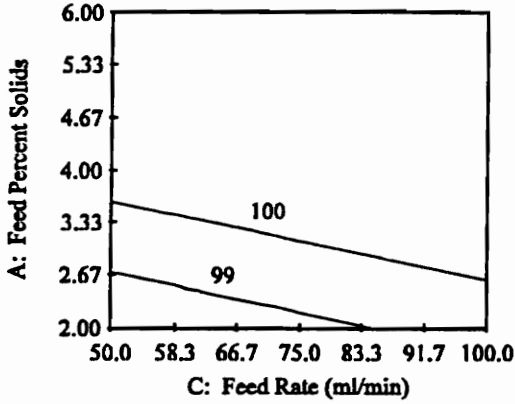
L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

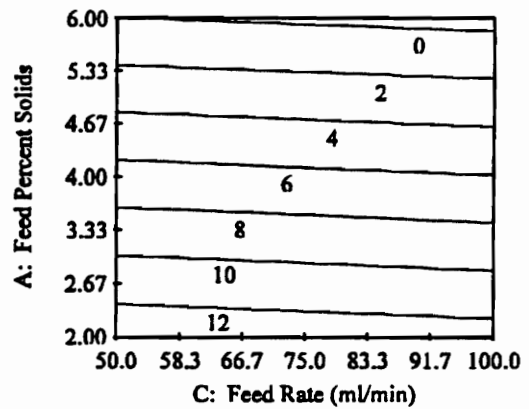
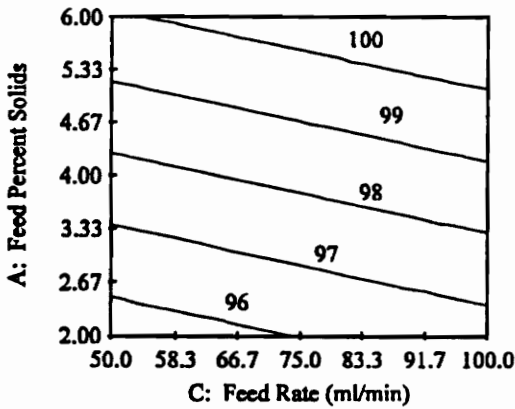
Response:
Combustible Recovery (%)

Response:
Separation Efficiency (%)

Actual Constants: Wash Water = 20 ml/min



Actual Constants: Wash Water = 60 ml/min



Actual Constants: Wash Water = 100 ml/min

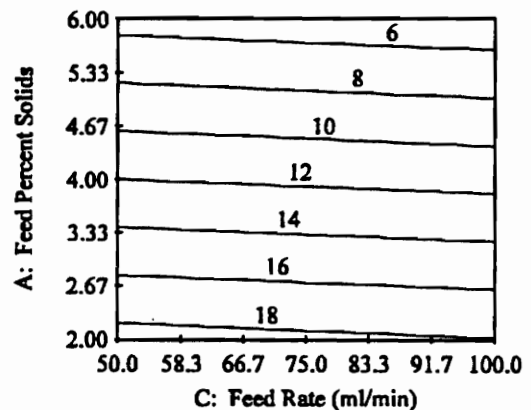
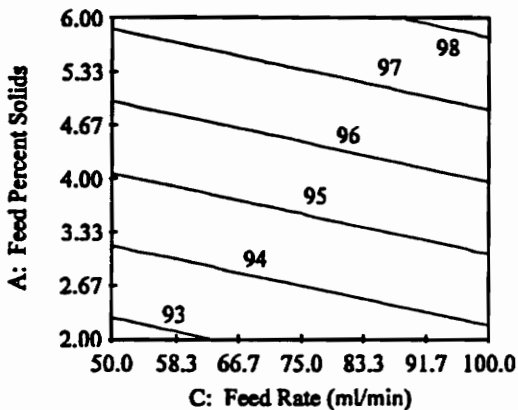


Figure 2.9 Combustible recovery and separation efficiency response surface plots obtained from the parametric testing of the Pittsburgh No. 8 coal.

Table 2.5 - Correlation matrix for the lamella thickener (Pittsburgh No. 8 coal).

Process Response	Effect of Increasing:		
	Feed Solids	Feed Rate	Water Rate
Clean Coal Yield	M↑	S↑	M↓
Combustible Recovery	S↑	S↑	M↓
Ash Rejection	L↓	S↓	L↑
Sulfur Rejection	M↓	S↓	M↑
Separation Efficiency	M↓	S↓	M↑

L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

responses. As with the Elkhorn No. 3 seam coal, the response surface plots indicate that the separation efficiency increased with a decrease in feed solids content. The decrease in performance can be attributed to the increased entrapment of ash-forming minerals within the hydrophobic coagula at higher solids concentrations. The plots also show that an increase in the wash water rate improved the separation efficiency. Higher wash water flows tended to expand the settled coagula bed and improve the removal of fine mineral matter. The significance of the wash water rate was greater for the Pittsburgh No. 8 seam coal than the Elkhorn No. 3 seam coal. However, unlike the Elkhorn No. 3 seam coal, increasing the feed rate increased the recovery of clean coal coagula and decreased the separation efficiency.

2.6 Centrifugal Sedimentation

The SHC process suffers from low throughput because of the relatively low settling rate of the small coagula. One technique for alleviating this problem would be to use a centrifuge to increase the gravitational force. This would significantly increase the sedimentation rate and enhance the separation of the coagula from dispersed mineral matter.

a) Experimental

A bench-scale Sharples semi-continuous centrifuge was initially set up for the centrifugal sedimentation experiments. However, it was noticed that the bowl was equipped with a metal cross beam which acted as an impeller at the feed entry point. Therefore, any coagula which was fed to the centrifuge was broken just before entering the bowl. Therefore, a batch rotor centrifuge (Figure 2.10) was used in place of the semi-continuous centrifuge. Preliminary tests were run on each coal seam to determine the key parameters and appropriate operating ranges for the experimental designs. The test data obtained from this work are provided in Appendix D. Tables 2.6 and 2.7 summarize the standard operating conditions and the parameter ranges examined in the preliminary tests.

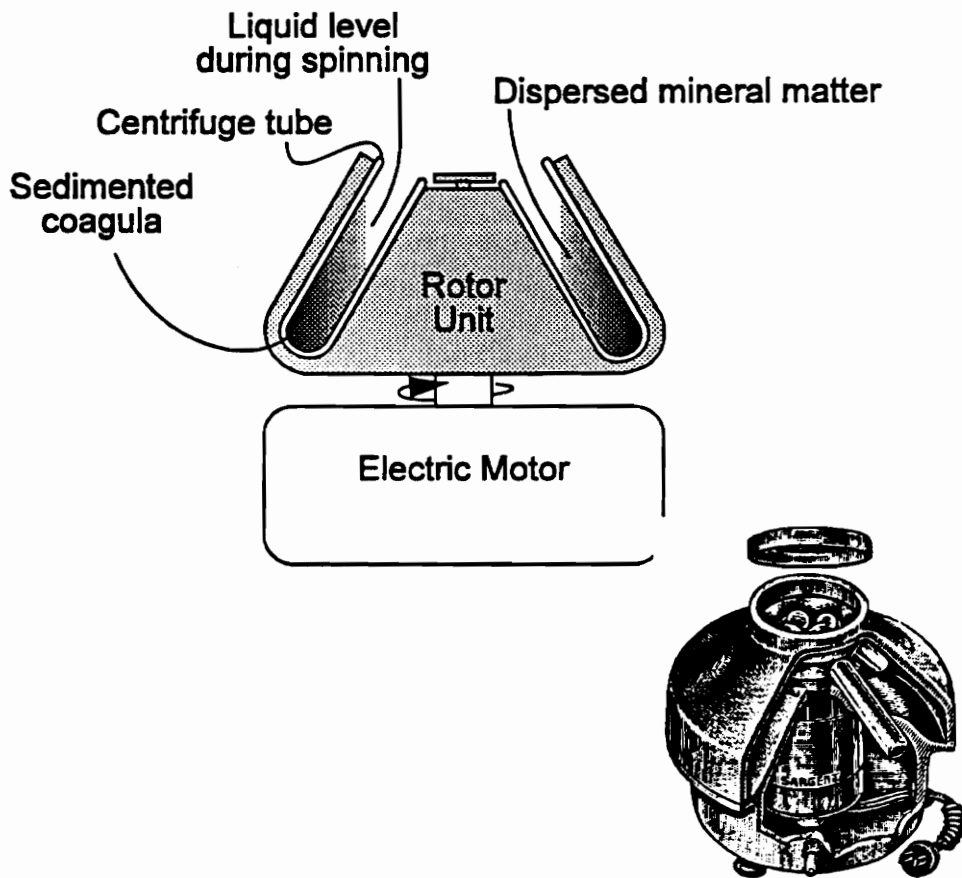


Figure 2.10 Schematic of the laboratory rotor centrifuge used in coagula recovery test work.

Table 2.6 - Standard operating conditions for the preliminary rotor centrifuge tests.

Parameter	Value
pH Condition Time	15 minutes
Chelating Agent	EDTA
Chelate Dosage	0.4 gm/liter
Chelate Condition Time	15 minutes
Feed Particle Size	-20 microns
Centrifuge Spin Time	5 minutes
Centrifuge Temperature	20°C

Table 2.7 - Experimental levels for the preliminary rotor centrifuge tests.

Coal Seam	Parameter	Units	Range
Elkhorn No. 3	Feed Solids	(%)	3, 5, 10, 14
	pH	---	7.5, 10
	Speed	(rpm)	500 - 1500
Pittsburgh No. 8	Feed Solids	(%)	4.5
	pH	---	7.5, 10
	Speed	(rpms)	500 - 2000

The feed for each rotor centrifuge experiment, approximately 265 ml, was split from a continually mixed sump containing conditioned slurry. All eight centrifuge tubes, 50 ml each, were equally filled from the sump plus a feed sample was collected to be assayed as the feed for the test. The centrifuge tubes were centrifuged immediately after filling, so as not to allow any sedimentation by gravity. After centrifuging, the dispersed matter was poured off each tube and combined, making a tailings sample. The sedimented matter from each tube was then combined as a product sample. The feed samples taken during each test were then used to balance the tailing assays, since some tailing samples were less than one gram in mass.

After the preliminary test work was completed for each coal, the results were used to determine the operating ranges for the experimental designs. Since the Pittsburgh No. 8 seam coal did not have favorable results in the preliminary test work, further centrifugal sedimentation tests were not conducted using this coal. Using the values from Table 2.8, a 3-level and 4-variable Box-Behnken experimental design was generated for the Elkhorn No. 3 seam coal using the Design-Expert software package. The experimental design required 27 individual centrifugal tests. None of the Box-Behnken tests used a chelating agent since previous experiments showed little advantage in using the EDTA. Assay results were balanced to provide a consistent set of data which the software used to perform the required regression analyses. A summary of the test results is given in Appendix D.

The regression analyses were used to develop empirical expressions relating the operating conditions to the separation performance. Initial data fitting was performed using standard linear, quadratic, and cubic regression models. The cubic model typically provided the best fit to the experimental data, while the linear model provided the worst. However, many of the coefficients in the cubic model were aliased (i.e., these terms could not be uniquely determined given the available data set). Thus, quadratic models were selected throughout this investigation since they minimized model complexity, provided good predictive correlations and reduced aliasing of the model coefficients. For

Table 2.8 - Experimental levels for the rotor centrifuge parametric tests.

Coal Seam	Parameter	Units	Parameter Range		
			Low (-1)	Normal (0)	High (1)
Elkhorn No. 3 (50% Ash)	Feed Solids	(%)	1	3	5
	pH	---	7.5	8.5	9.5
	Speed	(rpms)	750	1000	1250
	Spin Time	(minutes)	3	6	9

each model, the values of regression coefficients, standard errors, variances, t-tests, etc., were determined as a means of quantifying the statistical significance of each of the coefficients in the regression model.

After completing the statistical analyses, response surface plots were constructed for predicting combustible recovery, separation efficiency, and ash rejection for the Elkhorn No. 3 seam coal. Each set of data was plotted as functions of pH and rotor speed in the form of a contour diagram. Separate diagrams were constructed for each response to represent all possible combinations of high, normal, and low settings for the feed solids content and spin time. Thus, nine separate plots were constructed for each response.

b) Results and Discussion

The results of the statistical analyses for the Elkhorn No. 3 seam coal are provided in Appendix D. The corresponding response surface plots are shown in Figures 2.11-2.13 for combustible recovery, separation efficiency, and ash rejection, respectively. The parameter correlation chart for these tests is summarized in Table 2.9. The most significant parameters affecting the recovery of coagula were rotor speed (or effective G-force) and spin time. Increasing speed and spin time gave the smaller particles time to settle, which increased recovery. The pH also affected recovery, with the higher pH values providing higher recoveries for spin times greater than three minutes. At the higher pH values, the coagula are smaller and longer spin times are required to obtain a good recovery. Additionally, a higher feed solids content increased the specific gravity of the slurry. This lowered the particle settling rate and reduced the coagula recovery at the lower speeds and spin times.

The response surface plots indicate that the separation performance was most affected by the pH, with higher pH values giving higher separation efficiencies. As the pH decreased, there was more entrapment of gangue particles in the coagula. This reduced the separation efficiency and the rejection of ash. As determined from the batch

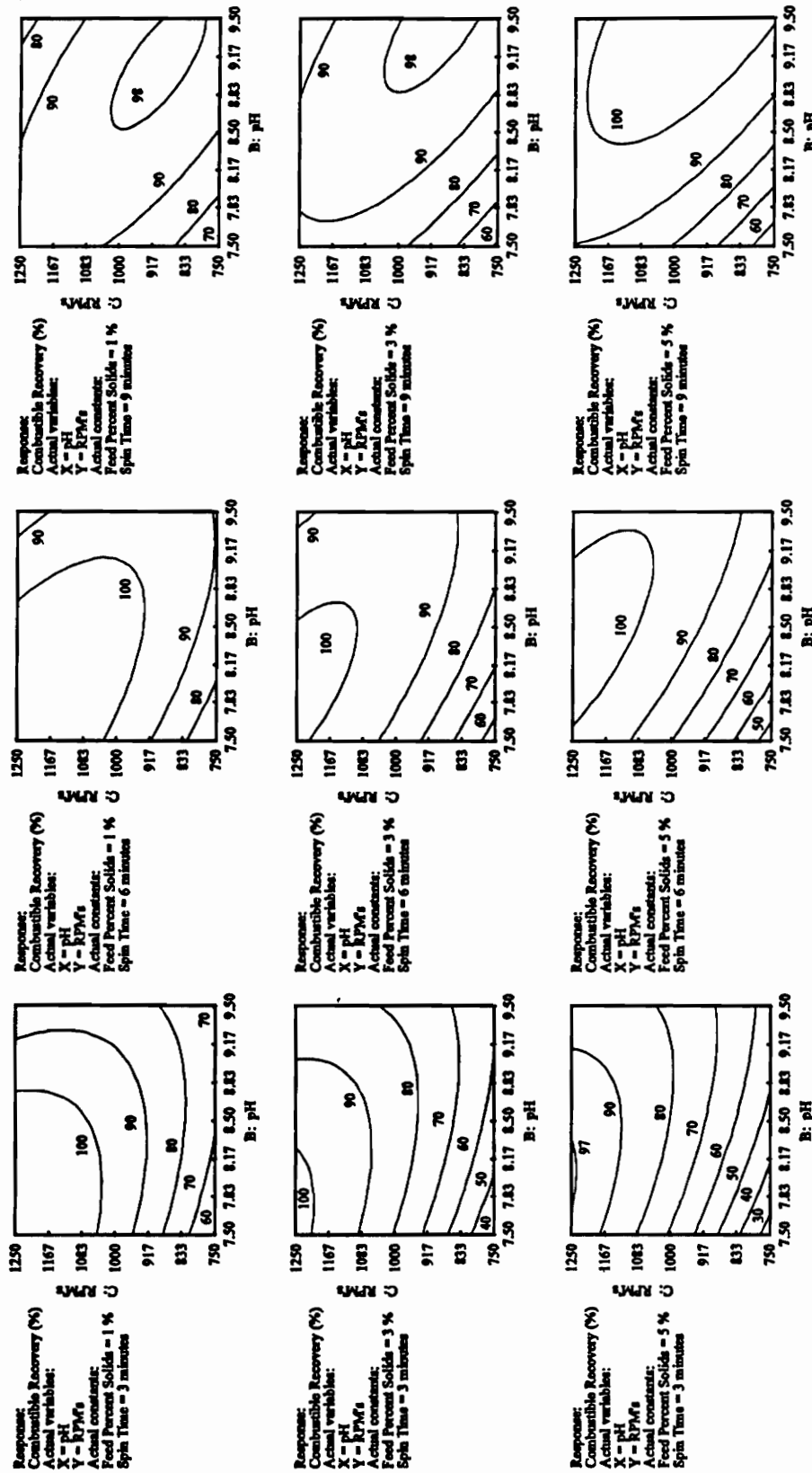


Figure 2.11 Combustible recovery response surface plots for centrifugal sedimentation of the Elkhorn No. 3 coal.

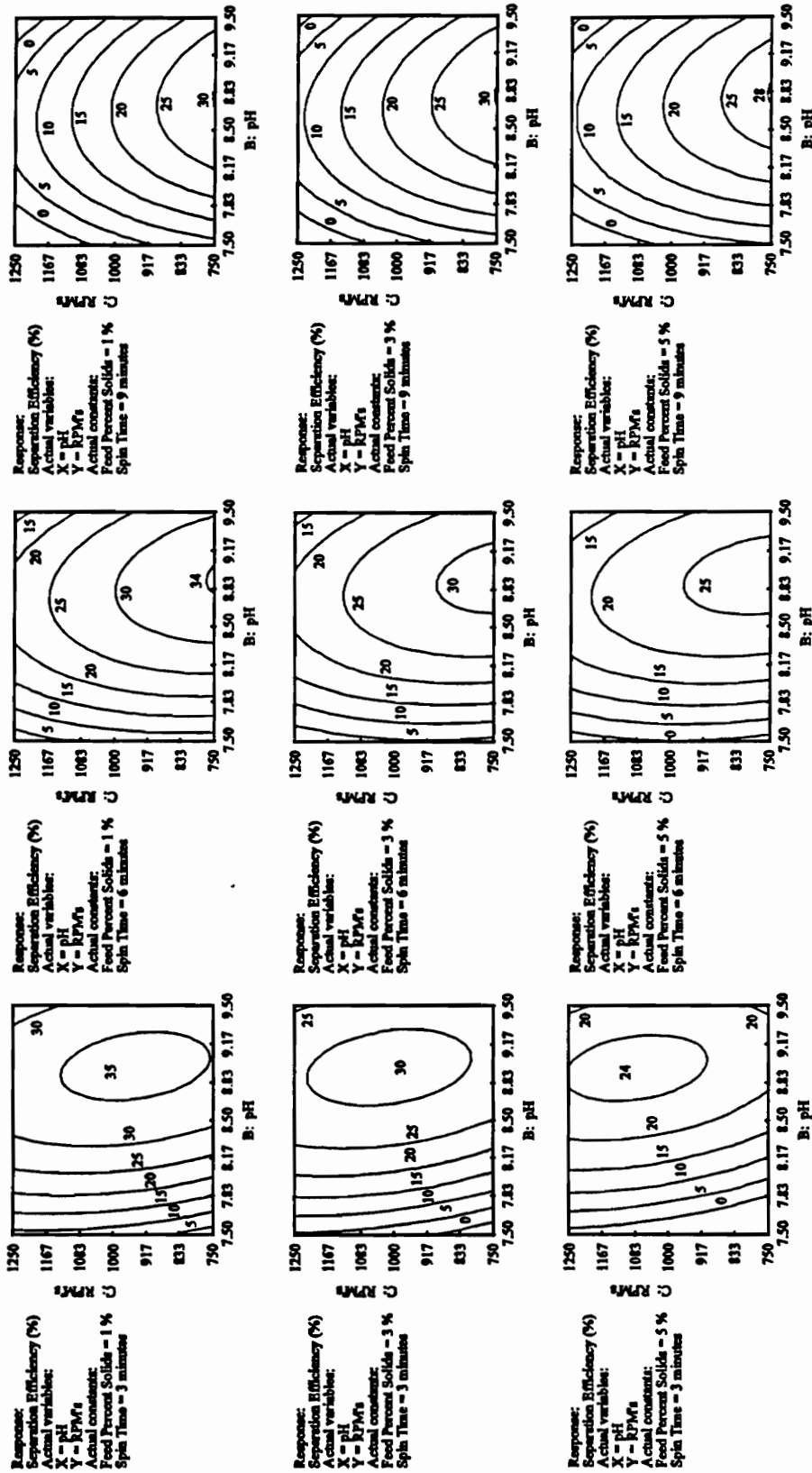


Figure 2.12 Separation efficiency response surface plots for centrifugal sedimentation of the Elkhorn No. 3 coal.

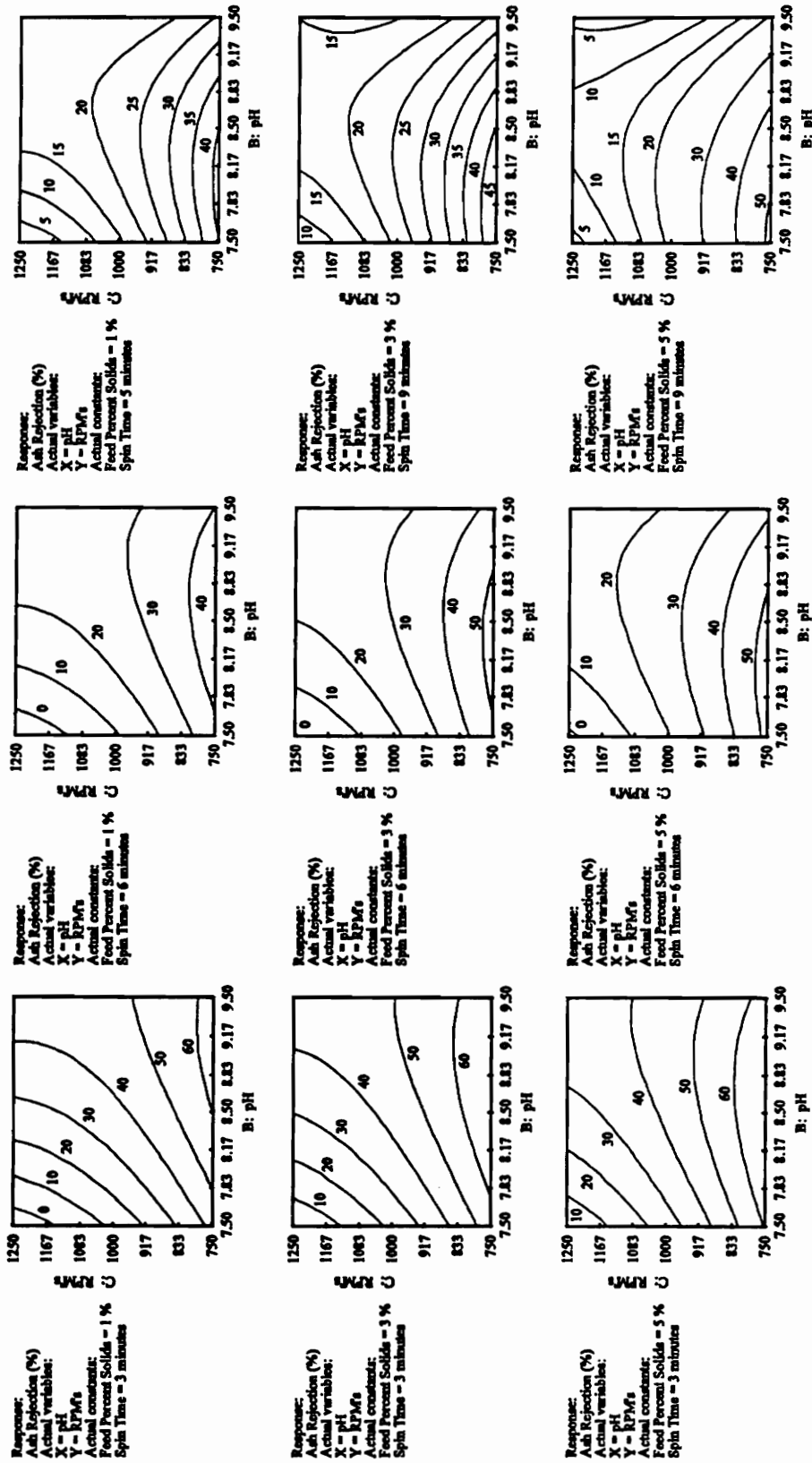


Figure 2.13 Ash rejection response surface plots for the centrifugal sedimentation tests conducted using the Elkhorn No. 3 coal.

Table 2.9 - Correlation matrix for the rotor centrifuge (Elkhorn No. 3 coal).

Process Response	Effect of Increasing:			
	Feed Solids	Spin Time	pH Level	Rotor rpm
Clean Coal Yield	S↓	M↑	S↓	L↑
Combustible Recovery	M↓	M↑	M↑	L↑
Ash Rejection	S↑	L↓	M↑	L↓
Sulfur Rejection	S↑	M↓	M↑	L↓
Separation Efficiency	M↓	M↓	L↑	M↓

L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

sedimentation tests, pH values from 8.5 to 9 were optimum for selective coagulation without EDTA. The recovery of the dispersed gangue minerals increased with increasing rotor speed and spin time. As a result, the separation efficiency and ash rejection also decreased as spin time and rotor speed increased. Finally, increasing feed solids decreased the separation efficiency due to increased entrapment at higher feed solid levels.

The response surface plots show that, like combustible recovery, ash rejection was most affected by rotor speed and spin time. Longer spin times and higher speeds increase the recovery of finer particles. In the SHC process the finer particles are comprised of dispersed gangue particles. So the recovery of the finer gangue particles decreased the ash rejection. The pH level also affected the ash rejection. Where higher pH values have higher ash rejection. The lower the pH value, the greater the entrapment of gangue minerals in the recovered coagula. The feed solids content had little affect on the ash rejection.

Centrifugal sedimentation tests were also attempted using the Pittsburgh No. 8 seam coal. The results of these preliminary tests are summarized in Figure 2.14. Unfortunately, the centrifuge did not produce favorable results with this particular coal. Thus, no further centrifuge work was undertaken using the Pittsburgh No. 8 coal.

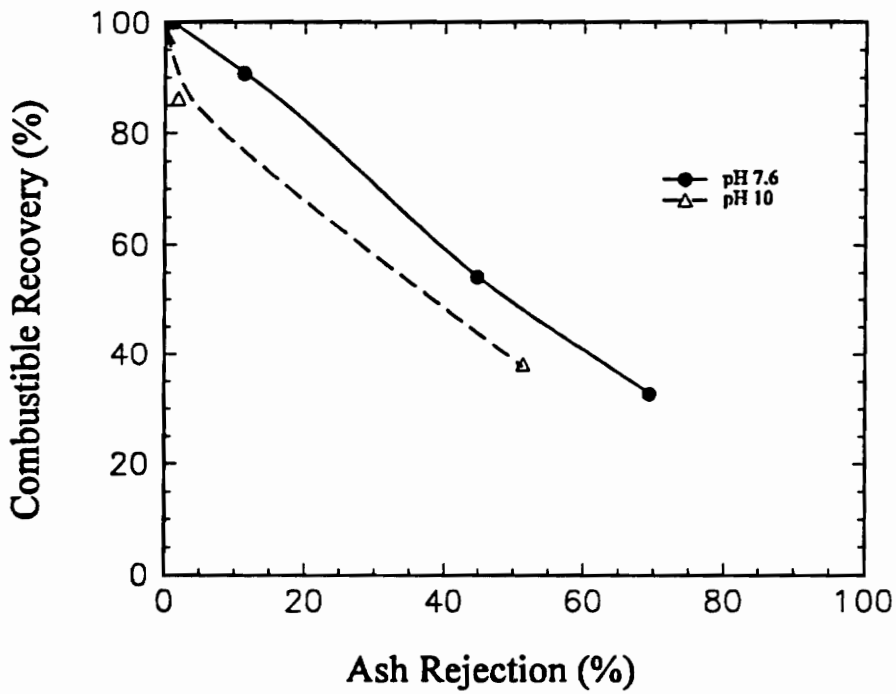


Figure 2.14 Grade-recovery curves for the centrifugal sedimentation tests conducted using the Pittsburgh No. 8 coal.

2.7 Vacuum Filtration

It was thought that vacuum filtration may have an advantage over the sedimentation techniques in that it is relatively unaffected by the slow sedimentation rate of the coagula. Therefore, a bench-scale test program was undertaken in the present work to evaluate the applicability of vacuum filtration for the separation of coal coagula from dispersed mineral matter.

a) Experimental

The effects of filter medium, applied vacuum, and feed percent solids were studied using both a filter leaf test kit and a bench-scale continuous filtration unit. The preliminary filter leaf tests were conducted prior to continuous filtration tests in order to determine appropriate operating ranges for the continuous unit (i.e. cycle time, cake form time, form vacuum, wash time, dry time, etc.). The filter leaf batch tests also examined nine different types of filter media. Five of these were selected for use in the continuous filtration unit test work. The continuous tests were conducted using a 100 mm x 200 mm diameter drum filter (Figure 2.15). The test unit was modified to permit the use of wash water to remove entrained mineral matter from the filter cake. At least 3 different settings were examined for each operating parameter.

Figure 2.16 shows a schematic of the filter leaf vacuum filtration tests. In these tests, feed slurry was prepared in a baffled 6-inch diameter (3-liter) mixing cell mounted on a magnetic stirrer. Approximately 1 liter of minus 20 micron slurry (Elkhorn No. 3 seam coal) was added to the mixing cell and conditioned for 15 minutes with EDTA. After conditioning, the pH was adjusted to 7.5 using hydrochloric acid (HCl) or sodium hydroxide (NaOH). The slurry was agitated for 15 minutes after reaching the 7.5 pH level. After the pH stabilized, the agitation rate of the magnetic stirrer was decreased to permit the coagula to grow for a period of approximately 5 minutes. While holding the coagula in suspension by gentle mixing, a 4-inch diameter filter leaf was submerged into the slurry. Tests were run with feed solids contents from 5% to 15%, submergence times

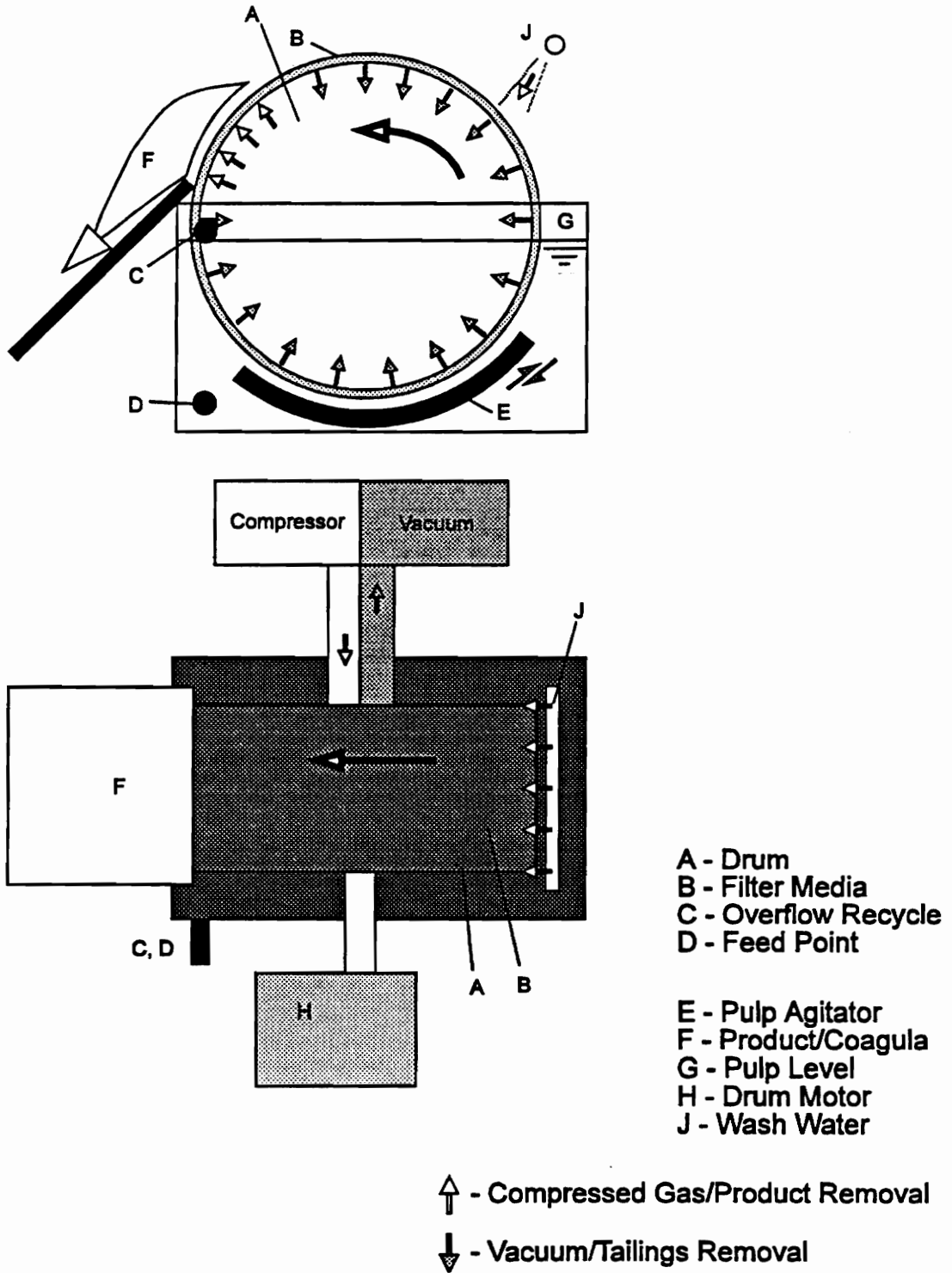


Figure 2.15 Schematic of the continuous vacuum filtration separation unit.

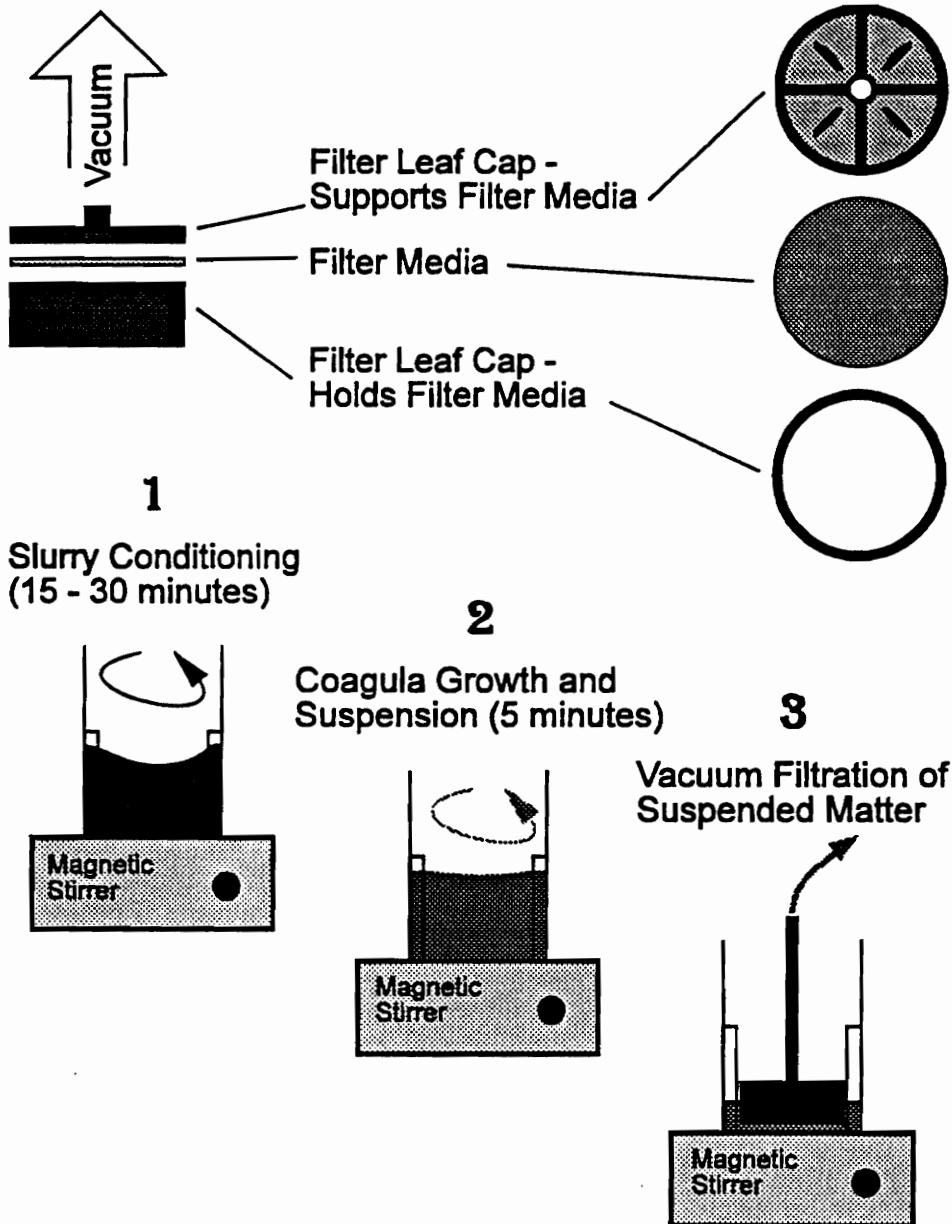


Figure 2.16 Schematic of the filter leaf batch test unit equipped with a vacuum pump.

as long as 3 minutes, and applied vacuums ranging from full to minimum. The first round of filter leaf tests was conducted using a laboratory vacuum pump.

The second round of filter leaf batch tests was run by feeding the conditioned slurry directly onto the filter medium. In these tests, the coagulated slurry was pumped into an inverted filter leaf (Figure 2.17) in place of the 3-liter mixing cell. The slurry was then filtered using a positive-displacement peristaltic pump operating at a volume flow of 50 ml/min. The peristaltic pump was used to decrease the applied vacuum to the filter leaf. The feed solids content was held constant at 5% solids while testing both the Elkhorn No. 3 and Pittsburgh No. 8 seam coals.

The continuous drum filter tests were attempted using only the Elkhorn No. 3 coal because of the poor results obtained during the filter leaf testing of the Pittsburgh No. 8 coal. In these tests, the pH was adjusted to 8.5 using hydrochloric acid (HCl) or sodium hydroxide (NaOH) and no EDTA was added. Five different filter media were examined. The applied vacuum and drum cycle time were independently varied from their maximum to minimum settings during each of the five filter media tests. Although higher feed solid contents would normally be preferred to promote the formation of the filter-cake, the solids content was kept at 5% to minimize the entrapment of gangue in the coagula.

b) Results and Discussion

The first round of filter leaf batch tests was performed using the Elkhorn No. 3 coal. The test data are summarized in Appendix E. Unfortunately, these tests did not produce favorable results. Both the dispersed and coagulated particles were consistently pulled through the filter medium and no filter cake was formed. In an attempt to correct these problems, a peristaltic pump was used in place of the vacuum pump to reduce the intensity of the applied vacuum. The second round of filter leaf batch tests also performed poorly despite using the lower vacuum. Separation efficiencies were less than 4% with one filtration pass. The feed slurry either passed through the filter media or the

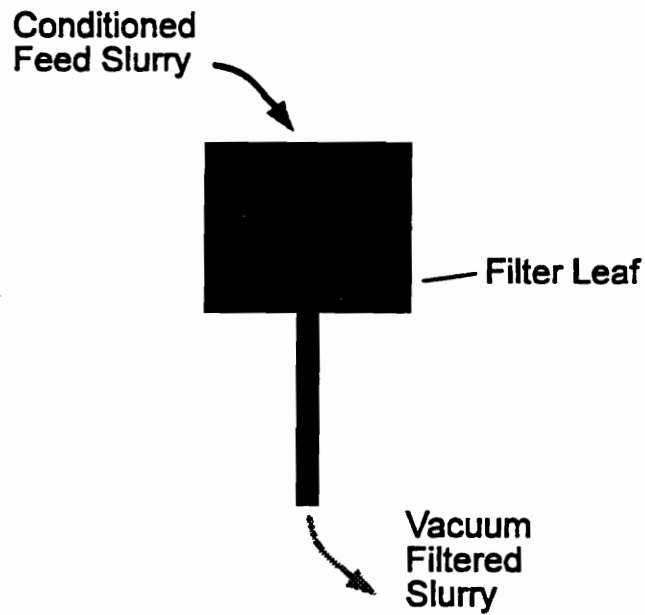
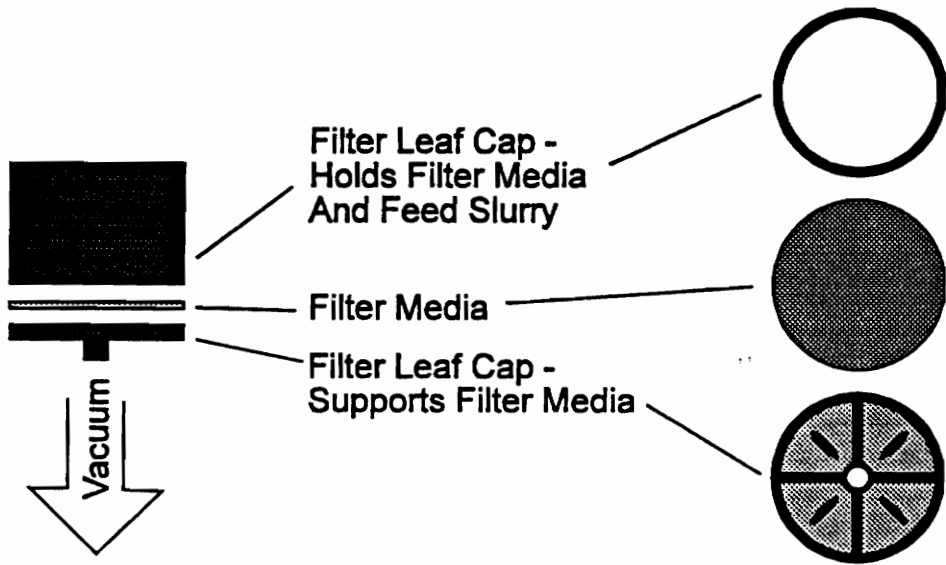


Figure 2.17 Schematic of the inverted filter leaf batch test unit equipped with a peristaltic pump.

media blinded and trapped the solids within the filter cake. In one test, the slurry which passed through the media was filtered twice without removing the filter cake. This produced the best filtration result with an overall separation efficiency of 12% at a 77% recovery of combustibles. The second round filter leaf tests are summarized in Appendix E.

The second round of filter leaf batch tests was conducted using the Pittsburgh No. 8 coal. As with the Elkhorn No. 3 coal, the separation efficiencies were very poor and did not exceed 1%. The filter media either formed a very thin cake (through which most of the solids passed) or the media became blinded (which retained all solids). The data for the filter leaf tests conducted using the Pittsburgh No. 8 coal are also given in Appendix E.

The continuous drum filter tests conducted with the Elkhorn No. 3 coal also did not produce encouraging results. There was no filter cake formation even at the lowest vacuum pump setting (< 1" of Hg). Furthermore, there was no improvement in separation performance when the pulp agitator was reduced to its lowest speed and the drum cycle time was increased to 3.5 minutes. In all cases, the coagulated feed slurry was pulled through the filter media without any separation occurring.

2.8 Froth Flotation

A unique characteristic of the SHC process is that the coagula formed during the process are hydrophobic. As a result, it is theoretically possible to recover the coagula by froth flotation. This would offer an attractive technique for overcoming the limitations associated with the low settling rates of the coal coagula. Furthermore, the use of hydrophobic coagulation should significantly increase the capacity of the froth flotation process since flotation rates increase sharply with increasing particle size. Therefore, a series of tests was conducted to evaluate the potential advantages of using froth flotation to separate the coal coagula from dispersed mineral matter.

a) Experimental

A 5 cm diameter and 150 cm tall high Microcel™ flotation column was used in the froth flotation tests (Figure 2.18). A column was employed because it provides a relatively quiescent environment which should minimize the break-up of coagula. In addition, the column wash water can be used to minimize fine particle entrainment in the froth product. The tests were conducted as a function of volumetric feed rate, feed percent solids, aeration rate, and wash water rate. Three different settings were examined for each operating parameter. Standard operating conditions for the flotation tests are summarized in Table 2.10. The test ranges for each operating parameter are given in Table 2.11. The feed slurry was introduced 114 cm above the bubble generator. The pH of the feed slurry and wash water was maintained at pH 7.5 throughout the test program. Both the Elkhorn No. 3 and Pittsburgh No. 8 coals were tested using this approach.

Separate 3-level and 4-variable factorial designs (with repeat tests at the central point) were generated for each coal using the Design-Expert software package. Each experimental design required 20 individual flotation tests. Assay values were material-balanced to provide a consistent set of data prior to using the software to perform the required regression analyses. A summary of the test results is given in Appendix F. The regression analyses were used to develop empirical expressions relating the operating

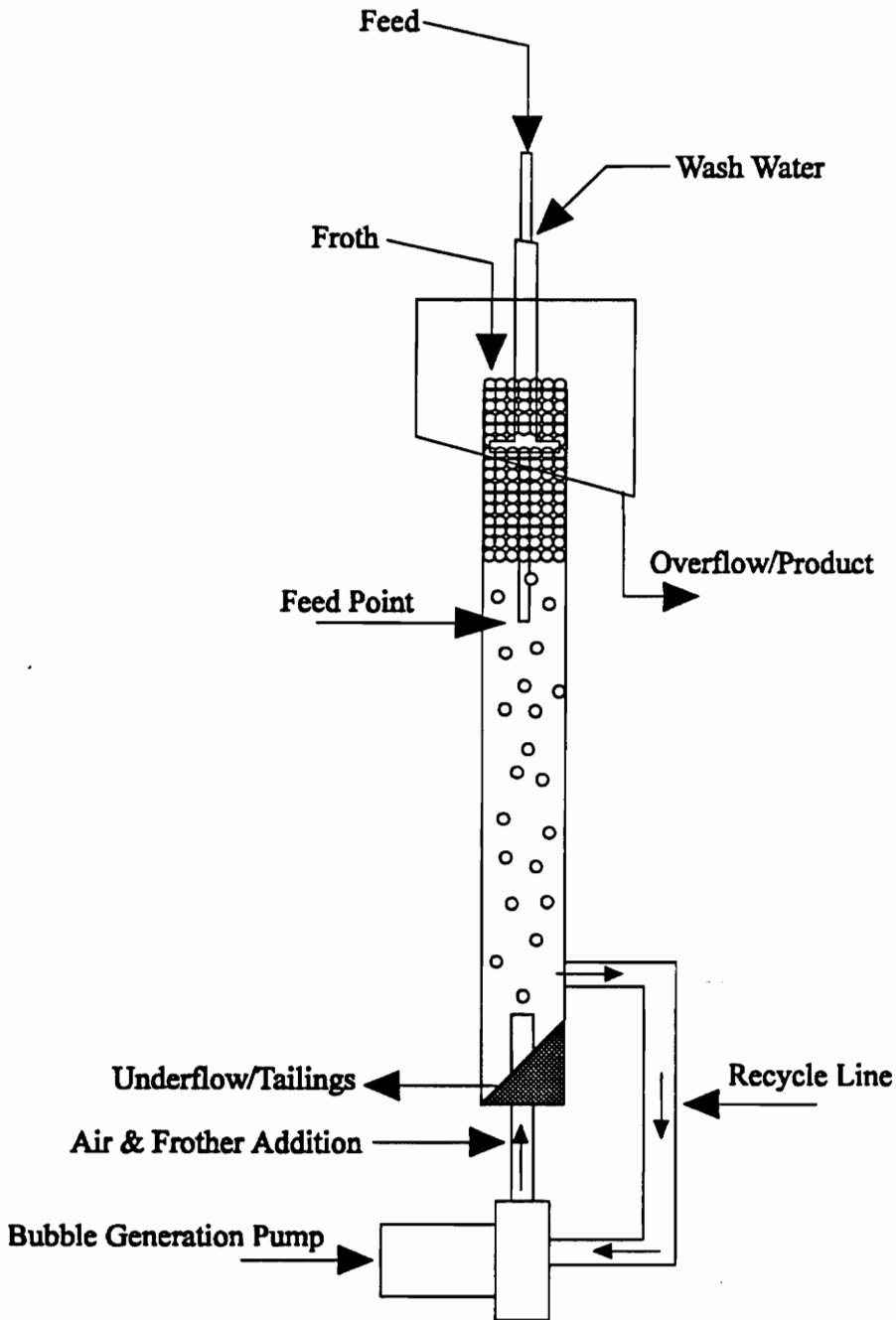


Figure 2.18 Schematic of laboratory flotation column used in coagula recovery test work.

Table 2.10 - Standard operating conditions for the flotation column tests.

Parameter	Value
pH	7.5
pH Condition Time	15 minutes
Chelating Agent	EDTA
Chelate Dosage	0.4 gm/liter
Chelate Condition Time	15 minutes
Frother Dosage	4 - 5 lb/ton
Feed Particle Size	-20 microns

Table 2.11 - Experimental levels for the column flotation parametric tests.

Coal Seam	Parameter	Units	Parameter Range		
			Low (-1)	Normal (0)	High (1)
Elkhorn No. 3 (50% Ash)	Feed Solids	(%)	4	7	10
	Wash Water	(ml/min)	200	400	600
	Feed Rate	(ml/min)	150	200	250
	Air Rate	(cc/min)	1000	1500	2000
Pittsburgh No. 8	Feed Solids	(%)	4	7	10
	Wash Water	(ml/min)	200	400	600
	Feed Rate	(ml/min)	150	200	250
	Air Rate	(cc/min)	1000	1500	2000

conditions to separator performance. Data fitting was performed using standard linear, quadratic, and cubic regression models. While the cubic model typically provided the best fits, many of the coefficients were aliased (i.e., these terms could not be uniquely determined given the available data set). Thus, quadratic models were selected since they minimized model complexity, provided good predictive correlations and reduced aliasing of the model coefficients. For each model, the values of regression coefficients, standard errors, variances, t-tests, etc., were determined as a means of quantifying the statistical significance of each of the coefficients in the regression model.

After completing the statistical analyses, response surface plots were constructed for predicting combustible recovery, concentrate ash, and concentrate sulfur for each of the two base coals. Each set of data were plotted as functions of feed flow rate and solids content in the form of a contour diagram. Thus, nine separate plots were constructed for each response to represent all possible combinations of high, normal, and low settings for the aeration rate and wash water rate.

b) Results and Discussion

The results of the statistical analyses for the Elkhorn No. 3 seam coal are provided in Appendix F. The corresponding response surface plots are shown in Figures 2.19-2.21 for combustible recovery, concentrate ash, and concentrate sulfur, respectively. The parameter correlation chart for these tests is summarized in Table 2.12. The test data indicate that wash water rate was by far the most significant parameter affecting the recovery of the hydrophobic coagula (Figure 2.19). Recovery typically decreased with increasing wash water rate, suggesting that the coagula formed by the SHC process are too weak to withstand the shear forces created in the froth by the wash water. The fact that flotation was unable to achieve recoveries greater than approximately 25-35% at high wash water rates demonstrates the difficulty in capturing fine particles using traditional flotation techniques. Therefore, the development of new flotation cells that are more

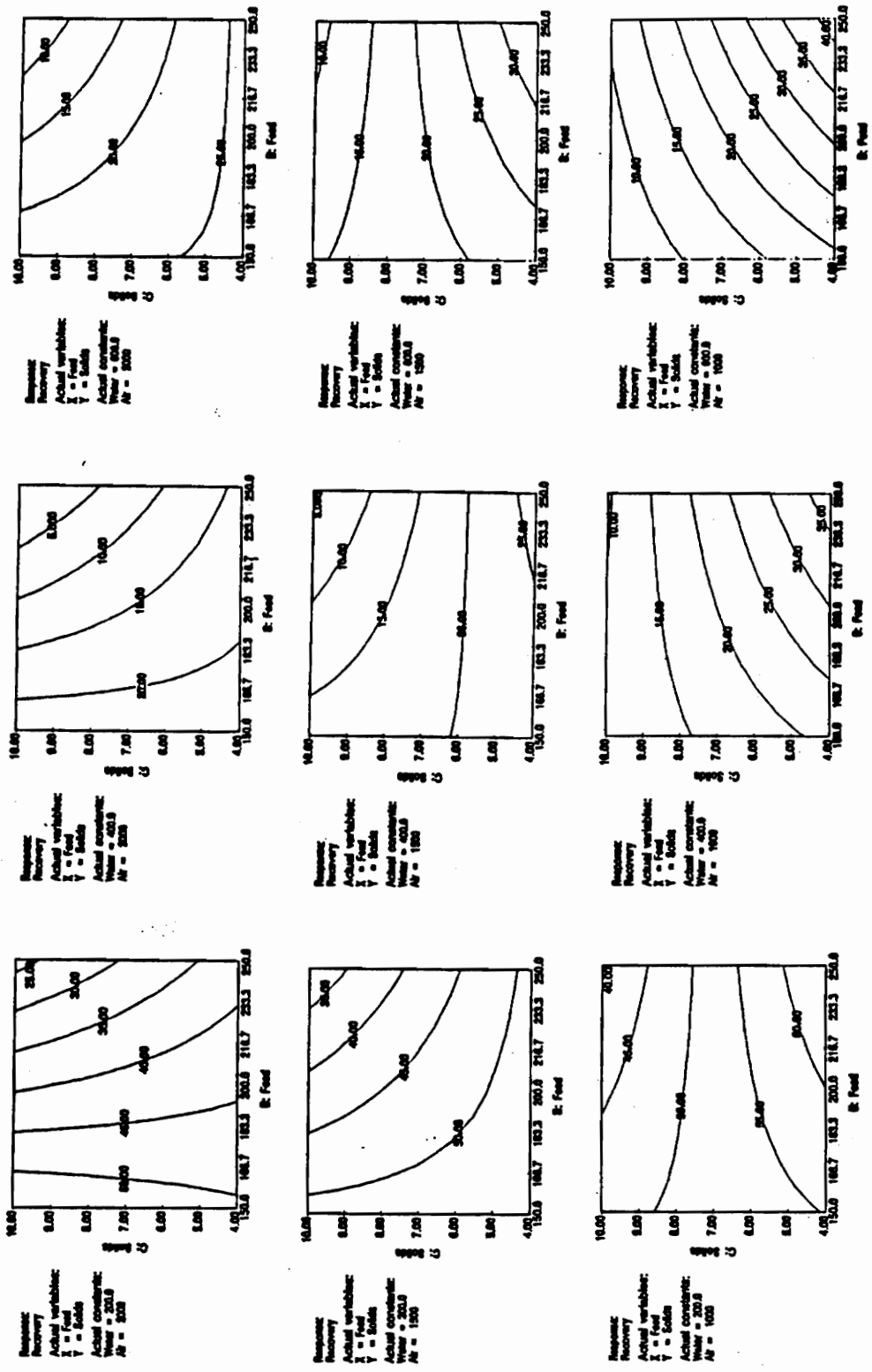


Figure 2.19 Combustible recovery response surface plots for column flotation from the parametric testing of the Elkhorn No. 3 coal.

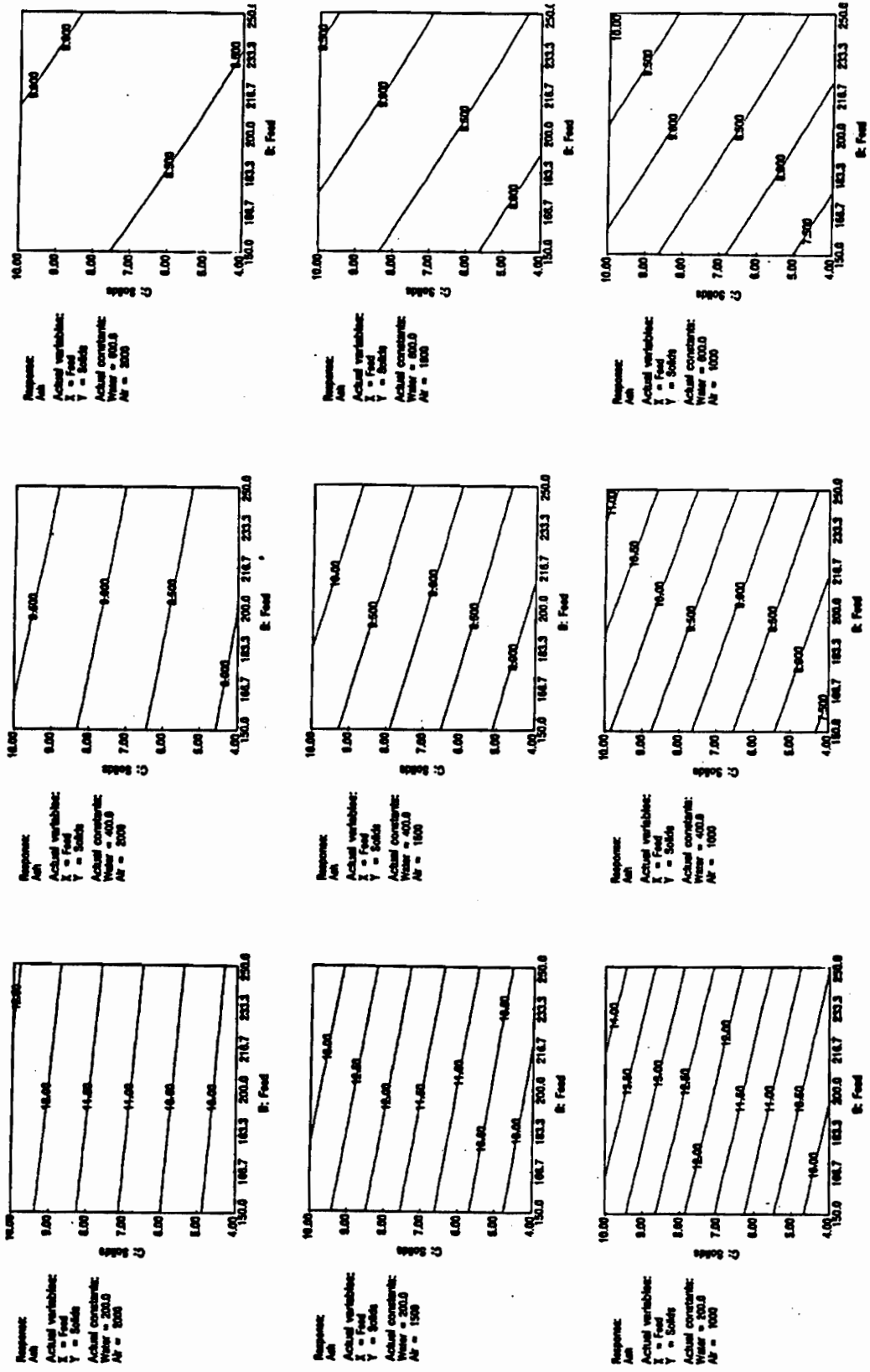


Figure 2.20 Ash content response surface plots for column flotation from the parametric testing of the Elkhorn No. 3 coal.

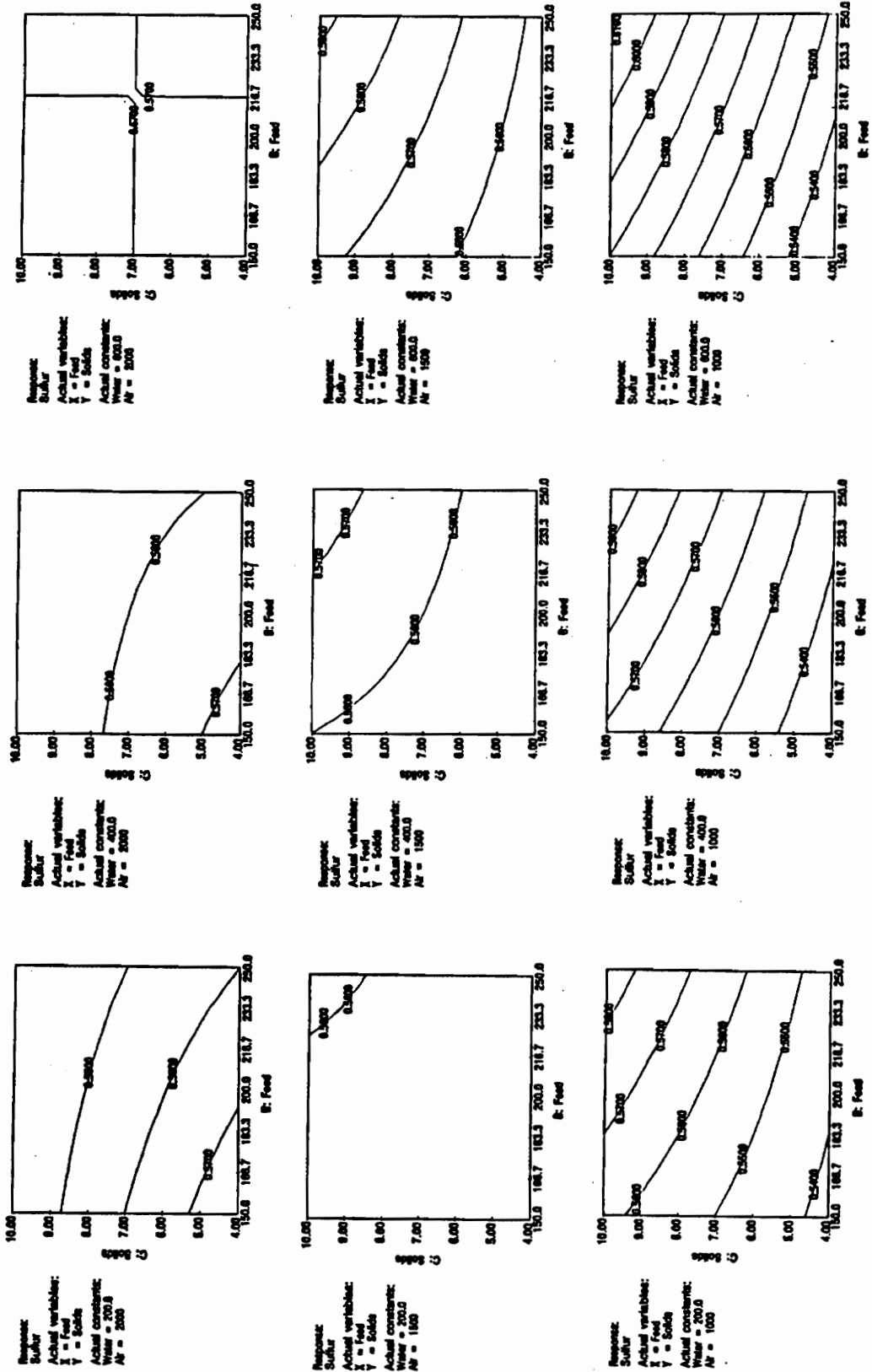


Figure 2.21 Sulfur content response surface plots for column flotation from the parametric testing of the Elkhorn No. 3 coal.

Table 2.12 - Correlation matrix for the flotation column (Elkhorn No. 3 coal).

Process Response	Effect of Increasing:			
	Feed Solids	Air Rate	Feed Rate	Water Rate
Clean Coal Yield	M↓	S↓	S↓	L↑
Combustible Recovery	M↓	S↓	S↓	L↓
Ash Rejection	S↑	S↑	S↑	M↑
Sulfur Rejection	M↑	S↑	S↑	L↑
Separation Efficiency	M↓	S↓	S↓	L↓

L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

quiescent and can take advantage of the particle size enlargement created by the SHC process may prove very beneficial.

Wash water rate was also found to produce the most significant change in the quality of the froth concentrate. High wash water rates provided the lowest ash contents (Figure 2.20) because of reduction in fine particle entrainment and a decrease in entrapment as a result of coagula breakup. This data suggests that, in addition to quiescent conditions, multiple stages of cleaner flotation may be required to effectively process the coagula formed by the SHC process. In contrast, little variation was observed in the sulfur contour plots (Figure 2.21) due to the relatively low sulfur content of the Elkhorn No. 3 seam coal.

Combustible recovery was generally found to decrease with increasing feed solids content (Figure 2.19). This finding was expected since higher solids loading often result in excessive crowding of the bubble surfaces and a corresponding loss of floatable particles. The same mechanism was responsible for the improved concentrate ash and sulfur values obtained at the higher solids contents (Figures 2.20 and 2.21). Next to wash water rate, feed solids content was found to be the most significant parameter in determining the recovery and quality of the froth concentrate.

Aeration rate was found to have a small negative impact on coagula recovery (Figure 2.19). This trend is opposite to that typically observed during the flotation of well-dispersed hydrophobic particles. In the present work, the higher gas rates are believed to create increased levels of turbulence that are detrimental to the formation of large, stable coagula. The increase in mixing turbulence with gas rate has been previously reported by researchers at Virginia Tech. Thus, high aeration rates and associated turbulence should be avoided when floating hydrophobic coagula. Aeration rate had little impact on either the ash or sulfur contents of the froth products (Figures 2.20 and 2.21).

Recovery tended to decrease with an increase in feed flow rate for most of the operating conditions examined in the present work (Figure 2.19). Similar responses were noted for the ash and sulfur contents of the froth concentrate (Figures 2.20 and 2.21).

The reduced recovery can be attributed to a decrease in retention time and increase in coagula breakage at the higher feed flow rates. The only exception to this trend was observed when low aeration rates were used together with low solids contents or high wash water flow rates. Since the lb/ton frother was held constant throughout the test program, the higher feed flow rates typically resulted in an increased concentration of frother in the flotation pulp. The additional frother reduced the bubble size and resulted in higher particle recoveries. This trend was not observed at low wash water rates since less dilution of the frother occurred under this condition. The incremental increase in frother concentration was too small to have an impact at the higher aeration rates.

A second set of statistical analyses was performed to examine the effects of flotation operating parameters on the recovery of hydrophobic coagula from the Pittsburgh No. 8 seam coal. Details related to the fitting of the various empirical models are provided in Appendix F. The response surface plots for combustible recovery, concentrate ash, and concentrate sulfur are shown in Figures 2.22-2.24. The parameter correlation chart for these tests is summarized in Table 2.13.

As with the Elkhorn No. 3 seam coal, the wash water rate was again found to have a strong adverse effect on the flotation recovery of the hydrophobic coagula (Figure 2.22). This data reinforces the hypothesis that the coagula formed by the SHC process are too weak to withstand the shear forces created in the froth by the wash water. Wash water rate also produced very large changes in the ash and sulfur contents of the froth concentrate (Figures 2.23 and 2.24). It is believed that the high wash water rates reduce fine particle entrainment and decrease entrapment as a result of coagula breakup. These data again suggest that the development of multi-stage quiescent flotation cells may be worth investigating.

Combustible recovery was found to be relatively insensitive to feed solids content, particularly when the air rate or wash water rate was high (Figure 2.22). The same trends were observed in the concentrate ash and sulfur contour plots (Figures 2.23

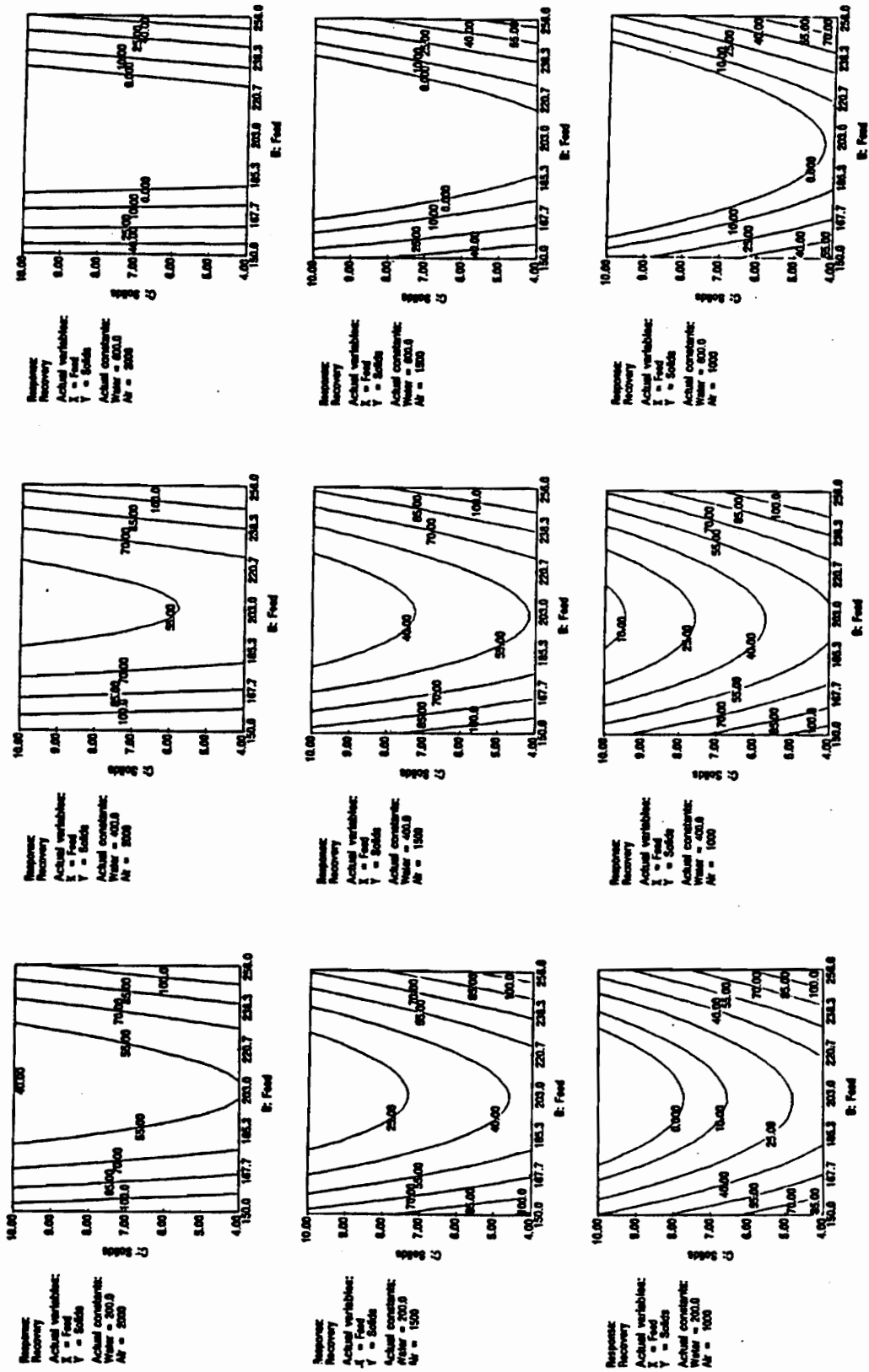


Figure 2.22 Combustible recovery response surface plots for column flotation from the parametric testing of the Pittsburgh No. 8 coal.

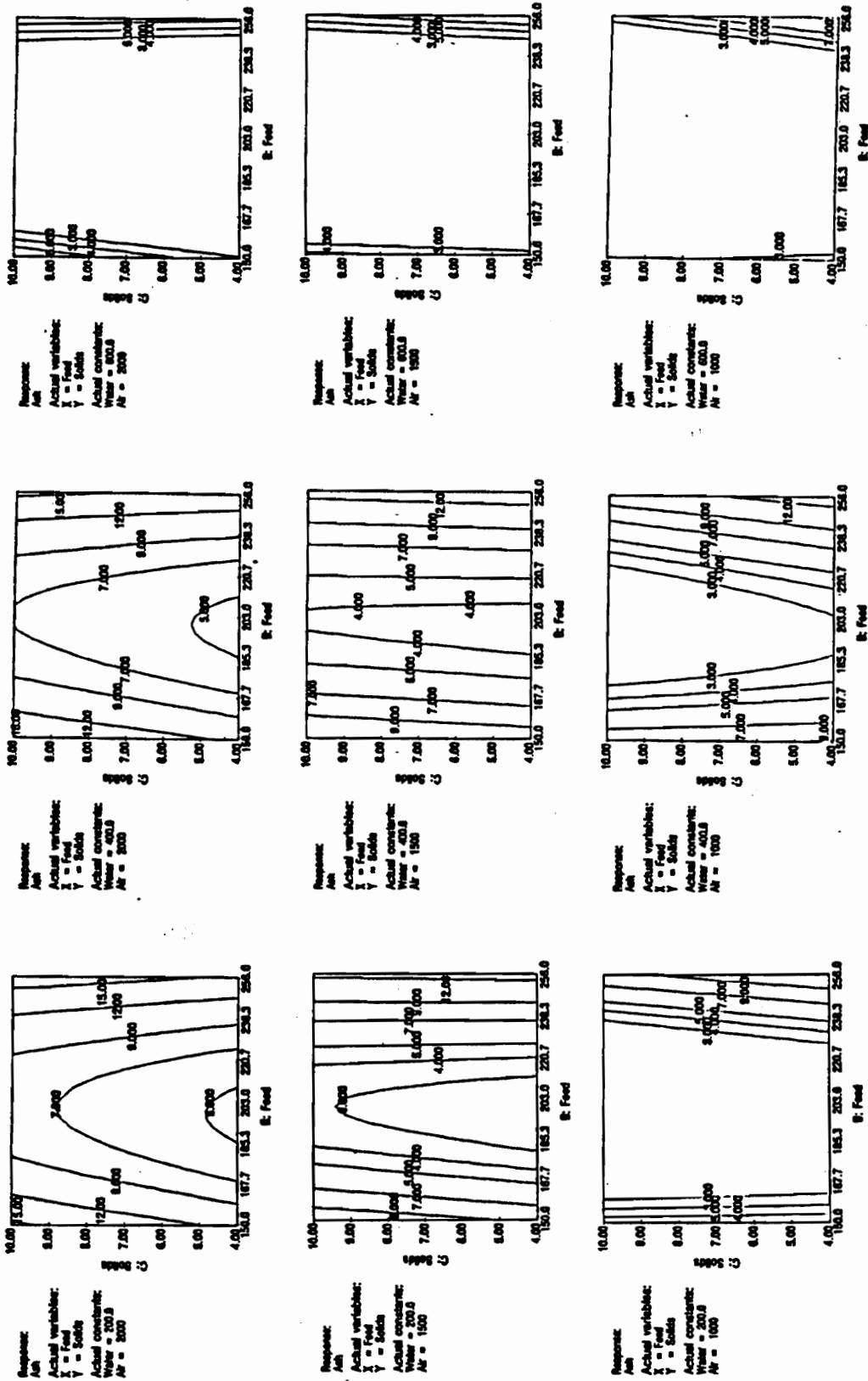


Figure 2.23 Ash content response surface plots for column flotation from the parametric testing of the Pittsburgh No. 8 coal.

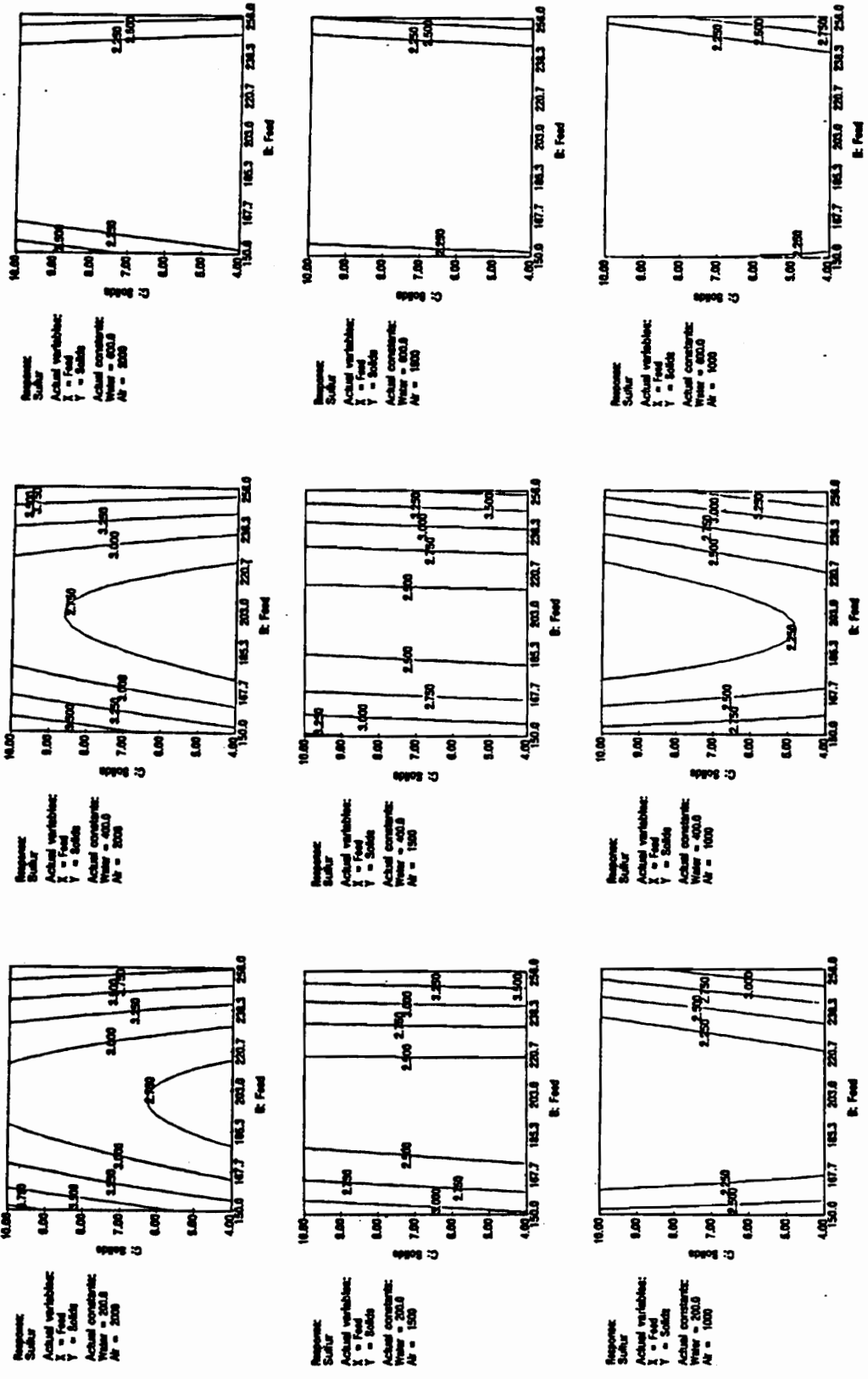


Figure 2.4 Sulfur content response surface plots for column flotation from the parametric testing of the Pittsburgh No. 8 coal.

Table 2.13 - Correlation matrix for the flotation column (Pittsburgh No. 8 coal).

Process Response	Effect of Increasing:			
	Feed Solids	Air Rate	Feed Rate	Water Rate
Clean Coal Yield	L↓	L↑	S↑	L↓
Combustible Recovery	L↓	M↑	S↑	L↓
Ash Rejection	S↑	L↓	M↓	L↑
Sulfur Rejection	M↑	L↓	M↓	L↑
Separation Efficiency	L↓	M↓	M↓	M↑

L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

and 2.24). These trends can be largely attributed to the overall poor response of the Pittsburgh No. 8 seam coal to selective coagulation, i.e., the Pittsburgh No. 8 coal either coagulated nonselectively or remained completely dispersed.

Aeration rate was found to have only a small impact on combustible recovery for the Pittsburgh No. 8 seam coal (Figure 2.22). In flotation, higher gas rates typically increase recovery by enhancing the flotation kinetics. However, in this case, higher gas rates can also result in higher coagula breakage rates which may reduce the recovery. For this particular coal, higher recoveries associated with enhanced flotation kinetics appear to be offset by the increase in coagula breakage rate with increasing gas rate. Similar trends are observed in the ash and sulfur contour plots (Figures 2.23 and 2.24).

The effect of feed flow rate on the recovery of the Pittsburgh No. 8 seam coal was completely unexpected (Figure 2.22). Typically, good recoveries were only achieved at very low or very high feed flow rates. Similarly, high ash and sulfur values were observed in the same operating regions (Figures 2.23 and 2.24). The batch sedimentation tests conducted in section 2.4 indicated that the coagulation of the Pittsburgh No. 8 seam coal was very nonselective, but much stronger than that observed for the Elkhorn No. 3 seam coal. Under conditions of low turbulence (i.e., low aeration, feed, and wash water flow rates), the strong coagulation tendency resulted in nearly 100% coal recoveries. Higher feed rates increased the level of turbulence, leading to a sharp increase in coagula breakup and corresponding loss of recovery. However, as discussed earlier, an increase in feed rate also increased the frother concentration. The higher frother addition rate resulted in the formation of smaller air bubbles, eventually leading to increased recoveries at the higher feed rates. These complex interactions, which have not yet been addressed by the research community, need to be investigated further in order to develop improved flotation and selective coagulation technologies.

Additional tests were conducted with each of the two coals to determine whether separation performance could be improved by using different pH levels. In general, low pH values cause greater coagulation of coal than do higher pH values. Therefore, tests

were conducted at pH values of 4, 6, 7.5 and 10 and feed flow rates of 100, 150 and 225 ml/min in order to assess whether coagulated coal floats faster and more cleanly than dispersed coal. Standard operating conditions for the flotation tests are summarized in Table 2.14. Test conditions, assay values, and various performance parameters for the tests on Elkhorn No. 3 coal are given in Appendix F. Plots of combustible recovery versus ash rejection and product rate versus feed rate are given in Figures 2.25 and 2.26. Similar information for the Pittsburgh No. 8 coal is given in Appendix F and Figures 2.27 and 2.28.

Figures 2.25 and 2.27 show that higher pH levels generally give cleaner products, presumably because the dispersion of both coal and clay is greatly enhanced above pH 7.5. This would be expected to lead to better flotation of coal and less entrapment of mineral matter in the froth. The product versus feed rate plots (Figures 2.26 and 2.28) also show that there are no significant differences in the rates of flotation of the coals at the different pH levels. Since both theory and experience indicate that larger coagula should float faster than dispersed particles, it must be concluded that the coagula present in the flotation feed are being destroyed by turbulence in the column. Analysis of the flow rates of water, particles, and bubbles in the pulp/froth transition zone of the typical flotation column gives Reynolds numbers of about 15,000. This is much higher than the Reynolds number range (100-3000) found to be necessary for acceptable coagula growth by researchers at Virginia Tech. Therefore, these analyses indicate that conventional column flotation is an unsuitable process for upgrading hydrophobic coagula.

Table 2.14 - Standard operating conditions for the flotation column pH tests.

Parameter	Value
pH	4, 6, 7.5, 10
pH Condition Time	15 minutes
Chelating Agent	EDTA
Chelate Dosage	0.4 gm/liter
Chelate Condition Time	15 minutes
Feed Particle Size	-20 microns
Feed Rate (ml/min)	100, 150, 225
Feed Solids Content	4 %
Wash Water	200 ml/min
Aeration Rate	1500 cc/min
Frother Dosage	3 - 4 lb/ton

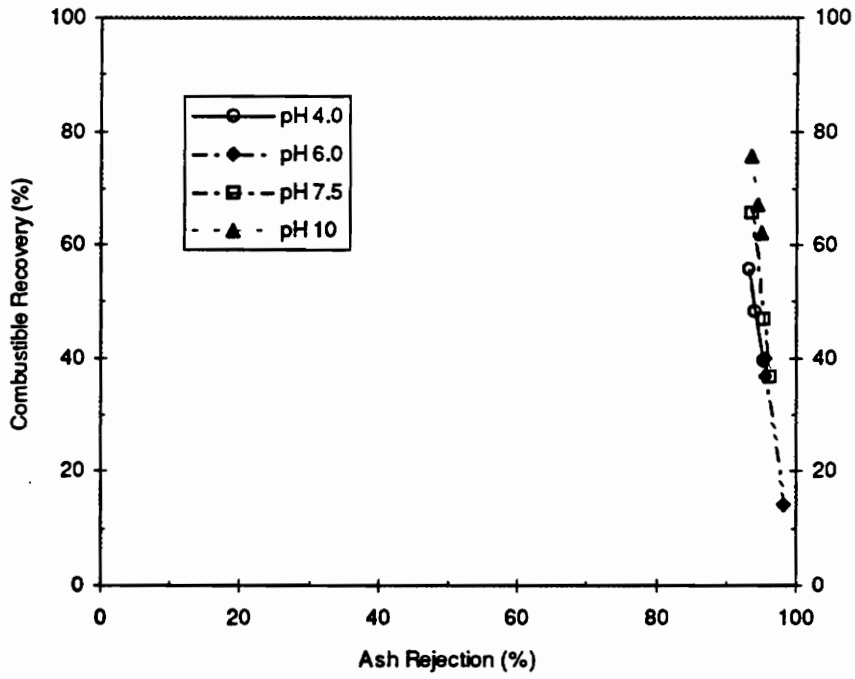


Figure 2.25 Combustible recovery versus ash rejection for the flotation of Elkhorn No. 3 coal at different pH values.

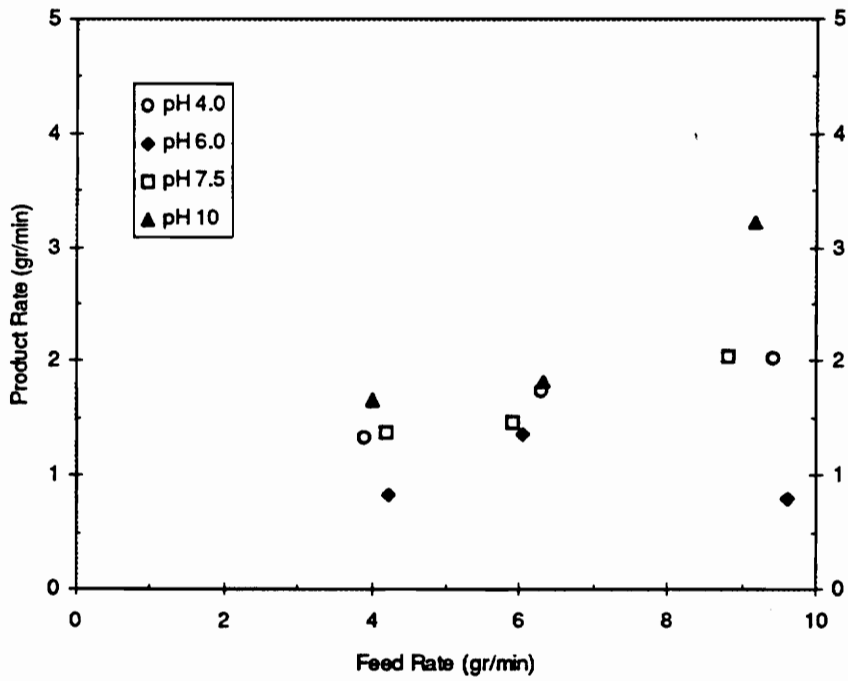


Figure 2.26 Product rate versus feed rate for the flotation of Elkhorn No. 3 coal at different pH values.

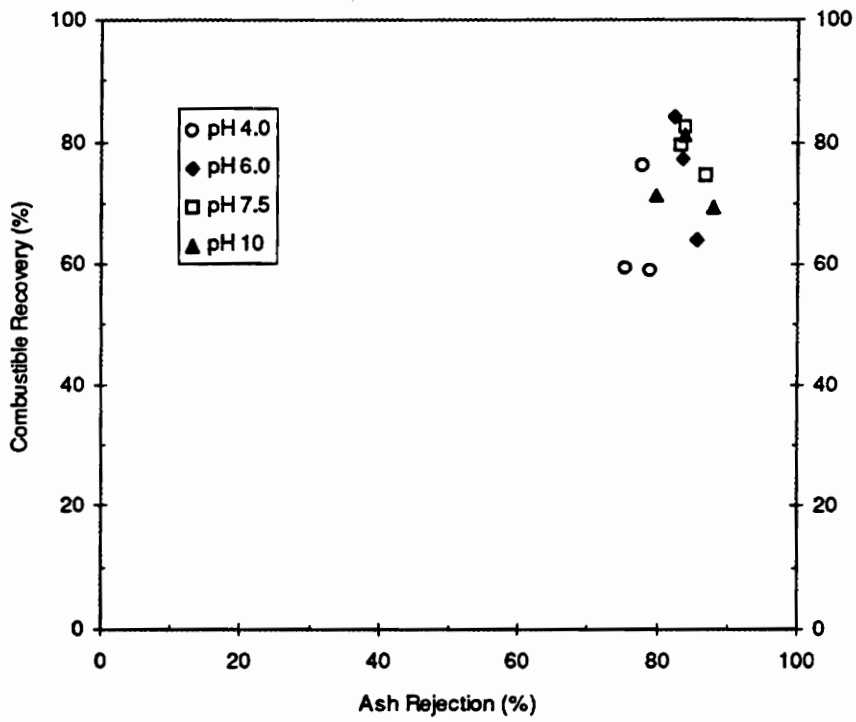


Figure 2.27 Combustible recovery versus ash rejection for the flotation of Pittsburgh No. 8 coal at different pH values.

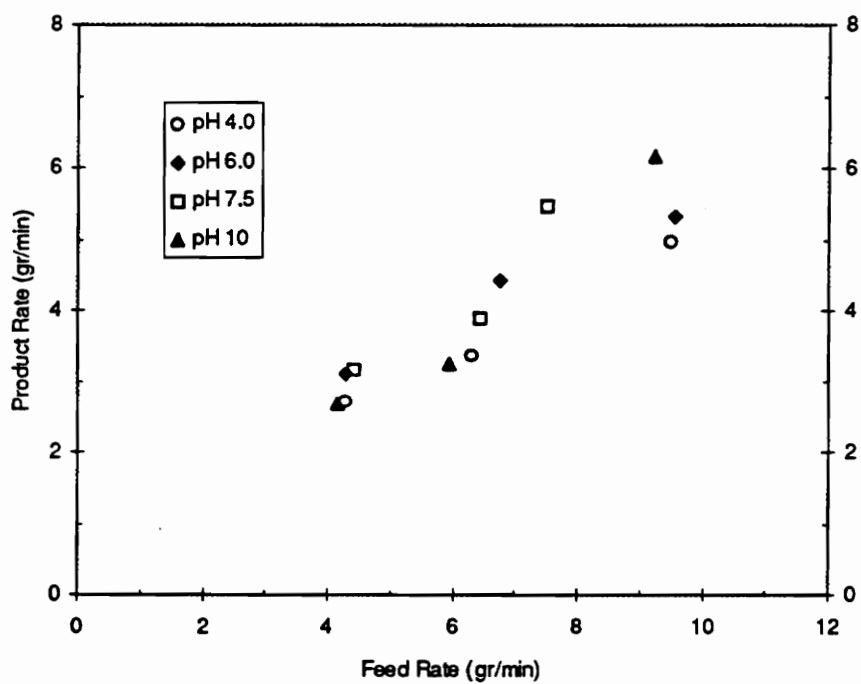


Figure 2.28 Product rate versus feed rate for the flotation of Pittsburgh No. 8 coal at different pH values.

2.9 Drum Screen and Low Oil Agglomeration (LOA)

All of the test work conducted to this point indicate that the coagula formed by the SHC process are too weak to be effectively recovered by centrifugation, flotation and filtration processes. Therefore, the use of low oil agglomeration was examined as a means of increasing coagula strength and size.

a) Experimental

A rotating drum screen (Figure 2.29) was used in the low oil agglomeration tests to separate the coal coagula from dispersed mineral matter. In these tests, micronized feed slurry was adjusted to a pH of 8.5 using hydrochloric acid or sodium hydroxide. After the pH stabilized, kerosene was added to the feed slurry as an emulsion and conditioned with the feed slurry for a minimum of 15 minutes. The feed slurry was then pumped through a static pipe into the interior of an eight-inch diameter drum covered with a 400 mesh nylon screen. Coagulated particles which were too large to pass through the screen were carried to the top of the drum by the slow drum rotation speed (1 rpm). At the top of the drum, wash water adjusted to the same pH (i.e., 8.5) as the feed slurry was used to flush the coal coagula off the screen and into a collection trough. Dispersed particles which flowed through the screen were collected separately as tailings. Variables examined in the test program are shown in Table 2.15 and include percent solids, kerosene dosage and feed rate. Standard operating conditions for the drum screen separation tests are summarized in Table 2.16. All of the drum screen tests were performed using an Elkhorn No. 3 coal having a 22% feed ash.

b) Results and Discussion

The test data obtained using the rotating drum screen and low oil agglomeration technique are shown in Figure 2.30. Test conditions, assay values, and various performance parameters are given in Appendix G. As shown, the tests produced poor results with less than 6% separation efficiencies. Figure 2.31 shows that increasing the

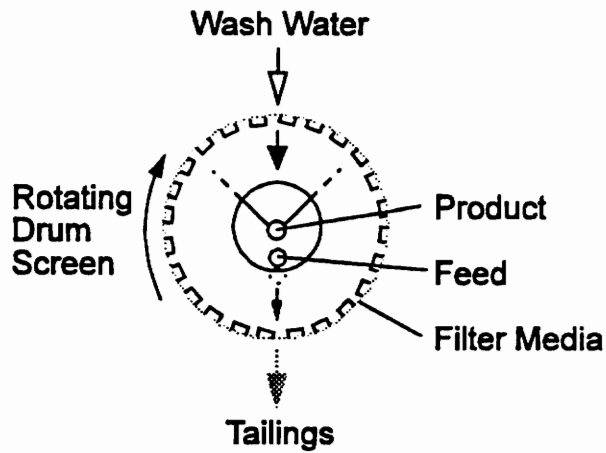
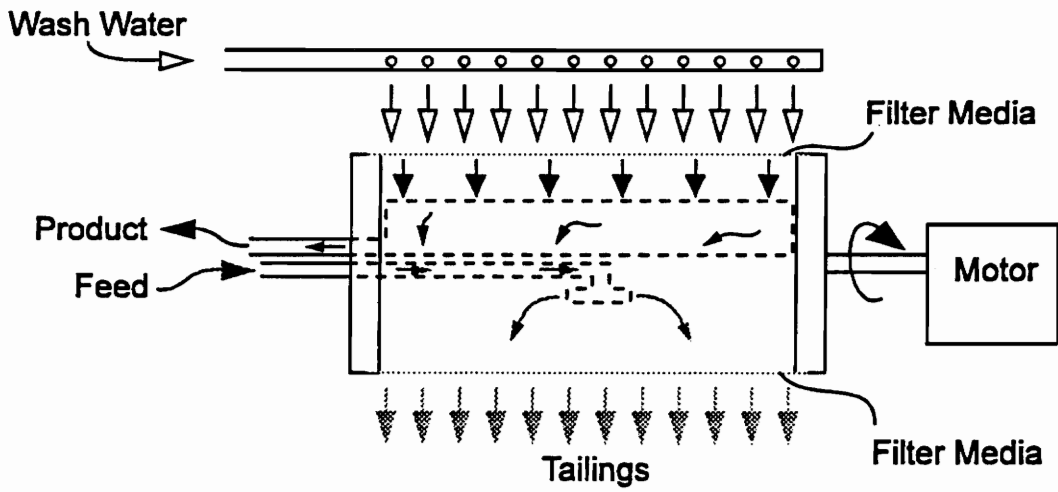


Figure 2.29 Schematic drawing of the rotating drum screen.

Table 2.15 - Operating conditions for the drum screen tests.

Parameter	Values
Feed Rate (ml/min)	40,70,100
Feed Solids Content (%)	3,5,7
Kerosene Dosage (lb/ton coal)	0, 3, 6, 12, 24, 48

Table 2.16 - Standard operating conditions for the drum screen tests.

Parameter	Value
pH	8.5
pH Condition Time	15 minutes
Chelating Agent	none
Kerosene Condition Time	15 minutes
Feed Particle Size	-20 microns
Filter Medium	nylon (400 mesh)
Pulp Depth	below drum
Drum Rotational Speed	0.95 rpm

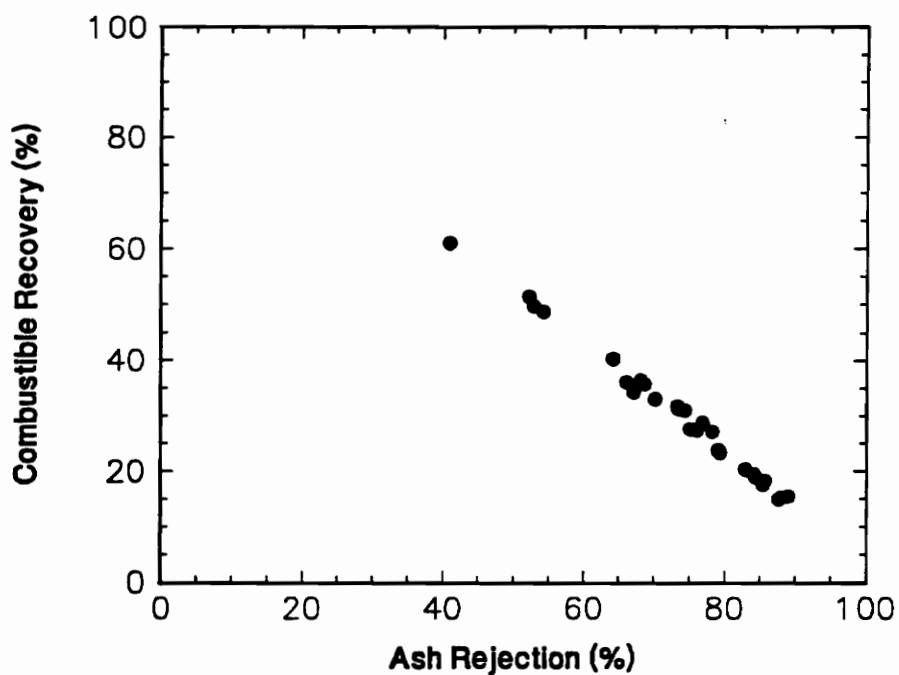


Figure 2.30 Separation results obtained using the drum screen.

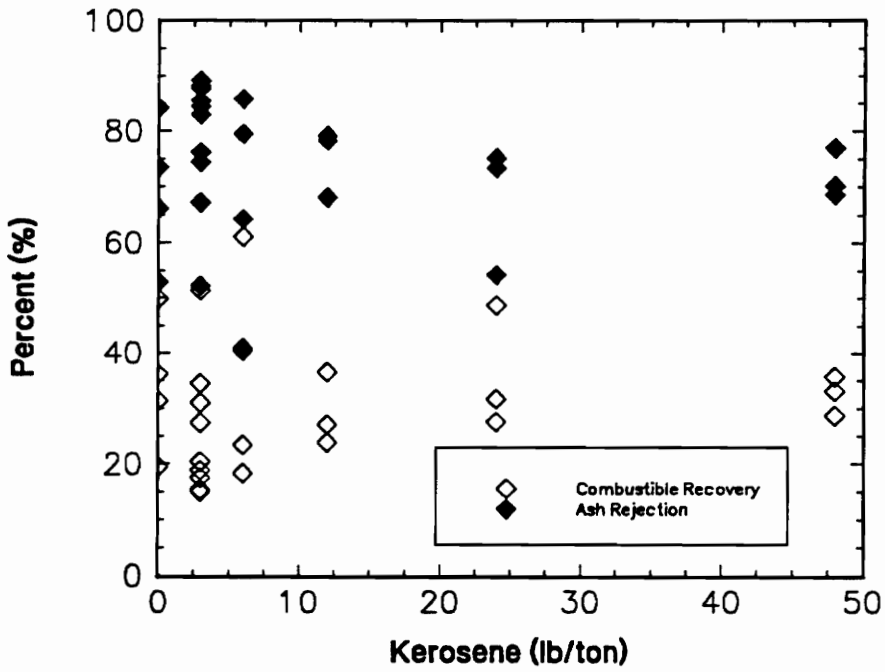


Figure 2.31 Effect of kerosene dosage on the performance of the drum screen.

kerosene dosage improved the coagula recovery. However, the ash rejection decreased at the higher kerosene dosages. This suggests that the higher kerosene dosages increased the coagula size which, in turn, increased coagula recovery and mineral matter entrapment.

2.10 Summary

Figure 2.32 shows the grade-recovery curves obtained using the various separation techniques evaluated in the present work for the upgrading of the Elkhorn No. 3 coal. As shown, the lamella thickener outperformed the other coagula separation techniques based on recovery of combustibles and separation efficiency. The lamella thickener obtained a 51% separation efficiency with a 92% recovery of combustibles.

The performance curves also indicate that conventional column flotation worked comparatively well for both coal seams tested. Unfortunately, the turbulence in the columns was found to disrupt coagula growth, making it an unsuitable separation method for selectively coagulated coal particles.

The use of vacuum filtration to separate coagulated coal particles from dispersed matter proved ineffective. The coagula were either broken by the applied vacuum, passing through the filter medium, or the medium became blinded and retained the feed solids. Either way, the filtration separation efficiencies were less than 4%.

As with filtration, the use of the drum screen and low oil agglomeration also proved ineffective in separating the coal coagula from dispersed mineral matter. The low oil agglomeration process did not prove to be selective enough while increasing coagula size and strength. Separation efficiencies with the drum screen were less than 6%.

The use of batch centrifugal sedimentation on the Elkhorn No. 3 seam coal produced high recovery results with moderate to low ash rejections. The best separation efficiency obtained was 40% with a 95% recovery of combustibles. The use of a continuous unit may provide results with slightly higher separation efficiencies, since the batch tests did not make accurate cuts between sedimented and dispersed particles. The

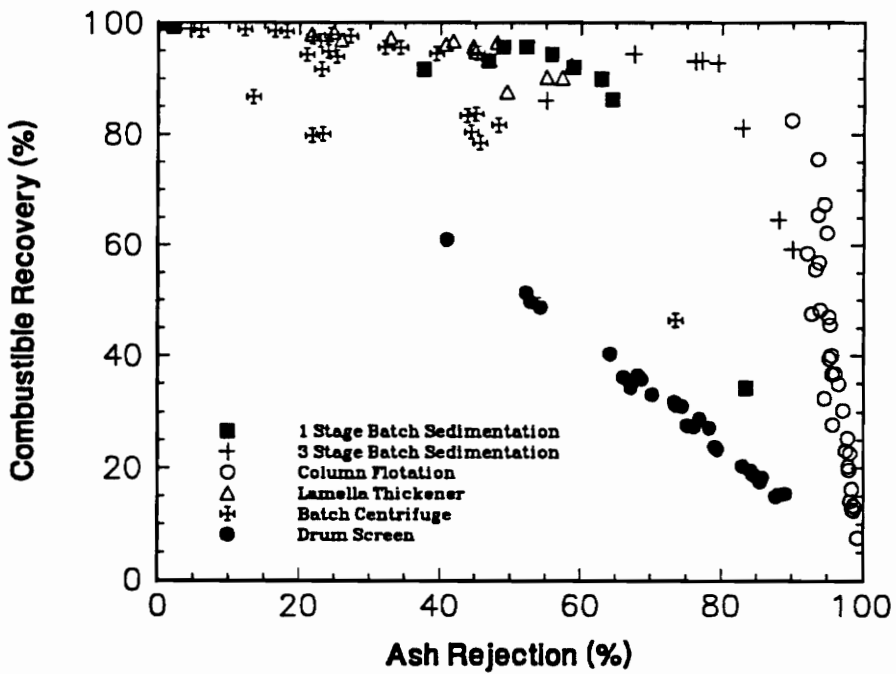


Figure 2.32 Summary of results obtained using the various separation devices for the Elkhorn No. 3 coal.

use of a continuous centrifuge may also reduce retention times and increase throughputs over the other sedimentation techniques examined since low feed solids provided a better separation efficiency.

Aside from froth flotation, the Pittsburgh No. 8 seam coal did not perform well using the other separation methods (Figure 2.33). The coagulation of the Pittsburgh No. 8 seam coal did not prove to be selective enough for upgrading. Sedimentation techniques rejected very little ash at suitable recovery values due to the heterocoagulation of gangue and coal particles. Further testing with dispersants and chelating agents may improve the selective coagulation of this coal, although insufficient resources were available in this project to investigate this possibility.

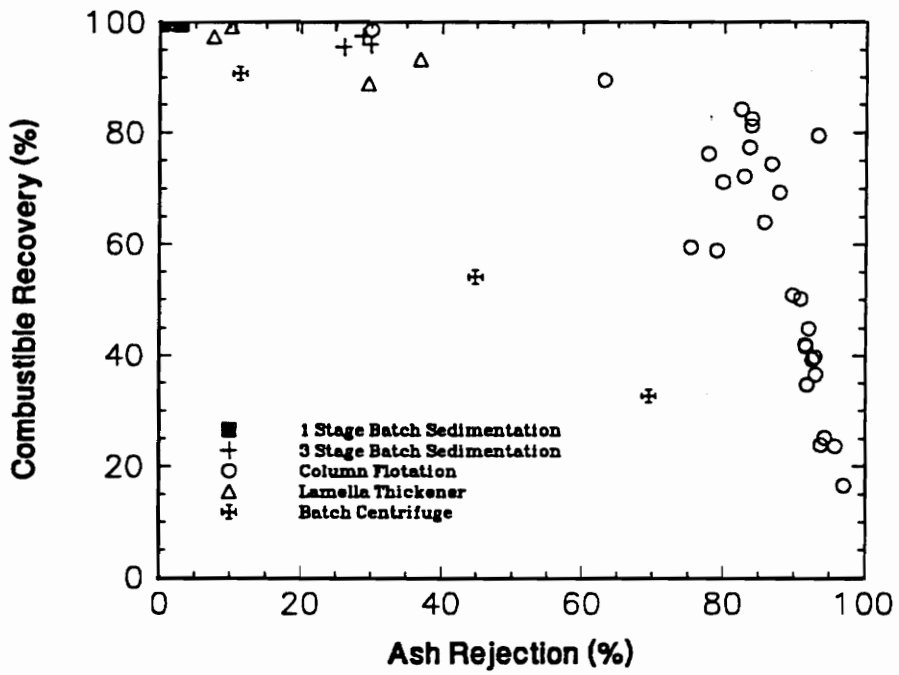


Figure 2.33 Summary of results obtained using the various separation devices for the Pittsburgh No. 8 coal.

Chapter 3 Bench-Scale Process Optimization

3.1 Preliminary Optimization

The objective of this task was to optimize the performance of the best combination of mixer and separator examined in the present work. It was concluded by researchers at Virginia Tech that any level of turbulence created by mixing was detrimental to the formation and subsequent growth of the hydrophobic coagula. However, some degree of mixing was found to be required between separation stages to redisperse mineral matter that may have become entrapped within the coal coagula. A simple mixing tank and impeller was deemed to be the most reliable and least costly agitator for this purpose. In section 2.10, it was concluded that the lamella thickener provided the only viable method of recovering coal coagula of the various separation processes tested. The lamella thickener was able to achieve a 51% separation efficiency with a 92% recovery of combustibles. Therefore, the optimization tests were conducted using multiple-stage cleaning with a simple mixer and lamella thickener separator.

a) Experimental

The first series of optimization tests were conducted using the Elkhorn No. 3 coal which had been wet-ground in a stirred ball mill to a topsize of 20 microns. In these tests, the lamella thickener was operated continuously with three cleaning stages. The standard operating parameters are shown in Table 3.1. The optimum operating conditions were chosen based on the test data obtained in section 2.5. The main criteria for selecting the best operating parameters were high separation efficiencies. In all tests, the wash water rate was set at 40 ml/min since higher wash water rates tended to increase the separation efficiency. EDTA was not utilized in these tests and the pH of the feed slurry was adjusted to 8.5 using hydrochloric acid (HCl) or sodium hydroxide (NaOH). This pH was found to be near the optimum pH for selective coagulation from the batch sedimentation tests reported in section 2.4. The middle feed rate of 100 ml/min was

Table 3.1 - Standard operating conditions for the 3-stage lamella thickener tests.

Parameter	Value
pH	8.5
pH Condition Time	15 minutes
Chelating Agent	none
Feed Particle Size	-20 microns
Feed Rate	100 ml/min
Wash Water Rate	40 ml/min
Mudline	constant
Lamella Plate Angle	45°
Cell Actual Area	42 inch ²
Cell Effective Area	170 inch ²
Cell Volume	11 liters

chosen since higher feed rates decreased recovery and lower feed rates decreased separation efficiency. The feed solids content in the first stage of cleaning was set at 4% solids. Because of dilution, the solids content was reduced to 3% and 2% in stages 2 and 3, respectively. The test data indicated that a higher separation efficiency could have been obtained in the first and second stages using a 2% feed solids content. However, a 4% feed solids content was used in the first stage to increase the feed mass flow rate to 0.5 lb/hr. The results of the multi-stage cleaning tests are summarized in Appendix H.

In the second series of optimization tests, an oil emulsion of kerosene and water was added to the feed slurry to improve the separation efficiency of the lamella thickener. Kerosene dosages of 0, 3, 6, 9, 12, 24 and 48 lbs/ton of coal were examined in these tests. Table 3.2 shows the standard operating conditions for the low-oil agglomeration lamella thickener tests. The lamella thickener was operated at a feed rate of 100 ml/min, a solids content of 4%, and a wash water rate of 40 ml/min. These settings were found to be near the optimum (based on separation efficiency) for the Elkhorn No. 3 coal. The test data obtained during this series of experiments are summarized in Appendix H.

b) Results and Discussion

Figure 3.1 shows the results of the lamella thickener tests conducted using the Elkhorn No. 3 coal. After three stages of cleaning, the 50% ash coal produced a final product of 16.9% ash. The overall combustible recovery and separation efficiency values were 84.1% and 64.2% respectively. A larger drop in combustible recovery was observed with the staged lamella thickener than with the batch sedimentation tests. For comparison, the three-stage batch sedimentation test conducted at 3% solids, pH 8.5 and no EDTA gave a combustible recovery of 93.3% and a separation efficiency of 70.5%. It is believed that the performance of the lamella thickener could have been improved by reducing the feed solids content in the first and second cleaning stages.

Table 3.2 - Standard operating conditions for the LOA lamella thickener tests.

Parameter	Value
pH	8.5
pH Condition Time	15 minutes
Chelating Agent	none
Kerosene Condition Time	30 minutes
Feed Rate	100 ml/min
Feed Solids Content	4%
Wash Water Rate	40 ml/min
Feed Particle Size	-20 microns
Mudline	constant
Lamella Plate Angle	45°
Cell Actual Area	42 inch ²
Cell Effective Area	170 inch ²
Cell Volume	11 liters

Figure 3.2 shows the results obtained during the low-oil agglomeration tests conducted using the lamella thickener. The addition of the kerosene emulsion into the feed slurry increased the ash rejection as much as 18% with only a 2.5% reduction in the combustible recovery. The separation efficiency peaked at 56% with a kerosene addition of 12 lbs/ton of feed coal.

The response surface plots shown in Figure 2.8 for the lamella type thickener show that the combustible recovery increased with increasing feed solids content and decreased with increasing feed rate. The increasing feed solids content increased the pulp density in the thickener which enhanced particle-particle collision and subsequent coagulation. The larger coagula settled faster, thereby increasing the coal recovery. The higher feed rates increased the superficial velocity which forced more coagula to the tailings overflow and decreased the recovery. The highest recovery values (99%) were obtained at high feed solids contents and low feed rates.

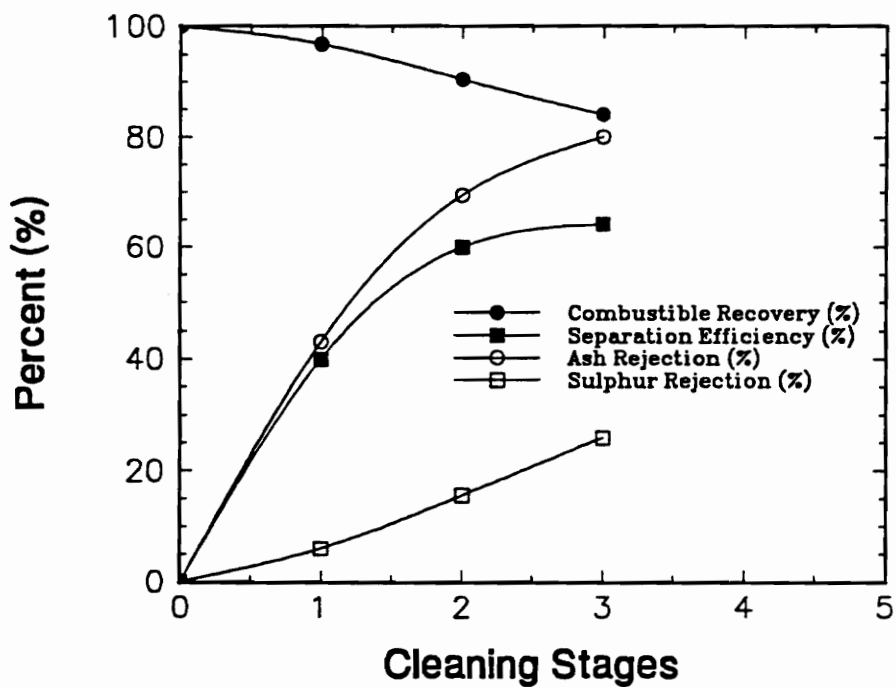


Figure 3.1 Test results obtained after 3 stages of cleaning using the lamella thickener.

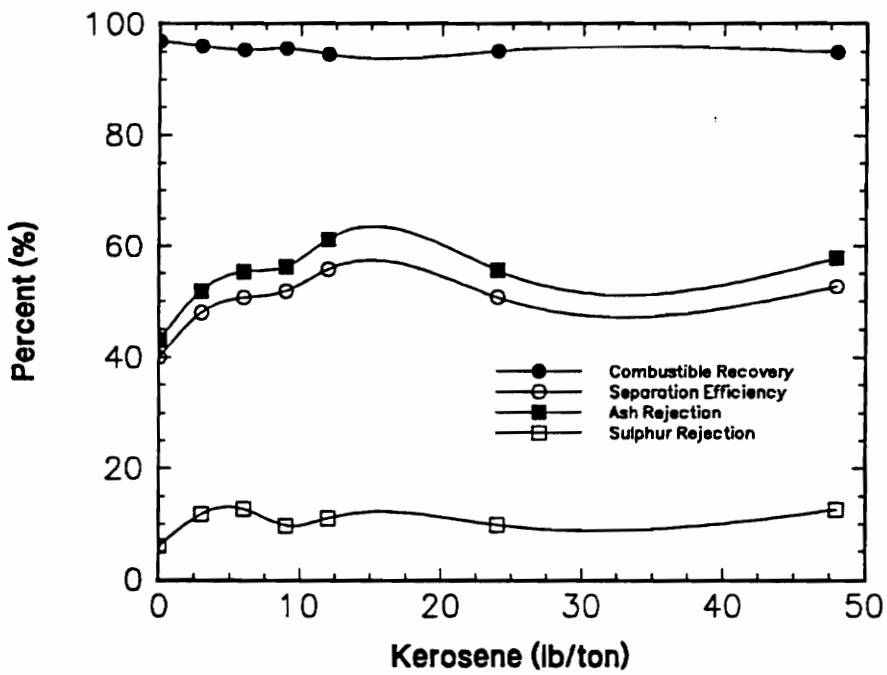


Figure 3.2 Effect of kerosene dosage on the recovery of combustibles, separation efficiency, and ash rejection obtained using the lamella thickener.

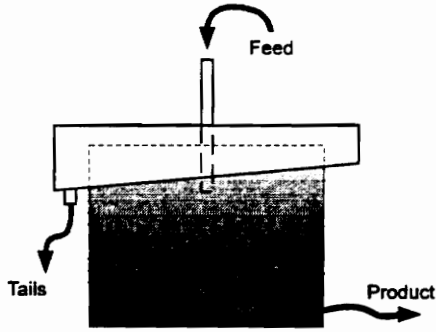
3.2 Comparison of Thickener Designs

The data obtained from the preliminary optimization tests indicate that the lamella thickener is capable of efficiently separating the coal coagula and dispersed mineral matter. However, this data does not indicate whether the lamella thickener design offers any advantages over the “open tank” design of traditional sedimentation processes. Therefore, an additional series of optimization tests was conducted in the present work to compare the performance of a conventional thickener with that of the lamella thickener.

a) Experimental

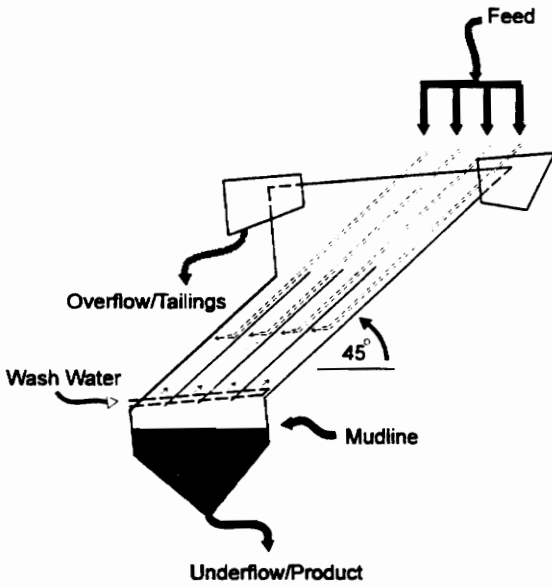
A 27 cm diameter conventional thickener with an effective area of 559 cm² was used in the comparison tests. Figure 3.3 compares the effective areas and volumes of the conventional and lamella thickeners. The two operating parameters studied in this series of tests were feed solids content and volumetric feed rate. Three different settings were examined for each operating parameter. The test ranges for each operating parameter are given in Table 3.3 while the standard operating parameters for these tests are given in Table 3.4. Using these values, a 3-level and 2-variable factorial design was generated for the Elkhorn No. 3 coal using the Design-Expert software package. The experimental design required nine individual tests. Appendix H provides a summary of the comparative test results.

After completing the experiments, regression analyses were used to develop empirical expressions relating the operating conditions to the separation performance. For each model, the values of regression coefficients, standard errors, variances, t-tests, etc., were determined as a means of quantifying the statistical significance of each of the coefficients in the regression model. Response surface plots were then constructed for predicting combustible recovery, ash rejection, and separation efficiency. Each set of data was plotted as a function of feed solids content and feed flow rate in the form of a contour diagrams. The results of the statistical analyses are provided in Appendix H.



Conventional Thickener

Diameter = 27 cm
 Actual Area = Effective Area = 559 cm²
 Volume = 10.5 liters



Lamella Thickener

Effective Area = $n \times a \times \cos(\alpha)$
 n = number of plates
 a = area of one plate
 α = angle of plates

Effective Area = $4 \times 387 \times \cos(45) = 1095 \text{ cm}^2$

Actual Area = 271 cm²
 Volume = 11 liters

Figure 3.3 Schematic of the conventional and lamella thickener test units.

Table 3.3 - Experimental levels for the conventional thickener parametric tests.

Coal Seam	Parameter	Units	Parameter Range		
			Low (-1)	Normal (0)	High (1)
Elkhorn No. 3 (50% Ash)	Feed Solids	(%)	2	3	4
	Feed Rate	(ml/min)	50	100	150

Table 3.4 - Standard operating conditions for the conventional thickener tests.

Parameter	Value
pH	8.5
pH Condition Time	15 minutes
Chelating Agent	none
Wash Water	none
Feed Point Depth	1.5 inches
Feed Particle Size	-20 microns
Mudline	constant
Cell Actual Area	87 inch ²
Cell Volume	10.5 liters
Cell Diameter	10.5 inches

b) Results and Discussion

The statistical correlations between the model predictions and the experimental results were found to be satisfactory with the exception of one test run. Test number 3 of the conventional thickener tests (Appendix H) was omitted as an outlier in the statistical analyses. Based on these statistical correlations, a correlation matrix (Table 3.5) was constructed to summarize the effects of the operating parameters on the separation performance. The corresponding response surface plots are shown in Figure 3.4 for combustible recovery, ash rejection, and separation efficiency. The trends observed in these plots are discussed in the following sections.

As with the lamella type thickener, increasing the feed rate also increased the separation efficiency for the conventional thickener design. However, the effect of increasing feed rate was greater for the conventional thickener than for the lamella thickener. The higher feed rates increased the superficial velocity which improved the removal of fine mineral matter. Unlike the lamella type thickener results, an increase in the feed solids content slightly increased the separation efficiency. This result was unexpected since all previous coagulation data showed a decreased rejection with increasing solids content. The decrease in rejection was attributed to an increase in the entrapment of mineral matter in the coagula at the higher solids contents. This reverse result for the conventional thickener could be due to the smaller range of feed solids examined and the effect of the omission of the outlier test #3. The maximum separation efficiency (30%) occurred at the moderate feed solids contents and feed rates.

The response surface plots indicate that the ash rejection was most affected by increasing the feed rate. The higher feed rates increased the superficial velocity and provided an increase in the rejection of mineral matter. The rejection of ash increased slightly with increasing feed solids content. This finding was unexpected since previous selective coagulation tests indicated a decrease in ash rejection with increasing feed solids content. As noted in the above paragraph, this unexpected result could be due to

Table 3.5 - Correlation matrix for the conventional thickener.

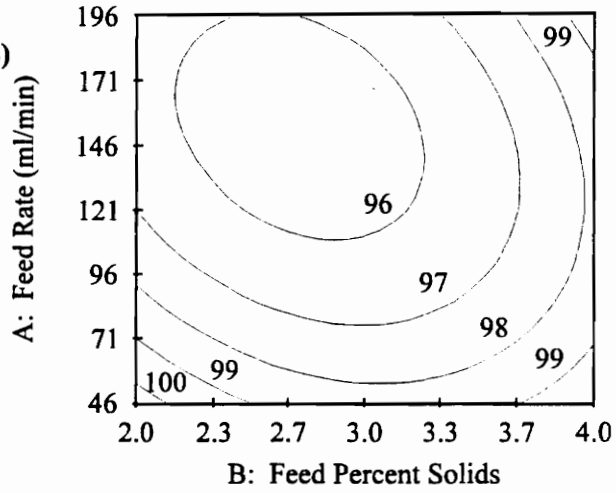
Process Response	Effect of Increasing:	
	Feed Solids	Feed Rate
Combustible Recovery	S↑	S↓
Ash Rejection	S↑	M↑
Separation Efficiency	S↑	M↑

L = Large, M = Medium, S = Small

↑ = Increasing Response, ↓ = Decreasing Response

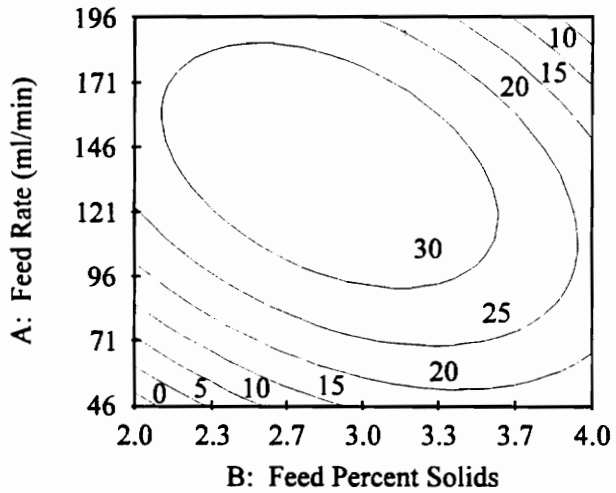
Response:
Combustible Recovery (%)

Actual variables:
X = Feed Percent Solids
Y = Feed Rate



Response:
Separation Efficiency (%)

Actual variables:
X = Feed Percent Solids
Y = Feed Rate



Response:
Ash Rejection (%)

Actual variables:
X = Feed Percent Solids
Y = Feed Rate

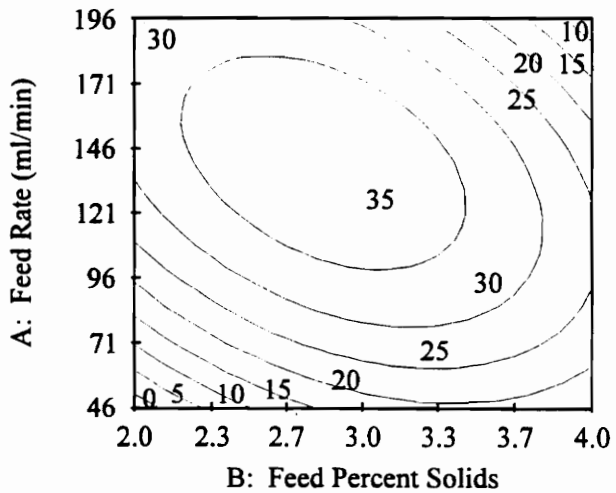


Figure 3.4 Response surface plots obtained from the statistical analysis of the conventional thickener test data.

the lack-of-fit of the regression model created by the omission of test run #3. The highest ash rejection (35%) was seen at moderate feed solids contents and feed rates.

3.3 Summary

Figure 3.5 shows the grade-recovery curves for the various thickener separation methods examined using the Elkhorn No. 3 seam coal. While all of the methods maintained high recoveries, the lamella thickener with low-oil agglomeration produced the best separation efficiencies with one stage of cleaning. The use of the lamella thickener without low-oil agglomeration had a slight drop in combustible recovery at comparable ash rejection values. The conventional thickener unit could not achieve ash rejections as high as the lamella thickener. The best single-stage thickener test employed low-oil agglomeration and produced a 95% recovery of combustibles at a 56% separation efficiency. The three-stage lamella thickener test without low-oil agglomeration achieved an overall recovery of 84% at a 64% separation efficiency.

The throughput for the lamella type thickener and the conventional thickener were compared using the results obtained from the lamella thickener, low-oil agglomeration thickener, and conventional thickener tests. Table 3.6 shows the tests with the maximum separation efficiencies from these experiments. Based on the calculated effective areas of the test units, the conventional thickener had the highest throughput of 10.4 lb/hr/m² with a separation efficiency of 34%. The lamella thickener provided only half the conventional thickener's effective area throughput with a maximum separation efficiency of 56%. However, when the comparison was based on the actual area of the test units, the lamella thickener had nearly double the throughput at 19.6 lb/hr/m² with separation efficiencies greater than the conventional thickener. Both units operated at about the same level when the throughput was compared based on the unit volume of the thickener, i.e., the conventional and the lamella thickener had throughputs of 0.055 and 0.048 lb/hr/liter respectively.

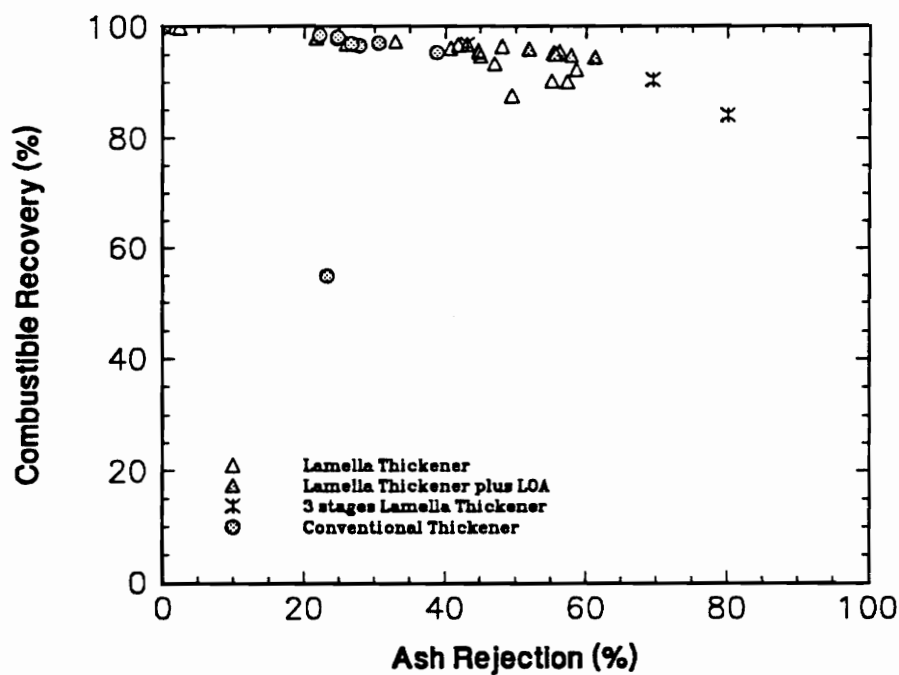


Figure 3.5 Summary of results obtained using the lamella and conventional thickener test units.

Table 3.6 - Comparison of the lamella and conventional thickeners.

	Thickener Design			
	Conventional	Lamella	Lamella	Lamella
<u>Test Conditions:</u>				
Feed Solids Content (%)	3.0	2.0	4.0	4.0
Feed Rate (ml/min)	150	50	100	100
Feed Rate (lb/hr)	0.58	0.13	0.53	0.53
Kerosene (lb/ton)	0.0	0.0	0.0	12.0
<u>Capacity Parameters:</u>				
Effective Area (lb/hr/m ²)	10.4	1.19	4.84	4.84
Actual Area (lb/hr/m ²)	10.4	4.80	19.6	19.6
Volume (lb/hr/liter)	0.055	0.012	0.048	0.048
<u>Separation Performance:</u>				
Combustible Recovery (%)	95	92	97	95
Separation Efficiency (%)	34	50	40	56
Ash Rejection (%)	39	60	43	61

3.4 Conclusions

The previous tests conducted at Virginia Tech employing the *Selective Hydrophobic Coagulation* (SHC) process to selectively coagulate and recover finely pulverized coal from dispersed mineral matter showed that the hydrophobic coagula were difficult to separate from dispersed mineral matter due to their small size and their susceptibility to breakage. In the present work, a variety of laboratory bench-scale tests were undertaken to determine the most effective methodology to recover the coal coagula. Several devices were evaluated in this study including lamella and conventional thickeners, a rotary centrifuge, a vacuum drum filter, a froth flotation column, and a rotating drum screen. Each unit was tested under optimum reagent conditions established from batch experiments. The test data indicate that the thickener was the best overall method for separating hydrophobic coagula from dispersed mineral matter based on overall process performance (e.g., recovery and grade), unit capacity, and engineering feasibility.

REFERENCES

- Allen, R.W. and Wheelock, T.D., 1993. "Effects of pH and Ionic Strength on Kinetics of Oil Agglomeration of Fine Coal," *Minerals Engineering*, Vol. 6, No. 1, pp. 87-97.
- Attia, Y.A., 1985. "Cleaning and Desulfurization of Coal Suspensions by Selective Flocculation," Processing and Utilization of High Sulfur Coals (Y.A. Attia, ed.), Elsevier Science Publishers B.V., Amsterdam, pp. 267-285.
- Attia, Y.A., Elzeky, M., Bavarian, F., Yu, S., 1990. "Ultrafine Coal Cleaning by Selective Flocculation and Froth Flotation," SME Annual Meeting, Salt Lake City, Utah, February - March, 1990.
- Capes, C.E. and Germain, R.J., 1982. "Selective Oil Agglomeration in Fine Coal Beneficiation," Physical Cleaning of Coal: Present and Developing Methods, (Y.A. Liu, ed.), Marcel Dekker, New York, pp 293-346.
- Capes, C.E., Coleman, R.D., Hazlett, J.D., Jonasson, K.A., and Thayer, W.L., 1988. "The recovery and utilization of fine coal," *CIM Bulletin*, Vol. 81, No. 918, October 1988, pp. 46-50.
- Chironis, N.P., ed., 1987. "Advanced Techniques Clean Ultrafine Coal at Homer City," *Coal Age*, December 1987, pp. 68-69.
- Fuerstenau, D.W., 1980. "Fine Particle Flotation," Fine Particle Processing, (P. Somasundaran, ed.), AIME, Inc., New York.
- Honaker, R.Q., 1988. Development of the Selective-Shear Coagulation Process for Ultrafine Coal Cleaning, M.S. Thesis, Department of Mining and Minerals Engineering, Virginia Polytechnic Institute and State University.
- Honaker, R.Q., 1992. A Fundamental Study of the Selective Hydrophobic Coagulation Process, Ph.D. Dissertation, Department of Mining and Minerals Engineering, Virginia Polytechnic Institute and State University.
- Killmeyer, R.P., 1985. "Selective agglomeration: Let's compare the emerging processes," *Coal Mining*, September 1985, pp. 45-49.
- Klemm, Robert, Whitt, Richard, and Kalb, G. William, 1990. "Pyrite Liberation Study Initial Development of Super-Centrifuge Continuous Flow Centrifugal Float/Sink Procedure," January 5.

- Kneller, W.A. and Maxwell, G.P., 1985. "Size, Shape, and Distribution of Microscopic Pyrite in Selected Ohio Coals," Processing and Utilization of High Sulfur Coals, (Y.A. Attia, ed.), Elsevier, Amsterdam, pp. 38-51.
- McCartney, J.T., O'Donnell, H.J., and Ergun, S., 1969. "Pyrite Size Distribution and Coal-Pyrite Particle Association in Steam Coals," USBM, R. I. 7231.
- Pugh, R.J. and Kitchener, J.A., 1971. "Theory of Selective Coagulation in Mixed Colloidal Suspensions," *Journal of Colloid and Interface Science*, Vol. 35, No. 4, pp. 656-664.
- Rao, T.C., 1988. "Recent advances in fine coal beneficiation," *Minerals and Metallurgical Processing*, August 1988, pp. 124-132.
- Ray, D.T., Hogg, R., 1987. "Agglomerate Breakage in Polymer-Flocculated Suspensions," *Journal of Colloid and Interface Science*, Vol. 116, No. 1, March, 1987, pp. 256-268.
- Robbins, G.A., Winschel, R.A., Amos, C.L., and Burke, F.P., 1992. "Agglomeration of low-rank coal as a pretreatment for direct coal liquefaction," *FUEL*, Vo. 71, September, pp. 1039-1046.
- Slaghuis, J.H. and Ferreira, L.C., 1987. "Selective spherical agglomeration of coal - An amended mechanism of agglomerate formation and growth and its effect on product quality," *FUEL*, Vol. 66, October, pp. 1427-1430.
- Venkatadri, R., Markuszewski, R., and Wheelock, T.D., 1988. "Oil Agglomeration of Weakly Hydrophobic Coals and Coal/Pyrite Mixtures," *Energy & Fuels*, Vol. 2, No. 2, pp. 145-150.
- Xu, Z., 1990. A Study of the Hydrophobic Interaction in Fine Particle Coagulation, Ph. D. Dissertation, Department of Materials Engineering Science, Virginia Polytechnic Institute and State University.
- Yarar, B., 1979. "Selective Flocculation as a Mineral Beneficiation Process," The Second World Filtration Congress, London, September, 1979, pp. 317-331.
- Yoon, R.H. and Luttrell, G.H., 1992. "Development of the Selective Coagulation Process - Final Report," DOE Contract No. DE-AC22-90PC90174.

APPENDIX A - ELZONE SIZE ANALYSIS DATA

**Wet Screen Size Data
Stirred Ball Mill Feed**

Pittsburgh No. 8 seam coal

<u>Size (microns)</u>	<u>Pass (%)</u>
150	83.31
106	67.65
75	55.03
53	42.63
38	31.67

Elkhorn No. 3 seam coal

<u>Size (microns)</u>	<u>Pass (%)</u>
150	90.92
106	79.69
75	69.62
53	57.74
38	48.01

**Elzone 80-XY Particle Size Data
Stirred Ball Mill Product
Elkhorn No. 3 seam coal**

Size (microns)	Pass (%)	Size (microns)	Pass (%)	Size (microns)	Pass (%)	Size (microns)	Pass (%)
0.660	0.000	1.500	21.114	3.770	73.817	8.460	93.409
0.670	0.532	1.530	21.916	3.850	75.022	8.640	93.537
0.680	2.051	1.560	22.802	3.930	76.075	8.830	93.811
0.700	2.932	1.600	23.634	4.020	77.014	9.020	93.957
0.710	3.634	1.630	24.539	4.110	77.759	9.210	93.957
0.730	4.205	1.670	25.538	4.190	78.598	9.410	94.289
0.750	4.669	1.710	26.505	4.280	79.612	9.610	94.289
0.760	5.131	1.740	27.593	4.380	80.277	9.820	94.665
0.780	5.536	1.780	28.564	4.470	80.972	10.030	94.866
0.800	5.910	1.820	29.707	4.570	81.503	10.240	95.294
0.810	6.290	1.860	30.958	4.660	82.171	10.460	95.522
0.830	6.626	1.900	32.454	4.760	82.922	10.690	95.764
0.850	6.973	1.940	33.844	4.870	83.631	10.920	95.764
0.870	7.319	1.990	34.983	4.970	84.415	11.150	96.041
0.890	7.671	2.030	36.358	5.080	85.092	11.390	96.041
0.910	8.026	2.080	37.761	5.190	85.676	11.640	96.041
0.930	8.404	2.120	38.896	5.300	86.121	11.890	96.041
0.950	8.786	2.170	40.336	5.410	86.500	12.140	96.041
0.970	9.189	2.220	41.677	5.530	87.003	12.410	96.420
0.990	9.587	2.270	42.985	5.650	87.542	12.670	96.420
1.010	10.058	2.320	44.733	5.770	87.924	12.940	96.420
1.030	10.539	2.370	46.143	5.890	88.169	13.220	96.420
1.060	11.032	2.420	47.846	6.020	88.559	13.510	96.420
1.080	11.540	2.470	49.446	6.150	89.021	13.800	96.420
1.100	12.101	2.530	50.697	6.280	89.515	14.090	96.420
1.130	12.601	2.580	52.042	6.410	89.777	14.400	97.013
1.150	13.161	2.640	53.354	6.550	90.280	14.710	97.013
1.180	13.692	2.700	54.515	6.690	90.519	15.020	97.013
1.200	14.248	2.760	55.953	6.840	90.582	15.350	97.732
1.230	14.851	2.820	56.931	6.980	90.718	15.680	97.732
1.260	15.438	2.880	58.495	7.130	91.223	16.010	98.550
1.290	16.039	2.940	59.923	7.290	91.300	16.360	98.550
1.310	16.752	3.250	61.527	7.440	91.546	16.710	98.550
1.340	17.432	3.320	62.561	7.600	91.721	17.070	98.550
1.370	18.127	3.390	66.271	7.770	91.814	17.430	98.550
1.400	18.883	3.460	68.171	7.930	92.610	17.810	98.550
1.430	19.535	3.540	69.463	8.110	92.716	18.190	98.550
1.470	20.365	3.610	70.865	8.280	93.168	18.580	98.550
		3.690	72.431			18.980	98.550

**Elzone 80-XY Particle Size Data
Stirred Ball Mill Product
Pittsburgh No. 8 seam coal**

Size (microns)	Pass (%)	Size (microns)	Pass (%)	Size (microns)	Pass (%)	Size (microns)	Pass (%)
0.660	0.000	1.530	3.655	3.850	37.863	8.830	80.174
0.670	0.033	1.560	3.916	3.930	39.326	9.020	80.713
0.680	0.139	1.600	4.196	4.020	40.942	9.210	80.856
0.700	0.193	1.630	4.472	4.110	42.455	9.410	81.315
0.710	0.241	1.670	4.759	4.190	43.936	9.610	81.967
0.730	0.283	1.710	5.111	4.280	45.506	9.820	82.662
0.750	0.323	1.740	5.507	4.380	47.055	10.030	82.662
0.760	0.356	1.780	5.877	4.470	48.749	10.240	83.452
0.780	0.387	1.820	6.312	4.570	50.393	10.460	84.083
0.800	0.421	1.860	6.689	4.660	51.962	10.690	84.755
0.810	0.459	1.900	7.172	4.760	53.507	10.920	84.994
0.830	0.495	1.940	7.610	4.870	55.195	11.150	84.994
0.850	0.530	1.990	8.176	4.970	56.122	11.390	85.536
0.870	0.563	2.030	8.831	5.080	57.227	11.640	85.826
0.890	0.605	2.080	9.368	5.190	58.485	11.890	86.443
0.910	0.647	2.120	9.965	5.300	59.825	12.140	87.430
0.930	0.692	2.170	10.687	5.410	60.991	12.410	87.430
0.950	0.738	2.220	11.358	5.530	62.261	12.670	88.178
0.970	0.792	2.270	12.084	5.650	63.288	12.940	88.178
0.990	0.850	2.320	12.778	5.770	64.594	13.220	88.178
1.010	0.907	2.370	13.629	5.890	65.985	13.510	88.630
1.030	0.980	2.420	14.442	6.020	67.225	13.800	88.630
1.060	1.041	2.470	15.338	6.150	68.079	14.090	88.630
1.080	1.126	2.530	16.101	6.280	69.535	14.400	90.273
1.100	1.214	2.580	17.023	6.410	70.553	14.710	92.026
1.130	1.308	2.640	18.048	6.550	71.327	15.020	92.026
1.150	1.417	2.700	19.037	6.690	71.987	15.350	92.689
1.180	1.531	2.760	20.096	6.840	72.690	15.680	92.689
1.200	1.635	2.820	21.175	6.980	73.316	16.010	92.689
1.230	1.749	2.880	22.295	7.130	74.048	16.360	92.689
1.260	1.879	2.940	23.354	7.290	74.474	16.710	93.546
1.290	2.030	3.250	24.383	7.440	75.005	17.070	93.546
1.310	2.191	3.320	25.579	7.600	75.972	17.430	93.546
1.340	2.350	3.390	29.024	7.770	76.573	17.810	93.546
1.370	2.519	3.460	30.626	7.930	77.033	18.190	93.546
1.400	2.720	3.540	32.142	8.110	77.716	18.580	94.724
1.430	2.937	3.610	33.401	8.280	78.238	18.980	97.235
1.470	3.176	3.690	35.095	8.460	78.682	19.390	98.573
1.500	3.396	3.770	36.521	8.640	79.038		

APPENDIX B - BATCH SEDIMENTATION DATA

Batch sedimentation SHC tests using Elkhorn No. 3 seam coal

Test Number	pH	EDTA (g/l)	Stage No.	% Solids			Yield (%)	Comb. Rec. (%)	Rejection		Separation Efficiency (%)
				Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulphur (%)	
1	4.0	0.4	1	3.05	3.05	0.00	100.00	100.00	0.00	0.00	0.00
2	5.0	0.4	1	3.05	3.05	0.00	100.00	100.00	0.00	0.00	0.00
3	6.0	0.4	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
4	6.5	0.4	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
5	6.6	0.4	1	2.93	2.93	0.00	100.00	100.00	0.00	0.00	0.00
6	7.0	0.4	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
7	7.5	0.4	1	2.77	7.56	0.91	72.74	93.22	46.87	13.85	40.08
8	7.5	0.4	1	2.98	8.09	1.13	73.02	96.08	49.52	11.49	45.60
9	7.6	0.4	1	2.97	7.94	1.12	72.61	96.31	50.37	11.49	46.69
10	8.0	0.4	1	2.78	7.35	1.09	71.39	95.76	52.23	12.97	47.99
11	9.0	0.4	1	2.77	8.29	1.30	63.58	89.96	62.91	16.78	52.87
12	10.0	0.4	1	2.78	0.00	2.78	0.00	0.00	100.00	100.00	0.00
13	7.5	0.4	1-3	2.86	5.23	0.54	65.35	94.40	67.55	16.78	61.95
14	8.5	0.4	1-3	2.90	5.54	0.57	57.85	92.76	79.44	19.23	72.21
15	9.6	0.4	1-3	2.89	5.49	0.82	35.86	59.26	90.08	47.80	49.34
16	7.6	0.4	1-2	2.87	6.00	0.74	62.60	91.53	64.55	19.34	56.07
16-1	7.5	0.4	1	2.87	7.19	1.01	73.42	94.11	45.90	11.54	40.02
16-2	7.7	0	2	2.23	6.00	0.46	85.17	97.44	34.61	5.49	32.05
17	7.6	0.4	1-3	2.92	5.41	0.58	57.70	93.06	76.22	19.05	69.28
17-1	7.6	0.4	1	2.92	7.15	1.09	71.89	95.85	51.02	11.66	46.87
17-2	7.7	0	2	2.10	6.13	0.41	86.33	98.22	35.94	5.49	34.16
17-3	7.7	0	3	1.88	5.41	0.19	92.94	98.90	24.21	2.94	23.11
18	7.5	0.4	1-5	2.90	4.73	0.39	54.75	91.70	81.95	21.37	73.65
18-1	7.5	0.4	1	2.90	7.76	1.04	74.68	96.28	46.98	10.42	43.25
18-2	7.5	0	2	2.20	6.22	0.42	86.50	97.97	34.39	5.77	32.36
18-3-5	7.5	0	3-5	0.80	4.73	0.14	85.12	97.35	47.95	7.80	45.30
1	7.0	0.2	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
2	7.5	0.2	1	2.77	7.15	0.93	77.04	91.76	37.70	12.82	29.46
3	8.0	0.2	1	2.82	7.44	1.13	67.92	94.37	55.82	13.95	50.19
4	9.0	0.2	1	2.79	8.18	1.42	61.22	86.32	64.45	19.28	50.77
5	10.0	0.2	1	2.80	0.00	2.80	0.00	0.00	100.00	100.00	0.00
6	7.5	0.2	1-3	2.86	5.07	0.58	67.08	86.11	55.04	20.72	41.15
7	9.5	0.2	1-3	2.89	5.18	0.79	40.02	65.67	88.01	44.56	53.68
1	4.0	0	1	3.05	3.05	0.00	100.00	100.00	0.00	0.00	0.00
2	5.0	0	1	3.05	3.05	0.00	100.00	100.00	0.00	0.00	0.00
3	6.0	0	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
4	7.0	0	1	2.82	2.82	0.00	100.00	100.00	0.00	0.00	0.00
5	7.5	0	1	2.77	6.22	0.08	98.68	99.37	2.01	4.63	1.38
6	8.3	0	1	2.78	6.99	1.04	72.86	95.74	48.93	11.23	44.67
7	9.0	0	1	2.79	7.71	1.25	66.53	92.01	58.93	15.87	50.94
8	9.0	0	1	3.05	8.34	1.41	65.82	92.91	59.44	16.30	52.35
9	9.1	0	1	2.98	7.76	1.37	66.65	92.35	58.21	13.99	50.56
10	9.0	0	1	2.98	7.22	1.41	72.72	94.08	50.03	12.31	44.11
11	10.0	0	1	2.78	9.06	2.37	25.34	34.31	83.49	65.41	17.80
12	10.5	0	1	2.93	0.00	2.93	0.00	0.00	100.00	100.00	0.00
13	7.5	0	1-3	2.87	3.32	0.08	97.35	99.09	4.54	18.71	3.63
14	8.5	0	1-3	2.90	5.55	0.58	59.33	93.31	77.15	18.46	70.46
15	9.5	0	1-3	2.89	5.31	0.71	50.63	81.15	82.95	30.77	64.10

Batch sedimentation SHC tests using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulphur (adjusted)		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	51.96	51.96	0.00	48.04	48.04	100.00	0.36	0.36	0.00
2	51.96	51.96	0.00	48.04	48.04	100.00	0.36	0.36	0.00
3	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
4	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
5	47.00	47.00	0.00	53.00	53.00	100.00	0.40	0.40	0.00
6	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
7	51.09	37.32	87.83	48.91	62.68	12.17	0.39	0.48	0.18
8	50.56	34.95	92.81	49.44	65.05	7.19	0.36	0.44	0.15
9	50.76	34.69	93.37	49.24	65.31	6.63	0.36	0.44	0.15
10	50.78	33.98	92.70	49.22	66.02	7.30	0.37	0.46	0.16
11	49.90	29.11	86.19	50.10	70.89	13.81	0.39	0.51	0.18
12	50.84	0.00	50.84	49.16	100.00	49.16	0.38	0.00	0.38
13	46.90	23.29	91.42	53.10	76.71	8.58	0.37	0.47	0.18
14	48.35	17.18	91.13	51.65	82.82	8.87	0.39	0.54	0.18
15	47.43	13.12	66.61	52.57	86.88	33.39	0.39	0.57	0.29
16	51.58	29.21	89.03	48.42	70.79	10.97	0.39	0.51	0.19
16-1	51.72	38.11	89.31	48.28	61.89	10.69	0.39	0.46	0.18
16-2	38.28	29.39	89.33	61.72	70.61	10.67	0.47	0.51	0.20
17	51.03	21.03	91.96	48.97	78.97	8.04	0.36	0.50	0.16
17-1	51.12	34.83	92.79	48.88	65.17	7.21	0.36	0.44	0.15
17-2	34.82	25.84	91.52	65.18	74.16	8.48	0.44	0.48	0.17
17-3	25.80	21.04	88.42	74.20	78.96	11.58	0.48	0.50	0.20
18	50.17	16.54	90.86	49.83	83.46	9.14	0.36	0.52	0.17
18-1	49.94	35.46	92.64	50.06	64.54	7.36	0.36	0.43	0.15
18-2	35.44	26.88	90.30	64.56	73.12	9.70	0.43	0.47	0.18
18-3-5	27.00	16.51	87.02	73.00	83.49	12.98	0.47	0.52	0.22
1	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
2	49.97	40.41	82.05	50.03	59.59	17.95	0.39	0.44	0.22
3	52.70	34.28	91.70	47.30	65.72	8.30	0.37	0.47	0.16
4	49.43	28.70	82.16	50.57	71.30	17.84	0.39	0.50	0.21
5	51.07	0.00	51.07	48.93	100.00	48.93	0.36	0.00	0.36
6	46.26	31.01	77.33	53.74	68.99	22.67	0.37	0.44	0.23
7	47.78	14.31	70.11	52.22	85.69	29.89	0.39	0.57	0.28
1	51.96	51.96	0.00	48.04	48.04	100.00	0.36	0.36	0.00
2	51.96	51.96	0.00	48.04	48.04	100.00	0.36	0.36	0.00
3	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
4	51.70	51.70	0.00	48.30	48.30	100.00	0.36	0.36	0.00
5	49.99	49.64	76.16	50.01	50.36	23.84	0.39	0.37	1.36
6	51.23	35.91	92.35	48.77	64.09	7.65	0.38	0.46	0.16
7	50.02	30.88	88.07	49.98	69.12	11.93	0.39	0.50	0.18
8	51.76	31.90	90.00	48.24	68.10	10.00	0.37	0.48	0.17
9	50.84	31.88	88.73	49.16	68.12	11.27	0.33	0.41	0.15
10	48.44	33.29	88.82	51.56	66.71	11.18	0.35	0.42	0.16
11	50.40	32.84	56.36	49.60	67.16	43.64	0.37	0.48	0.33
12	47.00	0.00	47.00	53.00	0.00	53.00	0.40	0.00	0.40
13	48.08	47.15	82.19	51.92	52.85	17.81	0.42	0.35	3.21
14	48.22	18.57	91.48	51.78	81.43	8.52	0.39	0.53	0.18
15	47.62	16.04	80.00	52.38	83.96	20.00	0.39	0.54	0.24

Batch Sedimentation SHC tests using Pittsburgh No. 8 seam coal

Test Number	pH	EDTA (g/l)	Stage No.	% Solids			Yield (%)	Comb. Rec. (%)	Rejection		Separation Efficiency (%)
				Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulphur (%)	
1	6.6	0.4	1	2.95	6.19	nes	99	100	nes	nes	7
2	7.5	0.4	1	2.86	6.26	nes	98	99	nes	nes	10
3	8.5	0.4	1	2.89	6.24	nes	98	99	nes	nes	6
4	6.5	0.4	1-3	2.89	6.29	0.05	95.89	99.12	19.38	9.13	18.50
5	7.6	0.4	1-3	2.87	6.44	0.06	94.64	98.73	24.79	11.15	23.52
6	8.5	0.4	1-3	2.90	5.59	0.11	91.82	95.61	26.05	16.09	21.65
7	9.5	0.4	1-3	2.90	3.44	0.30	106.77	106.94	-5.98	-5.32	0.96
8	7.5	0.4	1-6	2.95	6.60	0.05	92.82	97.54	28.73	16.08	26.27
9	7.5	0.2	1	2.86	6.59	nes	99	100	nes	nes	6
10	8.5	0.2	1	2.98	6.38	nes	99	99	nes	nes	4
11	6.5	0.0	1	2.95	5.81	nes	98	100	nes	nes	9
12	7.5	0.0	1	2.86	6.15	nes	99	100	nes	nes	5
13	8.5	0.0	1	2.98	6.63	0.04	99.12	99.54	2.94	nes	2.47
14	6.5	0.0	1-3	2.89	5.86	0.02	97.76	99.50	10.47	4.92	9.97
15	7.5	0.0	1-3	2.87	6.52	0.01	97.65	99.42	10.71	4.09	10.13
16	8.5	0.0	1-3	2.90	6.95	0.14	91.42	95.96	29.85	17.74	25.81
17	9.5	0.0	1	2.89	0.00	2.89	0.00	0.00	100.00	100.00	0.00

'nes' - Not Enough Sample

Batch Sedimentation SHC tests using Pittsburgh No. 8 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	17.96	16.88	nes	82.04	83.12	nes	4.15	4.08	nes
2	17.23	15.84	nes	82.77	84.16	nes	3.46	3.48	nes
3	16.87	16.06	nes	83.13	83.94	nes	3.27	3.29	nes
4	17.46	14.68	82.30	82.54	85.32	17.7	4.01	3.80	8.91
5	17.39	13.82	80.39	82.61	86.18	19.61	3.92	3.68	8.16
6	17.47	14.07	55.65	82.53	85.93	44.35	3.98	3.64	7.76
7	17.46	17.33	15.41	82.54	82.67	84.59	3.98	3.95	3.46
8	17.96	13.79	71.89	82.04	86.21	28.11	4.15	3.74	9.70
9	17.23	16.31	nes	82.77	83.69	nes	3.46	3.52	nes
10	16.87	16.33	nes	83.13	83.67	nes	3.27	3.32	nes
11	17.96	16.59	nes	82.04	83.41	nes	4.15	4.09	nes
12	17.23	16.48	nes	82.77	83.52	nes	3.46	3.53	nes
13	16.87	16.52	56.29	83.13	83.48	43.71	3.27	3.35	nes
14	17.46	15.99	81.66	82.54	84.01	18.34	4.01	3.90	8.81
15	17.39	15.9	79.42	82.61	84.10	20.58	3.92	3.85	6.834
16	17.58	13.49	61.17	82.42	86.51	38.83	3.95	3.55	8.27
17	17.46	0.00	17.46	82.54	0.00	82.54	4.01	0.00	4.01

'nes' - Not Enough Sample

APPENDIX C - LAMELLA THICKENER DATA

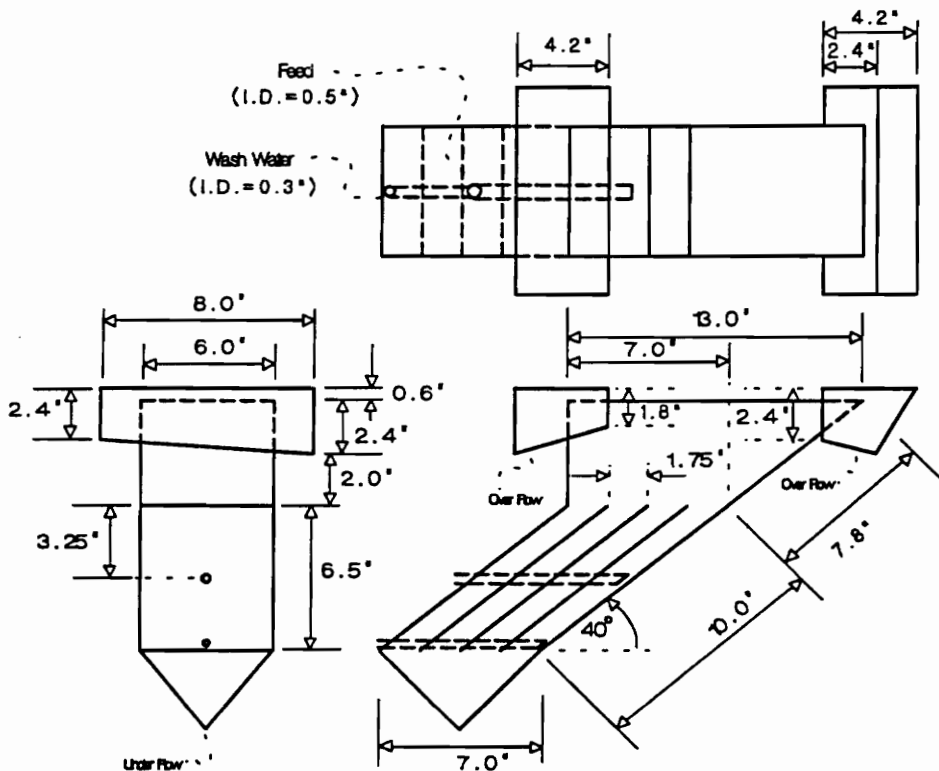


Plate Surface Area = $A = 60 \text{ inches}^2$

Number of plates = $n = 4$

Distance between plates = 1.75 inches

Cell Volume = 720 inches^3

Plate Angle (α) for the cell is adjustable from 30 to 55 degrees

- plate angle when using wash water = 45 to 55 degrees
- plate angle without wash water = 30 to 40 degrees (von Prof. Dr. sc.techn. Heinrich Schubert, *Aufbereitung fester mineralischer Rohstoffe Band III*. 1984).

Effective area = $n * A * \cos(\alpha)$

- effective area using wash water = $4 * (60) * \cos(50) = 154 \text{ inches}^2$
- effective area without wash water = $4 * (60) * \cos(35) = 197 \text{ inches}^2$

Lamella type Thickener

- shorter retention time than a conventional thickener
- underflow has a lower solids content than a conventional thickener; may apply low amplitude vibrations to provide solids compression and increased flow

Projected Output of Lamella type Thickener

- Feed rate = ?
- Feed solids content = ?
- Throughput = ? (0.5 - 1.0 lb/hr required)
- Underflow solids content = ?

Lamella Thickener retention time data using Elkhorn No. 3 coal seam

Number	Sample Time (minutes)	Flow Rates			% Solids			Mass Flow Rates		
		Wash Water (ml/min)	Feed (ml/min)	Product (ml/min)	Feed (%)	Product (%)	Tails (%)	Feed (g/min)	Product (g/min)	Tails (g/min)
1	20	0	100	21		8.66	0.51		2.02	0.40
2	53	0	100	19		9.91	0.71		1.10	0.56
3	90	0	100	19		9.96	0.81		1.11	0.62
4	127	0	100	20		9.92	0.84		1.65	0.63
5	157	0	100	21		10.49	0.85		2.17	0.62
6	186	0	100	20	2.87	10.41	0.89	3.28	2.16	0.65

Lamella Thickener retention time data using Elkhorn No. 3 coal seam

Number	Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)
			Ash (%)	Sulfur (%)	
1	70.25	92.23	51.86	12.70	44.09
2	70.86	92.60	50.96	12.70	43.56
3	73.16	93.84	47.48	11.49	41.32
4	73.07	94.22	47.81	10.37	42.03
5	74.17	95.04	46.71	10.37	41.74
6	72.69	94.75	49.02	11.29	43.76

Lamella Thickener retention time data using Elkhorn No. 3 coal seam

Number	% Ash (adjusted)			% Combustible (adjusted)			% Sulfur (adjusted)		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	49.86	34.17	86.91	50.14	65.83	13.09	0.36	0.44	0.16
2	49.92	34.55	87.29	50.08	65.45	12.71	0.36	0.44	0.16
3	50.06	35.94	88.54	49.94	64.06	11.46	0.36	0.44	0.15
4	50.32	35.94	89.34	49.68	64.06	10.66	0.36	0.44	0.14
5	49.99	35.92	90.39	50.01	64.08	9.61	0.36	0.44	0.14
6	50.40	35.35	90.46	49.60	64.65	9.54	0.36	0.45	0.14

Lamella thickener parametric test data using Elkhorn No. 3 seam coal

Test Number	Flow Rates				% Solids			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)	
	Wash Water (ml/min)	Feed (ml/min)	Product (lb/hr)	Tails (ml/min)	Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulfur (%)		
1	0	50	0.11	7	41	1.97	9.74	0.48	75.47	96.55	48.05	9.21	44.60
2	0	150	0.34	19	128	1.98	10.35	0.66	68.54	90.30	55.10	15.64	45.40
3	0	50	0.36	20	30	5.71	10.11	0.37	98.84	99.87	2.27	0.00	2.14
4*	0	148	1.12	50	98	5.81	10.81	1.02	88.40	98.06	21.72	4.17	19.78
5	40	50	0.11	7	83	1.90	11.12	0.33	67.97	92.40	58.59	15.32	50.99
6	40	149	0.33	19	170	1.86	8.64	0.55	67.92	90.21	57.33	16.07	47.54
7	40	49	0.31	20	70	5.49	10.40	0.44	87.33	98.39	24.86	3.67	23.25
8*	40	150	1.04	50	140	5.52	9.91	0.96	86.03	97.01	25.96	6.48	22.96
9	20	100	0.52	27	93	3.76	9.10	1.01	75.29	95.81	44.67	8.97	40.48
10	18	94	0.47	30	82	3.89	9.26	0.95	74.71	94.78	45.03	11.11	39.81
11	20	89	0.43	28	81	3.70	9.04	0.93	75.33	95.45	44.75	8.97	40.20

* - uncontrollable mudline

Lamella thickener parametric test data using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)			Superficial Velocity		Throughput		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Overflow (mm/min)	Per Plate (mm/min)	Eff. Area (lb/hr/sq.in.)	Actual Area (lb/hr/sq.in.)	Volume (lb/hr/litre)
1	47.26	32.53	92.58	52.74	67.47	7.42	0.38	0.46	0.14	8.15	15.13	0.0007	0.0027	0.0102
2	47.92	31.39	83.94	52.08	68.81	16.06	0.37	0.46	0.18	25.44	47.24	0.0020	0.0081	0.0310
3	47.92	47.38	94.09	52.08	52.62	5.91	0.37	0.37	0.09	5.96	11.07	0.0021	0.0087	0.0332
4*	48.86	43.27	91.45	51.14	56.73	8.55	0.36	0.39	0.13	19.47	36.17	0.0066	0.0266	0.1016
5	47.90	29.18	87.63	52.10	70.82	12.37	0.37	0.47	0.17	16.49	30.63	0.0006	0.0026	0.0098
6	46.88	29.45	83.79	53.12	70.55	16.21	0.38	0.47	0.19	33.78	62.74	0.0019	0.0078	0.0298
7	47.57	40.93	93.32	52.43	59.07	6.88	0.38	0.41	0.13	13.91	25.83	0.0018	0.0075	0.0285
8*	47.78	41.12	88.81	52.22	58.88	11.19	0.38	0.42	0.16	27.82	51.67	0.0061	0.0248	0.0949
9	50.70	37.26	91.64	49.30	62.74	8.36	0.35	0.42	0.13	18.48	34.32	0.0031	0.0124	0.0473
10	50.42	37.10	89.76	49.58	62.90	10.24	0.35	0.42	0.15	16.30	30.26	0.0028	0.0112	0.0429
11	50.05	36.71	90.78	49.95	63.29	9.22	0.35	0.42	0.13	16.10	29.89	0.0025	0.0102	0.0388

* - uncontrollable mudline

STATISTICAL DATA ANALYSIS

Description: Lamella Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Combustible Recovery (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed % Solids	(%)	2.000	6.000
B	Wash water	(ml/min)	0.000	40.000
C	Feed Rate	(ml/min)	50.000	150.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	100004.0	1	100004.0		
Linear	93.0	3	31.0	23.13	0.0005
Quadratic	8.6	6	1.4	1.718	0.5256
Cubic	0.8	1	0.8		
RESIDUAL	0.0	0			
TOTAL	100106.4	11			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	4	7	1.16	0.9084	0.8691	30.39
Quadratic	10	1	0.91	0.9919	0.9190	16825.88
Cubic	11	0				

ANOVA for Linear Model (Combustible Recovery/Lamella Thickener/Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	93.0	3	31.00	23.13	0.0005
RESIDUAL	9.4	7	1.34		
COR TOTAL	102.4	10			
ROOT MSE	1.16		R-SQUARED	0.9084	
DEP MEAN	95.35		ADJ R-SQUARED	0.8691	
C.V.	1.21%				

Predicted Residual Sum of Squares (PRESS) = 30.4

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	95.61	1	0.35	271.4	
A	3.25	1	0.45	7.253	0.0002
B	-0.71	1	0.41	-1.732	0.1268
C	-1.45	1	0.41	-3.533	0.0096

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 95.61 + (3.25 * A) - (0.71 * B) - (1.45 * C)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 92.73 + (1.6252 * \text{Feed \% Solids}) \\ & - (3.546E-02 * \text{Wash water}) \\ & - (2.905E-02 * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	96.55	94.53	2.02	0.462	2.385	1.221	5.096
3	99.87	100.54	-0.67	0.454	-0.781	0.127	-0.757
5	92.40	93.11	-0.71	0.448	-0.821	0.137	-0.800
7	98.39	98.79	-0.40	0.454	-0.473	0.047	-0.445
9	95.81	95.29	0.52	0.091	0.472	0.006	0.444
10	94.78	95.70	-0.92	0.093	-0.831	0.018	-0.811
11	95.45	95.45	0.00	0.096	0.004	0.000	0.003
2	90.30	91.62	-1.32	0.481	-1.583	0.581	-1.830
4	98.06	97.85	0.21	0.479	0.246	0.014	0.229
6	90.21	90.07	0.14	0.471	0.168	0.006	0.156
8	97.01	95.89	1.12	0.471	1.330	0.395	1.425

ANOVA for Quadratic Model (Comb. Recovery/Lamella Thickener/Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	101.6	9	11.28	13.60	0.2076
RESIDUAL	0.8	1	0.83		
COR TOTAL	102.4	10			
ROOT MSE	0.91		R-SQUARED	0.9919	
DEP MEAN	95.35		ADJ R-SQUARED	0.9190	
C.V.	0.96%				

Predicted Residual Sum of Squares (PRESS) = 16825.9

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	95.69	1	0.59	161.2	
A	7.23	1	3.78	1.914	0.3066
B	0.39	1	0.95	0.4081	0.7533
C	-1.84	1	0.46	-4.002	0.1559
A2	19.84	1	18.41	1.078	0.4761
B2	-16.22	1	33.82	-0.4796	0.7153
C2	-0.11	1	22.36	-5.00E-03	0.9968
AB	1.84	1	1.24	1.483	0.3778
AC	0.65	1	0.37	1.756	0.3295
BC	0.64	1	0.34	1.900	0.3085

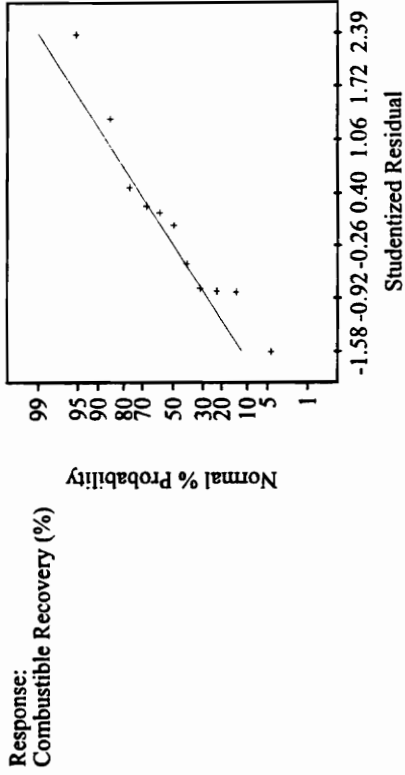
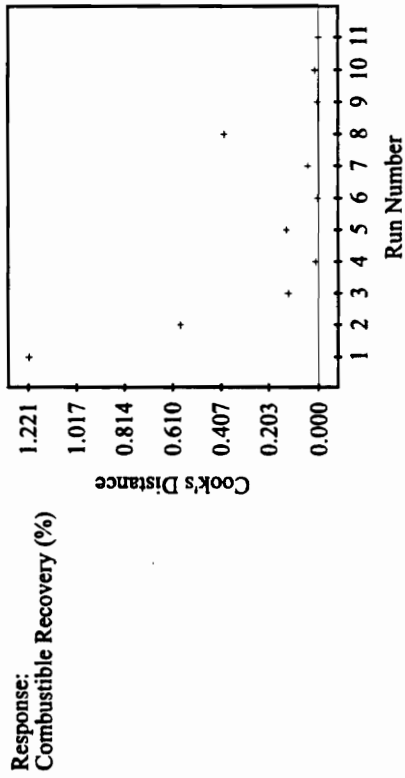
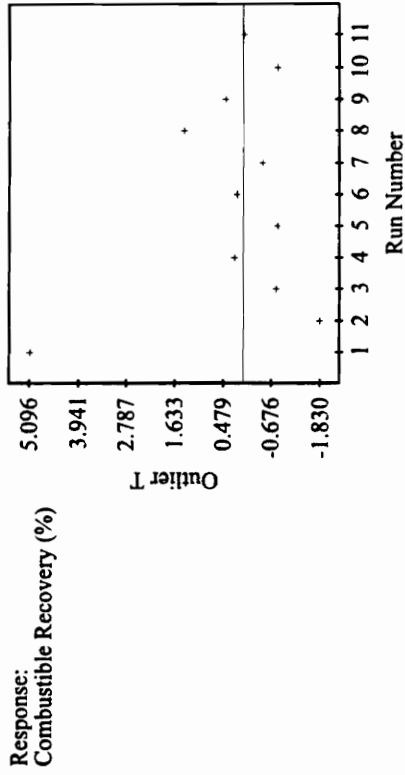
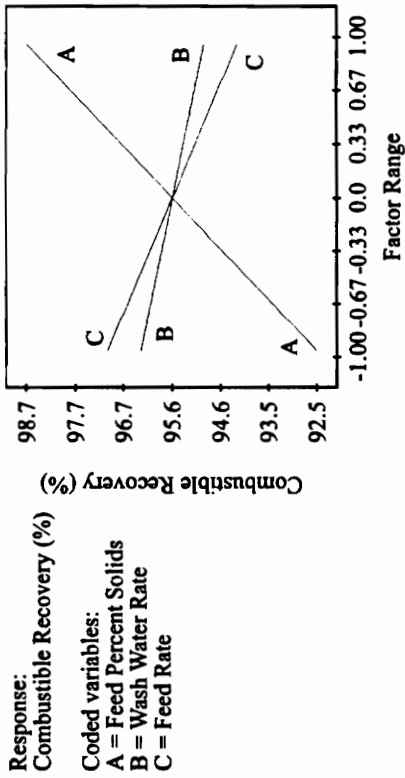
Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 95.69 + (7.23 * A) + (0.39 * B) - (1.84 * C) \\ & + (19.84 * A^2) - (16.22 * B^2) - (0.11 * C^2) \\ & + (1.84 * AB) + (0.65 * AC) + (0.64 * BC) \end{aligned}$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 154.81 - (37.646 * \text{Feed \% Solids}) \\ & + (1.3932 * \text{Wash water}) \\ & - (6.697E-02 * \text{Feed Rate}) \\ & + (4.9609 * \text{Feed \% Solids}^2) \\ & - (4.055E-02 * \text{Wash water}^2) \\ & - (4.472E-05 * \text{Feed Rate}^2) \\ & + (4.592E-02 * \text{Feed \% Solids} * \text{Wash water}) \\ & + (6.550E-03 * \text{Feed \% Solids} * \text{Feed Rate}) \\ & + (6.433E-04 * \text{Wash water} * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	96.55	96.56	-0.01	1.000	-1.000	640.752	0.000
3	99.87	99.82	0.05	0.997	1.000	31.372	0.000
5	92.40	92.37	0.03	0.999	1.000	122.169	0.000
7	98.39	98.41	-0.02	0.999	-1.000	177.336	0.000
9	95.81	95.16	0.65	0.494	1.000	0.098	0.000
10	94.78	95.41	-0.63	0.515	-1.000	0.106	0.000
11	95.45	95.47	-0.02	1.000	-1.000	204.641	0.000
2	90.30	90.28	0.02	1.000	1.000	218.746	0.000
4	98.06	98.08	-0.02	0.999	-1.000	137.479	0.000
6	90.21	90.26	-0.05	0.997	-1.000	38.354	0.000
8	97.01	97.00	0.01	1.000	1.000	456.072	0.000



Lamella thickener sedimentation separation of coagula formed using Elkhorn No. 3 (50% Ash) seam coal

STATISTICAL DATA ANALYSIS

Description: Lamella Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Separation Efficiency (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed % Solids	(%)	2.000	6.000
B	Wash water	(ml/min)	0.000	40.000
C	Feed Rate	(ml/min)	50.000	150.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	12931.1	1	12931.1		
Linear	1955.4	3	651.8	13.95	0.0024
Quadratic	326.0	6	54.3	55.82	0.1021
Cubic	1.0	1	1.0		
RESIDUAL	0.0	0			
TOTAL	15213.4	11			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	4	7	6.83	0.8567	0.7954	876.71
Quadratic	10	1	0.99	0.9996	0.9957	19740.06
Cubic	11	0				

ANOVA for Linear Model (Separation Efficiency/Lamella/Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	1955.4	3	651.79	13.95	0.0024
RESIDUAL	326.9	7	46.71		
COR TOTAL	2282.3	10			
ROOT MSE	6.83		R-SQUARED	0.8567	
DEP MEAN	34.29		ADJ R-SQUARED	0.7954	
C.V.	19.93%				

Predicted Residual Sum of Squares (PRESS) = 876.7

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	32.73	1	2.08	15.74	
A	-16.44	1	2.65	-6.215	0.0004
B	3.40	1	2.42	1.405	0.2027
C	1.60	1	2.43	0.6595	0.5307

Final Equation in Terms of Coded Factors:

$$\text{Separation Efficiency} = 32.73 - (16.44 * A) + (3.40 * B) + (1.60 * C)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 59.02 - (8.2214 * \text{Feed \% Solids}) \\ & + (0.16985 * \text{Wash water}) \\ & + (3.201E-02 * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	44.60	44.18	0.42	0.462	0.084	0.002	0.078
3	2.14	13.76	-11.62	0.454	-2.300	1.100	-4.311
5	50.99	50.97	0.02	0.448	0.004	0.000	0.004
7	23.25	22.20	1.05	0.454	0.209	0.009	0.194
9	40.48	34.38	6.10	0.091	0.937	0.022	0.928
10	39.81	33.02	6.79	0.093	1.043	0.028	1.051
11	40.20	34.85	5.35	0.096	0.824	0.018	0.803
2	45.40	47.38	-1.98	0.481	-0.402	0.037	-0.376
4	19.78	16.07	3.71	0.479	0.752	0.130	0.726
6	47.54	54.96	-7.42	0.471	-1.494	0.497	-1.675
8	22.96	25.40	-2.44	0.471	-0.490	0.054	-0.462

ANOVA for Quadratic Model (Separation Efficiency/Lamella/Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	2281.3	9	253.48	260.4	0.0481
RESIDUAL	1.0	1	0.97		
COR TOTAL	2282.3	10			
ROOT MSE	0.99		R-SQUARED	0.9996	
DEP MEAN	34.29		ADJ R-SQUARED	0.9957	
C.V.	2.88%				

Predicted Residual Sum of Squares (PRESS) = 19740.1

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	39.99	1	0.64	62.18	
A	-7.28	1	4.09	-1.778	0.3262
B	6.38	1	1.03	6.217	0.1015
C	0.49	1	0.50	0.9856	0.5046
A2	46.77	1	19.94	2.346	0.2565
B2	-8.52	1	36.63	-0.2327	0.8545
C2	-39.98	1	24.22	-1.651	0.3467
AB	5.86	1	1.34	4.363	0.1434
AC	2.77	1	0.40	6.853	0.0922
BC	-2.69	1	0.37	-7.343	0.0862

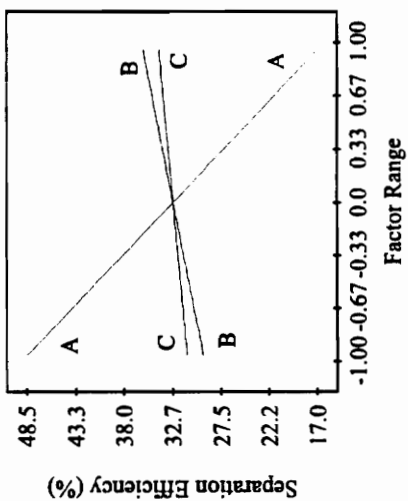
Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 39.99 - (7.28 * A) + (6.38 * B) + (0.49 * C) \\ & + (46.77 * A2) - (8.52 * B2) - (39.98 * C2) \\ & + (5.86 * AB) + (2.77 * AC) - (2.69 * BC) \end{aligned}$$

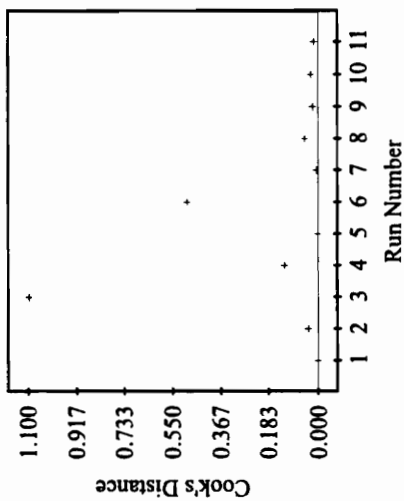
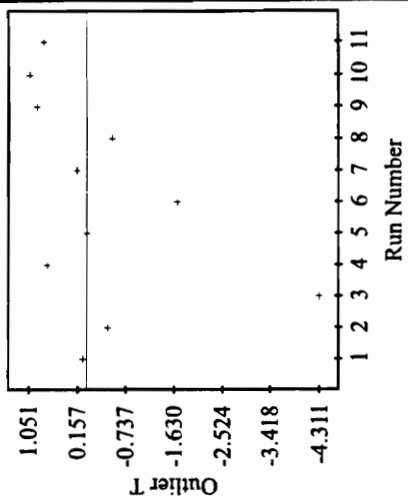
Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 83.20 - (102.87 * \text{Feed \% Solids}) \\ & + (0.85486 * \text{Wash water}) \\ & + (3.1517 * \text{Feed Rate}) \\ & + (11.693 * \text{Feed \% Solids}^2) \\ & - (2.130E-02 * \text{Wash water}^2) \\ & - (1.599E-02 * \text{Feed Rate}^2) \\ & + (0.14638 * \text{Feed \% Solids} * \text{Wash water}) \\ & + (2.768E-02 * \text{Feed \% Solids} * \text{Feed Rate}) \\ & - (2.694E-03 * \text{Wash water} * \text{Feed Rate}) \end{aligned}$$

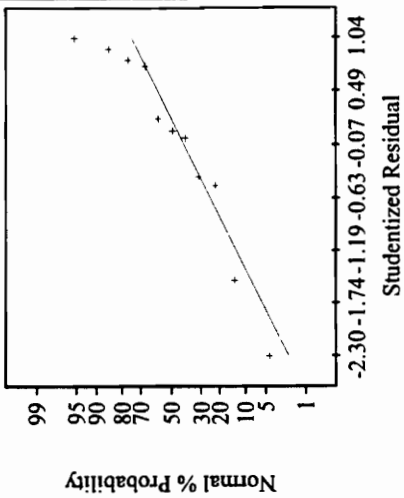
Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	44.60	44.59	0.01	1.000	1.000	640.752	0.000
3	2.14	2.20	-0.06	0.997	-1.000	31.372	0.000
5	50.99	51.02	-0.03	0.999	-1.000	122.169	0.000
7	23.25	23.23	0.02	0.999	1.000	177.336	0.000
9	40.48	41.18	-0.70	0.494	-1.000	0.098	0.000
10	39.81	39.12	0.69	0.515	1.000	0.106	0.000
11	40.20	40.18	0.02	1.000	1.000	204.641	0.000
2	45.40	45.42	-0.02	1.000	-1.000	218.746	0.000
4	19.78	19.75	0.03	0.999	1.000	137.479	0.000
6	47.54	47.49	0.05	0.997	1.000	38.354	0.000
8	22.96	22.97	-0.01	1.000	-1.000	456.072	0.000



Response: Separation Efficiency (%)



Response: Separation Efficiency (%)



Lamella thickener sedimentation separation of coagula formed using Pittsburgh No. 8 seam coal

Lamella thickener parametric test data using Pittsburgh No. 8 seam coal

Test Number	Flow Rates				% Solids			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)	
	Wash Water (ml/min)	Feed		Product (ml/min)	Tailings (ml/min)	Feed (%)	Product (%)			Tails (%)	Ash (%)		Sulphur (%)
1	20	50	0.12	7	63	1.99	10.52	0.01	97.64	99.23	10.13	0.00	9.36
2	100	50	0.12	140	10	2.12	9.44	0.09	85.71	88.94	29.59	21.53	18.53
3	20	50	0.35	30	40	5.81	8.66	nes	102.07	102.49	0.00	0.00	2.49
4	100	50	0.37	30	120	5.99	8.48	0.04	96.55	97.42	7.56	5.98	4.98
5	20	94	0.23	23	88	1.99	7.52	0.02	99.60	99.90	2.07	1.44	1.97
6	100	96	0.23	20	174	1.99	7.81	0.08	88.65	93.32	36.93	23.07	30.25
7*	20	100	0.72	90	30	6.02	5.86	nes	99.65	99.58	0.00	0.00	-0.42
8*	100	100	0.72	102	98	6.02	5.51	0.02	99.45	99.70	1.76	0.00	1.46
9	60	75	0.36	42	90	3.98	6.48	0.02	99.70	99.86	1.05	0.51	0.91
10	61	75	0.36	42	90	3.99	7.01	0.02	100.15	100.07	-0.56	0.00	-0.49
11	62	75	0.38	43	94	4.05	6.37	0.02	99.18	99.60	2.77	1.10	2.37

* - uncontrollable mudline

'nes' - Not Enough Sample

Lamella thickener parametric test data using Pittsburgh No. 8 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)			Superficial Velocity		Throughput		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Overflow (mm/min)	Plate (mm/min)	Eff. Area (lb/hr/sq.in.)	Actual Area (lb/hr/sq.in.)	Volume (lb/hr/litre)
1	16.95	15.60	72.92	83.05	84.40	27.08	3.96	3.82	nes	3.97	7.38	0.0007	0.0028	0.0106
2	17.42	14.31	36.08	82.58	85.69	63.92	4.26	3.90	6.42	19.87	36.91	0.0007	0.0028	0.0106
3	16.76	16.42	nes	83.24	83.58	100.00	4.27	4.24	nes	3.97	7.38	0.0021	0.0084	0.0319
4	17.37	16.63	38.10	82.63	83.37	61.90	4.30	4.20	6.87	19.87	36.91	0.0022	0.0088	0.0335
5	15.53	15.27	79.50	84.47	84.73	20.50	3.55	3.52	8.49	3.97	7.38	0.0013	0.0054	0.0208
6	15.42	10.97	50.18	84.58	89.03	49.82	3.57	3.10	7.22	19.87	36.91	0.0013	0.0054	0.0208
7*	17.25	17.31	nes	82.75	82.69	100.00	3.69	3.67	nes	3.97	7.38	0.0043	0.0172	0.0658
8*	17.25	17.04	55.11	82.75	82.96	44.89	3.69	3.62	nes	19.87	36.91	0.0043	0.0172	0.0658
9	17.32	17.19	60.55	82.68	82.81	39.45	4.04	4.03	7.76	11.92	22.14	0.0021	0.0087	0.0330
10	17.38	17.45	63.59	82.62	82.55	36.41	4.03	4.03	7.21	12.12	22.51	0.0021	0.0087	0.0330
11	17.84	17.49	60.08	82.16	82.51	39.92	3.80	3.78	7.26	12.32	22.88	0.0022	0.0090	0.0342

* - uncontrollable mudline

'nes' - Not Enough Sample

STATISTICAL DATA ANALYSIS

Description: Lamella Thickener
 Coal Sample: Pittsburgh No. 8
 Response: Combustible Recovery (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed % Solids	(%)	2.000	6.000
B	Wash Water	(ml/min)	20.000	100.000
C	Feed Rate	(ml/min)	50.000	100.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	106058.0	1	106058.0		
Linear	100.3	3	33.4	5.246	0.0328
Quadratic	44.6	6	7.4	162.1	0.0601
Cubic	0.0	1	0.0		
RESIDUAL	0.0	0			
TOTAL	106202.9	11			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	4	7	2.52	0.6922	0.5602	139.79
Quadratic	10	1	0.21	0.9997	0.9968	
Cubic	11	0				

ANOVA for Linear Model (Combustible Recovery/Lamella/Pittsburgh)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	100.3	3	33.43	5.246	0.0328
RESIDUAL	44.6	7	6.37		
COR TOTAL	144.9	10			
ROOT MSE	2.52		R-SQUARED	0.6922	
DEP MEAN	98.19		ADJ R-SQUARED	0.5602	
C.V.	2.57%				

Predicted Residual Sum of Squares (PRESS) = 139.8

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	98.23	1	0.76	128.9	
A	2.22	1	0.91	2.441	0.0447
B	-2.76	1	0.89	-3.092	0.0175
C	0.55	1	0.94	0.5892	0.5742

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 98.23 + (2.22 * A) - (2.76 * B) + (0.55 * C)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 96.27 + (1.1115 * \text{Feed \% Solids}) \\ & - (6.900E-02 * \text{Wash Water}) \\ & + (2.211E-02 * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	99.23	98.22	1.01	0.453	0.544	0.061	0.514
3	102.49	102.44	0.05	0.467	0.028	0.000	0.026
2	88.94	92.81	-3.87	0.452	-2.069	0.884	-3.074
4	97.42	97.14	0.28	0.486	0.154	0.006	0.143
9	99.86	98.23	1.63	0.091	0.677	0.011	0.648
10	100.07	98.16	1.91	0.091	0.793	0.016	0.770
11	99.60	98.20	1.40	0.092	0.580	0.008	0.551
5	99.90	99.19	0.71	0.445	0.379	0.029	0.354
7	99.58	103.77	-4.19	0.487	-2.315	1.272	-4.430
6	93.32	93.71	-0.39	0.467	-0.213	0.010	-0.198
8	99.70	98.25	1.45	0.469	0.790	0.138	0.766

ANOVA for Quadratic Model (Combustible Recovery/Lamella/Pittsburgh)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	144.8	9	16.09	351.3	0.0414
RESIDUAL	0.0	1	0.05		
COR TOTAL	144.9	10			
ROOT MSE	0.21		R-SQUARED	0.9997	
DEP MEAN	98.19		ADJ R-SQUARED	0.9968	
C.V.	0.22%				

Case(s) with leverage of 1.0000: PRESS statistic not defined.

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	100.01	1	0.16	637.1	
A	-2.62	1	3.54	-0.7395	0.5946
B	-3.27	1	0.43	-7.512	0.0843
C	6.12	1	3.97	1.542	0.3662
A2	-26.80	1	17.25	-1.554	0.3641
B2	-16.73	1	13.50	-1.240	0.4322
C2	44.64	1	33.04	1.351	0.4057
AB	3.33	1	1.29	2.580	0.2354
AC	-4.98	1	3.11	-1.601	0.3555
BC	0.00	1	0.86	1.29E-03	0.9992

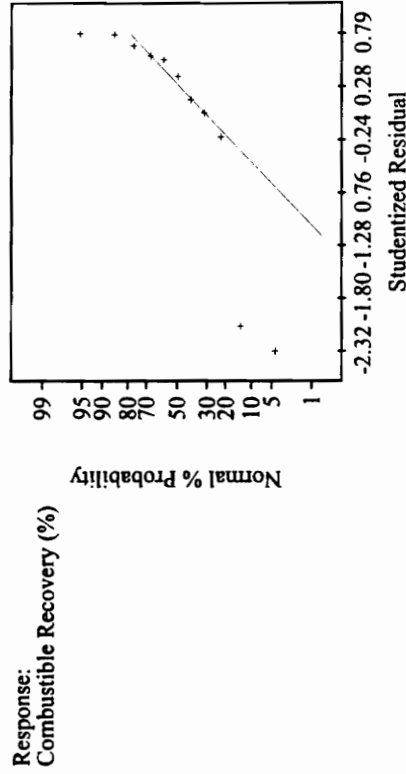
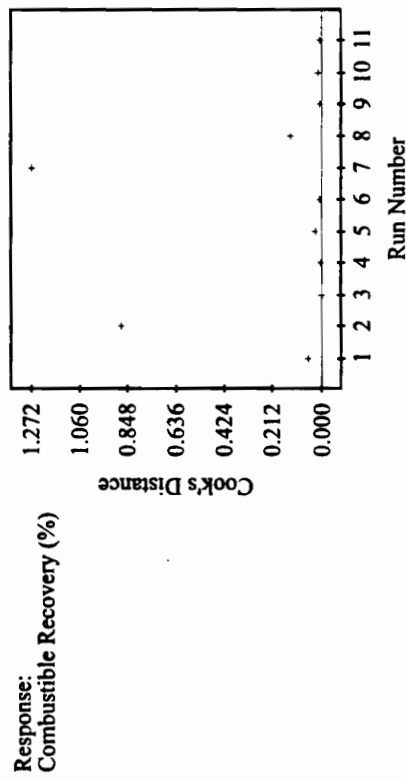
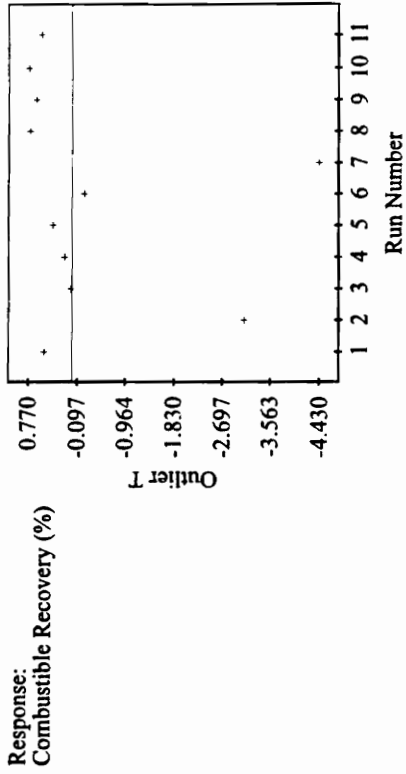
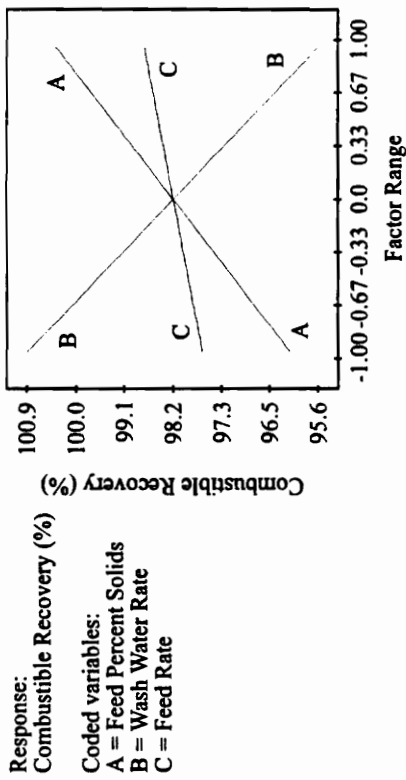
Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 100.01 - (2.62 * A) - (3.27 * B) + (6.12 * C) \\ & - (26.80 * A^2) - (16.73 * B^2) + (44.64 * C^2) \\ & + (3.33 * AB) - (4.98 * AC) + (0.00 * BC) \end{aligned}$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 328.80 + (57.256 * \text{Feed \% Solids}) \\ & + (1.0067 * \text{Wash Water}) \\ & - (10.070 * \text{Feed Rate}) \\ & - (6.6989 * \text{Feed \% Solids}^2) \\ & - (1.046E-02 * \text{Wash Water}^2) \\ & + (7.142E-02 * \text{Feed Rate}^2) \\ & + (4.157E-02 * \text{Feed \% Solids} * \text{Wash Water}) \\ & - (0.09956 * \text{Feed \% Solids} * \text{Feed Rate}) \\ & + (1.107E-06 * \text{Wash Water} * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	99.23	99.23	0.00	1.000			
3	102.49	102.49	-7.254E-07	1.000			
2	88.94	88.94	-0.00	1.000			
4	97.42	97.42	0.00	1.000			
9	99.86	100.01	-0.15	0.538	-1.000	0.116	0.000
10	100.07	99.91	0.16	0.465	1.000	0.087	0.000
11	99.60	99.61	-0.01	0.997	-1.000	38.004	0.000
5	99.90	99.90	-0.00	1.000			
7	99.58	99.58	0.00	1.000			
6	93.32	93.32	-0.00	1.000			
8	99.70	99.70	-0.00	1.000			



Lamella thickener sedimentation separation of coagula formed using Pittsburgh No. 8 seam coal

 STATISTICAL DATA ANALYSIS

Description: Lamella Thickener
 Coal Sample: Pittsburgh No. 8
 Response: Separation Efficiency (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed % Solids	(%)	2.000	6.000
B	Wash Water	(ml/min)	20.000	100.000
C	Feed Rate	(ml/min)	50.000	100.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	463.6	1	463.6		
Linear	560.4	3	186.8	3.574	0.0750
Quadratic	364.5	6	60.8	45.33	0.1132
Cubic	1.3	1	1.3		
RESIDUAL	0.0	0			
TOTAL	1389.9	11			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	4	7	7.23	0.6050	0.4357	1064.68
Quadratic	10	1	1.16	0.9986	0.9855	
Cubic	11	0				

ANOVA for Linear Model (Separation Efficiency/Lamella/Pittsburgh)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	560.4	3	186.81	3.574	0.0750
RESIDUAL	365.9	7	52.27		
COR TOTAL	926.3	10			
ROOT MSE	7.23		R-SQUARED	0.6050	
DEP MEAN	6.49		ADJ R-SQUARED	0.4357	
C.V.	111.37%				

Predicted Residual Sum of Squares (PRESS) = 1064.7

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	6.44	1	2.18	2.953	
A	-6.67	1	2.61	-2.558	0.0377
B	5.31	1	2.56	2.077	0.0765
C	-0.31	1	2.69	-0.1154	0.9114

Final Equation in Terms of Coded Factors:

$$\text{Separation Efficiency} = 6.44 - (6.67 * A) + (5.31 * B) - (0.31 * C)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 12.76 - (3.3360 * \text{Feed \% Solids}) \\ & + (0.13271 * \text{Wash Water}) \\ & - (1.240E-02 * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	9.36	8.12	1.24	0.453	0.232	0.011	0.216
3	2.49	-4.56	7.05	0.467	1.336	0.391	1.432
2	18.53	18.40	0.13	0.452	0.024	0.000	0.022
4	4.98	5.39	-0.41	0.486	-0.079	0.001	-0.073
9	0.91	6.44	-5.53	0.091	-0.803	0.016	-0.780
10	-0.49	6.58	-7.07	0.091	-1.025	0.026	-1.030
11	2.37	6.38	-4.01	0.092	-0.581	0.009	-0.552
5	1.97	7.57	-5.60	0.445	-1.040	0.217	-1.048
7	-0.42	-5.85	5.43	0.487	1.048	0.260	1.056
6	30.25	18.16	12.09	0.467	2.290	1.149	4.231
8	1.46	4.77	-3.31	0.469	-0.628	0.087	-0.599

ANOVA for Quadratic Model (Separation Efficiency/Lamella/Pittsburgh)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	925.0	9	102.78	76.69	0.0884
RESIDUAL	1.3	1	1.34		
COR TOTAL	926.3	10			
ROOT MSE	1.16		R-SQUARED	0.9986	
DEP MEAN	6.49		ADJ R-SQUARED	0.9855	
C.V.	17.83%				

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	0.12	1	0.85	0.1448	
A	23.48	1	19.13	1.228	0.4352
B	8.12	1	2.35	3.455	0.1794
C	-35.56	1	21.45	-1.658	0.3456
A2	211.19	1	93.29	2.264	0.2648
B2	49.30	1	73.00	0.6753	0.6219
C2	-271.49	1	178.72	-1.519	0.3706
AB	-17.50	1	6.97	-2.511	0.2413
AC	23.49	1	16.82	1.396	0.3957
BC	10.31	1	4.65	2.216	0.2698

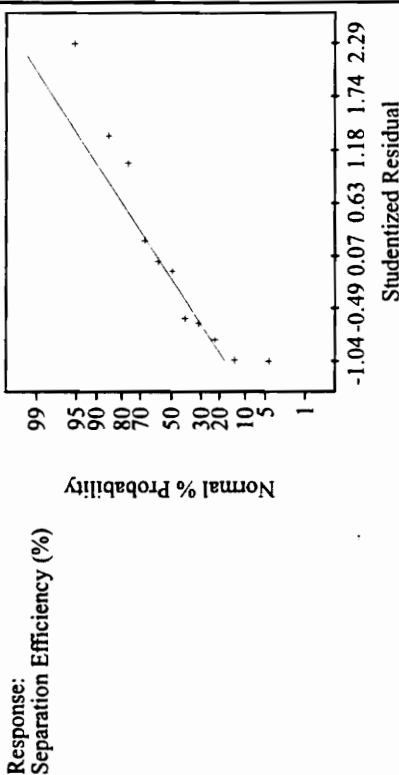
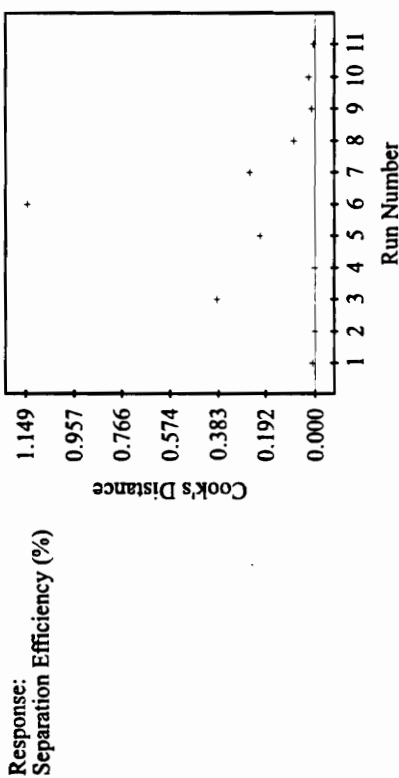
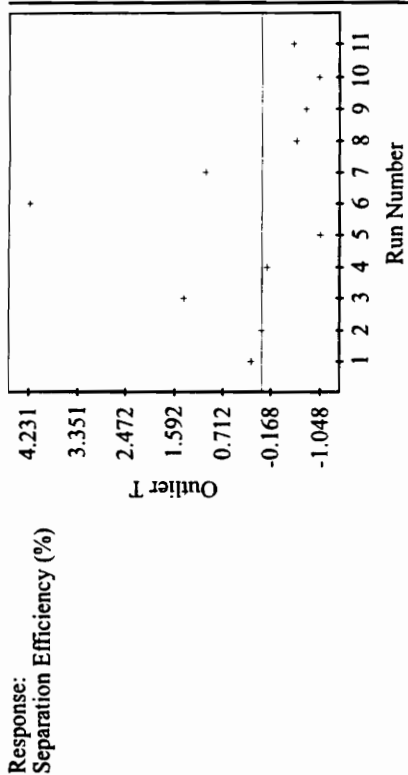
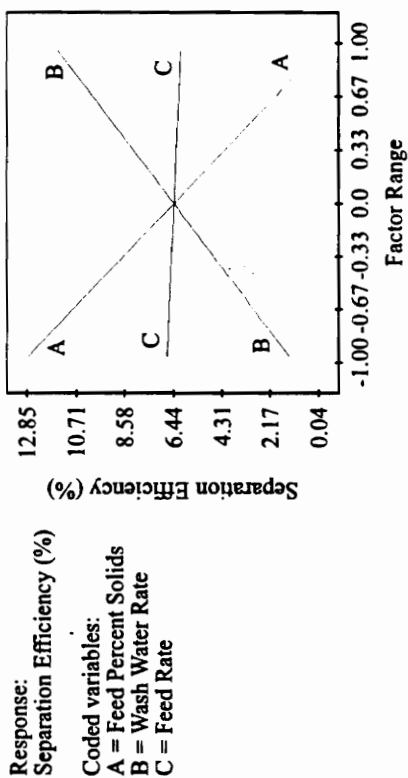
Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 0.12 + (23.48 * A) + (8.12 * B) - (35.56 * C) \\ & + (211.19 * A^2) + (49.30 * B^2) \\ & - (271.49 * C^2) - (17.50 * AB) \\ & + (23.49 * AC) + (10.31 * BC) \end{aligned}$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Separation Efficiency} = & -1305.31 - (32.73 * \text{Feed \% Solids}) \\ & - (3.3926 * \text{Wash Water}) \\ & + (61.238 * \text{Feed Rate}) \\ & + (52.797 * \text{Feed \% Solids}^2) \\ & + (3.081E-02 * \text{Wash Water}^2) \\ & - (0.43439 * \text{Feed Rate}^2) \\ & - (0.21878 * \text{Feed \% Solids} * \text{Wash Water}) \\ & + (0.46970 * \text{Feed \% Solids} * \text{Feed Rate}) \\ & + (1.031E-02 * \text{Wash Water} * \text{Feed Rate}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
1	9.36	9.36	-0.00	1.000			
3	2.49	2.49	0.00	1.000			
2	18.53	18.52	0.01	1.000			
4	4.98	4.98	-0.00	1.000			
9	0.91	0.12	0.79	0.538	1.000	0.116	0.000
10	-0.49	0.36	-0.85	0.465	-1.000	0.087	0.000
11	2.37	2.31	0.06	0.997	1.000	38.004	0.000
5	1.97	1.97	0.00	1.000			
7	-0.42	-0.41	-0.01	1.000			
6	30.25	30.25	0.00	1.000			
8	1.46	1.46	0.00	1.000			



Lamella thickener sedimentation separation of coagula formed using Pittsburgh No. 8 seam coal

APPENDIX D - ROTOR CENTRIFUGE DATA

Rotor centrifuge preliminary test data using Elkhorn No. 3 seam coal

Test Number	pH	RPM's	Effective G's	% Solids		Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Separation Efficiency (%)	% Ash			% Combustible		
				Feed (%)	Tails (%)					Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	7.5	500	0.99	4.99	*nm	31.47	36.08	73.16	9.24	49.95	42.61	53.32	50.05	57.39	46.68
2	7.5	750	2.24	5.01	2.20	52.72	61.61	56.17	17.78	49.98	41.55	59.38	50.02	58.45	40.62
3	7.5	1000	3.98	4.99	*nm	81.18	93.10	30.77	23.87	49.95	42.60	81.65	50.05	57.40	18.35
4	7.5	1250	6.21	5.01	0.05	88.77	98.08	20.56	18.64	49.98	44.73	91.47	50.02	55.27	8.53
5	7.5	1500	8.95	4.99	*nm	87.28	98.08	23.53	21.61	49.95	43.76	92.43	50.05	56.24	7.57
6	7.5	500	0.99	10.29	9.40	10.65	11.16	89.84	1.00	50.93	48.58	51.21	49.07	51.42	48.79
7	7.5	750	2.24	10.29	6.23	45.58	47.81	56.57	4.38	50.93	48.53	52.94	49.07	51.47	47.06
8	7.5	1000	3.98	10.29	3.89	70.86	75.31	33.42	8.73	50.93	47.85	58.42	49.07	52.15	41.58
9	7.5	1250	6.21	10.29	2.59	78.78	83.60	25.86	9.46	50.93	47.93	62.07	49.07	52.07	37.93
10	7.5	1500	8.95	10.29	0.42	93.69	99.03	11.46	10.50	50.93	48.13	92.48	49.07	51.87	7.52
11	7.5	500	0.99	13.55	12.48	11.83	12.03	88.36	0.39	50.78	49.96	50.89	49.22	50.04	49.11
12	7.5	800	2.55	13.55	9.72	15.00	14.95	84.95	-0.10	50.78	50.95	50.75	49.22	49.05	49.25
13	7.5	1000	3.98	13.55	8.79	76.92	76.97	76.97	23.12	50.78	50.75	50.88	49.22	49.25	49.12
14	7.5	1250	6.21	13.55	3.06	83.95	85.06	17.12	2.18	50.78	50.13	54.18	49.22	49.87	45.82
15	7.5	1500	8.95	13.55	0.00	98.33	99.63	2.93	2.56	50.78	50.13	89.12	49.22	49.87	10.88
16	10.0	500	0.99	5.01	3.74	17.56	18.70	83.57	2.28	50.26	47.02	50.95	49.74	52.98	49.05
17	10.0	1000	3.98	5.01	1.06	75.09	84.36	34.08	18.44	50.26	44.12	68.77	49.74	55.88	31.23
18	10.0	1500	8.95	5.01	0.49	83.16	90.94	24.53	15.47	50.26	45.61	73.23	49.74	54.39	26.77
19	10.0	2000	15.91	5.01	0.02	87.92	94.35	18.45	12.80	50.26	46.62	76.74	49.74	53.38	23.26
20	7.5	550	1.20	2.87	1.87	32.71	39.64	73.93	13.57	51.09	40.72	56.13	48.91	59.28	43.87
21	7.5	750	2.24	2.88	0.70	77.04	90.33	35.72	26.05	51.04	42.59	79.39	48.96	57.41	20.61
22	7.5	1000	3.98	2.89	0.51	83.34	96.20	29.01	25.21	51.00	43.44	88.83	49.00	56.56	11.17
23	7.5	1200	5.73	2.89	0.31	87.71	98.23	22.49	20.72	50.78	44.88	92.92	49.22	55.13	7.08
24	7.5	1500	8.95	2.89	0.23	91.19	98.70	16.04	14.74	50.98	46.94	92.79	49.02	53.06	7.21
25	10.0	500	0.99	2.87	1.62	36.36	42.40	69.32	11.72	51.51	43.46	56.11	48.49	56.54	43.89
26	10.0	1000	3.98	2.92	0.51	71.16	79.19	36.53	15.72	51.11	45.59	64.73	48.89	54.41	35.27
27	10.0	1500	8.95	2.90	0.43	79.75	87.88	27.98	15.86	51.27	46.30	70.84	48.73	53.70	29.16

*nm - not measured

Rotor centrifuge preliminary test data using Pittsburgh No. 8 seam coal

Test Number	pH	RPM's	Effective G's	% Solids		Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Separation Efficiency (%)	% Ash			% Combustible		
				Feed (%)	Tails (%)					Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	7.6	500	0.99	4.48	3.86	54.39	54.22	44.78	-1.00	16.97	17.23	16.66	83.03	82.77	83.34
2	7.6	750	2.24	4.50	1.52	32.39	32.77	69.45	2.22	16.88	15.92	17.34	83.12	84.08	82.66
3	7.6	1000	3.98	4.48	0.22	90.42	90.77	11.30	2.06	16.86	16.54	19.88	83.14	83.46	80.12
4	7.6	1250	6.21	4.47	0.00	99.81	100.00	1.14	1.14	16.89	16.73	100.00	83.11	83.27	0.00
5	10.0	500	0.99	4.51	3.50	40.10	38.27	51.34	-10.39	17.56	21.31	15.05	82.44	78.69	84.95
6	10.0	1000	3.98	4.51	0.47	88.32	86.20	1.89	-11.91	17.77	19.74	2.88	82.23	80.26	97.12
7	10.0	1500	8.95	4.49	0.00	109.19	110.95	-1.11	9.83	17.84	16.52	2.16	82.16	83.48	97.84
8	10.0	2000	15.91	4.52	0.00	97.80	97.39	0.26	-2.35	17.70	18.05	2.12	82.30	81.95	97.88

Rotor centrifuge parametric test data using Elkhorn No. 3 seam coal

Test Number	pH	Spin Time (min)	RPM's	Effective G's	Feed		Comb. Rec. (%)	Rejection		Separation Efficiency (%)	% Ash (Adjusted)			% Sulphur (Adjusted)		
					Solids (%)	Yield (%)		Ash (%)	Sulphur (%)		Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	7.6	6	1050	4.38	1.07	96.38	98.78	5.93	0.00	4.71	51.00	49.78	83.46	0.40	0.40	0.40
2	7.6	6	1100	4.81	4.71	78.97	79.75	21.79	0.00	1.55	50.59	50.10	52.43	0.41	0.41	0.41
3	9.5	6	1000	3.98	1.08	84.15	93.96	25.34	0.82	19.29	50.83	45.10	81.25	0.39	0.46	0.02
4	9.6	6	1100	4.81	4.67	85.27	94.87	24.09	6.50	18.30	50.63	45.07	82.82	0.41	0.46	0.16
5	8.5	3	750	2.24	2.75	36.13	46.47	73.48	55.68	19.86	51.84	38.05	59.64	0.43	0.54	0.37
6	8.5	3	1250	6.21	2.85	84.90	97.69	27.19	4.59	24.88	51.42	44.10	92.57	0.43	0.48	0.13
7	8.5	9	750	2.24	2.76	66.05	81.67	48.25	22.48	29.11	52.22	40.92	74.20	0.43	0.50	0.29
8	8.6	9	1250	6.21	2.84	93.17	98.96	12.23	0.00	11.19	51.71	48.71	92.66	0.42	0.46	0.00
9	8.6	3	1000	3.98	0.70	74.64	94.61	45.25	16.24	39.86	50.09	36.74	89.39	0.39	0.44	0.25
10	8.5	3	1000	3.98	4.68	66.31	78.50	45.61	19.76	24.00	50.56	41.47	68.45	0.41	0.47	0.27
11	8.5	9	1000	3.98	0.70	87.76	97.51	21.83	5.53	19.34	50.40	44.89	89.92	0.41	0.44	0.18
12	8.5	9	1000	3.98	4.70	91.15	98.76	16.48	0.00	15.24	49.94	45.76	93.01	0.39	0.44	0.00
13	7.5	6	750	2.24	2.77	47.69	49.38	53.87	0.00	3.43	51.83	50.13	53.38	0.42	0.42	0.41
14	9.5	6	750	2.24	2.87	69.38	83.33	43.80	16.23	27.45	51.41	41.64	73.55	0.42	0.50	0.23
15	7.5	6	1250	6.21	2.78	98.91	100.00	2.10	1.12	1.95	51.93	51.40	100.00	0.43	0.43	0.31
16	9.5	6	1250	6.21	2.86	84.02	91.70	23.14	7.60	14.84	51.77	47.36	74.95	0.43	0.47	0.20
17	8.5	6	750	2.24	0.69	77.41	94.48	39.43	5.06	33.91	50.34	39.39	87.86	0.42	0.51	0.09
18	8.5	6	750	2.24	4.72	68.10	80.43	44.41	20.72	26.19	49.64	40.52	69.11	0.43	0.50	0.28
19	8.5	6	1250	6.21	0.70	86.84	96.84	22.93	0.00	19.77	50.59	44.90	88.14	0.41	0.49	0.00
20	8.5	6	1250	6.21	4.70	90.22	98.60	18.11	0.00	16.71	50.10	45.47	92.83	0.41	0.46	0.00
21	7.5	3	1000	3.98	2.76	86.67	86.77	13.43	0.00	0.91	51.67	51.61	52.06	0.43	0.43	0.44
22	9.5	3	1000	3.98	2.85	68.70	83.67	45.07	20.72	27.83	52.11	41.67	75.02	0.43	0.50	0.28
23	7.5	9	1000	3.98	2.76	78.29	80.06	23.35	20.53	3.42	52.00	50.91	55.93	0.43	0.44	0.41
24	9.5	9	1000	3.98	2.85	86.44	94.31	21.05	6.76	15.36	51.23	46.79	79.53	0.43	0.47	0.22
25	8.5	6	1000	3.98	2.81	81.28	95.65	32.12	7.06	27.77	51.75	43.22	88.78	0.43	0.49	0.16
26	8.6	6	1000	3.98	2.84	86.44	97.99	24.44	5.52	22.43	51.48	45.00	92.79	0.43	0.47	0.17
27	8.5	6	1000	3.98	2.80	80.14	95.60	34.34	5.94	29.94	51.65	42.32	89.29	0.43	0.50	0.13

STATISTICAL DATA ANALYSIS
Description: Centrifugal SedimentationCoal Sample: Elkhorn No. 3 (50% Ash)Response: Combustible Recovery (%)**Design Summary**

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	% Solids	(%)	1.000	5.000
B	pH		7.500	9.500
C	RPM's	(rpm)	750.000	1250.000
D	time	(min)	3.000	9.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	211619.5	1	211619.5		
Linear	2518.4	4	629.6	5.869	0.0023
Quadratic	1817.1	10	181.7	4.017	0.0130
Cubic	498.0	8	62.2	5.541	0.0578
RESIDUAL	44.9	4	11.2		
TOTAL	216497.8	27			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	2356.3	20	117.8	63.17	0.0157
Quadratic	539.2	10	53.9	28.91	0.0339
Cubic	41.2	2	20.6	11.05	0.0830
PURE ERR	3.7	2	1.9		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	22	10.36	0.5162	0.4283	3627.06
Quadratic	15	12	6.73	0.8887	0.7589	3113.95
Cubic	23	4	3.35	0.9908	0.9401	5941.35

ANOVA for Linear Model (Combustible Recovery / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	2518.4	4	629.59	5.869	0.0023
RESIDUAL	2360.0	22	107.27		
Lack Of Fit	2356.3	20	117.81	63.17	0.0157
Pure Error	3.7	2	1.87		
COR TOTAL	4878.4	26			
ROOT MSE	10.36		R-SQUARED	0.5162	
DEP MEAN	88.53		ADJ R-SQUARED	0.4283	
C.V.	11.70%				

Predicted Residual Sum of Squares (PRESS) = 3627.1

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0	PROB > t
Intercept	88.53	1	1.99	44.42	
A	-3.77	1	2.99	-1.262	0.2203
B	3.92	1	2.99	1.313	0.2028
C	12.34	1	2.99	4.126	0.0004
D	5.30	1	2.99	1.772	0.0903

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 88.53 - (3.77 * A) + (3.92 * B) + (12.34 * C) + (5.30 * D)$$

Final Equation in Terms of Actual Factors:

$$\text{Combustible Recovery} = 0.89 - (1.8862 * \% \text{ Solids}) + (3.9250 * \text{pH}) + (4.934\text{E-}02 * \text{RPM's}) + (1.7656 * \text{time})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	98.78	88.38	10.40	0.204	1.125	0.065	1.133	15
2	79.75	80.83	-1.08	0.204	-0.117	0.001	-0.115	24
3	93.96	96.23	-2.27	0.204	-0.245	0.003	-0.240	11
4	94.87	88.68	6.19	0.204	0.669	0.023	0.661	25
5	46.47	70.90	-24.43	0.204	-2.643	0.357	-3.126	7
6	97.69	95.57	2.12	0.204	0.229	0.003	0.224	23
7	81.67	81.49	0.18	0.204	0.019	0.000	0.019	16
8	98.96	106.16	-7.20	0.204	-0.779	0.031	-0.772	18
9	94.61	87.01	7.60	0.204	0.823	0.035	0.816	5
10	78.50	79.46	-0.96	0.204	-0.104	0.001	-0.102	22
11	97.51	97.60	-0.09	0.204	-0.010	0.000	-0.010	4
12	98.76	90.06	8.70	0.204	0.942	0.045	0.939	3
13	49.38	72.27	-22.89	0.204	-2.477	0.314	-2.849	26
14	83.33	80.12	3.21	0.204	0.347	0.006	0.340	6
15	100.00	96.94	3.06	0.204	0.331	0.006	0.324	12
16	91.70	104.79	-13.09	0.204	-1.417	0.103	-1.452	8
17	94.48	79.97	14.51	0.204	1.570	0.126	1.628	14
18	80.43	72.42	8.01	0.204	0.866	0.038	0.861	19
19	96.84	104.64	-7.80	0.204	-0.844	0.036	-0.838	1
20	98.60	97.09	1.51	0.204	0.163	0.001	0.159	13
21	86.77	79.31	7.46	0.204	0.807	0.033	0.801	20
22	83.67	87.16	-3.49	0.204	-0.378	0.007	-0.370	10
23	80.06	89.90	-9.84	0.204	-1.065	0.058	-1.068	27
24	94.31	97.75	-3.44	0.204	-0.373	0.007	-0.365	2
25	95.65	88.53	7.12	0.037	0.700	0.004	0.692	17
26	97.99	88.53	9.46	0.037	0.931	0.007	0.928	21
27	95.60	88.53	7.07	0.037	0.696	0.004	0.687	9

ANOVA for Quadratic Model (Combustible Recovery / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	4335.5	14	309.68	6.845	0.0010
RESIDUAL	542.9	12	45.24		
Lack Of Fit	539.2	10	53.92	28.91	0.0339
Pure Error	3.7	2	1.87		
COR TOTAL	4878.4	26			
ROOT MSE	6.73		R-SQUARED	0.8887	
DEP MEAN	88.53		ADJ R-SQUARED	0.7589	
C.V.	7.60%				

Predicted Residual Sum of Squares (PRESS) = 3114.0

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	96.41	1	3.88	24.83	
A	-3.77	1	1.94	-1.943	0.0759
B	3.93	1	1.94	2.021	0.0661
C	12.34	1	1.94	6.353	0.0001
D	5.30	1	1.94	2.728	0.0183
A2	2.63	1	2.91	0.9043	0.3836
B2	-6.18	1	2.91	-2.122	0.0553
C2	-8.31	1	2.91	-2.853	0.0145
D2	-5.88	1	2.91	-2.019	0.0664
AB	4.99	1	3.36	1.482	0.1640
AC	3.95	1	3.36	1.175	0.2627
AD	4.34	1	3.36	1.290	0.2212
BC	-10.56	1	3.36	-3.141	0.0085
BD	4.34	1	3.36	1.290	0.2214
CD	-8.48	1	3.36	-2.522	0.0268

Final Equation in Terms of Coded Factors:

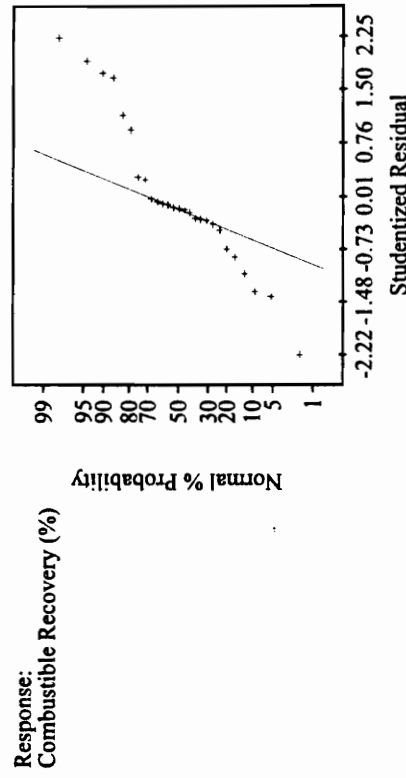
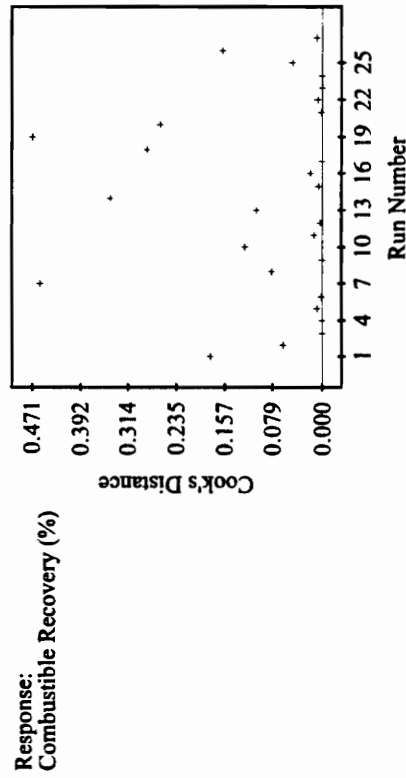
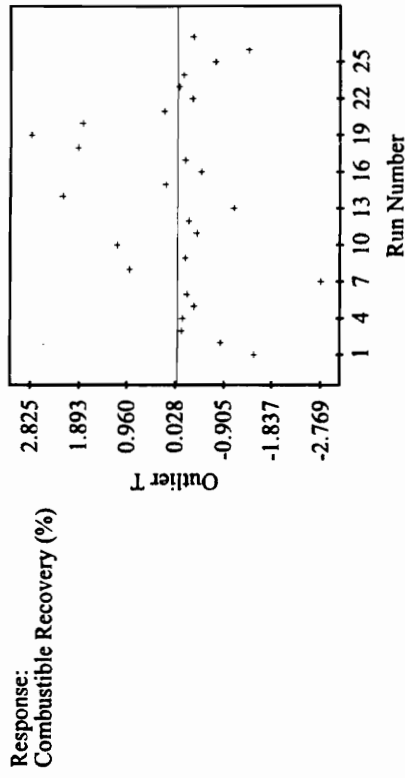
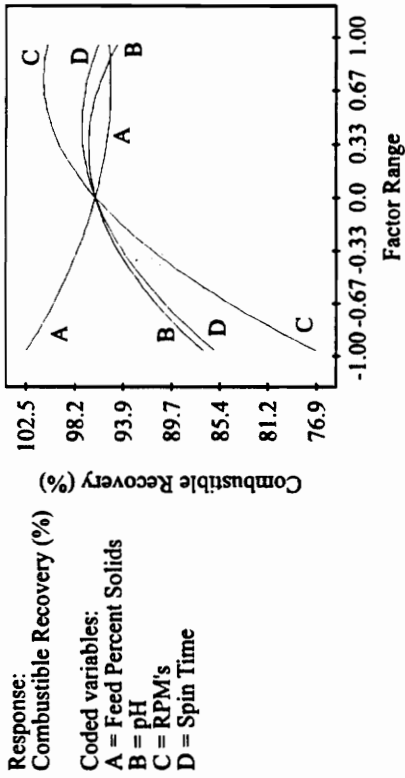
$$\begin{aligned} \text{Combustible Recovery} = & 96.41 - (3.77 * A) + (3.93 * B) + (12.34 * C) \\ & + (5.30 * D) + (2.63 * A^2) - (6.18 * B^2) \\ & - (8.31 * C^2) - (5.88 * D^2) + (4.99 * AB) \\ & + (3.95 * AC) + (4.34 * AD) - (10.56 * BC) \\ & + (4.34 * BD) - (8.48 * CD) \end{aligned}$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & - 841.22 - (39.268 * \% \text{ Solids}) + (135.08 * \text{pH}) \\ & + (0.71849 * \text{RPM's}) + (6.4560 * \text{time}) \\ & + (0.65844 * \% \text{ Solids}^2) - (6.1800 * \text{pH}^2) \\ & - (1.329\text{E-}04 * \text{RPM's}^2) - (0.65333 * \text{time}^2) \\ & + (2.4925 * \% \text{ Solids} * \text{pH}) \\ & + (7.905\text{E-}03 * \% \text{ Solids} * \text{RPM's}) \\ & + (0.72333 * \% \text{ Solids} * \text{time}) \\ & - (4.225\text{E-}02 * \text{pH} * \text{RPM's}) + (1.4458 * \text{pH} * \text{time}) \\ & - (1.131\text{E-}02 * \text{RPM's} * \text{time}) \end{aligned}$$

ANOVA for Quadratic Model (Combustible Recovery / Centrifuge / Elkhorn)

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	98.78	97.70	1.08	0.583	0.249	0.006	0.239	15
2	79.75	80.18	-0.43	0.583	-0.100	0.001	-0.096	24
3	93.96	95.58	-1.62	0.583	-0.373	0.013	-0.359	11
4	94.87	98.00	-3.13	0.583	-0.722	0.049	-0.707	25
5	46.47	56.11	-9.64	0.583	-2.220	0.460	-2.769	7
6	97.69	97.75	-0.06	0.583	-0.013	0.000	-0.012	23
7	81.67	83.67	-2.00	0.583	-0.460	0.020	-0.445	16
8	98.96	91.37	7.59	0.583	1.747	0.285	1.937	18
9	94.61	95.98	-1.37	0.583	-0.316	0.009	-0.304	5
10	78.50	79.76	-1.26	0.583	-0.290	0.008	-0.278	22
11	97.51	97.90	-0.39	0.583	-0.089	0.001	-0.085	4
12	98.76	99.03	-0.27	0.583	-0.062	0.000	-0.060	3
13	49.38	55.10	-5.72	0.583	-1.318	0.162	-1.364	26
14	83.33	84.08	-0.75	0.583	-0.172	0.003	-0.165	6
15	100.00	100.90	-0.90	0.583	-0.207	0.004	-0.198	12
16	91.70	87.62	4.08	0.583	0.939	0.082	0.934	8
17	94.48	86.13	8.35	0.583	1.924	0.345	2.215	14
18	80.43	70.68	9.75	0.583	2.246	0.471	2.825	19
19	96.84	102.89	-6.05	0.583	-1.394	0.181	-1.458	1
20	98.60	103.25	-4.65	0.583	-1.072	0.107	-1.079	13
21	86.77	79.47	7.30	0.583	1.682	0.264	1.841	20
22	83.67	78.64	5.03	0.583	1.158	0.125	1.176	10
23	80.06	81.39	-1.33	0.583	-0.306	0.009	-0.294	27
24	94.31	97.91	-3.60	0.583	-0.830	0.064	-0.818	2
25	95.65	96.41	-0.76	0.333	-0.139	0.001	-0.133	17
26	97.99	96.41	1.58	0.333	0.287	0.003	0.276	21
27	95.60	96.41	-0.81	0.333	-0.148	0.001	-0.142	9



The batch centrifugal separation of coagula formed using Elkhorn No. 3 (50% Ash) seam coal

STATISTICAL DATA ANALYSIS

Description: Centrifugal SedimentationCoal Sample: Elkhorn No. 3 (50% Ash)Response: Separation Efficiency (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	% Solids	(%)	1.000	5.000
B	pH		7.500	9.500
C	RPM's	(rpm)	750.000	1250.000
D	time	(min)	3.000	9.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	9231.1	1	9231.1		
Linear	1429.8	4	357.4	5.185	0.0043
Quadratic	1323.2	10	132.3	8.205	0.0006
Cubic	141.5	8	17.7	1.359	0.4070
RESIDUAL	52.0	4	13.0		
TOTAL	12177.6	27			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	1486.8	20	74.3	4.977	0.1804
Quadratic	163.6	10	16.4	1.096	0.5676
Cubic	22.2	2	11.1	0.7419	0.5741
PURE ERR	29.9	2	14.9		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	22	8.30	0.4853	0.3917	2275.84
Quadratic	15	12	4.02	0.9343	0.8577	1009.79
Cubic	23	4	3.61	0.9823	0.8852	3258.69

ANOVA for Linear Model (Separation Efficiency / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	1429.8	4	357.44	5.185	0.0043
RESIDUAL	1516.7	22	68.94		
Lack Of Fit	1486.8	20	74.34	4.977	0.1804
Pure Error	29.9	2	14.94		
COR TOTAL	2946.4	26			
ROOT MSE	8.30		R-SQUARED	0.4853	
DEP MEAN	18.49		ADJ R-SQUARED	0.3917	
C.V.	44.90%				

Predicted Residual Sum of Squares (PRESS) = 2275.8

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	18.49	1	1.60	11.57	
A	-2.91	1	2.40	-1.213	0.2380
B	8.92	1	2.40	3.724	0.0012
C	-4.22	1	2.40	-1.760	0.0924
D	-3.64	1	2.40	-1.519	0.1431

Final Equation in Terms of Coded Factors:

$$\text{Separation Efficiency} = 18.49 - (2.91 * A) + (8.92 * B) - (4.22 * C) - (3.64 * D)$$

Final Equation in Terms of Actual Factors:

$$\text{Separation Efficiency} = -28.86 - (1.4537 * \% \text{ Solids}) + (8.9250 * \text{pH}) - (1.687\text{E-}02 * \text{RPM's}) - (1.2133 * \text{time})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	4.71	12.47	-7.76	0.204	-1.048	0.056	-1.050	15
2	1.55	6.66	-5.11	0.204	-0.689	0.024	-0.681	24
3	19.29	30.32	-11.03	0.204	-1.489	0.113	-1.534	11
4	18.30	24.51	-6.21	0.204	-0.838	0.036	-0.832	25
5	19.86	26.35	-6.49	0.204	-0.876	0.039	-0.871	7
6	24.88	17.91	6.97	0.204	0.940	0.045	0.938	23
7	29.11	19.07	10.04	0.204	1.355	0.094	1.383	16
8	11.19	10.63	0.56	0.204	0.075	0.000	0.073	18
9	39.86	25.04	14.82	0.204	2.001	0.205	2.161	5
10	24.00	19.22	4.78	0.204	0.645	0.021	0.636	22
11	19.34	17.76	1.58	0.204	0.214	0.002	0.209	4
12	15.24	11.94	3.30	0.204	0.445	0.010	0.437	3
13	3.43	13.78	-10.35	0.204	-1.397	0.100	-1.430	26
14	27.45	31.63	-4.18	0.204	-0.565	0.016	-0.556	6
15	1.95	5.35	-3.40	0.204	-0.459	0.011	-0.450	12
16	14.84	23.20	-8.36	0.204	-1.128	0.065	-1.135	8
17	33.91	25.62	8.29	0.204	1.120	0.064	1.126	14
18	26.19	19.80	6.39	0.204	0.862	0.038	0.857	19
19	19.77	17.18	2.59	0.204	0.350	0.006	0.342	1
20	16.71	11.37	5.34	0.204	0.721	0.027	0.713	13
21	0.91	13.21	-12.30	0.204	-1.659	0.141	-1.733	20
22	27.83	31.06	-3.23	0.204	-0.435	0.010	-0.427	10
23	3.42	5.93	-2.51	0.204	-0.338	0.006	-0.331	27
24	15.36	23.78	-8.42	0.204	-1.136	0.066	-1.144	2
25	27.77	18.49	9.28	0.037	1.139	0.010	1.147	17
26	22.43	18.49	3.94	0.037	0.484	0.002	0.475	21
27	29.94	18.49	11.45	0.037	1.405	0.015	1.439	9

ANOVA for Quadratic Model (Separation Efficiency / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	2752.9	14	196.64	12.19	0.0001
RESIDUAL	193.5	12	16.13		
Lack Of Fit	163.6	10	16.36	1.096	0.5676
Pure Error	29.9	2	14.94		
COR TOTAL	2946.4	26			
ROOT MSE	4.02		R-SQUARED	0.9343	
DEP MEAN	18.49		ADJ R-SQUARED	0.8577	
C.V.	21.72%				

Predicted Residual Sum of Squares (PRESS) = 1009.8

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	26.71	1	2.32	11.52	
A	-2.91	1	1.16	-2.508	0.0275
B	8.92	1	1.16	7.699	0.0001
C	-4.22	1	1.16	-3.638	0.0034
D	-3.64	1	1.16	-3.140	0.0085
A2	-0.96	1	1.74	-0.5523	0.5909
B2	-13.44	1	1.74	-7.729	0.0001
C2	-2.16	1	1.74	-1.241	0.2383
D2	-1.94	1	1.74	-1.118	0.2854
AB	0.54	1	2.01	0.2702	0.7916
AC	1.16	1	2.01	0.5802	0.5725
AD	2.94	1	2.01	1.464	0.1688
BC	-2.78	1	2.01	-1.386	0.1910
BD	-3.74	1	2.01	-1.865	0.0868
CD	-5.73	1	2.01	-2.856	0.0145

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Separation Efficiency} = & 26.71 - (2.91 * A) + (8.92 * B) - (4.22 * C) \\ & - (3.64 * D) - (0.96 * A^2) - (13.44 * B^2) \\ & - (2.16 * C^2) - (1.94 * D^2) + (0.54 * AB) \\ & + (1.16 * AC) + (2.94 * AD) - (2.78 * BC) \\ & - (3.74 * BD) - (5.73 * CD) \end{aligned}$$

Final Equation in Terms of Actual Factors:

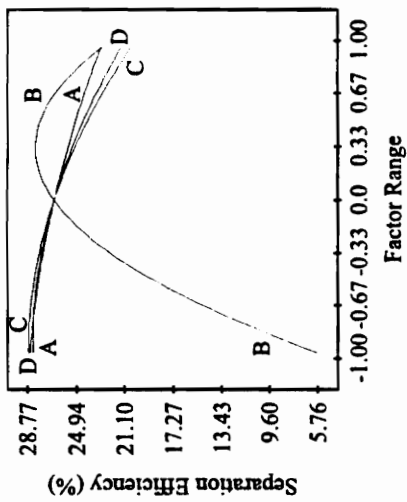
$$\begin{aligned} \text{Separation Efficiency} = & -1217.51 - (7.5888 * \% \text{ Solids}) + (255.20 * \text{pH}) \\ & + (0.18568 * \text{RPM's}) + (18.166 * \text{time}) \\ & - (0.24010 * \% \text{ Solids}^2) - (13.439 * \text{pH}^2) \\ & - (3.453\text{E-}05 * \text{RPM's}^2) - (0.21602 * \text{time}^2) \\ & + (0.27125 * \% \text{ Solids} * \text{pH}) \\ & + (2.330\text{E-}03 * \% \text{ Solids} * \text{RPM's}) \\ & + (0.49000 * \% \text{ Solids} * \text{time}) \\ & - (1.113\text{E-}02 * \text{pH} * \text{RPM's}) \\ & - (1.2483 * \text{pH} * \text{time}) \\ & - (7.647\text{E-}03 * \text{RPM's} * \text{time}) \end{aligned}$$

ANOVA for Quadratic Model (Separation Efficiency / Centrifuge / Elkhorn)

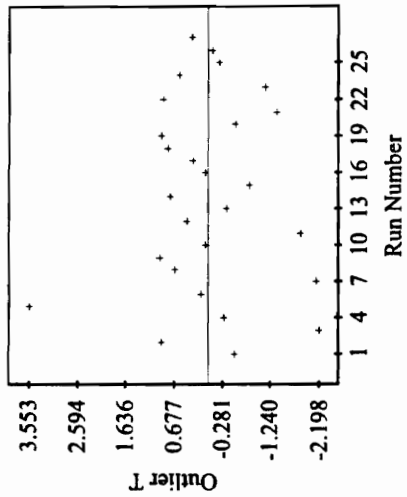
Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	4.71	6.84	-2.13	0.583	-0.821	0.063	-0.809	15
2	1.55	-0.06	1.61	0.583	0.622	0.036	0.605	24
3	19.29	23.60	-4.31	0.583	-1.664	0.258	-1.817	11
4	18.30	18.87	-0.57	0.583	-0.221	0.005	-0.212	25
5	19.86	24.73	-4.87	0.583	-1.880	0.330	-2.143	7
6	24.88	27.77	-2.89	0.583	-1.114	0.116	-1.127	23
7	29.11	28.92	0.19	0.583	0.072	0.000	0.069	16
8	11.19	9.02	2.17	0.583	0.838	0.065	0.826	18
9	39.86	33.30	6.56	0.583	2.532	0.598	3.553	5
10	24.00	21.60	2.40	0.583	0.925	0.080	0.919	22
11	19.34	20.14	-0.80	0.583	-0.307	0.009	-0.295	4
12	15.24	20.20	-4.96	0.583	-1.914	0.342	-2.198	3
13	3.43	3.63	-0.20	0.583	-0.076	0.001	-0.073	26
14	27.45	27.04	0.41	0.583	0.158	0.002	0.151	6
15	1.95	0.76	1.19	0.583	0.461	0.020	0.445	12
16	14.84	13.04	1.80	0.583	0.694	0.045	0.678	8
17	33.91	31.88	2.02	0.583	0.781	0.057	0.768	14
18	26.19	23.74	2.45	0.583	0.945	0.083	0.941	19
19	19.77	21.12	-1.35	0.583	-0.521	0.025	-0.504	1
20	16.71	17.63	-0.92	0.583	-0.357	0.012	-0.343	13
21	0.91	2.30	-1.39	0.583	-0.536	0.027	-0.520	20
22	27.83	27.64	0.19	0.583	0.073	0.001	0.070	10
23	3.42	2.51	0.91	0.583	0.351	0.012	0.338	27
24	15.36	12.87	2.49	0.583	0.961	0.086	0.957	2
25	27.77	26.71	1.06	0.333	0.322	0.003	0.310	17
26	22.43	26.71	-4.28	0.333	-1.306	0.057	-1.350	21
27	29.94	26.71	3.23	0.333	0.984	0.032	0.983	9

Response:
Separation Efficiency (%)

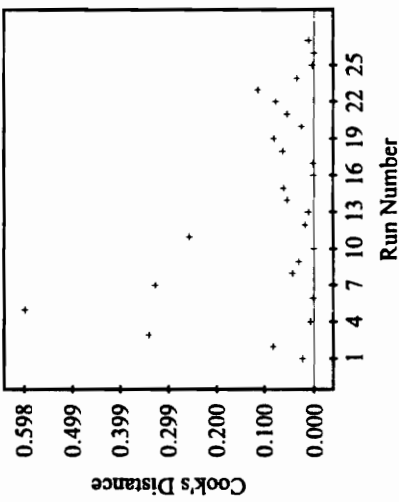
Coded variables:
A = Feed Percent Solids
B = pH
C = RPM's
D = Spin Time



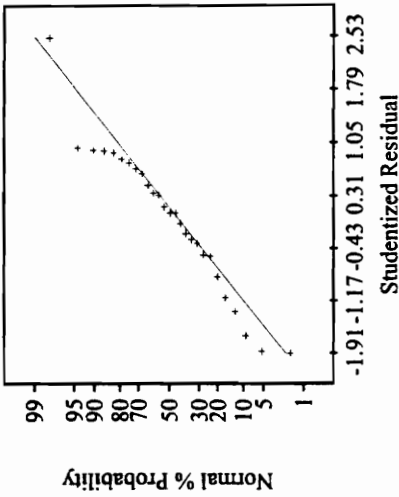
Response:
Separation Efficiency (%)



Response:
Separation Efficiency (%)



Response:
Separation Efficiency (%)



The batch centrifugal separation of coagula formed using Elkhorn No. 3 (50% Ash) seam coal

STATISTICAL DATA ANALYSIS

Description: Centrifugal SedimentationCoal Sample: Elkhorn No. 3 (50% Ash)Response: Ash Rejection (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	% Solids	(%)	1.000	5.000
B	pH		7.500	9.500
C	RPM's	(rpm)	750.000	1250.000
D	time	(min)	3.000	9.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	24243.6	1	24243.6		
Linear	4531.6	4	1132.9	11.34	0.0001
Quadratic	1609.3	10	160.9	3.277	0.0277
Cubic	498.2	8	62.3	2.735	0.1733
RESIDUAL	91.1	4	22.8		
TOTAL	30973.9	27			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	2144.7	20	107.2	3.974	0.2201
Quadratic	535.4	10	53.5	1.984	0.3814
Cubic	37.1	2	18.6	0.6879	0.5925
PURE ERR	54.0	2	27.0		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	22	10.00	0.6733	0.6139	3440.37
Quadratic	15	12	7.01	0.9124	0.8103	3205.19
Cubic	23	4	4.77	0.9865	0.9120	5467.73

ANOVA for Linear Model (Ash Rejection / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	4531.6	4	1132.90	11.34	0.0001
RESIDUAL	2198.7	22	99.94		
Lack Of Fit	2144.7	20	107.23	3.974	0.2201
Pure Error	54.0	2	26.99		
COR TOTAL	6730.3	26			
ROOT MSE	10.00		R-SQUARED	0.6733	
DEP MEAN	29.97		ADJ R-SQUARED	0.6139	
C.V.	33.36%				

Predicted Residual Sum of Squares (PRESS) = 3440.4

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0	PROB > t
Intercept	29.97	1	1.92	15.58	
A	0.82	1	2.89	0.2824	0.7803
B	5.17	1	2.89	1.791	0.0871
C	-16.46	1	2.89	-5.704	0.0001
D	-8.90	1	2.89	-3.085	0.0054

Final Equation in Terms of Coded Factors:

$$\text{Ash Rejection} = 29.97 + (0.82 * A) + (5.17 * B) - (16.46 * C) - (8.90 * D)$$

Final Equation in Terms of Actual Factors:

$$\text{Ash Rejection} = 68.47 + (0.40750 * \% \text{ Solids}) + (5.1683 * \text{pH}) - (6.585E-02 * \text{RPM's}) - (2.9678 * \text{time})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	5.93	23.98	-18.05	0.204	-2.024	0.210	-2.191	15
2	21.79	25.61	-3.82	0.204	-0.428	0.009	-0.420	24
3	25.34	34.32	-8.98	0.204	-1.006	0.052	-1.007	11
4	24.09	35.95	-11.86	0.204	-1.329	0.090	-1.354	25
5	73.48	55.33	18.15	0.204	2.035	0.212	2.206	7
6	27.19	22.41	4.78	0.204	0.536	0.015	0.527	23
7	48.25	37.52	10.73	0.204	1.202	0.074	1.215	16
8	12.23	4.60	7.63	0.204	0.855	0.037	0.850	18
9	45.25	38.05	7.20	0.204	0.807	0.033	0.800	5
10	45.61	39.68	5.93	0.204	0.664	0.023	0.656	22
11	21.83	20.25	1.58	0.204	0.177	0.002	0.174	4
12	16.48	21.88	-5.40	0.204	-0.605	0.019	-0.596	3
13	53.87	41.26	12.61	0.204	1.414	0.102	1.449	26
14	43.80	51.60	-7.80	0.204	-0.874	0.039	-0.869	6
15	2.10	8.34	-6.24	0.204	-0.699	0.025	-0.691	12
16	23.14	18.67	4.47	0.204	0.501	0.013	0.492	8
17	39.43	45.61	-6.18	0.204	-0.693	0.025	-0.685	14
18	44.41	47.24	-2.83	0.204	-0.317	0.005	-0.311	19
19	22.93	12.69	10.24	0.204	1.148	0.067	1.157	1
20	18.11	14.32	3.79	0.204	0.425	0.009	0.417	13
21	13.43	33.70	-20.27	0.204	-2.272	0.264	-2.538	20
22	45.07	44.04	1.03	0.204	0.116	0.001	0.113	10
23	23.35	15.89	7.46	0.204	0.836	0.036	0.830	27
24	21.05	26.23	-5.18	0.204	-0.581	0.017	-0.572	2
25	32.12	29.97	2.15	0.037	0.220	0.000	0.215	17
26	24.44	29.97	-5.53	0.037	-0.563	0.002	-0.554	21
27	34.34	29.97	4.37	0.037	0.446	0.002	0.438	9

ANOVA for Quadratic Model (Ash Rejection / Centrifuge / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	6140.9	14	438.64	8.931	0.0003
RESIDUAL	589.3	12	49.11		
Lack Of Fit	535.4	10	53.54	1.984	0.3814
Pure Error	54.0	2	26.99		
COR TOTAL	6730.3	26			
ROOT MSE	7.01		R-SQUARED	0.9124	
DEP MEAN	29.97		ADJ R-SQUARED	0.8103	
C.V.	23.39%				

Predicted Residual Sum of Squares (PRESS) = 3205.2

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	30.30	1	4.05	7.489	
A	0.82	1	2.02	0.4029	0.6941
B	5.17	1	2.02	2.555	0.0252
C	-16.46	1	2.02	-8.137	0.0001
D	-8.90	1	2.02	-4.401	0.0009
A2	-3.67	1	3.03	-1.210	0.2494
B2	-7.20	1	3.03	-2.374	0.0352
C2	6.04	1	3.03	1.992	0.0697
D2	4.08	1	3.03	1.344	0.2037
AB	-4.28	1	3.50	-1.221	0.2456
AC	-2.45	1	3.50	-0.6992	0.4978
AD	-1.43	1	3.50	-0.4074	0.6909
BC	7.78	1	3.50	2.220	0.0465
BD	-8.48	1	3.50	-2.422	0.0322
CD	2.57	1	3.50	0.7327	0.4778

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Ash Rejection} = & 30.30 + (0.82 * A) + (5.17 * B) - (16.46 * C) \\ & - (8.90 * D) - (3.67 * A^2) - (7.20 * B^2) \\ & + (6.04 * C^2) + (4.08 * D^2) - (4.28 * AB) \\ & - (2.45 * AC) - (1.43 * AD) + (7.78 * BC) \\ & - (8.48 * BD) + (2.57 * CD) \end{aligned}$$

Final Equation in Terms of Actual Factors:

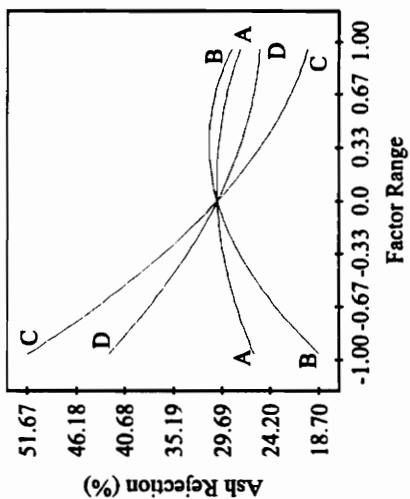
$$\begin{aligned} \text{Ash Rejection} = & -279.67 + (30.424 * \% \text{ Solids}) + (119.90 * \text{pH}) \\ & - (0.52953 * \text{RPM}'s) + (12.925 * \text{time}) - (0.91833 * \% \text{ Solids}^2) \\ & - (7.2033 * \text{pH}^2) + (9.671\text{E-}05 * \text{RPM}'s^2) + (0.45324 * \text{time}^2) \\ & - (2.1388 * \% \text{ Solids} * \text{pH}) - (4.900\text{E-}03 * \% \text{ Solids} * \text{RPM}'s) \\ & - (0.23792 * \% \text{ Solids} * \text{time}) + (3.111\text{E-}02 * \text{pH} * \text{RPM}'s) \\ & - (2.8283 * \text{pH} * \text{time}) + (3.423\text{E-}03 * \text{RPM}'s * \text{time}) \end{aligned}$$

ANOVA for Quadratic Model (Ash Rejection / Centrifuge / Elkhorn)

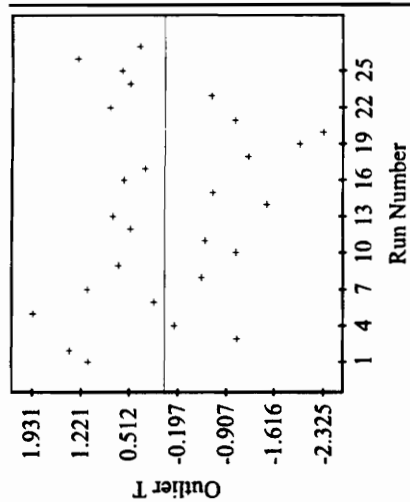
Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	5.93	9.16	-3.23	0.583	-0.715	0.048	-0.699	15
2	21.79	19.35	2.44	0.583	0.540	0.027	0.523	24
3	25.34	28.05	-2.71	0.583	-0.600	0.034	-0.583	11
4	24.09	21.13	2.96	0.583	0.655	0.040	0.638	25
5	73.48	68.36	5.12	0.583	1.133	0.120	1.148	7
6	27.19	30.30	-3.11	0.583	-0.687	0.044	-0.671	23
7	48.25	45.41	2.84	0.583	0.627	0.037	0.610	16
8	12.23	17.63	-5.40	0.583	-1.193	0.133	-1.216	18
9	45.25	37.37	7.88	0.583	1.743	0.283	1.931	5
10	45.61	41.85	3.76	0.583	0.831	0.064	0.819	22
11	21.83	22.41	-0.58	0.583	-0.129	0.002	-0.124	4
12	16.48	21.19	-4.71	0.583	-1.041	0.101	-1.045	3
13	53.87	48.21	5.66	0.583	1.251	0.146	1.284	26
14	43.80	42.99	0.81	0.583	0.178	0.003	0.171	6
15	2.10	-0.27	2.37	0.583	0.523	0.026	0.507	12
16	23.14	25.63	-2.49	0.583	-0.549	0.028	-0.533	8
17	39.43	45.87	-6.44	0.583	-1.423	0.189	-1.494	14
18	44.41	52.40	-7.99	0.583	-1.766	0.291	-1.965	19
19	22.93	17.84	5.09	0.583	1.124	0.118	1.138	1
20	18.11	14.57	3.54	0.583	0.782	0.057	0.768	13
21	13.43	22.43	-9.00	0.583	-1.989	0.369	-2.325	20
22	45.07	49.73	-4.66	0.583	-1.031	0.099	-1.034	10
23	23.35	21.59	1.76	0.583	0.389	0.014	0.375	27
24	21.05	14.96	6.09	0.583	1.347	0.169	1.400	2
25	32.12	30.30	1.82	0.333	0.318	0.003	0.306	17
26	24.44	30.30	-5.86	0.333	-1.024	0.035	-1.026	21
27	34.34	30.30	4.04	0.333	0.706	0.017	0.690	9

Response:
Ash Rejection (%)

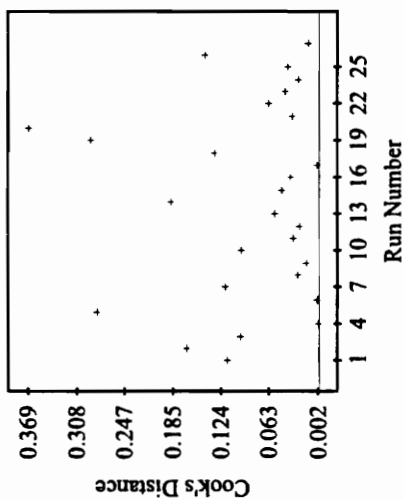
Coded variables:
A = Feed Percent Solids
B = pH
C = RPM's
D = Spin Time



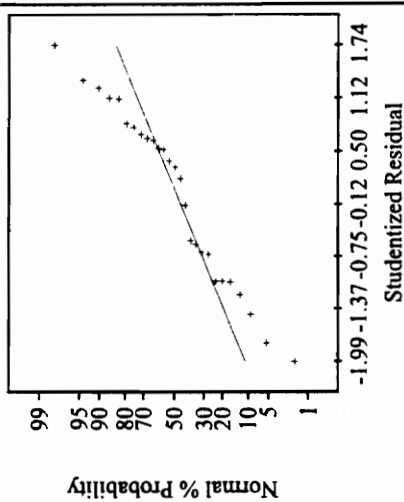
Response:
Ash Rejection (%)



Response:
Ash Rejection (%)



Response:
Ash Rejection (%)



The batch centrifugal separation of coagula formed using Elkhorn No. 3 (50% Ash) seam coal

APPENDIX E - FILTER LEAF BATCH DATA

Round 1 filter leaf tests using Elkhorn No. 3 seam coal

Test Number	Filter Medium	Cycle Time (sec)	Stage Number	pH	EDTA	% Solids		Form		Dry		Wash Water			Dischrg Press (psi)	Total air thru cycle (CFM)
						Feed (%)	Tails (%)	Time (sec)	Vacuum (mm Hg)	Time (sec)	Vacuum (mm Hg)	Time (sec)	Rate (ml/min)	Vacuum (mm Hg)		
1	PO-801 HF	180	1	7.5	0.4	15.0		60	80			0				
2	NY-413	180	1	7.5	0.4	15.0		60	80			0				
3	PO-801 HF	180	1	7.5	0.4	10.0		60	60	90		0				
4	PO-801 RF	180	1	7.6	0.4	10.0		60	60			0				
5	SA-601	180	1	7.5	0.4	10.0		60	70			0				
6	POPR-852 F	180	1	7.5	0.4	9.8		60	80	90	50	0				
		180	2	7.5	0			60	90	90	50	0				
7	PO-8-1 RF	180	1	7.5	0.4	5.0		60	60			0				
8	NY-413	180	1	7.5	0.4	10.0		60	60			0				
9	NY-413	180	1	7.5	0.4	5.0		60		90		0				
10	PO-801 HF	180	1	7.5	0.4	5.0		60	55	90		0				
11	SA-601	180	1	7.5	0.4	5.0		60	60	90		0				
12	POPR-852 F	?	1	7.6	0.4	4.5		?	60	90		0				
								(ml/min)		(ml/min)						
13	POPR-852 F		1	7.5	0.4	4.7	4.2	470	50		50	0				
14	POPR-852 F		1	7.7	0.4	4.7	4.3	180	50		50	0				
15	POPR-852 F		1	7.6	0.4	5.0	4.0	180	50		50	0				

Round 1 filter leaf tests using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)			Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Sulfur Rejection (%)	Separation Efficiency (%)
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)					
1														
2														
3														
4														
5														
6	47.88	45.85	47.91	52.12	54.15	52.09				1.46	1.51	1.51	0.12	
	47.91	46.92	47.48	52.09	53.08	52.52				-76.79	-78.25	-78.25	-3.05	
7														
8														
9														
10														
11														
12	47.78	62.46	49.80	52.22	37.54	50.20				-15.96	-11.47	-11.47	9.39	
13	51.39	66.85	50.58	48.61	33.15	49.42				4.98	3.40	3.40	-3.08	
14	51.46	59.22	50.88	48.54	40.78	49.12				6.95	5.84	5.84	-2.16	
15	51.60		54.86	48.40		45.14								

Round II filter leaf tests using Elkhorn No. 3 seam coal

Test Number	Filter Media	%Solids		Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Separation Efficiency (%)	% Ash			% Combustible		
		Feed (%)	Tails (%)					Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1-step	POPR-852F	4.73	3.89	19.16	20.45	82.08	2.53	51.17	47.80	51.88	48.83	52.20	48.12
2	POPR-852F	4.65	0.78	83.56	85.11	18.16	3.27	51.04	46.33	52.27	48.96	53.67	47.73
3-twice	POPR-852F	4.65	1.51	70.56	76.51	35.17	11.68	51.04	46.79	60.84	48.96	53.21	39.16
4	PO 801 RF		4.67		0.00	100.00	0.00						
5	SA 601		4.77		0.00	100.00	0.00						
6	NY 413		4.72		0.00	100.00	0.00						
7	PO 801 HF		4.57		0.00	100.00	0.00						
8	P 851	4.69	4.03	8.47	9.30	92.37	1.68	49.23	44.38	49.78	50.77	55.62	50.22
9	P 854 F	4.69	2.28	51.68	52.63	49.30	1.93	49.23	48.18	50.11	50.77	51.82	49.89
10	P 858	4.59	3.42	22.05	23.91	79.88	3.78	49.17	44.80	50.30	50.83	55.20	49.70
11	P 859	4.59						49.17			50.83		

Round II filter leaf tests using Pittsburgh No. 8 seam coal

Test Number	Filter Media	%Solids		Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Separation Efficiency (%)	% Ash			% Combustible		
		Feed (%)	Tails (%)					Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)
1	POPR-852F		1.01	81.65	81.71	18.65	0.36	17.60	17.54	17.89	82.40	82.46	82.11
2	P 851	5.13	3.30	36.45	36.39	63.30	-0.31	17.48	17.60	17.41	82.52	82.40	82.59
3	SA 601		4.88	0.00	0.00	100.00	0.00	17.51	0.00	17.51	82.49	0.00	82.49
4	P 859		4.65	2.55	2.54	97.44	-0.02	17.57	17.69	17.57	82.43	82.31	82.43
5	P 854 F		0.01	99.79	99.80	0.27	0.07	17.59	17.58	22.22	82.41	82.42	77.78
6	P 858	5.07	1.53	70.82	70.78	29.00	-0.21	17.56	17.60	17.45	82.44	82.40	82.55
7	PO 801 HF		2.42	55.31	55.30	44.65	-0.05	17.65	17.66	17.63	82.35	82.34	82.37
8	PO 801 RF	5.17	4.46	10.38	10.23	88.89	-0.89	17.59	18.83	17.45	82.41	81.17	82.55
9	NY 413		4.97	0.00	0.00	100.00	0.00	17.54	0.00	17.54	82.46	0.00	82.46

APPENDIX F - COLUMN FLOTATION DATA

Column flotation parametric test data using Elkhorn No. 3 seam coal

Test No.	Flow Rates					Bias Factor	Bias Rate (cm/sec)	% Solids		Mass Flow Rates		
	Wash Water (ml/min)	Feed (ml/min)	Aeration (cc/min)	Frother (ml/min)	Tailings (ml/min)			Feed (%)	Tails (%)	Feed (g/min)	Product (g/min)	Tails (g/min)
1	600	150	1000	15	740	0.983	0.485	3.70	0.66	5.55	0.43	4.63
2	200	150	2000	38	280	0.650	0.107	10.00	3.95	15.00	4.04	10.00
3	200	250	1000	25	350	0.500	0.082	3.70	1.51	9.25	2.53	4.55
4	600	250	2000	25	710	0.767	0.378	3.70	0.96	9.25	1.67	6.45
5	600	250	2000	63	830	0.967	0.477	10.00	2.72	25.00	2.13	22.00
6	400	200	1500	35	600	1.000	0.329	6.75	1.86	13.81	1.18	10.52
7	200	250	2000	63	440	0.950	0.156	10.00	4.71	25.00	3.34	19.52
8	200	150	1000	15	340	0.950	0.156	3.70	1.17	8.62	1.54	3.58
9	200	250	1000	63	400	0.750	0.123	9.47	5.15	23.68	4.92	19.16
10	400	200	1500	35	590	0.975	0.321	6.73	2.27	13.81	1.18	12.49
11	600	250	1000	63	840	0.983	0.485	9.47	2.80	23.68	1.24	23.04
12	400	200	1500	35	590	0.975	0.321	6.75	1.87	13.81	1.14	10.39
13	200	150	1000	38	210	0.300	0.049	9.40	4.32	14.30	4.26	9.84
14	600	250	1000	25	750	0.833	0.411	3.70	0.91	9.25	0.80	6.57
15	400	200	1500	35	580	0.950	0.312	6.75	1.86	13.81	1.31	10.27
16	600	150	2000	15	680	0.883	0.436	3.70	0.66	8.62	0.85	4.26
17	600	150	1000	38	710	0.933	0.461	9.47	2.05	14.66	0.99	13.69
18	200	250	2000	25	460	1.050	0.173	3.65	nm	9.44	1.33	8.09
19	600	150	2000	38	650	0.833	0.411	9.40	2.10	14.30	1.32	13.35
20	200	150	2000	15	320	0.850	0.140	3.70	1.15	8.62	1.83	3.35

nm - Not Measured

Column flotation parametric test data using Elkhorn No. 3 seam coal

Test No.	% Ash (Adjusted)			% Combustible (Adjusted)			% Sulfur (Adjusted)			Combustibility		Rejection		Separation Efficiency (%)
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Yield (%)	Recovery (%)	Ash (%)	Sulfur (%)	
1	49.04	7.11	54.99	50.96	92.89	45.01	0.34	0.52	0.32	12.43	22.65	98.20	84.71	20.85
2	50.43	12.04	69.37	49.57	87.96	30.63	0.35	0.54	0.26	33.04	58.62	92.11	50.41	50.74
3	50.06	10.90	83.73	49.94	89.10	16.27	0.35	0.53	0.19	46.23	82.48	89.93	28.74	72.42
4	50.26	8.86	60.04	49.74	91.14	39.96	0.35	0.56	0.30	19.11	35.01	96.63	69.23	31.65
5	51.08	9.06	52.85	48.92	90.94	47.15	0.36	0.57	0.35	4.04	7.51	99.28	92.80	6.80
6	50.66	8.94	54.70	49.34	91.06	45.30	0.36	0.56	0.34	8.83	16.29	98.44	85.86	14.74
7	49.63	12.99	56.65	50.37	87.01	43.35	0.35	0.54	0.31	16.08	27.77	95.79	73.17	23.57
8	50.57	9.39	64.22	49.43	90.61	35.78	0.35	0.55	0.29	24.90	45.64	95.38	63.74	41.01
9	49.87	14.17	58.22	50.13	85.83	41.78	0.41	0.59	0.36	18.96	32.46	94.61	68.72	27.07
10	50.18	9.02	53.45	49.82	90.98	46.55	0.35	0.57	0.34	7.36	13.44	98.68	92.92	12.12
11	49.13	9.90	52.14	50.87	90.10	47.86	0.40	0.61	0.38	7.13	12.62	98.56	86.74	11.19
12	50.26	9.00	55.22	49.74	91.00	44.78	0.35	0.56	0.33	10.73	19.63	98.08	86.09	17.71
13	50.50	13.35	64.37	49.50	86.65	35.63	0.36	0.56	0.28	27.19	47.59	92.81	55.56	40.40
14	49.71	8.23	57.96	50.29	91.77	42.04	0.34	0.56	0.30	16.59	30.27	97.25	74.66	27.53
15	50.13	9.15	55.28	49.87	90.85	44.72	0.35	0.55	0.33	11.16	20.34	97.96	85.71	18.30
16	50.79	8.06	57.49	49.21	91.94	42.51	0.35	0.58	0.32	13.55	25.32	97.85	80.88	23.17
17	49.61	9.19	52.60	50.39	90.81	47.40	0.40	0.58	0.39	6.89	12.41	98.72	92.37	11.14
18	49.60	9.25	55.54	50.40	90.75	44.46	0.37	0.56	0.34	12.83	23.11	97.61	79.36	20.71
19	50.65	8.65	53.88	49.35	91.35	46.12	0.36	0.57	0.34	7.14	13.22	98.78	86.23	12.00
20	50.74	10.03	69.18	49.26	89.97	30.82	0.35	0.56	0.26	31.17	56.94	93.84	52.00	50.78

Column flotation parametric test data using Pittsburgh No. 8 seam coal

Test No.	Flow Rates						Bias Factor	Bias Rate (cm/sec)	% Solids		Mass Flow Rates		
	Wash Water (ml/min)	Feed (ml/min)	Aeration (cc/min)	Frother (lb/ton)	(ml/min)	Tailings (ml/min)			Feed (%)	Tails (%)	Feed (g/min)	Product (g/min)	Tails (g/min)
1	600	250	1000	5	31	680	0.72	0.35	3.61	0.76	9.00	4.00	5.33
2	200	256	2000	4	63	0	-1.28	-0.21	9.75	0.00	25.13	23.42	0.00
3	600	150	1000	5	47	690	0.90	0.44	9.97	1.94	15.00	4.00	13.13
4	600	250	1000	5	78	800	0.92	0.45	9.97	2.92	26.77	4.81	22.37
5	200	158	2000	4	38	0	-0.79	-0.13	9.80	0.00	15.77	15.06	0.00
6	200	250	1000	5	78	380	0.65	0.11	9.97	5.71	25.00	6.01	19.72
7	600	150	2000	5	47	520	0.62	0.30	10.00	2.16	15.00	6.43	10.88
8	200	250	1000	5	31	150	-0.50	-0.08	3.61	0.41	9.02	9.24	0.29
9	600	250	2000	5	78	540	0.48	0.24	10.00	3.66	25.00	7.63	19.71
10	600	150	1000	5	19	700	0.92	0.45	3.61	0.51	6.24	2.36	3.55
11	400	200	1500	5	44	480	0.70	0.23	6.81	2.10	14.47	6.56	9.20
12	200	250	2000	4	25	0	-1.25	-0.21	3.93	0.00	10.09	11.16	0.00
13	400	200	1500	5	44	440	0.60	0.20	6.81	2.13	14.47	5.24	9.33
14	400	200	1500	5	44	490	0.73	0.24	6.81	2.11	14.47	5.29	9.18
15	200	150	1000	5	47	250	0.50	0.08	9.97	4.18	15.00	5.62	10.29
16	600	250	2000	4	25	760	0.85	0.42	3.93	0.87	10.09	3.35	6.53
17	200	150	1000	5	19	150	0.00	0.00	3.61	0.42	6.24	6.96	0.48
18	400	200	1500	5	44	480	0.65	0.21	6.81	2.10	14.47	5.34	9.10
19	600	158	2000	4	15	720	0.94	0.46	3.93	0.54	6.44	2.50	3.81
20	200	168	2000	4	15	260	0.46	0.08	3.93	0.93	6.68	4.16	2.29

Column flotation parametric test data using Pittsburgh No. 8 seam coal

Test No.	% Ash (Adjusted)			% Combustible (Adjusted)			% Sulfur (Adjusted)			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulfur (%)	
1	14.70	3.34	23.96	85.30	96.66	76.04	3.51	2.30	4.50	44.91	50.89	89.80	70.51	40.68
2	16.81	16.26	0.00	83.19	83.74	0.00	4.01	3.87	0.00	99.34	100.00	0.00	0.00	0.00
3	16.67	3.37	20.10	83.33	96.63	79.90	3.54	2.29	3.86	20.50	23.77	95.86	86.81	19.63
4	16.72	3.44	18.93	83.28	96.56	81.07	3.57	2.31	3.78	14.27	16.54	97.06	90.76	13.61
5	16.78	16.14	0.00	83.22	83.86	0.00	4.04	3.89	0.00	99.24	100.00	0.00	0.00	0.00
6	16.64	4.88	19.76	83.36	95.12	80.24	3.59	2.54	3.87	20.97	23.93	93.85	85.10	17.78
7	16.00	3.45	21.88	84.00	96.55	78.12	3.62	2.36	4.22	31.90	36.67	93.12	78.97	29.79
8	14.58	10.81	78.77	85.42	89.19	21.23	3.54	3.20	9.25	94.45	98.62	29.97	14.68	28.59
9	15.98	4.06	19.36	84.02	95.94	80.64	3.63	2.44	3.97	22.09	25.23	94.39	85.06	19.61
10	14.62	3.00	23.84	85.38	97.00	76.16	3.53	2.25	4.55	44.24	50.26	90.92	71.73	41.18
11	16.80	3.64	23.58	83.20	96.36	76.42	3.73	2.42	4.40	34.00	39.38	92.63	78.05	32.01
12	16.11	13.17	0.00	83.89	86.83	0.00	4.22	3.45	0.00	96.61	100.00	0.00	0.00	0.00
13	16.62	3.80	23.94	83.38	96.20	76.06	3.77	2.43	4.54	36.35	41.93	91.69	76.48	33.62
14	16.73	3.87	24.10	83.27	96.13	75.90	3.75	2.42	4.51	36.43	42.06	91.57	76.53	33.63
15	16.64	4.45	21.96	83.36	95.55	78.04	3.59	2.38	4.12	30.38	34.83	91.87	79.81	26.70
16	16.68	3.41	23.62	83.32	96.59	76.38	4.09	2.37	5.00	34.34	39.81	92.98	79.95	32.79
17	14.57	6.58	50.72	85.43	93.42	49.28	3.54	2.76	7.10	81.90	89.56	63.01	36.05	52.57
18	16.75	3.86	24.03	83.25	96.14	75.97	3.74	2.40	4.50	36.09	41.68	91.68	76.78	33.36
19	16.56	3.38	24.93	83.44	96.62	75.07	4.03	2.31	5.11	38.84	44.97	92.07	77.89	37.05
20	16.50	4.47	37.09	83.50	95.53	62.91	4.01	2.59	6.43	63.12	72.21	82.90	59.30	55.11

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Elkhorn No. 3
 Response: Combustible Recovery

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	18199.8	1	18199.8		
Linear	3804.7	4	951.2	4.364	0.0154
Quadratic	2111.5	7	301.6	2.084	0.1625
Cubic	879.7	4	219.9	3.163	0.1454
RESIDUAL	278.1	4	69.5		
TOTAL	25273.7	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	3238.8	12	269.9	26.52	0.0103
Quadratic	1127.3	5	225.5	22.16	0.0142
Cubic	247.6	1	247.6	24.33	0.0160
PURE ERR	30.5	3	10.2		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	14.76	0.5378	0.4146	6037.87
Quadratic	12	8	12.03	0.8363	0.6113	11597.81
Cubic	16	4	8.34	0.9607	0.8133	63437.37

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	3804.7	4	951.17	4.364	0.0154
RESIDUAL	3269.3	15	217.95		
Lack Of Fit	3238.8	12	269.90	26.52	0.0103
Pure Error	30.5	3	10.18		
COR TOTAL	7074.0	19			
ROOT MSE	14.76		R-SQUARED	0.5378	
DEP MEAN	30.17		ADJ R-SQUARED	0.4146	
C.V.	48.94%				

Predicted Residual Sum of Squares (PRESS) = 6037.9

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	30.17	1	3.30	9.138	
A	-13.48	1	3.69	-3.651	0.0024
B	-1.95	1	3.69	-0.5277	0.6055
C	-6.83	1	3.69	-1.850	0.0842
D	-2.41	1	3.69	-0.6540	0.5230

Final Equation in Terms of Coded Factors:

$$\text{Recovery} = 30.17 - 13.48 * A - 1.95 * B - 6.83 * C - 2.41 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Recovery} = 88.08 - 6.738E-02 * \text{Water} - 3.895E-02 * \text{Feed} - 2.2754 * \text{Solids} - 4.828E-03 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	45.64	54.83	-9.19	0.300	-0.744	0.047	-0.732	8
2	22.65	27.88	-5.23	0.300	-0.423	0.015	-0.411	1
3	82.48	50.93	31.55	0.300	2.554	0.559	3.282	3
4	30.27	23.98	6.29	0.300	0.509	0.022	0.496	14
5	47.59	41.18	6.41	0.300	0.519	0.023	0.506	13
6	12.41	14.23	-1.82	0.300	-0.147	0.002	-0.142	17
7	32.46	37.28	-4.82	0.300	-0.390	0.013	-0.379	9
8	12.62	10.33	2.29	0.300	0.185	0.003	0.179	11
9	56.94	50.00	6.94	0.300	0.562	0.027	0.549	20
10	25.32	23.05	2.27	0.300	0.184	0.003	0.178	16
11	23.11	46.11	-23.00	0.300	-1.862	0.297	-2.051	18
12	35.01	19.16	15.85	0.300	1.284	0.141	1.314	4
13	58.62	36.35	22.27	0.300	1.803	0.279	1.968	2
14	13.22	9.40	3.82	0.300	0.309	0.008	0.300	19
15	27.77	32.45	-4.68	0.300	-0.379	0.012	-0.368	7
16	7.51	5.50	2.01	0.300	0.162	0.002	0.157	5
17	16.29	30.17	-13.88	0.050	-0.964	0.010	-0.962	6
18	13.44	30.17	-16.73	0.050	-1.162	0.014	-1.177	10
19	19.63	30.17	-10.54	0.050	-0.732	0.006	-0.720	12
20	20.34	30.17	-9.83	0.050	-0.683	0.005	-0.670	15

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	5916.1	11	537.83	3.716	0.0365
RESIDUAL	1157.8	8	144.73		
Lack Of Fit	1127.3	5	225.46	22.16	0.0142
Pure Error	30.5	3	10.18		
COR TOTAL	7074.0	19			
ROOT MSE	12.03		R-SQUARED	0.8363	
DEP MEAN	30.17		ADJ R-SQUARED	0.6113	
C.V.	39.88%				

Predicted Residual Sum of Squares (PRESS) = 11597.8

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	17.43	1	6.02	2.897	
A	-13.48	1	3.01	-4.480	0.0021
B	-1.95	1	3.01	-0.6475	0.5354
C	-6.83	1	3.01	-2.270	0.0529
D	-2.41	1	3.01	-0.8026	0.4454
A2	15.93	1	6.73	2.368	0.0454
B2	ALIASED	0			
C2	ALIASED	0			
D2	ALIASED	0			
AB	3.42	1	3.01	1.138	0.2879
AC	-1.61	1	3.01	-0.5353	0.6070
AD	2.80	1	3.01	0.9318	0.3787
BC	-4.49	1	3.01	-1.492	0.1740
BD	-5.64	1	3.01	-1.875	0.0976
CD	2.67	1	3.01	0.8873	0.4008

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Recovery} = & 17.43 - 13.48 * A - 1.95 * B - 6.83 * C - 2.41 * D \\ & + 15.93 * A2 + 0.00 * B2 + 0.00 * C2 + 0.00 * D2 \\ & + 3.42 * AB - 1.61 * AC + 2.80 * AD - 4.49 * BC \\ & - 5.64 * BD + 2.67 * CD \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	45.64	55.17	-9.53	0.687	-1.417	0.368	-1.532	8
2	22.65	18.99	3.66	0.687	0.544	0.054	0.519	1
3	82.48	64.68	17.80	0.687	2.646	1.284	7.010	3
4	30.27	42.20	-11.93	0.687	-1.773	0.577	-2.129	14
5	47.59	48.38	-0.79	0.687	-0.117	0.003	-0.109	13
6	12.41	5.75	6.66	0.687	0.990	0.180	0.988	17
7	32.46	39.94	-7.48	0.687	-1.112	0.227	-1.131	9
8	12.62	11.01	1.61	0.687	0.239	0.010	0.225	11
9	56.94	50.68	6.26	0.687	0.931	0.159	0.922	20
10	25.32	25.71	-0.39	0.687	-0.058	0.001	-0.054	16
11	23.11	37.63	-14.52	0.687	-2.160	0.855	-3.128	18
12	35.01	26.36	8.65	0.687	1.287	0.304	1.352	4
13	58.62	54.56	4.06	0.687	0.604	0.067	0.578	2
14	13.22	23.15	-9.93	0.687	-1.476	0.400	-1.619	19
15	27.77	23.56	4.21	0.687	0.625	0.072	0.600	7
16	7.51	5.85	1.66	0.687	0.247	0.011	0.232	5
17	16.29	17.43	-1.14	0.250	-0.109	0.000	-0.102	6
18	13.44	17.43	-3.99	0.250	-0.382	0.004	-0.361	10
19	19.63	17.43	2.20	0.250	0.212	0.001	0.199	12
20	20.34	17.43	2.91	0.250	0.280	0.002	0.263	15

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Elkhorn No. 3
 Response: Concentrate Ash

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	1946.56	1	1946.56		
Linear	55.05	4	13.76	17.48	0.0001
Quadratic	10.56	7	1.51	9.603	0.0024
Cubic	0.87	4	0.22	2.276	0.2227
RESIDUAL	0.38	4	0.10		
TOTAL	2013.43	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	11.79	12	0.98	125.5	0.0010
Quadratic	1.23	5	0.25	31.51	0.0085
Cubic	0.36	1	0.36	46.01	0.0066
PURE ERR	0.02	3	0.01		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	0.887	0.8234	0.7763	21.462
Quadratic	12	8	0.396	0.9812	0.9554	12.665
Cubic	16	4	0.310	0.9943	0.9728	92.202

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	55.05	4	13.764	17.48	0.0001
RESIDUAL	11.81	15	0.787		
Lack Of Fit	11.79	12	0.982	125.5	0.0010
Pure Error	0.02	3	0.008		
COR TOTAL	66.87	19			
ROOT MSE	0.887		R-SQUARED	0.8234	
DEP MEAN	9.865		ADJ R-SQUARED	0.7763	
C.V.	8.99%				

Predicted Residual Sum of Squares (PRESS) = 21.46

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	9.865	1	0.198	49.72	
A	-1.443	1	0.222	-6.502	0.0001
B	0.345	1	0.222	1.555	0.1407
C	1.095	1	0.222	4.936	0.0002
D	-0.205	1	0.222	-0.9241	0.3701

Final Equation in Terms of Coded Factors:

$$\text{Ash} = 9.865 - 1.443 * A + 0.345 * B + 1.095 * C - 0.205 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Ash} = 9.431 - 7.213E-03 * \text{Water} + 6.900E-03 * \text{Feed} + 0.36500 * \text{Solids} - 4.100E-04 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	9.390	10.073	-0.683	0.300	-0.920	0.073	-0.915	8
2	7.110	7.188	-0.078	0.300	-0.105	0.001	-0.102	1
3	10.900	10.763	0.137	0.300	0.185	0.003	0.178	3
4	8.230	7.878	0.352	0.300	0.474	0.019	0.462	14
5	13.350	12.263	1.087	0.300	1.464	0.184	1.528	13
6	9.190	9.378	-0.188	0.300	-0.253	0.005	-0.245	17
7	14.170	12.953	1.217	0.300	1.639	0.230	1.748	9
8	9.900	10.068	-0.168	0.300	-0.226	0.004	-0.219	11
9	10.040	9.663	0.377	0.300	0.508	0.022	0.495	20
10	8.060	6.778	1.282	0.300	1.727	0.256	1.864	16
11	9.250	10.353	-1.103	0.300	-1.486	0.189	-1.554	18
12	8.860	7.468	1.392	0.300	1.875	0.301	2.070	4
13	12.050	11.853	0.197	0.300	0.265	0.006	0.257	2
14	8.650	8.968	-0.318	0.300	-0.428	0.016	-0.416	19
15	12.990	12.543	0.447	0.300	0.602	0.031	0.589	7
16	9.060	9.658	-0.598	0.300	-0.805	0.056	-0.796	5
17	8.940	9.865	-0.925	0.050	-1.070	0.012	-1.076	6
18	9.020	9.865	-0.845	0.050	-0.978	0.010	-0.976	10
19	9.000	9.865	-0.865	0.050	-1.001	0.011	-1.001	12
20	9.150	9.865	-0.716	0.050	-0.827	0.007	-0.818	15

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	65.61	11	5.964	37.98	0.0001
RESIDUAL	1.26	8	0.157		
Lack Of Fit	1.23	5	0.247	31.51	0.0085
Pure Error	0.02	3	0.008		
COR TOTAL	66.87	19			
ROOT MSE	0.396		R-SQUARED	0.9812	
DEP MEAN	9.865		ADJ R-SQUARED	0.9554	
C.V.	4.02%				

Predicted Residual Sum of Squares (PRESS) = 12.66

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	9.027	1	0.198	45.56	
A	-1.443	1	0.099	-14.56	0.0001
B	0.345	1	0.099	3.483	0.0083
C	1.095	1	0.099	11.05	0.0001
D	-0.205	1	0.099	-2.069	0.0723
A2	1.047	1	0.222	4.729	0.0015
B2	ALIASED	0			
C2	ALIASED	0			
D2	ALIASED	0			
AB	0.035	1	0.099	0.3533	0.7330
AC	-0.528	1	0.099	-5.325	0.0007
AD	0.230	1	0.099	2.322	0.0488
BC	0.015	1	0.099	0.1514	0.8834
BD	-0.175	1	0.099	-1.767	0.1153
CD	-0.277	1	0.099	-2.801	0.0232

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Ash} = & 9.027 - 1.443 * A + 0.345 * B + 1.095 * C - 0.205 * D \\ & + 1.047 * A2 + 0.000 * B2 + 0.000 * C2 + 0.000 * D2 \\ & + 0.035 * AB - 0.528 * AC + 0.230 * AD + 0.015 * BC \\ & - 0.175 * BD - 0.277 * CD \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	9.390	9.582	-0.192	0.687	-0.869	0.138	-0.854	8
2	7.110	7.223	-0.113	0.687	-0.508	0.047	-0.483	1
3	10.900	10.522	0.377	0.687	1.704	0.532	1.997	3
4	8.230	8.302	-0.073	0.687	-0.327	0.020	-0.308	14
5	13.350	13.353	-0.003	0.687	-0.011	0.000	-0.011	13
6	9.190	8.883	0.307	0.687	1.388	0.353	1.490	17
7	14.170	14.352	-0.183	0.687	-0.824	0.124	-0.806	9
8	9.900	10.022	-0.122	0.687	-0.553	0.056	-0.527	11
9	10.040	9.617	0.422	0.687	1.907	0.667	2.416	20
10	8.060	8.177	-0.117	0.687	-0.530	0.052	-0.505	16
11	9.250	9.857	-0.607	0.687	-2.742	1.379	-10.485	18
12	8.860	8.557	0.302	0.687	1.366	0.342	1.459	4
13	12.050	12.278	-0.228	0.687	-1.027	0.193	-1.031	2
14	8.650	8.727	-0.078	0.687	-0.350	0.022	-0.330	19
15	12.990	12.578	0.412	0.687	1.862	0.636	2.314	7
16	9.060	9.167	-0.108	0.687	-0.485	0.043	-0.461	5
17	8.940	9.027	-0.087	0.250	-0.255	0.002	-0.239	6
18	9.020	9.027	-0.007	0.250	-0.022	0.000	-0.020	10
19	9.000	9.027	-0.028	0.250	-0.080	0.000	-0.075	12
20	9.150	9.027	0.122	0.250	0.357	0.004	0.337	15

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Elkhorn No. 3
 Response: Concentrate Sulfur

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	6.316880	1	6.316880		
Linear	0.001750	4	0.000437	1.030	0.4237
Quadratic	0.005220	7	0.000746	5.188	0.0169
Cubic	0.000550	4	0.000138	0.9167	0.5326
RESIDUAL	0.000600	4	0.000150		
TOTAL	6.325000	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	0.006170	12	0.000514	7.713	0.0594
Quadratic	0.000950	5	0.000190	2.850	0.2089
Cubic	0.000400	1	0.000400	6.000	0.0917
PURE ERR	0.000200	3	0.000067		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	0.02061	0.2155	0.0063	0.01280
Quadratic	12	8	0.01199	0.8584	0.6636	0.01008
Cubic	16	4	0.01225	0.9261	0.6490	0.10276

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	0.001750	4	0.00044	1.030	0.4237
RESIDUAL	0.006370	15	0.00042		
Lack Of Fit	0.006170	12	0.00051	7.713	0.0594
Pure Error	0.000200	3	0.00007		
COR TOTAL	0.008120	19			
ROOT MSE	0.02061		R-SQUARED	0.2155	
DEP MEAN	0.56200		ADJ R-SQUARED	0.0063	
C.V.	3.67%				

Predicted Residual Sum of Squares (PRESS) = 0.012799

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	0.56200	1	0.00461	122.0	
A	0.00625	1	0.00515	1.213	0.2438
B	0.00375	1	0.00515	0.7279	0.4779
C	0.00750	1	0.00515	1.456	0.1661
D	-1.164E-13	1	0.00515	-2.26E-11	1.0000

Final Equation in Terms of Coded Factors:

$$\text{Sulfur} = 0.56200 + 0.00625 * A + 0.00375 * B + 0.00750 * C - 1.164E-13 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Sulfur} = 0.51700 + 3.125E-05 * \text{Water} + 7.500E-05 * \text{Feed} + 2.500E-03 * \text{Solids} - 2.327E-16 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	0.55000	0.54450	0.00550	0.300	0.319	0.009	0.309	8
2	0.52000	0.55700	-0.03700	0.300	-2.146	0.395	-2.490	1
3	0.53000	0.55200	-0.02200	0.300	-1.276	0.140	-1.306	3
4	0.56000	0.56450	-0.00450	0.300	-0.261	0.006	-0.253	14
5	0.56000	0.55950	0.00050	0.300	0.029	0.000	0.028	13
6	0.58000	0.57200	0.00800	0.300	0.464	0.018	0.452	17
7	0.59000	0.56700	0.02300	0.300	1.334	0.153	1.373	9
8	0.61000	0.57950	0.03050	0.300	1.769	0.268	1.921	11
9	0.57000	0.54450	0.02550	0.300	1.479	0.187	1.546	20
10	0.58000	0.55700	0.02300	0.300	1.334	0.153	1.373	16
11	0.57000	0.55200	0.01800	0.300	1.044	0.093	1.047	18
12	0.56000	0.56450	-0.00450	0.300	-0.261	0.006	-0.253	4
13	0.54000	0.55950	-0.01950	0.300	-1.131	0.110	-1.142	2
14	0.57000	0.57200	-0.00200	0.300	-0.116	0.001	-0.112	19
15	0.54000	0.56700	-0.02700	0.300	-1.566	0.210	-1.654	7
16	0.57000	0.57950	-0.00950	0.300	-0.551	0.026	-0.538	5
17	0.56000	0.56200	-0.00200	0.050	-0.100	0.000	-0.096	6
18	0.57000	0.56200	0.00800	0.050	0.398	0.002	0.387	10
19	0.56000	0.56200	-0.00200	0.050	-0.100	0.000	-0.096	12
20	0.55000	0.56200	-0.01200	0.050	-0.597	0.004	-0.584	15

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	0.006970	11	0.00063	4.408	0.0224
RESIDUAL	0.001150	8	0.00014		
Lack Of Fit	0.000950	5	0.00019	2.850	0.2089
Pure Error	0.000200	3	0.00007		
COR TOTAL	0.008120	19			
ROOT MSE	0.01199		R-SQUARED	0.8584	
DEP MEAN	0.56200		ADJ R-SQUARED	0.6636	
C.V.	2.13%				

Predicted Residual Sum of Squares (PRESS) = 0.010084

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	0.56000	1	0.00599	93.41	
A	0.00625	1	0.00300	2.085	0.0705
B	0.00375	1	0.00300	1.251	0.2462
C	0.00750	1	0.00300	2.502	0.0368
D	-1.389E-13	1	0.00300	-4.63E-11	1.0000
A2	0.00250	1	0.00670	0.3730	0.7188
B2	ALIASED	0			
C2	ALIASED	0			
D2	ALIASED	0			
AB	0.00250	1	0.00300	0.8341	0.4284
AC	0.00625	1	0.00300	2.085	0.0705
AD	0.00125	1	0.00300	0.4170	0.6876
BC	0.00375	1	0.00300	1.251	0.2462
BD	-0.00625	1	0.00300	-2.085	0.0705
CD	-0.01500	1	0.00300	-5.004	0.0010

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Sulfur} = & 0.56000 + 0.00625 * A + 0.00375 * B + 0.00750 * C \\ & - 1.389E-13 * D + 0.00250 * A^2 + 0.00000 * B^2 \\ & + 0.00000 * C^2 + 0.00000 * D^2 + 0.00250 * AB \\ & + 0.00625 * AC + 0.00125 * AD + 0.00375 * BC \\ & - 0.00625 * BD - 0.01500 * CD \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	0.55000	0.53750	0.01250	0.687	1.865	0.638	2.320	8
2	0.52000	0.53000	-0.01000	0.687	-1.492	0.408	-1.643	1
3	0.53000	0.54500	-0.01500	0.687	-2.238	0.918	-3.424	3
4	0.56000	0.54750	0.01250	0.687	1.865	0.638	2.320	14
5	0.56000	0.56250	-0.00250	0.687	-0.373	0.026	-0.352	13
6	0.58000	0.58000	-8.136E-13	0.687	-0.000	0.000	-0.000	17
7	0.59000	0.58500	0.00500	0.687	0.746	0.102	0.723	9
8	0.61000	0.61250	-0.00250	0.687	-0.373	0.026	-0.352	11
9	0.57000	0.57750	-0.00750	0.687	-1.119	0.230	-1.140	20
10	0.58000	0.57500	0.00500	0.687	0.746	0.102	0.723	16
11	0.57000	0.56000	0.01000	0.687	1.492	0.408	1.643	18
12	0.56000	0.56750	-0.00750	0.687	-1.119	0.230	-1.140	4
13	0.54000	0.54250	-0.00250	0.687	-0.373	0.026	-0.352	2
14	0.57000	0.56500	0.00500	0.687	0.746	0.102	0.723	19
15	0.54000	0.54000	-7.638E-13	0.687	-0.000	0.000	-0.000	7
16	0.57000	0.57250	-0.00250	0.687	-0.373	0.026	-0.352	5
17	0.56000	0.56000	-5.613E-13	0.250	-0.000	0.000	-0.000	6
18	0.57000	0.56000	0.01000	0.250	0.963	0.026	0.958	10
19	0.56000	0.56000	-5.613E-13	0.250	-0.000	0.000	-0.000	12
20	0.55000	0.56000	-0.01000	0.250	-0.963	0.026	-0.958	15

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Pittsburgh No. 8
 Response: Combustible Recovery

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	57495.7	1	57495.7		
Linear	10059.6	4	2514.9	6.797	0.0025
Quadratic	4642.5	8	580.3	4.478	0.0316
Cubic	902.4	4	225.6	141.0	0.0010
RESIDUAL	4.8	3	1.6		
TOTAL	73105.0	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	5544.9	12	462.1	288.8	0.0003
Quadratic	902.4	4	225.6	141.0	0.0010
Cubic	0.0	0			
PURE ERR	4.8	3	1.6		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	19.2	0.6445	0.5497	10750.4
Quadratic	13	7	11.4	0.9419	0.8422	15478.2
Cubic	17	3	1.3	0.9997	0.9981	

Case(s) with leverage of 1.0000: PRESS statistic not defined.

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	10059.6	4	2514.9	6.797	0.0025
RESIDUAL	5549.7	15	370.0		
Lack Of Fit	5544.9	12	462.1	288.8	0.0003
Pure Error	4.8	3	1.6		
COR TOTAL	15609.3	19			
ROOT MSE	19.2		R-SQUARED	0.6445	
DEP MEAN	53.6		ADJ R-SQUARED	0.5497	
C.V.	35.87%				

Predicted Residual Sum of Squares (PRESS) = 10750.4

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	53.6	1	4.3	12.45	
A	-20.7	1	4.8	-4.299	0.0006
B	0.1	1	5.0	2.43E-02	0.9809
C	-11.6	1	4.8	-2.408	0.0293
D	8.1	1	4.8	1.692	0.1112

Final Equation in Terms of Coded Factors:

$$\text{Recovery} = 53.6 - 20.7 * A + 0.1 * B - 11.6 * C + 8.1 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Recovery} = 97.1 - 0.10342 * \text{Water} + 2.419\text{E-}03 * \text{Feed} - 3.8609 * \text{Solids} + 1.630\text{E-}02 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	89.6	77.6	12.0	0.309	0.747	0.050	0.736	17
2	50.3	36.2	14.0	0.301	0.872	0.065	0.864	10
3	98.6	77.9	20.8	0.300	1.290	0.142	1.322	8
4	50.9	36.5	14.4	0.308	0.900	0.072	0.894	1
5	34.8	54.4	-19.6	0.305	-1.223	0.131	-1.245	15
6	23.8	13.1	10.7	0.297	0.663	0.037	0.650	3
7	23.9	54.7	-30.8	0.304	-1.916	0.320	-2.130	6
8	16.5	13.3	3.2	0.312	0.202	0.004	0.195	4
9	72.2	94.0	-21.7	0.278	-1.330	0.136	-1.368	20
10	45.0	52.6	-7.6	0.293	-0.469	0.018	-0.457	19
11	100.0	94.1	5.9	0.287	0.360	0.010	0.350	12
12	39.8	52.8	-13.0	0.295	-0.803	0.054	-0.793	16
13	100.0	70.8	29.2	0.297	1.813	0.278	1.982	5
14	36.7	29.4	7.3	0.310	0.457	0.019	0.444	7
15	100.0	71.0	29.0	0.306	1.810	0.289	1.978	2
16	25.2	29.6	-4.4	0.298	-0.272	0.006	-0.264	9
17	39.4	53.6	-14.2	0.050	-0.759	0.006	-0.748	11
18	41.9	53.6	-11.7	0.050	-0.623	0.004	-0.610	13
19	42.1	53.6	-11.6	0.050	-0.616	0.004	-0.603	14
20	41.7	53.6	-11.9	0.050	-0.636	0.004	-0.623	18

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	14702.1	12	1225.2	9.454	0.0032
RESIDUAL	907.2	7	129.6		
Lack Of Fit	902.4	4	225.6	141.0	0.0010
Pure Error	4.8	3	1.6		
COR TOTAL	15609.3	19			
ROOT MSE	11.4		R-SQUARED	0.9419	
DEP MEAN	53.6		ADJ R-SQUARED	0.8422	
C.V.	21.23%				

Predicted Residual Sum of Squares (PRESS) = 15478.2

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	41.3	1	5.7	7.249	
A	-22.1	1	3.1	-7.138	0.0002
B	-5.0	1	5.0	-1.015	0.3438
C	-14.5	1	3.7	-3.936	0.0056
D	11.8	1	4.1	2.908	0.0227
A2	-36.5	1	39.1	-0.9356	0.3806
B2	55.6	1	41.4	1.343	0.2212
C2	ALIASED	0			
D2	ALIASED	0			
AB	0.2	1	3.7	5.17E-02	0.9602
AC	1.9	1	2.9	0.6481	0.5376
AD	-9.0	1	3.1	-2.889	0.0234
BC	-2.3	1	3.1	-0.7215	0.4940
BD	-4.0	1	5.0	-0.8029	0.4485
CD	9.2	1	3.7	2.501	0.0409

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Recovery} = & 41.3 - 22.1 * A - 5.0 * B - 14.5 * C + 11.8 * D \\ & - 36.5 * A2 + 55.6 * B2 + 0.0 * C2 + 0.0 * D2 \\ & + 0.2 * AB + 1.9 * AC - 9.0 * AD - 2.3 * BC \\ & - 4.0 * BD + 9.2 * CD \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	89.6	86.4	3.2	0.835	0.686	0.184	0.657	17
2	50.3	55.8	-5.6	0.720	-0.927	0.170	-0.916	10
3	98.6	88.4	10.2	0.751	1.791	0.743	2.253	8
4	50.9	58.7	-7.8	0.698	-1.242	0.275	-1.303	1
5	34.8	39.6	-4.7	0.698	-0.759	0.102	-0.733	15
6	23.8	16.6	7.2	0.751	1.259	0.367	1.325	3
7	23.9	32.5	-8.6	0.720	-1.429	0.405	-1.572	6
8	16.5	10.3	6.2	0.835	1.341	0.702	1.440	4
9	72.2	82.1	-9.9	0.739	-1.694	0.625	-2.042	20
10	45.0	33.6	11.4	0.685	1.785	0.533	2.238	19
11	100.0	103.5	-3.5	0.686	-0.548	0.051	-0.519	12
12	39.8	37.9	1.9	0.680	0.303	0.015	0.282	16
13	100.0	89.3	10.7	0.713	1.750	0.585	2.160	5
14	36.7	48.7	-12.1	0.817	-2.479	2.113	-6.565	7
15	100.0	97.3	2.7	0.768	0.489	0.061	0.461	2
16	25.2	26.5	-1.3	0.901	-0.358	0.090	-0.334	9
17	39.4	41.3	-1.9	0.250	-0.191	0.001	-0.177	11
18	41.9	41.3	0.7	0.250	0.068	0.000	0.063	13
19	42.1	41.3	0.8	0.250	0.081	0.000	0.075	14
20	41.7	41.3	0.4	0.250	0.042	0.000	0.039	18

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Pittsburgh No. 8
 Response: Concentrate Ash

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	705.2	1	705.2		
Linear	200.7	4	50.2	4.950	0.0096
Quadratic	107.3	8	13.4	2.100	0.1721
Cubic	44.7	4	11.2	989.1	0.0001
RESIDUAL	0.0	3	0.0		
TOTAL	1057.9	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	152.0	12	12.7	1122	0.0001
Quadratic	44.7	4	11.2	989.1	0.0001
Cubic	0.0	0			
PURE ERR	0.0	3	0.0		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	3.18	0.5689	0.4540	291.14
Quadratic	13	7	2.53	0.8732	0.6559	762.22
Cubic	17	3	0.11	0.9999	0.9994	

Case(s) with leverage of 1.0000: PRESS statistic not defined.

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	200.7	4	50.17	4.950	0.0096
RESIDUAL	152.0	15	10.14		
Lack Of Fit	152.0	12	12.67	1122	0.0001
Pure Error	0.0	3	0.01		
COR TOTAL	352.7	19			
ROOT MSE	3.18		R-SQUARED	0.5689	
DEP MEAN	5.94		ADJ R-SQUARED	0.4540	
C.V.	53.61%				

Predicted Residual Sum of Squares (PRESS) = 291.1

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	5.90	1	0.71	8.282	
A	-3.02	1	0.80	-3.788	0.0018
B	0.91	1	0.82	1.102	0.2878
C	0.53	1	0.80	0.6617	0.5182
D	1.45	1	0.80	1.825	0.0881

Final Equation in Terms of Coded Factors:

$$\text{Ash} = 5.90 - 3.02 * A + 0.91 * B + 0.53 * C + 1.45 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Ash} = 2.72 - 1.508E-02 * \text{Water} + 1.813E-02 * \text{Feed} + 0.17557 * \text{Solids} + 2.908E-03 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	6.56	6.03	0.53	0.309	0.200	0.004	0.194	17
2	3.00	-0.00	3.00	0.301	1.128	0.109	1.139	10
3	10.76	7.84	2.92	0.300	1.095	0.103	1.103	8
4	3.34	1.81	1.53	0.308	0.577	0.030	0.564	1
5	4.45	7.08	-2.63	0.305	-0.992	0.086	-0.992	15
6	3.37	1.05	2.32	0.297	0.868	0.064	0.861	3
7	4.88	8.90	-4.02	0.304	-1.512	0.199	-1.586	6
8	3.44	2.86	0.58	0.312	0.218	0.004	0.211	4
9	4.47	9.26	-4.79	0.278	-1.772	0.241	-1.925	20
10	3.38	3.05	0.33	0.293	0.123	0.001	0.119	19
11	12.77	10.75	2.02	0.287	0.751	0.046	0.740	12
12	3.41	4.72	-1.31	0.295	-0.490	0.020	-0.477	16
13	16.01	10.14	5.87	0.297	2.200	0.409	2.583	5
14	3.45	3.96	-0.51	0.310	-0.193	0.003	-0.187	7
15	16.24	11.91	4.33	0.306	1.632	0.235	1.738	2
16	4.06	5.77	-1.71	0.298	-0.642	0.035	-0.629	9
17	3.64	5.90	-2.26	0.050	-0.729	0.006	-0.717	11
18	3.80	5.90	-2.10	0.050	-0.677	0.005	-0.665	13
19	3.87	5.90	-2.03	0.050	-0.655	0.005	-0.642	14
20	3.86	5.90	-2.04	0.050	-0.658	0.005	-0.645	18

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	308.0	12	25.67	4.018	0.0371
RESIDUAL	44.7	7	6.39		
Lack Of Fit	44.7	4	11.17	989.1	0.0001
Pure Error	0.0	3	0.01		
COR TOTAL	352.7	19			
ROOT MSE	2.53		R-SQUARED	0.8732	
DEP MEAN	5.94		ADJ R-SQUARED	0.6559	
C.V.	42.56%				

Predicted Residual Sum of Squares (PRESS) = 762.2

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	3.79	1	1.26	3.001	
A	-3.19	1	0.69	-4.632	0.0024
B	0.23	1	1.10	0.2061	0.8426
C	0.14	1	0.82	0.1745	0.8664
D	1.92	1	0.90	2.132	0.0705
A2	-4.31	1	8.67	-0.4975	0.6341
B2	7.42	1	9.19	0.8066	0.4464
C2	ALIASED	0			
D2	ALIASED	0			
AB	-0.32	1	0.83	-0.3848	0.7118
AC	-0.27	1	0.65	-0.4103	0.6939
AD	-1.50	1	0.69	-2.182	0.0654
BC	-0.52	1	0.70	-0.7385	0.4842
BD	-0.40	1	1.10	-0.3651	0.7258
CD	1.08	1	0.82	1.321	0.2282

Final Equation in Terms of Coded Factors:

$$\begin{aligned} \text{Ash} = & 3.79 - 3.19 * A + 0.23 * B + 0.14 * C + 1.92 * D \\ & - 4.31 * A2 + 7.42 * B2 + 0.00 * C2 + 0.00 * D2 \\ & - 0.32 * AB - 0.27 * AC - 1.50 * AD - 0.52 * BC \\ & - 0.40 * BD + 1.08 * CD \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	6.56	5.87	0.69	0.835	0.674	0.177	0.645	17
2	3.00	3.67	-0.67	0.720	-0.500	0.050	-0.471	10
3	10.76	8.80	1.96	0.751	1.552	0.558	1.773	8
4	3.34	5.32	-1.98	0.698	-1.426	0.362	-1.568	1
5	4.45	5.56	-1.11	0.698	-0.796	0.113	-0.773	15
6	3.37	2.29	1.08	0.751	0.858	0.171	0.840	3
7	4.88	6.42	-1.54	0.720	-1.155	0.264	-1.188	6
8	3.44	1.87	1.57	0.835	1.527	0.911	1.732	4
9	4.47	7.21	-2.74	0.739	-2.125	0.984	-3.302	20
10	3.38	0.96	2.42	0.685	1.707	0.488	2.068	19
11	12.77	12.68	0.09	0.686	0.067	0.001	0.062	12
12	3.41	3.18	0.23	0.680	0.160	0.004	0.148	16
13	16.01	13.13	2.88	0.713	2.128	0.866	3.315	5
14	3.45	6.09	-2.64	0.817	-2.444	2.055	-5.914	7
15	16.24	16.47	-0.23	0.768	-0.190	0.009	-0.177	2
16	4.06	4.07	-0.01	0.901	-0.009	0.000	-0.009	9
17	3.64	3.79	-0.15	0.250	-0.070	0.000	-0.065	11
18	3.80	3.79	0.01	0.250	0.003	0.000	0.003	13
19	3.87	3.79	0.08	0.250	0.035	0.000	0.033	14
20	3.86	3.79	0.07	0.250	0.031	0.000	0.029	18

STATISTICAL DATA ANALYSIS

Description: Froth Flotation
 Coal Sample: Pittsburgh No. 8
 Response: Combustible Sulfur

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Water	ml/min	200.000	600.000
B	Feed	ml/min	150.000	250.000
C	Solids	%	4.000	10.000
D	Air	ml/min	1000.000	2000.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	141.246	1	141.246		
Linear	3.371	4	0.843	6.067	0.0041
Quadratic	1.469	8	0.184	2.093	0.1732
Cubic	0.614	4	0.153	969.2	0.0001
RESIDUAL	0.000	3	0.000		
TOTAL	146.701	20			

Lack of Fit Tests

MODEL	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
Linear	2.083	12	0.174	1096	0.0001
Quadratic	0.614	4	0.153	969.2	0.0001
Cubic	0.000	0			
PURE ERR	0.000	3	0.000		

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	5	15	0.373	0.6180	0.5162	4.013
Quadratic	13	7	0.296	0.8874	0.6943	10.396
Cubic	17	3	0.013	0.9999	0.9994	

Case(s) with leverage of 1.0000: PRESS statistic not defined.

ANOVA for Linear Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	3.371	4	0.843	6.067	0.0041
RESIDUAL	2.084	15	0.139		
Lack Of Fit	2.083	12	0.174	1096	0.0001
Pure Error	0.000	3	0.000		
COR TOTAL	5.455	19			
ROOT MSE	0.373		R-SQUARED	0.6180	
DEP MEAN	2.657		ADJ R-SQUARED	0.5162	
C.V.	14.02%				

Predicted Residual Sum of Squares (PRESS) = 4.013

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	2.653	1	0.083	31.80	
A	-0.387	1	0.093	-4.148	0.0009
B	0.112	1	0.096	1.167	0.2614
C	0.048	1	0.093	0.5144	0.6145
D	0.206	1	0.093	2.204	0.0436

Final Equation in Terms of Coded Factors:

$$\text{Sulfur} = 2.653 - 0.387 * A + 0.112 * B + 0.048 * C + 0.206 * D$$

Final Equation in Terms of Actual Factors:

$$\text{Sulfur} = 2.248 - 1.933\text{E-}03 * \text{Water} + 2.247\text{E-}03 * \text{Feed} + 1.598\text{E-}02 * \text{Solids} + 4.113\text{E-}04 * \text{Air}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	2.770	2.674	0.096	0.309	0.311	0.009	0.301	17
2	2.250	1.900	0.350	0.301	1.122	0.108	1.132	10
3	3.220	2.898	0.322	0.300	1.031	0.091	1.033	8
4	2.300	2.125	0.175	0.308	0.564	0.028	0.551	1
5	2.380	2.770	-0.390	0.305	-1.254	0.138	-1.280	15
6	2.280	1.996	0.284	0.297	0.908	0.070	0.902	3
7	2.540	2.994	-0.454	0.304	-1.461	0.186	-1.524	6
8	2.310	2.221	0.089	0.312	0.288	0.008	0.279	4
9	2.590	3.125	-0.535	0.278	-1.690	0.220	-1.815	20
10	2.310	2.330	-0.020	0.293	-0.063	0.000	-0.061	19
11	3.560	3.310	0.250	0.287	0.796	0.051	0.785	12
12	2.370	2.536	-0.166	0.295	-0.532	0.024	-0.519	16
13	3.920	3.199	0.721	0.297	2.308	0.450	2.777	5
14	2.360	2.408	-0.048	0.310	-0.154	0.002	-0.149	7
15	3.880	3.419	0.461	0.306	1.485	0.195	1.553	2
16	2.440	2.632	-0.192	0.298	-0.616	0.032	-0.603	9
17	2.420	2.653	-0.233	0.050	-0.641	0.004	-0.628	11
18	2.430	2.653	-0.223	0.050	-0.614	0.004	-0.601	13
19	2.420	2.653	-0.233	0.050	-0.641	0.004	-0.628	14
20	2.400	2.653	-0.253	0.050	-0.697	0.005	-0.684	18

ANOVA for Quadratic Model

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	4.840	12	0.403	4.596	0.0260
RESIDUAL	0.614	7	0.088		
Lack Of Fit	0.614	4	0.153	969.2	0.0001
Pure Error	0.000	3	0.000		
COR TOTAL	5.455	19			
ROOT MSE	0.296		R-SQUARED	0.8874	
DEP MEAN	2.657		ADJ R-SQUARED	0.6943	
C.V.	11.15%				

Predicted Residual Sum of Squares (PRESS) = 10.396

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR HO COEFFICIENT=0	PROB > t
Intercept	2.417	1	0.148	16.32	
A	-0.406	1	0.081	-5.033	0.0015
B	0.036	1	0.129	0.2779	0.7891
C	0.005	1	0.096	5.29E-02	0.9593
D	0.258	1	0.106	2.444	0.0445
A2	-0.480	1	1.016	-0.4725	0.6509
B2	0.827	1	1.078	0.7673	0.4680
C2	ALIASED	0			
D2	ALIASED	0			
AB	-0.034	1	0.098	-0.3443	0.7408
AC	-0.015	1	0.076	-0.2010	0.8464
AD	-0.185	1	0.081	-2.293	0.0555
BC	-0.058	1	0.082	-0.7037	0.5044
BD	-0.050	1	0.129	-0.3892	0.7087
CD	0.134	1	0.096	1.392	0.2065

Final Equation in Terms of Coded Factors:

$$\begin{aligned}
 \text{Sulfur} = & 2.417 - 0.406 * A + 0.036 * B + 0.005 * C + 0.258 * D \\
 & - 0.480 * A2 + 0.827 * B2 + 0.000 * C2 + 0.000 * \\
 \text{D2} & - 0.034 * AB - 0.015 * AC - 0.185 * AD - 0.058 * BC \\
 & - 0.050 * BD + 0.134 * CD
 \end{aligned}$$

Final Equation in Terms of Actual Factors:

Not available for ALIASED models.

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t	Run Ord
1	2.770	2.663	0.107	0.835	0.887	0.307	0.871	17
2	2.250	2.319	-0.069	0.720	-0.439	0.038	-0.412	10
3	3.220	3.018	0.202	0.751	1.363	0.430	1.472	8
4	2.300	2.539	-0.239	0.698	-1.471	0.385	-1.639	1
5	2.380	2.552	-0.172	0.698	-1.056	0.199	-1.067	15
6	2.280	2.146	0.134	0.751	0.907	0.190	0.893	3
7	2.540	2.676	-0.136	0.720	-0.870	0.150	-0.853	6
8	2.310	2.136	0.174	0.835	1.448	0.819	1.602	4
9	2.590	2.922	-0.332	0.739	-2.194	1.049	-3.633	20
10	2.310	2.055	0.255	0.685	1.532	0.393	1.740	19
11	3.560	3.536	0.024	0.686	0.143	0.003	0.133	12
12	2.370	2.317	0.053	0.680	0.319	0.017	0.297	16
13	3.920	3.557	0.363	0.713	2.290	1.003	4.231	5
14	2.360	2.660	-0.300	0.817	-2.366	1.926	-4.897	7
15	3.880	3.935	-0.055	0.768	-0.387	0.038	-0.362	2
16	2.440	2.448	-0.008	0.901	-0.091	0.006	-0.084	9
17	2.420	2.417	0.003	0.250	0.010	0.000	0.009	11
18	2.430	2.417	0.013	0.250	0.049	0.000	0.045	13
19	2.420	2.417	0.003	0.250	0.010	0.000	0.009	14
20	2.400	2.417	-0.017	0.250	-0.068	0.000	-0.063	18

Column flotation pH test data using Elkhorn No. 3 seam coal

Test Number	pH	Flow Rates						Bias		% Solids		Mass Flow Rates		
		Wash Water (ml/min)	Feed (ml/min)	Aeration (cc/min)	Frother (ml/min)	(lb/ton)	Tailings (ml/min)	Factor (%)	Rate (cm/sec)	Feed (%)	Tails (%)	Feed (g/min)	Product (g/min)	Tails (g/min)
1	4.0	200	105	1500	11	3.55	290	0.925	0.152	3.76	0.91	3.90	1.32	2.58
2	4.0	200	162	1500	15	3.14	350	0.940	0.155	3.80	1.27	6.29	1.73	4.26
3	4.0	200	242	1500	22	3.08	440	0.990	0.163	3.77	1.71	9.43	2.03	7.31
4	6.0	200	108	1500	10	3.14	295	0.935	0.154	3.93	1.13	4.23	0.82	3.26
5	6.0	200	156	1500	15	3.26	300	0.720	0.118	3.89	1.41	6.04	1.35	4.43
6	6.0	200	244	1500	24	3.34	450	1.030	0.169	4.07	1.96	9.61	0.80	8.71
7	7.5	196	106	1500	10	3.20	280	0.888	0.143	3.86	0.97	4.20	1.38	2.72
8	7.5	196	152	1500	15	3.35	340	0.959	0.155	3.90	1.27	5.91	1.47	4.26
9	7.5	196	224	1500	24	3.64	410	0.949	0.153	3.87	1.68	8.81	2.05	6.80
10	10.0	196	104	1500	10	3.26	280	0.898	0.145	3.83	0.91	4.01	1.66	2.51
11	10.0	196	158	1500	15	3.22	315	0.801	0.129	3.91	1.28	6.31	1.82	3.03
12	10.0	196	234	1500	24	3.48	360	0.643	0.104	3.90	1.65	9.17	3.23	5.84

Column flotation pH test data using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adjusted)			% Sulfur (adjusted)			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulfur (%)	
1	51.35	11.24	68.98	48.65	88.76	31.02	0.33	0.50	0.25	30.53	55.71	93.32	51.52	49.02
2	50.36	11.14	64.85	49.64	88.86	35.15	0.35	0.52	0.28	26.98	48.29	94.03	56.67	42.33
3	50.19	10.77	61.37	49.81	89.23	38.63	0.35	0.54	0.30	22.09	39.58	95.26	67.86	34.84
4	50.13	9.97	61.58	49.87	90.03	38.42	0.33	0.51	0.28	22.19	40.05	95.59	66.40	35.64
5	50.57	10.85	60.74	49.43	89.15	39.26	0.33	0.51	0.28	20.38	36.77	95.63	66.40	32.39
6	50.80	11.53	54.13	49.20	88.47	45.87	0.34	0.51	0.32	7.82	14.06	98.23	84.21	12.28
7	49.80	8.77	72.95	50.20	91.23	27.05	0.35	0.54	0.24	36.07	65.55	93.65	43.43	59.20
8	49.55	9.08	63.85	50.45	90.92	36.15	0.35	0.53	0.28	26.11	47.05	95.22	57.60	42.27
9	50.00	9.48	60.34	50.00	90.52	39.66	0.36	0.56	0.31	20.33	36.81	96.15	68.89	32.95
10	49.54	7.64	79.00	50.46	92.36	21.00	0.34	0.54	0.20	41.28	75.56	93.63	34.60	69.20
11	49.92	7.45	74.21	50.08	92.55	25.79	0.35	0.55	0.23	36.38	67.24	94.57	41.07	61.81
12	50.28	7.57	71.78	49.72	92.43	28.22	0.36	0.57	0.25	33.48	62.25	94.96	45.57	57.21

Column flotation pH test data using Pittsburgh No. 8 seam coal

Test No.	pH	Flow Rates					Bias		% Solids		Mass Flow Rates		
		Wash Water (ml/min)	Feed (ml/min)	Aeration (cc/min)	Frother (ml/min) (lb/ton)	Tails (ml/min)	Factor (%)	Rate (cm/sec)	Feed (%)	Tails (%)	Feed (g/min)	Product (g/min)	Tails (g/min)
1	4.0	196	104	1500	11 3.59	240	0.694	0.112	4.02	0.60	4.29	2.70	1.38
2	4.0	196	152	1500	15 3.35	275	0.628	0.101	4.02	1.04	6.31	3.35	2.86
3	4.0	196	232	1500	24 3.51	315	0.423	0.068	4.06	1.40	9.51	4.96	4.31
4	6.0	196	102	1500	10 3.33	248	0.745	0.120	4.03	0.49	4.29	3.09	1.22
5	6.0	196	160	1500	15 3.18	250	0.459	0.074	4.04	0.82	6.76	4.42	2.13
6	6.0	196	234	1500	24 3.48	305	0.362	0.058	4.03	1.35	9.55	5.33	4.19
7	7.5	196	106	1500	10 3.20	225	0.607	0.098	4.07	0.55	4.41	3.16	1.32
8	7.5	196	154	1500	15 3.30	255	0.515	0.083	4.10	0.92	6.43	3.89	2.42
9	7.5	198	185	1500	16 2.93	210	0.126	0.021	3.98	0.90	7.50	5.46	2.26
10	10.0	198	107	1500	10 3.17	240	0.672	0.109	3.88	0.50	4.15	2.68	1.23
11	10.0	198	154	1500	10 2.20	260	0.535	0.087	4.94	0.89	5.95	3.24	2.38
12	10.0	198	235	1500	19 2.74	185	-0.253	-0.041	3.92	1.31	9.25	6.16	2.73

Column flotation pH test data using Pittsburgh No. 8 seam coal

Test No.	% Ash (adjusted)			% Combustible (adjusted)			% Sulfur (adjusted)			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulfur (%)	
1	16.53	5.44	39.33	83.47	94.56	60.67	3.67	2.66	5.75	67.28	76.21	77.86	51.21	54.07
2	16.49	6.56	27.55	83.51	93.44	72.45	3.67	2.74	4.72	52.69	58.96	79.04	60.41	38.00
3	16.60	7.63	27.00	83.40	92.37	73.00	3.55	2.80	4.43	53.69	59.47	75.32	57.42	34.79
4	17.05	4.11	51.80	82.95	95.89	48.20	3.63	2.44	6.84	72.87	84.23	82.44	50.96	66.67
5	17.27	4.23	43.53	82.73	95.77	56.47	3.75	2.41	6.45	66.82	77.35	83.63	57.05	60.99
6	17.04	4.37	32.88	82.96	95.63	67.12	3.59	2.44	5.04	55.56	64.04	85.75	62.10	49.80
7	16.40	3.67	48.54	83.60	96.33	51.46	3.65	2.26	7.16	71.63	82.54	83.97	55.65	66.51
8	16.60	3.41	40.35	83.40	96.59	59.65	3.70	2.23	6.34	64.29	74.46	86.79	61.29	61.25
9	16.12	3.87	43.77	83.88	96.13	56.23	3.51	2.29	6.28	69.30	79.42	83.36	54.71	62.78
10	15.06	3.39	44.28	84.94	96.61	55.72	3.47	2.16	6.75	71.46	81.28	83.91	55.52	65.19
11	15.08	3.01	33.67	84.92	96.99	66.33	3.57	2.10	5.82	60.63	69.25	87.90	64.42	57.15
12	15.27	4.84	33.29	84.73	95.16	66.71	3.54	2.35	5.59	63.34	71.14	79.92	58.00	51.06

APPENDIX G - DRUM SCREEN DATA

Drum screen test data using Elkhorn No. 3 (22% ash) seam coal

Test Number	Flow Rates				Kerosene (lb/ton)	% Solids			
	Feed (ml/min)	Feed (lb/hr)	Product (ml/min)	Tails (ml/min)		Wash Water (ml/min)	Feed (%)	Product (%)	Tails (%)
1	70	0.290	212	246	388	0	3.13	0.23	0.67
2	70	0.267	232	256	418	6	2.88	0.17	0.68
3	70	0.583	296	160	386	0	6.28	0.73	1.39
4	70	0.574	336	148	414	6	6.18	0.72	1.51
5	40	0.238	244	190	394	0	4.49	0.23	0.63
6	40	0.248	314	134	408	6	4.67	0.25	0.72
7	100	0.608	284	200	384	0	4.58	0.66	1.25
8	100	0.606	268	261	429	6	4.57	0.50	1.40
9	40	0.163	176	246	382	3	3.08	0.12	0.41
10	40	0.343	200	228	388	3	6.48	0.31	0.90
11	100	0.403	180	318	398	3	3.04	0.24	0.88
12	100	0.858	220	290	410	3	6.47	1.44	1.15
13	70	0.442	182	288	400	3	4.76	0.35	1.01
14	70	0.445	198	278	406	3	4.80	0.36	1.02
15	70	0.448	198	259	387	3	4.83	0.35	1.05

Drum screen test data using Elkhorn No. 3 (22% ash) seam coal

Test Number	Yield (%)	Comb. Rec. (%)	Ash Rej. (%)	Sep. Eff. (%)	% Ash		
					Feed (%)	Product (%)	Tails (%)
1	18.68	19.50	84.23	3.73	22.09	18.65	22.88
2	17.38	18.28	85.79	4.08	22.13	18.09	22.98
3	49.20	49.80	52.94	2.73	21.88	20.93	22.80
4	60.56	61.00	40.98	1.98	21.93	21.37	22.79
5	30.26	31.31	73.52	4.83	21.78	19.06	22.96
6	39.33	40.33	64.23	4.56	22.02	20.03	23.31
7	35.67	36.18	66.11	2.29	22.02	20.92	22.63
8	22.78	23.41	79.45	2.86	22.17	20.00	22.81
9	14.44	15.46	89.07	4.53	22.42	16.97	23.34
10	19.62	20.38	83.03	3.40	22.14	19.15	22.87
11	14.50	15.25	88.17	3.43	22.10	18.03	22.79
12	34.13	34.49	67.15	1.64	22.24	21.41	22.67
13	18.11	18.86	84.51	3.37	22.16	18.95	22.87
14	16.84	17.52	85.53	3.05	22.27	19.13	22.90
15	14.36	14.97	87.75	2.71	22.37	19.09	22.92

APPENDIX H - PRELIMINARY OPTIMIZATION DATA

Lamella thickener 3 cleaning stages data using Elkhorn No. 3 seam coal

Test Number	Flow Rates					% Solids			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)
	Wash Water (ml/min)	Feed (ml/min)	(lb/hr)	Product (ml/min)	Tails (ml/min)	Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulphur (%)	
I	42	102	0.527	33	112	3.98	8.19	1.00	78.35	96.91	43.10	6.01	40.01
II	42	101	0.437	33	110	3.30	7.62	0.44	86.27	97.28	37.82	5.12	35.10
III	41	101	0.283	18	124	2.15	7.13	0.14	93.06	98.35	26.33	3.51	24.68
I						3.98	8.19	1.00	78.35	96.91	43.10	6.01	40.01
I-II						3.98	7.62	0.44	62.72	90.53	69.43	15.62	59.97
I-III						3.98	7.13	0.14	54.34	84.14	60.10	25.97	64.24

Lamella thickener 3 cleaning stages data using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)			Superficial Velocity		Throughput		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Overflow (mm/min)	Plate (mm/min)	Eff. Area (lb/hr/sq.in.)	Actual Area (lb/hr/sq.in.)	Volume (lb/hr/litre)
I	46.39	33.69	92.35	53.61	66.31	7.65	0.46	0.53	0.15	22.26	41.33	0.0031	0.0126	0.0480
II	31.37	22.61	86.39	68.63	77.39	13.61	0.54	0.59	0.21	21.86	40.60	0.0026	0.0104	0.0398
III	21.46	16.99	81.38	78.54	83.01	18.62	0.6	0.63	0.26	24.64	45.76	0.0017	0.0067	0.0258
I	46.39	33.69	92.35	53.61	66.31	7.65	0.46	0.53	0.15					
I-II	46.39	22.61	86.39	53.61	77.39	13.61	0.46	0.59	0.21					
I-III	46.39	16.99	81.38	53.61	83.01	18.62	0.46	0.63	0.26					

Lamella thickener and LOA data using Elkhorn No. 3 seam coal

Test Number	Flow Rates				Kerosene (lb/ton)	% Solids			Yield (%)	Combustible Recovery (%)	Rejection		Separation Efficiency (%)	
	Wash Water (ml/min)	Feed (ml/min)	Product (lb/hr)	Tails (ml/min)		Feed (%)	Product (%)	Tails (%)			Ash (%)	Sulphur (%)		
0	42	103	0.527	33	112	0	3.98	8.19	1.00	78.35	96.91	43.10	6.01	40.01
3	41	100	0.534	35	106	3	3.98	7.29	0.99	73.70	96.11	51.89	11.75	48.00
6	39	100	0.547	34	105	6	4.07	7.41	0.98	71.28	95.37	55.34	12.71	50.71
9	41	101	0.520	34	108	9	3.87	6.93	0.98	70.82	96.65	56.23	9.78	51.88
12	42	102	0.530	33	111	12	3.92	7.13	0.99	66.48	94.54	61.25	11.06	56.79
24	40	103	0.504	32	111	24	3.72	6.81	1.04	69.85	95.16	56.61	9.88	50.77
48	40	103	0.500	33	110	48	3.70	6.74	1.05	68.18	94.88	57.84	12.57	52.73

Lamella thickener and LOA data using Elkhorn No. 3 seam coal

Test Number	% Ash (adjusted)			% Combustible (adj.)			% Sulfur (adjusted)			Superficial Velocity		Throughput		
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)	Overflow (mm/min)	Plate (mm/min)	Eff. Area (lb/hr/sq.in.)	Actual Area (lb/hr/sq.in.)	Volume (lb/hr/litre)
0	46.39	33.69	92.35	53.81	66.31	7.65	0.46	0.53	0.15	22.26	41.33	0.0031	0.0126	0.0480
3	46.89	30.48	92.12	53.31	69.52	7.88	0.48	0.59	0.20	21.06	39.12	0.0031	0.0127	0.0486
6	47.50	29.76	91.53	52.50	70.24	8.47	0.46	0.58	0.19	20.87	38.75	0.0032	0.0130	0.0498
9	47.86	29.58	92.22	52.14	70.42	7.78	0.45	0.56	0.16	21.46	39.86	0.0031	0.0124	0.0473
12	50.30	29.32	91.91	49.70	70.68	8.09	0.41	0.56	0.13	22.06	40.97	0.0031	0.0126	0.0482
24	49.86	31.89	91.95	50.14	68.31	8.06	0.41	0.52	0.14	22.06	40.97	0.0030	0.0120	0.0459
48	50.64	31.31	92.06	49.36	68.69	7.94	0.40	0.51	0.16	21.86	40.60	0.0029	0.0119	0.0454

Conventional thickener parametric test data using Elkhorn No. 3 seam coal

Test Number	Flow Rates				% Solids			Yield (%)	Combustible Recovery (%)	Ash Rejection (%)	Separation Efficiency (%)
	Feed (ml/min)	(lb/hr)	Product (ml/min)	Tails (ml/min)	Feed (%)	Product (%)	Tails (%)				
1	46	0.197	10	36	3.00	6.78	0.68	86.29	97.92	24.61	22.53
2	50	0.131	35	15	1.98	1.81	0.29	104.05	100.68	-7.72	-7.05
3	49	0.269	37	12	3.98	3.34		66.12	54.97	23.31	-21.73
4	196	0.520	76	120	2.00	2.88	0.73	84.94	96.66	27.77	24.44
5	151	0.582	64	87	2.94	5.01	1.16	78.52	95.43	38.75	34.18
6	100	0.264	46	54	1.99	4.46	0.63	83.90	97.07	30.54	27.61
7	100	0.400	34	66	2.98	4.10	1.11	84.82	96.98	26.68	23.67
8	100	0.523	60	40	3.94	6.18		88.59	98.46	22.16	20.63
9	152	0.794	124	28		5.54	1.12	86.82	98.19	24.73	22.92

Conventional thickener parametric test data using Elkhorn No. 3 seam coal

Test Number	% Ash			% Combustible			Superficial Velocity (mm/min)	Throughput	
	Feed (%)	Product (%)	Tails (%)	Feed (%)	Product (%)	Tails (%)		Area (lb/hr/sq.in.)	Volume (lb/hr/litre)
1	51.64	45.12	92.67	48.36	54.88	7.33	6.44	0.0023	0.0188
2	47.92	49.61	91.32	52.08	50.39	8.68	2.68	0.0015	0.0125
3	51.32	59.53	35.30	48.68	40.47	64.70	2.15	0.0031	0.0256
4	47.97	40.79	88.47	52.03	59.21	11.53	21.47	0.0060	0.0495
5	49.46	38.58	89.24	50.54	61.42	10.76	15.56	0.0067	0.0554
6	47.71	39.50	90.48	52.29	60.50	9.52	9.66	0.0030	0.0252
7	51.40	44.43	90.34	48.60	55.57	9.66	11.81	0.0046	0.0381
8	47.87	42.06	92.97	52.13	57.94	7.03	7.16	0.0060	0.0498
9	49.60	43.00	93.07	50.40	57.00	6.93	5.01	0.0091	0.0756

STATISTICAL DATA ANALYSIS

Description: Conventional Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Combustible Recovery (%)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed Rate	(ml/min)	50.000	150.000
B	Feed % Solids	(%)	2.000	4.000

Analysis Summary

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	77722.0	1	77722.0		
Linear	439.3	2	219.6	1.100	0.3919
Quadratic	677.6	3	225.9	1.301	0.4170
Cubic	521.0	3	173.7		
RESIDUAL	0.0	0			
TOTAL	79359.9	9			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	3	6	14.13	0.2682	0.0243	3331.51
Quadratic	6	3	13.18	0.6819	0.1518	10547.20
Cubic	9	0				

ANOVA for Linear Model (Combustible Recovery / Conventional / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	439.3	2	219.65	1.100	0.3919
RESIDUAL	1198.6	6	199.76		
COR TOTAL	1637.9	8			
ROOT MSE	14.13		R-SQUARED	0.2682	
DEP MEAN	92.93		ADJ R-SQUARED	0.0243	
C.V.	15.21%				

Predicted Residual Sum of Squares (PRESS) = 3331.5

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	92.55	1	4.73	19.55	
A	3.91	1	4.77	0.8194	0.4439
B	-6.55	1	5.81	-1.126	0.3032

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 92.55 + (3.91 * A) - (6.55 * B)$$

Final Equation in Terms of Actual Factors:

$$\text{Combustible Recovery} = 104.37 + (7.811E-02 * \text{Feed Rate}) - (6.5459 * \text{Feed \% Solids})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	100.68	95.19	5.49	0.455	0.526	0.077	0.492
6	97.07	99.09	-2.02	0.285	-0.169	0.004	-0.155
4	96.66	106.59	-9.93	0.596	-1.105	0.600	-1.130
1	97.92	88.33	9.59	0.269	0.794	0.077	0.766
7	96.98	92.55	4.43	0.112	0.333	0.005	0.307
5	95.43	96.53	-1.10	0.208	-0.087	0.001	-0.080
3	54.97	82.02	-27.05	0.384	-2.439	1.237	-23.877
8	98.46	86.00	12.46	0.278	1.037	0.138	1.045
9	98.19	90.06	8.13	0.413	0.751	0.132	0.720

ANOVA for Quadratic Model (Combustible Rec. / Conventional / Elkhorn)

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	1116.9	5	223.39	1.286	0.4451
RESIDUAL	521.0	3	173.65		
COR TOTAL	1637.9	8			
ROOT MSE	13.18		R-SQUARED	0.6819	
DEP MEAN	92.93		ADJ R-SQUARED	0.1518	
C.V.	14.18%				

Predicted Residual Sum of Squares (PRESS) = 10547.2

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	97.31	1	8.89	10.94	
A	6.98	1	5.21	1.340	0.2728
B	-7.83	1	5.73	-1.366	0.2652
A2	-0.53	1	6.19	-8.63E-02	0.9367
B2	-5.34	1	9.49	-0.5623	0.6132
AB	9.69	1	6.48	1.496	0.2316

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 97.31 + (6.98 * A) - (7.83 * B) - (0.53 * A^2) - (5.34 * B^2) + (9.69 * AB)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 114.79 - (0.39896 * \text{Feed Rate}) \\ & + (4.8218 * \text{Feed \% Solids}) \\ & - (2.137E-04 * \text{Feed Rate}^2) \\ & - (5.3379 * \text{Feed \% Solids}^2) \\ & + (0.19374 * \text{Feed Rate} * \text{Feed \% Solids}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	100.68	101.98	-1.30	0.785	-0.213	0.028	-0.175
6	97.07	99.80	-2.73	0.711	-0.385	0.061	-0.323
4	96.66	92.63	4.03	0.951	1.380	6.149	1.866
1	97.92	89.15	8.77	0.624	1.086	0.326	1.138
7	96.98	97.31	-0.33	0.455	-0.034	0.000	-0.028
5	95.43	103.87	-8.44	0.471	-0.881	0.115	-0.835
3	54.97	66.59	-11.62	0.740	-1.730	1.421	-26.767
8	98.46	84.14	14.32	0.444	1.458	0.283	2.204
9	98.19	100.89	-2.70	0.818	-0.481	0.173	-0.409

STATISTICAL DATA ANALYSIS

Description: Conventional Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Combustible Recovery (%) (Minus Test Number 3)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed Rate	(ml/min)	50.000	150.000
B	Feed % Solids	(%)	2.000	4.000

Analysis Summary (Minus Test Number 3)

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	76321.29	1	76321.29		
Linear	6.47	2	3.24	1.552	0.2990
Quadratic	8.97	3	2.99	4.124	0.2013
Cubic	1.45	2	0.73		
RESIDUAL	0.00	0			
TOTAL	76338.18	8			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	3	5	1.444	0.3831	0.1363	30.289
Quadratic	6	2	0.852	0.9141	0.6995	72.457
Cubic	8	0				

ANOVA for Linear Model (Combustible Recovery / Conventional / Elkhorn)
Minus Test Number 3

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	6.47	2	3.235	1.552	0.2990
RESIDUAL	10.42	5	2.084		
COR TOTAL	16.89	7			
ROOT MSE	1.444		R-SQUARED	0.3831	
DEP MEAN	97.674		ADJ R-SQUARED	0.1363	
C.V.	1.48%				

Predicted Residual Sum of Squares (PRESS) = 30.29

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	97.901	1	0.533	183.7	
A	-0.929	1	0.527	-1.762	0.1384
B	0.050	1	0.655	7.70E-02	0.9416

Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 97.901 - (0.929 * A) + (0.050 * B)$$

Final Equation in Terms of Actual Factors:

$$\text{Combustible Recovery} = 99.607 - (1.858E-02 * \text{Feed Rate}) + (5.042E-02 * \text{Feed \% Solids})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	100.680	98.779	1.901	0.466	1.801	0.943	2.719
6	97.070	97.850	-0.780	0.286	-0.640	0.055	-0.597
4	96.660	96.067	0.593	0.689	0.737	0.401	0.698
1	97.920	98.904	-0.984	0.363	-0.854	0.138	-0.826
7	96.980	97.901	-0.921	0.136	-0.686	0.025	-0.645
5	95.430	96.953	-1.523	0.208	-1.186	0.123	-1.251
8	98.460	97.951	0.509	0.398	0.454	0.046	0.415
9	98.190	96.985	1.205	0.454	1.129	0.353	1.170

ANOVA for Quadratic Model (Combustible Rec. / Conventional / Elkhorn)
Minus Test Number 3

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	15.44	5	3.088	4.259	0.2010
RESIDUAL	1.45	2	0.725		
COR TOTAL	16.89	7			
ROOT MSE	0.852		R-SQUARED	0.9141	
DEP MEAN	97.674		ADJ R-SQUARED	0.6995	
C.V.	0.87%				

Predicted Residual Sum of Squares (PRESS) = 72.46

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	96.253	1	0.576	167.1	
A	-1.281	1	0.457	-2.807	0.1069
B	0.322	1	0.479	0.6710	0.5714
A2	0.677	1	0.403	1.680	0.2349
B2	1.730	1	0.668	2.590	0.1223
AB	0.616	1	0.538	1.144	0.3710

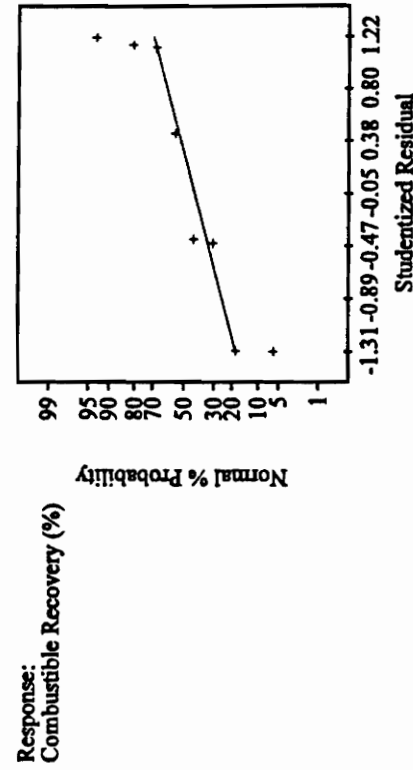
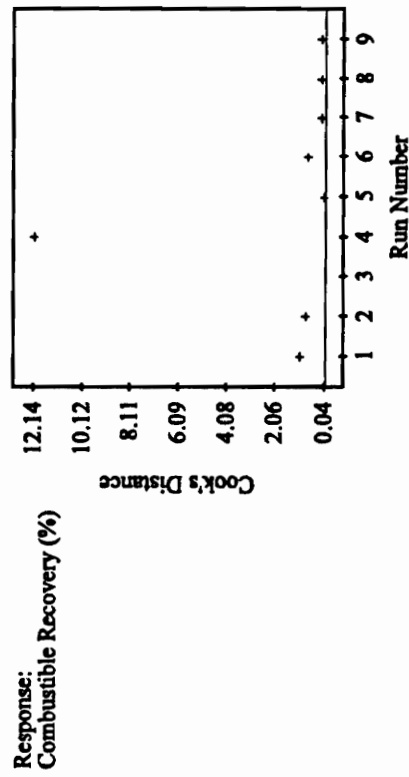
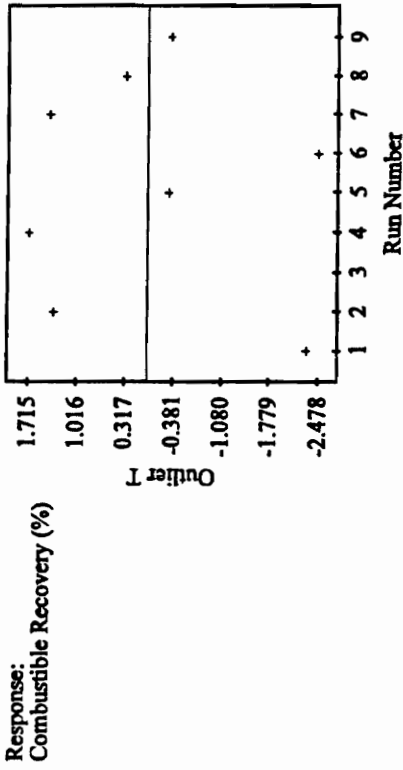
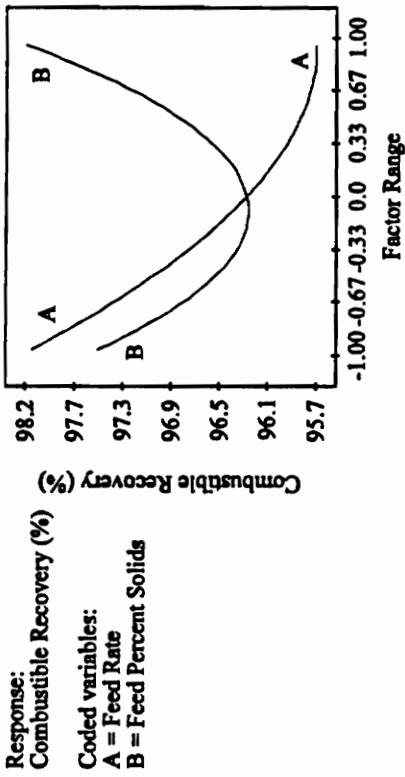
Final Equation in Terms of Coded Factors:

$$\text{Combustible Recovery} = 96.253 - (1.281 * A) + (0.322 * B) + (0.677 * A^2) + (1.730 * B^2) + (0.616 * AB)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Combustible Recovery} = & 119.825 - (0.11673 * \text{Feed Rate}) \\ & - (11.291 * \text{Feed \% Solids}) \\ & + (2.707E-04 * \text{Feed Rate}^2) \\ & + (1.7300 * \text{Feed \% Solids}^2) \\ & + (1.232E-02 * \text{Feed Rate} * \text{Feed \% Solids}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	100.680	100.236	0.444	0.791	1.142	0.823	1.368
6	97.070	97.662	-0.592	0.719	-1.311	0.735	-2.478
4	96.660	96.513	0.147	0.980	1.222	12.141	1.715
1	97.920	98.426	-0.506	0.790	-1.298	1.055	-2.308
7	96.980	96.253	0.727	0.457	1.159	0.189	1.429
5	95.430	95.650	-0.220	0.601	-0.409	0.042	-0.303
8	98.460	98.305	0.155	0.831	0.442	0.160	0.329
9	98.190	98.345	-0.155	0.831	-0.442	0.160	-0.329



STATISTICAL DATA ANALYSIS

Description: Conventional Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Separation Efficiency (%) (Minus Tests Number 3)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed Rate	(ml/min)	50.000	150.000
B	Feed % Solids	(%)	2.000	4.000

Analysis Summary (Minus Test Number 3)

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	3567.2	1	3567.2		
Linear	371.2	2	185.6	1.410	0.3268
Quadratic	423.8	3	141.3	1.206	0.4832
Cubic	234.2	2	117.1		
RESIDUAL	0.0	0			
TOTAL	4596.4	8			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	3	5	11.47	0.3607	0.1049	1982.21
Quadratic	6	2	10.82	0.7724	0.2035	9963.13
Cubic	8	0				

**ANOVA for Linear Model (Separation Efficiency / Conventional / Elkhorn)
Minus Test Number 3**

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	371.2	2	185.61	1.410	0.3268
RESIDUAL	658.0	5	131.60		
COR TOTAL	1029.2	7			
ROOT MSE	11.47		R-SQUARED	0.3607	
DEP MEAN	21.12		ADJ R-SQUARED	0.1049	
C.V.	54.33%				

Predicted Residual Sum of Squares (PRESS) = 1982.2

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	20.08	1	4.24	4.741	
A	6.24	1	4.19	1.489	0.1966
B	3.58	1	5.20	0.6872	0.5225

Final Equation in Terms of Coded Factors:

$$\text{Separation Efficiency} = 20.08 + (6.24 * A) + (3.58 * B)$$

Final Equation in Terms of Actual Factors:

$$\text{Separation Efficiency} = - 3.13 + (0.12479 * \text{Feed Rate}) + (3.5765 * \text{Feed \% Solids})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	-7.05	10.27	-17.32	0.466	-2.065	1.239	-4.813
6	27.61	16.50	11.11	0.286	1.146	0.175	1.193
4	24.44	28.48	-4.04	0.689	-0.632	0.295	-0.590
1	22.53	13.34	9.19	0.363	1.003	0.191	1.004
7	23.67	20.08	3.59	0.136	0.337	0.006	0.305
5	34.18	26.45	7.73	0.208	0.758	0.050	0.720
8	20.63	23.66	-3.03	0.398	-0.340	0.026	-0.308
9	22.92	30.15	-7.23	0.454	-0.852	0.201	-0.825

ANOVA for Quadratic Model (Separation Eff. / Conventional / Elkhorn)
Minus Test Number 3

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	795.0	5	159.00	1.358	0.4756
RESIDUAL	234.2	2	117.11		
COR TOTAL	1029.2	7			
ROOT MSE	10.82		R-SQUARED	0.7724	
DEP MEAN	21.12		ADJ R-SQUARED	0.2035	
C.V.	51.25%				

Predicted Residual Sum of Squares (PRESS) = 9963.1

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	31.21	1	7.32	4.264	
A	8.18	1	5.80	1.410	0.2939
B	2.15	1	6.09	0.3527	0.7580
A2	-5.78	1	5.12	-1.129	0.3760
B2	-9.46	1	8.49	-1.114	0.3811
AB	-6.26	1	6.84	-0.9144	0.4571

Final Equation in Terms of Coded Factors:

$$\text{Separation Efficiency} = 31.21 + (8.18 * A) + (2.15 * B) - (5.78 * A^2) - (9.46 * B^2) - (6.26 * AB)$$

Final Equation in Terms of Actual Factors:

$$\begin{aligned} \text{Separation Efficiency} = & - 137.40 + (1.0015 * \text{Feed Rate}) \\ & + (71.415 * \text{Feed \% Solids}) \\ & - (2.312E-03 * \text{Feed Rate}^2) \\ & - (9.4585 * \text{Feed \% Solids}^2) \\ & - (0.12515 * \text{Feed Rate} * \text{Feed \% Solids}) \end{aligned}$$

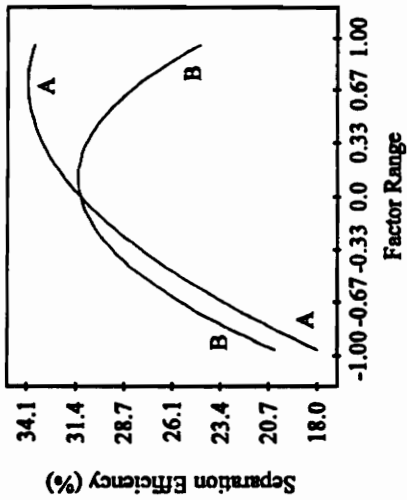
Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	-7.05	-0.62	-6.43	0.791	-1.301	1.068	-2.342
6	27.61	19.60	8.01	0.719	1.397	0.834	6.359
4	24.44	26.02	-1.58	0.980	-1.029	8.620	-1.061
1	22.53	15.63	6.90	0.790	1.391	1.213	5.460
7	23.67	31.21	-7.54	0.457	-0.946	0.126	-0.899
5	34.18	33.54	0.64	0.601	0.093	0.002	0.066
8	20.63	23.90	-3.27	0.831	-0.734	0.440	-0.607
9	22.92	19.65	3.27	0.831	0.734	0.440	0.607

Response:
Separation Efficiency (%)

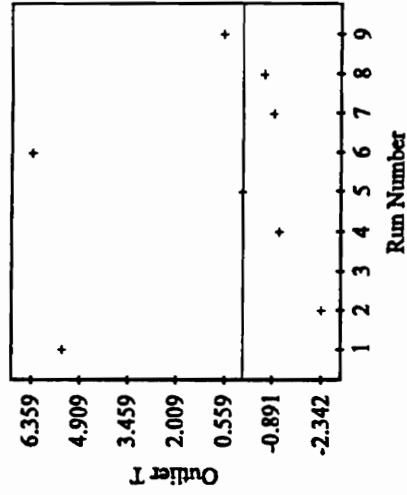
Coded variables:

A = Feed Rate

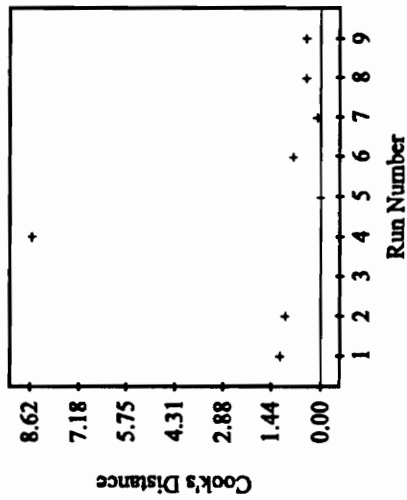
B = Feed Percent Solids



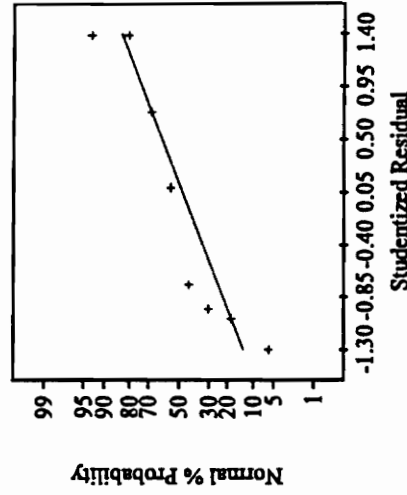
Response:
Separation Efficiency (%)



Response:
Separation Efficiency (%)



Response:
Separation Efficiency (%)



STATISTICAL DATA ANALYSIS

Description: Conventional Thickener
 Coal Sample: Elkhorn No. 3 (50% Ash)
 Response: Ash Rejection (%) (Minus Test Number 3)

Design Summary

FAC	FACTOR	UNITS	-1 LEVEL	+1 LEVEL
A	Feed Rate	(ml/min)	50.000	150.000
B	Feed % Solids	(%)	2.000	4.000

Analysis Summary (Minus Test Number 3)

Sequential Model Sum of Squares

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MEAN	4395.5	1	4395.5		
Linear	464.8	2	232.4	1.409	0.3271
Quadratic	553.1	3	184.4	1.357	0.4509
Cubic	271.7	2	135.9		
RESIDUAL	0.0	0			
TOTAL	5685.1	8			

ANOVA Summary Statistics of Models Fit

SOURCE	UNALIASED TERMS	RESID DF	ROOT MSE	R-SQR	ADJ R-SQR	PRESS
Linear	3	5	12.84	0.3604	0.1046	2483.71
Quadratic	6	2	11.66	0.7893	0.2626	11717.66
Cubic	8	0				

**ANOVA for Linear Model (Ash Rejection / Conventional / Elkhorn)
Minus Test Number 3**

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	464.8	2	232.42	1.409	0.3271
RESIDUAL	824.8	5	164.96		
COR TOTAL	1289.7	7			
ROOT MSE	12.84		R-SQUARED	0.3604	
DEP MEAN	23.44		ADJ R-SQUARED	0.1046	
C.V.	54.79%				

Predicted Residual Sum of Squares (PRESS) = 2483.7

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	22.18	1	4.74	4.677	
A	7.17	1	4.69	1.528	0.1872
B	3.52	1	5.83	0.6047	0.5718

Final Equation in Terms of Coded Factors:

$$\text{Ash Rejection} = 22.18 + (7.17 * A) + (3.52 * B)$$

Final Equation in Terms of Actual Factors:

$$\text{Ash Rejection} = -2.72 + (0.14330 * \text{Feed Rate}) + (3.5238 * \text{Feed \% Solids})$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	-7.72	11.49	-19.21	0.466	-2.046	1.216	-4.537
6	30.54	18.65	11.89	0.286	1.095	0.160	1.124
4	27.77	32.41	-4.64	0.689	-0.648	0.310	-0.606
1	24.61	14.44	10.17	0.363	0.992	0.187	0.990
7	26.68	22.18	4.50	0.136	0.377	0.007	0.342
5	38.75	29.49	9.26	0.208	0.810	0.057	0.778
8	22.16	25.70	-3.54	0.398	-0.356	0.028	-0.322
9	24.73	33.15	-8.42	0.454	-0.887	0.218	-0.865

ANOVA for Quadratic Model (Ash Rejection / Conventional / Elkhorn)
Minus Test Number 3

SOURCE	SUM OF SQUARES	DF	MEAN SQUARE	F VALUE	PROB > F
MODEL	1018.0	5	203.59	1.499	0.4465
RESIDUAL	271.7	2	135.86		
COR TOTAL	1289.7	7			
ROOT MSE	11.66		R-SQUARED	0.7893	
DEP MEAN	23.44		ADJ R-SQUARED	0.2626	
C.V.	49.73%				

Predicted Residual Sum of Squares (PRESS) = 11717.7

INDEPENDENT VARIABLE	COEFFICIENT ESTIMATE	DF	STANDARD ERROR	t FOR H0 COEFFICIENT=0	PROB > t
Intercept	34.95	1	7.88	4.433	
A	9.46	1	6.25	1.514	0.2691
B	1.82	1	6.56	0.2778	0.8073
A2	-6.45	1	5.51	-1.171	0.3623
B2	-11.19	1	9.14	-1.224	0.3456
AB	-6.86	1	7.37	-0.9313	0.4500

Final Equation in Terms of Coded Factors:

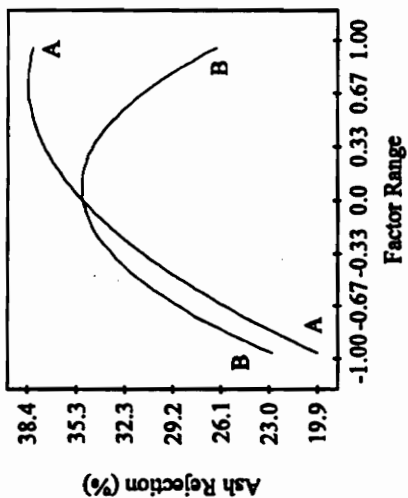
$$\text{Ash Rejection} = 34.95 + (9.46 * A) + (1.82 * B) - (6.45 * A2) - (11.19 * B2) - (6.86 * AB)$$

Final Equation in Terms of Actual Factors:

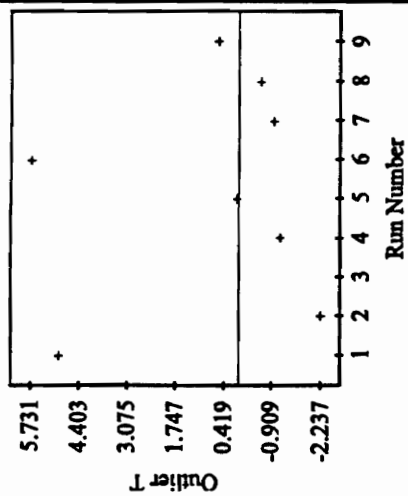
$$\begin{aligned} \text{Ash Rejection} = & - 157.15 + (1.1174 * \text{Feed Rate}) \\ & + (82.688 * \text{Feed \% Solids}) \\ & - (2.581\text{E-}03 * \text{Feed Rate}^2) \\ & - (11.189 * \text{Feed \% Solids}^2) \\ & - (0.13729 * \text{Feed Rate} * \text{Feed \% Solids}) \end{aligned}$$

Obs Ord	ACTUAL VALUE	PREDICTED VALUE	RESIDUAL	LEVER	STUDENT RESID	COOK'S DIST	OUTLIER t
2	-7.72	-0.84	-6.88	0.791	-1.291	1.053	-2.237
6	30.54	21.94	8.60	0.719	1.393	0.829	5.731
4	27.77	29.50	-1.73	0.980	-1.045	8.892	-1.098
1	24.61	17.20	7.41	0.790	1.387	1.205	4.987
7	26.68	34.95	-8.27	0.457	-0.963	0.130	-0.930
5	38.75	37.89	0.86	0.601	0.117	0.003	0.083
8	22.16	25.58	-3.42	0.831	-0.714	0.416	-0.584
9	24.73	21.31	3.42	0.831	0.714	0.416	0.584

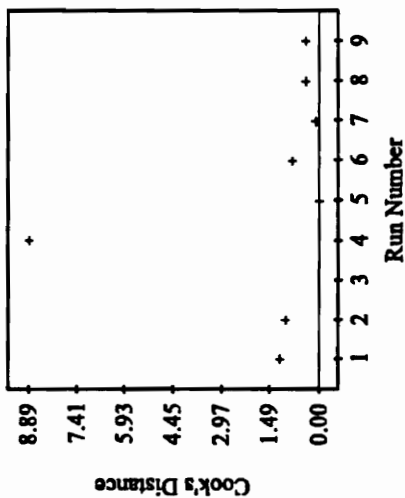
Response:
Ash Rejection (%)
Coded variables:
A = Feed Rate
B = Feed Percent Solids



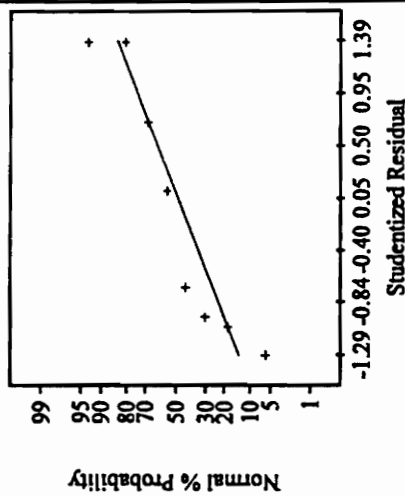
Response:
Ash Rejection (%)



Response:
Ash Rejection (%)



Response:
Ash Rejection (%)



VITA

Jason R. Pyecha was born during 1968 in Charleston, South Carolina. One year later his family moved to Raleigh, North Carolina, where his father began his graduate studies at North Carolina State University in Nuclear Engineering. In September of 1978 the family move to Lynchburg, Virginia. While in Lynchburg, Jason was graduated from Brookville High School in the top 10% of his class. He was accepted and enrolled at Virginia Polytechnic Institute and State University in 1986. After being in *General Engineering* for two years, he enrolled in the Mining and Minerals Engineering Department. He was graduated from Virginia Tech in 1991 with a B.S. degree in Mining and Minerals Engineering and a minor in Geology. After obtaining his B.S. degree, Jason entered the graduate program in the Department of Mining and Minerals Engineering at Virginia Tech. Upon completion of his M.S. thesis, he plans to begin his professional career working for Cominco Alaska at the Red Dog Mine Site.



Jason R. Pyecha