Manganese Contaminated Discharges: Treatment and Prevention

by

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<u>Abstract</u> Pen Coal Corporation has developed an intensive program involving several water treatment strategies on past surface mining operations and designed and fully implemented a Toxic Materials Handling Plan (TMHP) to prevent water quality problems on active surface operations. The development of the treatment strategies involved the assessment of existing treatments and a detailed raw water quality study including treatability studies. The most prevalent contaminant was manganese. The most practical and cost effective systems were evaluated and installed. The prevention of water quality problems on active operations involved a detailed hydrogeological and geochemical investigation of the affected overburden and coal. This investigation resulted in the development of a (TMHP) that has yielded significant reductions in manganese in the mine water regime.

Key Words: Treatability, Calcium Oxide, Sludge Disposal, Hydrologic Flow Paths, Cumulative Manganese, Neutralization, Leachability, Dilution

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3 Publication in this proceedings does not prevent authors from publishing their manuscripts, whole or in part, in other publication outlets.

Introduction

Pen Coal Corporation has an extensive mining operation located near Dunlow, in southern Wayne County, West Virginia. The operation consists of two active underground mines in the Coalburg Seam, one active underground mine in the 5-Block Seam, an active surface mine in the Coalburg Seam, an active surface mine in the 5-Block Seam, a preparation plant, a refuse fill and an impoundment. These operations are located in the watersheds of the East Fork and West Fork of Twelvepole Creek-

Mining operations began at the Honey Branch Surface Mine in September 1987, this operation consisted of the contour mining of the Coalburg Seam. During the summer of 1988, Pen Coal began mining operations at the Frank Branch Surface Mine which involved contour mining and point removal of the 5-Block Seam in the watershed of the East Fork of Twelvepole Creek, which is the major tributary of East Lynn Lake, a U.S. Army Corps of Engineers project.

The mining operations involving the 5 Block Seam have continued to expand to also involve the drainage areas of Kiah Creek, also a tributary of the East Fork of Twelvepole Creek-

Some minor water quality problems were detected during 1990, but were easily treated and corrected. As mining has progressed the elevation of the 5-Block Seam has continued to drop closer to drainage. This created some operational problems due to the lack of spoil storage areas. This also caused an increase in the quantity of surface water which encountered the mining area. During 1993 the water quality problem associated with the surface mining of the 5-Block Seam became more pronounced and has required a more intensive effort to control and abate. Pen Coal began an extensive "Water Quality Improvement Plan" in February 1994 to determine the most cost effective methods for treatment of the existing problems and methods to prevent or minimize future problems.

Statement of Purpose

The purpose of this paper is to share the experience of Pen Coal Corporation with the coal mining industry. The writers would like to specifically address the following points of significance:

- The most difficult contaminant to remove in AMD treatment is manganese. This dictates the type of treatments utilized on an operation where manganese concentrations exceed allowable regulatory limits.
- The development of a TMHP (Toxic Materials Handling Plan) to prevent the formation of acid or alkaline discharges which are contaminated with manganese, require a thorough understanding of the sources of the manganese and the post mining hydrologic regime.
- The success of water treatment and an on-going TMHP requires the cooperation and understanding of the scope of the problems by a wide range of individuals. This group consists of the company's upper management, supervisors, operators, and also involves the members of the regulatory agencies involved in the situation.

Water Treatment

The first phase of the "Water Quality Improvement Plan" was to utilize respected environmental consultants, to evaluate and recommend improvements in water treatment at our existing facilities. As a result of these evaluations it was determined that the most problematic contaminated waters were located on our Queens Fork surface mine. The discharges from this reclaimed operation flow into Kiah Creek which is a tributary of the East Fork of Twelvepole Creek. <u>Queens Fork</u> <u>Typical Raw Water Quality</u>

рН	Iron	Manganese	Aluminum	Flow
2.9-6.0	10-80	14-450	1-100	5-850 gpm
pH Iron 2.9-6.0 10-8	Manganese 14-450	Aluminum Flow 1-100 5-850 gpm		

A brief review of the above data illustrates that these discharges will require a good dose of alkalinity to counter the low pH's and high metals. The most difficult contaminate to remove based on bench scale treatability studies and actual field experiences was the manganese.

The most cost effective methods for the removal of manganese appeared to be elevation of the pH to the 10.0 range. Three chemicals were evaluated to determine their effectiveness: 20% sodium hydroxide, anhydrous ammonia, and calcium oxide. A typical analysis for one Queens Fork outlet is presented below:

Chemical	Treat. pH	Final pH	Final Mn	Sludge	\$/gal
NaOH	10.20	9.46	1.20	0.50"	\$.000238
NHs	10.15	9.36	1.30	0.20"	\$.000203
CaO	10.40	9.33	0.60	0.25"	\$.000119

A brief review of the data will show that the lowest final pH and chemical consumption cost was provided by utilizing calcium oxide. The calcium oxide cost was only half as much as sodium hydroxide or anhydrous ammonia. The sludge volume generated by calcium oxide was only half as much as sodium hydroxide and about the same as anhydrous ammonia.

The main cost considerations when evaluating a treatment strategy are fisted below:

- 1. Equipment cost
- 2. Chemical cost
- 3. Sludge disposal cost
- 4. Labor

A more detailed water quality inventory was conducted on the entire operation which indicated water quality problems of varying degrees existed at approximately twenty five sites associated with the surface mining of the 5 Block Seam. The degrees of contamination ranged from a manganese concentration of 3 to 5 mg/I to sites with severe acidity and metals problems. The treatment plan developed for each site was based on manganese removal since it appeared to be the primary contaminant at all the sites.

After careful review by Pen Coal Corporation and the participating regulatory agencies a plan was developed which is listed below:

- 1. Calcium oxide would be utilized as the treatment chemical of choice.
- 2. Aqua-Fix units would be purchased to dispense the calcium oxide.
- 3. A mobile dredge and pumping system would be purchased to aid in sludge disposal.

- 4. The sodium hydroxide treatment systems already in place on Queens Fork would remain as backup systems.
- 5. Additional ponds would be constructed where possible to increase retention time.
- 6. A staff of one supervisor and four hourly positions would be created to monitor and operate the systems.
- 7. An intensive effort would be made to improve access to all the sites to insure more effective treatment and monitoring of the sites.

By the end of 1994 approximately five miles of access roads were constructed, ten additional ponds were built, twenty-three aqua-fix units were purchased and installed, a dredging and pumping system was acquired, and three on-bench cells were constructed for sludge disposal.

Water Treatment Concerns

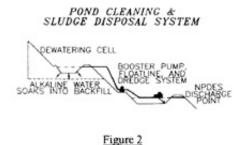
The primary problems associated with the use of calcium oxide are solubility. A staged treatment system was developed, where possible. at some sites. Treatment occurs at three different locations. See figure 1:

TACED WATER BACKEILL Figure 1

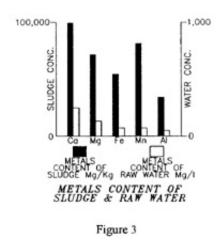
- 1. Initial dose of calcium oxide consumes acidity, and precipitates some iron, aluminum, and manganese.
- 2. Second dose of calcium oxide elevates pH > 10, and precipitates remaining iron, aluminum, and some manganese.
- 3. Sludge deposited in pond consists mainly of calcium, magnesium, iron, aluminum, some manganese, and pH drops to a range of 8 to 9 at pond discharge.
- 4. Third dose of calcium oxide elevates pH in excess of 10 to precipitate remaining manganese.
- 5. Sludge deposited in pond consists mainly of calcium, magnesium, iron, manganese, and aluminum.
- 6. The pH drops to a range of 9 to 10, for discharge into receiving stream, calcium content approximately 100 mg/I higher than raw inflow, manganese is typically less than 2 mg/l.

Pond Cleaning and Sludge Disposal

The precipitated metals are metal hydroxides which produce a gelatinous mixture with only a 5 to 10% solids content. This material is most effectively removed by pumping and dewatering. See figure below:



The content of the dried sludge was interesting -See figure below comparing raw water quality.



Water Treatment Costs Reduction

During the first six months of utilizing calcium oxide and Aqua-Fix units successfully to treat our discharges, we noticed several areas where improvements could be made to increase efficiency and reduce costs. The major problem incurred was delivery of the calcium oxide to the treatment units. We were currently utilizing calcium oxide delivered in 50 lb. bags on pallets. This system was time consuming and created a disposal problem with bags and pallets. We were currently using approximately two tons per day of calcium oxide which involved the handling of 80 bags per day or 2400 bags per month.

Pen Coal decided to determine the feasibility of going to a mobile bulk system. Since daily consumption was spread out over a dozen different sites, the installation of a large bin at each site would be very costly. We decided on the installation of a 55 ton bulk bin (approximately one month consumption) and a delivery truck of 2-1/2 ton capacity (approximately one days consumption). The delivery truck consisted of a bulk delivery bin similar to a small ANFO system used on surface mines. This bin was mounted on a four-wheel drive truck for easy access to our units.

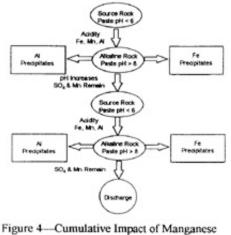
The cost bulk calcium oxide was approximately 60% cheaper than the bagged material. This system was completed in February 1996 and has resulted in considerable savings in chemical costs.

Prevention

The second phase of the "Water Quality improvement Plan" was to identify the stratagraphical sources of the acidity and metals, and to develop and implement a prevention plan on our present and future operations.

A much more intensive overburden drilling program was initiated with a more detailed geochemical analysis of the overburden. Areas in the stratagraphic column were identified as being potential sources of the acidity, but it was somewhat unclear the source of the manganese. The typical AMD situation results in low pH water with the associated elevated iron levels, but the manganese content seems to be more variable and the most persistent contaminant.

Pen Coal has determined that in certain areas the manganese contaminant is similar in character to sulfate concentration. The manganese is **an indicator** of acidic conditions somewhere in the hydrologic flow path. Like sulfate, the manganese tends to be cumulative, with the only reduction in concentration due more to dilution rather than actual precipitation.



Dissolution

A procedure was developed in conjunction with Pen Coal and REIC Laboratories to determine the availability and the concentration of manganese in the coals and overburden material. The results of this analysis indicated that there were certain sensitive units which could easily liberate excessive concentrations of manganese.

A series of extractions were performed to first determine the total manganese content of the sample. Then two more extractions were performed to determine the leachability of the manganese. One extraction utilized 20 milliliters of pH 3 sulfuric acid solution and the other extraction utilized a 1 molar ammonium acetate/0.2% hydroquinone solution. Refer to the appendix for the detailed procedure and analysis.

The results of the analysis were very interesting. Total manganese values ranged form 3.25 mg/kg to 4,350 mg/kg while the pH 3 sulfuric acid extraction yielded values from 0.00(ND) to 16.4 mg/kg and the I molar ammonium acetate/0.2% hydroquinone extraction yielded values from 0.00(ND) to 340 mg/kg. There where some samples that yielded high values for both extractions and what was particularly interesting was that these samples had relatively low total manganese values. For example, sample No. 47 had a total manganese value of 77.6 mg/kg with a pH 3 sulfuric acid extractable manganese of 11.8 mg/kg and a I molar

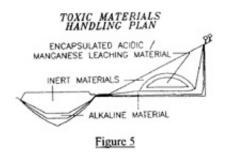
ammonium acetate/0.2% hydroquinone extraction manganese of 13.5 mg/kg. Sample No. 70 had a total manganese of 5 10 mg/kg pH 3 sulfuric acid extractable manganese of only 1.02 mg/kg and a I molar ammonium acetate/0.2% hydroquinone extractable manganese of only 6.41 mg/kg.

By utilizing data such as this, along with the acid base accounting data, a materials handling plan was developed with the prevention of manganese solublization being a main concern.

Development of a Toxic/Alkaline Materials Handling Plan

Based on the results of the detailed analysis of the 5-Block overburden sequence, a plan was developed to minimize hydrologic contact with acid producing/manganese leaching material, and to maximize hydrologic contact with alkaline high pH material.

As a result of the previously discussed overburden analysis, effort was made to place the alkaline overburden adjacent to highwalls, on pit floors and as blankets under hollow fills to induce alkalinity into the hydrologic flow paths. The acidic material and other material prone to easily leachable manganese would be placed high above the pit floor and away from hydrologic flow paths. The most acidic of this material would be placed in specially designated areas known as encapsulation cells. These encapsulation cells would then be covered with approximately eight feet of alkaline shales and siltstones to reduce permeability and then topsoiled and revegetated.



The water that does come in contact with the overburden that has a large amount of easily leachable manganese should be minimized. The small contaminated amount of water that does occur in the vicinity of the encapsulation cells should be adequately diluted with other alkaline water as it finally reaches the major hydrologic flow parts.

Conclusions

The effective treatment of manganese contaminated discharges requires a thorough understanding of all the variables such as site accessibility, flow rates, and degree of contamination. The success of the treatment also requires a commitment by the mining company to supply the needed manpower and equipment to perform the work effectively.

The development and implementation of a materials handling, plan to prevent manganese contaminated discharges requires the ability to locate the sources of the contamination and to be able to hydrologically isolate this material during active mining and reclamation.

The success of both the treatment and prevention of manganese contaminated discharges requires a great degree of cooperation between the mining company and associated regulatory agencies.

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APPENDIX A

<u>Principle</u>

The presence of manganese in AMD is of concern to both generators and regulatory agencies. We were hoping to gain insight into the process that leaches manganese into drainage water, and perhaps reduce the amount of manganese loading that takes place by special handling units that were visually obvious, and proven in the laboratory to leach high levels of manganese.

Overburden in the backfill consists predominately of sandstone and shale. It is one assumption that the shales contain manganese as manganese carbonates, whereas the manganese in sandstone is present as manganese oxides. To differentiate between the sources of manganese, overburden samples were extracted by dilute oxidation and by dilute reduction. Oxidation was accomplished by leaching samples in dilute sulfuric acid in an attempt to free manganese present as manganese carbonate. Reduction was attempted by using a 0.2% hydroquinone solution to reduce manganese oxides.

Total manganese was determined and compared to the result of the two leaching procedures to establish the percentage of manganese leached- The results of the leachings were then compared to the core description in an attempt to determine if units that leached high level of manganese could be identified in a highwall by visual cues.

Total Manganese by EPA SW-846 for manganese - see methodology shown below.

I. Method 3050 - Acid Digestion of Sediments, Sludges and Soils for Total Metals Analysis

II. Scope and Purpose

Method 3050 is an acid digestion procedure used to prepare sediments, sludges, soils, and many other solid samples for total metals analysis by flame or graphite furnace atomic absorption spectroscopy (FAA and GFAA, respectively) or by inductively coupled argon plasma emission spectroscopy (ICP).

III. Method Summary

A representative 2.0 g sample is digested in nitric acid (HNO_3) and hydrogen peroxide (H_2O_2) . The digestate is then refluxed with either HNO_3 or HCl, depending on which instrumentation will be used for the analysis. HCl is used as the final reflux acid where FAA or ICP analysis is to be used. HNO_3 is employed as the final dilution acid when GFAA analysis is required. Filtration is often required after final volume adjustment to remove remaining insoluble material.

IV. Apparatus and Materials

- A. Whatman #41 filter paper, filter funnels, and filtration rack
- B. Heavy-walled, 250 ml beakers
- C. Laboratory Balance, with readout to 0.01 g
- D. Hot plates, with variable temperature controls
- E. Disposable plastic pipettes, graduated
- F. Ribbed watch glasses
- G. Electronic timer
- H. 125 ml polyethylene bottles
- I. 1 ml Eppendorf pipette, with tips

Note: All glassware must be thoroughly detergent washed and rinsed with ASTM Type 11 mineralized water. Acid washing may be necessary, depending on analysis results for reagent blanks.

V. Reagents

- A. Type II demineralized water
- B. Concentrated nitric acid, reagent grade
- C. Concentrated hydrochloric acid, reagent grade
- D. Hydrogen peroxide (301/o), high purity
- E. Appropriate analyte spiking solutions

VI. Procedure

- A. Mix sample thoroughly to achieve homogeneity. Weigh out 2.0 g (to nearest 0.01 g) of sample and transfer to a heavy-walled beaker,
- B. Add 10 ml 1: 1 HNO₃ (5 ml Type 11 water and 5 ml concentrated HNO₃ mix, and cover with a watch glass. Heat sample to 95 degrees C (setting of 3.5) and reflux for 10 to 15 minutes without boiling. Cool sample and add 5 ML of concentrated HNO₃ cover and reflux for 30 minutes. Add another 5 ML of concentrated HNO₃, and evaporate to around 5 ML (without boiling or drying

the sample). Some samples may require additional HNO_3 for complete digestion. Cool the sample.

- C. Add 2 ML of Type 11 water and 3 ML 30% H202. Cover the beaker and return to the hot plate. Care must be taken to prevent losses of sample and analyte during the peroxide reaction. Heat until the reaction subsides and then Cool.
- D. Continue adding H202 in I ML aliquots with warming until the peroxide reaction is minimal or until the sample's appearance remains unchanged. Do not add more than a total of 10 ML of 30% H202.
- E. If the sample is being prepared for analysis via...
 - 1. FAA or ICP, then add 5 ML concentrated HCl and 10 ML Type II water, cover, and return beaker to the hot plate. Reflux for 15 more minutes. Cool and dilute to final volume (usually 5 ML) in a 125 ML polyethylene bottle.
 - 2. GFAA, cover the beaker and continue heating the digestate until the volume has been reduced to approximately 5 ML. Cool and dilute to final volume (usually 50 ML) in a 125 ML polyethylene bottle.
- F. Filtration is generally necessary to remove remaining insoluble material. Filtration should be carried out after final volume adjustment to minimize analyte losses. Filtering of the sample prior to volume adjustment can lead to reduced analyte recovery due to entrapment by the filter.
- G. Calculate the dilution factor for the sample using the following formula:

Dilution Factor (DF) =

wt of total sample after volume adjustment wt of initial sample

For a 2.0 g sample and a final volume of 50 ml (50 g), the dilution factor will be 25: 1. Label the sample container with job #, sample #, dilution factor, and digestion method (3050). If the sample was digested for FAA or ICP analysis, label the bottle as "3050-flame". For GFAA analysis, label as "3050-fur". Analytes to be quantified should also be included in the labeling.

Note: Quite often, sample sizes other than 2.0 g and final volumes other than 50 ml (50 g) will be necessary. Possible reasons for these deviations include...

- 1. Very small amounts of sample (less than 2.0 g)
- 2. The need for lower quantifying/detection levels-Thus the need for a lower dilution factor. Lower dilution factors can be achieved by using a sample weight higher than 2.0 g and/or using a lower final volume.
- 3. The need to minimize matrix complexity, interference, etc. This is generally accomplished by increasing the dilution factor, i.e. Using a larger final volume or a

VII. Quality Control

- A. Reagent Blanks: For each batch of samples, a reagent blank is prepared. The reagent blank consists of a 50 nil aliquot of Type 11 demineralized water carried through the entire digestion procedure, i.e., treated just as a sample. Labeling for blanks should include the word "Blank," the digestion method (3050), flame or furnace (fur) analysis, and the job #'s to which the blank applies.
- B. Duplicates: Duplicate samples are processed on a 10% basis. Samples with uncommon or questionable matrices are given priority for duplication. Jobs with three or more samples are also given priority for duplication. Duplicates are denoted by an "x" after the sample
- C. Spikes:
 - 1. A matrix digestion spike and a distilled water digestion spike are to be prepared for each batch of samples digested. Spiking is accomplished by adding one or more I ml aliquots of appropriate spiking solutions to a duplicate of the sample to be spiked The spike must contain all of the analytes that are to be measured for that sample. A 5 ml aliquot of Type II water is also to be spiked Oust as the matrix spike) and processed just like a sample.
 - 2. Samples with uncommon or questionable matrices are to be given priority for spiking. Jobs with three or more samples are also given priority for spiking.
 - 3. Several different multielement spiking solutions have to be made up and maintained to cover the wide range of analyte combinations that are possible. Two different concentrations of each spiking solution have to be made up and maintained. Which concentration (high or low level) of spiking solution to be used is dependent on which instrument is to be used for the analysis. The low level spiking solutions will be used for GFAA work and the high level solutions for FAA and ICP work
 - 4. Clearly mark all spiked samples as to how they were spiked. Spiked samples are denoted by an "S" after the sample #.

VIII. Corrective Action

Corrective action, usually in the form of redigestion, will be necessary if any of the quality control (reagent blanks, duplicates, spikes) fails during the analysis.

X. References

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Easily Reducible Manganese - determined by using method 18-3.5 from Methods of Soil Analysis part 2, 2nd edition (Sherman et al., 1942; Adams, 1965)

For the study the sample weight was changed from 10 grams to 5 grams, but the 10: 1 ratio of milliliters of leaching solution to gram of sample was maintained.-We also modified the method by using a stir bar and stir plate set at 4(on a scale of 1-10) instead of a mechanical shaker. We stiffed the solution for 30 minutes.

I. Method

Add 50 milliliters of ammonium acetate-hydroquinone solution buffered to a pH of 7 to 5 grams of solute. Stir on stir plate at a speed of 4 out of 10 for 30 minutes. Filter through a Whatman #41 cellulose filter and run on ICP or AA.

II. Reagents

- 1. Type II distilled water
- 2. Neutral ammonium acetate (I N), containing 0.2% hydroquinone

III. Apparatus

- 1. Beakers
- 2. Stir plate & stir bar
- 3. Funnels
- 4. Whatman #41 filter paper

Manganese solubility in varying pH solutions

Manganese was determined by adding 10 grams of <2 millimeters material to 50 milliliters of digestion solution (pH from 8 to 2) and stiffing at medium speed for 30 minutes. To achieve pH 8 calcium oxide was carefully added to distilled water while being monitored by a pH meter. For pH 7 through 2 sulfuric acid was carefully added until the desired pH was reached. This is a very exacting procedure because the distilled water lacks buffering. After spinning for 30 minutes the solution is filtered through a 0.45 micron filter, acidified, and 0. 5 milliliters of concentration nitric acid (15 M) and run on the AA or ICP.

I. Reagents

- 1. Distilled water
- 2. Sulfuric acid
- 3. Calcium Oxide

II. Apparatus

REIC LABORATORY

Acid-Base Account ~

10. O	Date: October 17 1995				Calcium Carbo	Calcium Carbonate Equivalent In Tons/1000Tons of Material	Tons/1000Tons	of Material			By/Bw	
ate: d	Date: October 17, 1999 DEIC 14h #- 0805-36981 and 0995-37716	and 0995-	37716	-								1100
	000 LO 0000 LO 000	THICKNESS	ROCK	* SULFUR	POTENTIAL	NEUTRALIZATION	NET NEUTRALIZERS	ALIZERS	PASTE	TOTAL	NCIO	EASILY
SAMPLE	INTERVAL		TYPE	Pyrtte	ACIDITY	POTENTIAL	DEPOENCY	EXCESS	Ŧ	MANOANESE	LEACHABLE NN"	KEDUCIBLE MN
	0.00-6.00	6,00	Wa	0.02*	0.63	4.29	4,92		4.2	73.7	0.75	8.35
	6.00-12.00	6.00	SS	0.02	0.63	1.77		1.15	5.7	720	8.74	340
	12.00-16.95	4.95	SS	0.19	5.94	9,85		3.91	7.2	642	3.65	6.56
4	16.95-22.00	5.05	SS	0.21	6.56	13.01		6.45	7.5	424	3.53	10.7
	22.00-27.05	5.05	ss	0.04	1.25	18.95		17.70	8.2	538	322	24
	27.05-31.40	4.35	ss	0.03	0.94	13.52		12.58	8.5	375	1.84	14.5
-	31.40-35.70	4.30	ss	0.11	3.44	227.50		224.06	8.5	1100	0.42	31.1
	36 70-36 10	0.40	ss	0.27	8.44	210.00		201.56	8.1	4350	0.54	36.3
	36 10.37 00	080	SSMS	.60.0	2.81	1.27	1.54		1.7	43.2	0.36	32
	37 00.37 66	0.55	COAL	3,38	105.63	-23.48	129.11		3.3	16.4	1.89	2.87
2	01.10-01.10	0.36	H	0.75*	23.44	-12.62	36,06		4.7	37.7	3.52	6.96
=	04.00-00-00	010	HS	0.13*	4.06	1.01	3.05		1.7	28.4	Q	1.52
12	04/04-02/10		en/En	+	406	5.31		1.25	7.9	50.8	Q	1.34
13	40,40-42.05	9		+	a c	430		1.49	8.4	192	0.38	0.99
14	42.65-44.85	2.20	5	0.09	10.7			5 08	86	527	QN	2.33
15	44.85-48.85	4,00	R	0.12	3.75	8.13		000		300	030	141
16	48,85-51.35	2.50	HS	0.10	3.13	7.08		955	0'0	007	200	5
	51.35-56.10	4.75	ŝŝ	0.04	1.25	10.74		9.49	8.4	516	877	3
	56 10-58 70	2.60	ss	0.07	2.19	2.28		0.09	7.8	31	1.78	2.51
2 2	58.70-62.00	3.30	ss	0.04*	1.25	-1.00	2.25		7.0	94.0	3.43	3.04
2				-		0.04	1810		6.0	162	10.5	12.3

As Reference in EPA manual: EPA-600/2-78-054; Fiel

Research, Environmental and Industrial Consultants INC.

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- pH 3 Sulfuric Acid Manganese Extraction
 - 1M Ammonlum Acetate/0.2% Hydroquinone Manganese Extractant

- 1. Beakers
- Stir plate and stir bars
 pH meter

Acid-Base Account ~

REIC LABORATORY

Company: PEN COAL COMPANY Site: CORE 95-21

1995 Calcium Carbonate E -36981 and 0995-37716	STB: C	018. COTE 83-61											
JOD #f. 0995-36981 and 0995-37716 Merrestant Merrestant </th <th>Date:</th> <th>October 17, 1995</th> <th></th> <th></th> <th></th> <th>Calcium Carbo</th> <th>wate Equivalent in</th> <th>Tons/1000Tons</th> <th>of Material</th> <th></th> <th></th> <th>Bybu</th> <th></th>	Date:	October 17, 1995				Calcium Carbo	wate Equivalent in	Tons/1000Tons	of Material			Bybu	
Survet Theores Rock Rule Note Rule Net Rule Note	REIC J	lob #: 0995-36981	and 0995-	37716								-	0.000
NITENNIL Three Image Londry Monthul, DEFENDIO EXERCISE Diff Mundhuld EX15-63-00 5.25 SS 0.04* 1.25 0.26 0.89 2.14 7.1 1330 Ex15-63-00 3.50 SS 0.00* 2.81 7.05 2.14 7.1 1330 F80-72-80 3.50 SS 0.00 2.81 7.06 2.84 7.1 1330 F30-72-80 1.50 SS 0.00 2.84 0.26 0.66 8.4 7.1 1330 7.40-75-80 1.50 SHEM 0.04 1.25 1.52 1.52 1.52 1.53 1.50 7.80-74-80 1.20 SHEM 0.04 1.55 1.52 1.52 1.52 1.52 1.53 1.50 7.40-76-80 2.50 SS 0.04 1.55 1.52 1.52 1.52 1.52 1.52 1.52 1.42 2.76 2.78 1.50 1.42	SAMPLE	SAMPLE	THICKNESS	ROCK	% SULFUR	POTENTIAL	NEUTRALIZATION	NET NEUT	RALIZERS	PASTE	TOTAL	ege Vege	EASILY
E3156840 5.5 0.0r 1.25 0.26 1.25 0.26 0.29 5.31 7.46 6.2 81.9 7.1 1330 664,066300 0.39 55 0.07 5.31 7.46 8.4 306 72,0-73,00 1.50 55 0.03 0.38 0.03 2.81 7.06 8.2 8.19 306 72,0-74,00 1.50 55 0.38 0.39 0.38 0.12 9.50 1.75 1330 74,0-76,00 2.40 1.50 9.38 0.12 9.38 0.12 9.50 1.75 133 74,0-76,00 1.20 2.41 2.50 5.3 0.07 1.25 153 156 1	NUMBER	INTERVAL		The	Pyrtte	ACIDITY	POTENTIAL	DEFICIENCY	EXCESS -	H	MANDANESE	LEACHABLE MN"	REDUCIBLE MN
0 050 SS 017 5.31 7.45 2.14 7.1 1300 1 0530-7280 350 SS 0.09 281 7.06 4.26 8.4 306 1 72.80-7430 150 SS 0.09 281 7.06 8.4 306 1 72.80-7430 0.10 SH 0.37 0.36 0.39 9.39 0.12 9.50 1.50 1.50 1 74.00-76.00 2.40 LOSS - - - - 1.50 1.53 1.53 1 74.00-76.00 2.40 LOS 2.51 2.51 2.51 1.55 1.53 1.53 1 74.00-76.00 2.40 LOS 2.442 2.15 1.53 1.53 1 76.00-75.00 2.50 S1 0.25 1.55 1.55 1.55 1.55 1 2.60 2.61 2.52 1.55 1.55 1.55 1.55	5	63,15-68.40	5.25	SS	0.04*	125	0.26	665'0		8.2	81.9	1.66	120
(63.30-72.80) (35) (53) (56) (27) (26) (27) (26) (26)	a	68.40-69.30	0:00	SS	0.17*	5.31	7,45		2.14	1.1	1330	15.3	18.5
7280-74.30 150 53 0.03° 0.94 0.26 0.69 62 150 133 160 74.30-74.40 0.10 SH 0.30° 9.38 -0.12 950 7.5 133 133 74.30-76.40 0.10 SH 0.30° 9.38 -0.12 9.50 7.5 133 133 76.00-76.00 1.20 SHEM 0.04° 1.25 4.42 7.5 7.6 7.8 7.19 76.00-76.00 1.20 SH 0.04' 1.25 4.42 7.1 8.1 210 76.00-80.00 2.80 0.04 1.25 4.42 7.1 8.1 210 80.004330 2.90 SH 0.04 1.25 4.42 8.1 210 80.00530 2.91 0.06 2.50 3.92 3.14 0.1 2.1 2.1 80.0068.00 2.81 0.06 2.50 3.92 9.3 3.0 1.42 1.42 <td< td=""><td>8</td><td>69.30-72.80</td><td>3.50</td><td>SS</td><td>0.09</td><td>2.81</td><td>7.08</td><td></td><td>4.26</td><td>8.4</td><td>306</td><td>4.84</td><td>5.26</td></td<>	8	69.30-72.80	3.50	SS	0.09	2.81	7.08		4.26	8.4	306	4.84	5.26
7430-7440 010 SH 030' 9.38 -0.12 9.50 75 133 7440-7660 2.40 LOSS - - - - - - - - - 133 74.40-7660 2.40 LOSS - 1.55 1.52 -<	24	72.80-74.30	1.50	SS	0.03*	0.94	0.26	0.68		8.2	180	16.4	49.0
7440-7680 240 LOSS ·	52	74,30-74.40	0.10	SH	0:30*	9.38	-0.12	9,50		7.5	133	0.94	14.1
7680-7800 1.20 8HEM 0.04' 1.25 1.52 0.27 7.6 27.8 7800-80.80 2.80 8H 0.06' 2.81 417 1.36 8.0 318 7800-80.80 2.80 8H 0.06' 2.81 4.17 8.1 2.10 318 8080-83.30 2.80 8H 0.01' 0.31 0.51 0.20 8.1 210 8080-88.20 1.40 8H 0.01 0.31 0.51 0.20 8.1 4.30 86.80-88.20 1.40 8H 0.26 3.82 8.4 1.32 8.1 210 86.80-88.20 1.40 8H 1.25 315.00 313.75 8.1 4.30 86.20-85.20 3.20 8.2 0.04 1.25 315.00 8.3 352 86.20-85.20 3.20 8.2 0.04 1.25 315.00 8.3 352 92.20-95.20 3.20 8.2 0.04 1.1.		74.40-76.80	2.40	LOSS			•						
78.00-80.80 2.80 8.1 0.09* 2.81 4.17 1.36 8.0 318 80.80-80.30 2.50 55 0.04 1.25 4.42 3.17 8.1 210 80.80-80.30 2.50 55 0.04 1.25 4.42 3.17 8.1 210 80.80-80.30 3.50 51 0.01 0.31 0.51 0.51 3.17 8.1 210 86.80-80.20 1.40 51 0.03 3.92 3.92 1.42 8.7 183 86.80-80.20 1.40 51 0.03 3.92 3.92 8.7 183 86.80-80.20 2.75 55 0.04 1.256 3.1500 8.2 1610 86.20-80.20 3.20 55 0.03 1.256 3.1500 9.3 36.2 96.20-96.20 3.20 55 0.03 0.04 1.256 8.8 35.2 96.20-96.20 3.20 5 0.20 3.37.	36	76.80-78.00	1.20	SHIEM		1.25	1.52		0.27	7.6	27.8	0.34	4.37
90.000330 2.50 SS 0.04 1.25 4.42 3.17 8.1 210 80.0008330 3.50 SH 0.01* 0.31 0.51 0.51 8.7 8.7 133 86.0086300 1.40 SH 0.08 2.50 3.92 3.92 8.7 193 86.0086300 1.40 SH 0.08 2.50 3.92 3.92 8.7 193 86.20-80.20 2.75 SS 0.03 0.94 12.26 313.75 8.8 1610 92.20-95.20 2.75 SS 0.04 1.25 315.00 313.75 8.8 7610 92.20-95.20 3.20 SS 0.01 1.25 315.00 5.8 5.9 5.9 92.20-95.20 3.20 SS 0.01 0.23 313.75 8.8 7610 761 92.20-95.20 3.20 SS 0.01 0.23 7711 8.3 352 92.20-96.20 SS	27	78.00-80.80	2.80	FS	0.09*	2.81	4.17		1.36	8.0	318	0.34	2.49
83.30-86.80 3.50 SH 0.01* 0.31 0.51 0.20 8.3 4.30 86.80-86.20 1.40 SH 0.08 2.50 3.92 3.92 8.7 193 86.80-86.20 1.40 SH 0.08 2.50 3.92 1.42 8.7 193 88.20-99.25 1.05 SS 0.03 1.256 315.00 11.32 8.2 1610 99.256-92.00 2.75 SS 0.03 1.256 315.00 9.33.75 8.8 362 99.256-96.20 3.70 SS 0.04 1.25 315.00 0.93 313.75 8.8 362 99.250-96.20 3.70 SS 0.01 3.31 10.23 8.7 313.75 8.8 362 95.20-96.20 3.00 SS 0.01 3.31 10.23 10.23 150 352 96.20-96.40 1.20 SS 0.69 0.51 0.43 36.4 36.4	38	80.80-83.30	2.50	SS	0.04	1.25	4.42		3.17	8.1	210	0.58	1.53
86.80-86.20 1.40 SH 0.06 2.50 3.82 1.42 8.7 183 86.80-86.20 1.05 SS 0.03 0.94 12.26 11.32 8.2 1610 88.20-89.25 1.05 SS 0.03 0.94 1.256 315.00 8.2 8.8 352 89.25-92.00 2.75 SS 0.04 1.255 315.00 0.93 8.3 36.2 1610 99.25-99.20 3.20 SS 0.01 3.13 10.23 8.8 8.8 352 95.20-99.20 3.20 SS 0.01 3.13 10.23 7.11 8.3 1560 96.20-99.20 1.20 SH 0.02 0.94 0.01 9.93 36.2 36.2 98.20-99.40 1.20 SH 0.02 0.94 0.71 8.3 36.4 98.40-101.00 1.60 SH 0.51 0.43 2.81 8.4 36.4 101.00-103.65	8	83,30-86,80	3.50	HS	0.01	0.31	0.51		0.20	8,3	4.30	Q	96:0
88.20-89.25 1.05 SS 0.03 0.94 12.26 11.32 8.2 1610 89.25-92.00 2.75 SS 0.04 1.25 315.00 313.75 8.8 352 92.00-95.20 3.20 SS 0.04 1.25 315.00 0.93 8.8 352 92.00-95.20 3.20 SS 0.03 0.94 0.01 0.93 8.8 352 92.00-95.20 3.20 SS 0.03 0.94 0.01 0.93 7.11 8.3 352 92.00-95.20 3.00 SS 0.03 0.043 7.11 8.3 352 98.20-98.40 1.20 SH 0.02 0.94 0.51 0.43 8.4 36.4 98.20-98.40 1.20 SH 0.25 9.66 11.87 0.43 8.4 36.4 98.40-101.00 1.60 SH 0.63 0.65 9.06 11.87 0.43 8.5 20.8 1010.	8	86,80-88,20	1.40	ЧS	0.08	2.50	3.92		1.42	8.7	193	Q	1.41
89.25-92.00 2.75 SS 0.04 1.25 315.00 313.75 8.8 352 92.00-95.20 3.20 SS 0.03 0.94 0.01 0.93 16.3 352 95.20-96.20 3.00 SS 0.10 3.13 10.23 7.11 8.3 36.2 95.20-96.20 3.00 SS 0.10 3.13 10.23 7.11 8.3 1590 95.20-96.20 1.20 SH 0.03 0.94 0.51 0.43 8.4 36.4 96.20-96.40 1.20 SH 0.02 9.06 11.87 0.43 8.4 36.4 96.20-96.40 1.60 3.16 0.51 0.43 8.4 36.4 96.20-96.40 1.60 3.16 0.51 8.4 36.4 36.4 101.00-103.65 2.65 SH 0.63 1.165 7.8 35.2 101.00-103.65 6.00 SS 0.63 0.56 0.56 36.4	31	88.20-89.25	1.05	ss	0.03	0.94	12.26		11.32	8.2	1610	1.02	3.02
92.00-95.20 3.20 SS 0.03* 0.94 0.01 0.93 8.3 38.2 95.20-96.20 3.00 SS 0.10 3.13 10.23 7.11 8.3 1590 96.20-96.20 3.00 SS 0.10 3.13 10.23 7.11 8.3 1590 96.20-96.40 1.20 SH 0.03* 0.94 0.51 0.43 8.4 36.4 96.20-96.40 1.20 SH 0.29 9.06 11.87 0.43 8.4 36.4 97.10.100 1.60 SH 0.29 9.06 11.87 2.81 8.5 20.8 101.00-103.65 2.65 SH 0.23 0.63 1.65 7.8 35.2 101.00-103.65 6.00 SS 0.63 0.63 0.56 7.8 35.2 103.65-109.55 6.00 SS 0.63 0.36 0.25 7.8 36.6	3	89.25-92.00	2.75	ss	0.04	1.25	315.00		313.75	8.8	352	Q	5.33
95:20-96:20 3:00 S:1 10.23 7.11 8.3 1590 98:20-99:40 1.20 SH 0.03* 0.94 0.51 0.43 8.4 36.4 98:20-99:40 1.20 SH 0.03* 0.94 0.51 0.43 8.4 36.4 98:20-90:40 1.60 SH 0.29 9.06 11.87 8.1 8.5 20.8 99:40-101.00 1.60 SH 0.29 9.06 11.87 2.81 8.5 20.8 101.00-103.65 2.65 SH 0.63 1.65 1.02 8.3 35.2 101.00-103.65 6.00 SS 0.63 0.58 0.63 36.4 36.4 101.00-103.65 6.00 SS 0.63 0.58 0.63 35.2 20.8 101.00-103.65 6.00 SS 0.63 0.58 33.6 33.6	8	92.00-95.20	3.20	ss	0.03*	0.94	0.01	0.93		8.3	38.2	0.63	1.14
98.20-99.40 1.20 SH 0.03* 0.94 0.51 0.43 8.4 36.4 99.40-10100 1.60 SH 0.29 9.06 11.87 2.81 8.5 20.8 99.40-10100 1.60 SH 0.29 9.06 11.87 2.81 8.5 20.8 101.00-103.65 2.65 SH 0.02* 0.63 1.65 1.02 8.3 35.2 101.00-103.65 5.00 SS 0.03* 0.38 0.25 8.3 35.2 103.65-109.55 6.00 SS 0.03* 0.38 0.25 7.8 38.6	R	95.20-98.20	3.00	ss	0.10	3.13	10.23		7.11	8.3	1590	1.28	3.17
99.40-101.00 1.60 SH 0.29 9.06 11.87 2.81 8.5 20.8 101.00-103.65 2.65 SH 0.02* 0.63 1.65 1.02 8.3 35.2 101.00-103.65 6.00 SS 0.03 0.58 0.53 35.2 35.2 103.65-109.55 6.00 SS 0.03 0.38 0.25 7.8 33.6	8	98.20-99.40	1.20	R	0.03*	0.94	0.51	0.43		8.4	36.4	Q	Q
101.00-103.65 2.65 SH 0.02* 0.63 1.65 1.02 8.3 35.2 103.65-109.55 6.00 SS 0.03 0.38 0.25 7.8 38.6 103.65-109.55 6.00 SS 0.03 0.38 0.25 7.8 38.6	æ	99.40-101.00	1.60	HS	0.29	9.06	11.87		2.81	8.5	20.8	QN	Q
103.65-109.55 6.00 SS 0.02 ⁺ 0.63 0.38 0.25 7.8 33.6	37	101.00-103.65	2.65	SH	0.02*	0.63	1.65		1.02	8.3	35.2	QN	Q
	8	103.65-109.55	6.00	SS	0.02*	0.63	0.38	0.25		7.8	38.6	0.40	Q
	1	400 EE 410 M	236	55	0.02	0.63	1.14		0.51	7.8	90.4	0.72	0.66

~As Reference in EPA manual: EPA-600/2-78-054; Field and Laboratory Methods Applicable to Overburden and Minesolis.

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** - 1M Ammonium Acetate/0.2% Hydroquinone Manganese Extractant pH 3 Sulfuric Acid Manganese Extraction

Acid-Base Account ~

REIC LABORATORY

Company: PEN COAL CORPORATION Site: CORE 95-21

EIC J	REIC Job #: 0895-36981 and 0995-37715	and USBS	OL LO		POTENTIAL	NELTERAL CATION	NET NEUT	NET NEUTRALCERS	PASTE	TOTAL	VCID .	EASTY
SAMPLE	SAMPLE	THECKNESS	KOCK	SULFUR	MININ	NEUTRALIAN	NET NEAL	MAKENS				
NUMBER	INTERVAL		TYPE	Preto	NOIDHY	FOTENTIAL	DEFICIENCY	DXCESS	Ł	MANAAMESE	LEACHARLE MN	MEDUCIBLE MIN
40	112.00-115.50	3.50	SS	0.03	0.94	58.75		57.81	8.5	326	0.50	20.4
4	115.50-120.00	4.50	SS	-20.0	0.63	1.27		0.64	7.8	54.5	0.77	0.73
42	120.00-125.05	5.05	SS	0.03	0.94	3.29		2.35	7.9	69.0	1.02	0.91
\$	125.05-130.00	4.95	SS	0.01	0.31	-0.63	0.94		7.8	38.0	12	0.62
4	130.00-135.00	5.00	SS	0.02*	0.63	-0.50	1.13		8.1	40.2	1.48	0.53
\$	135.00-140.00	5.00	SS	0.04*	1.25	0.38	0.87		7.6	109	1.32	0.83
\$	140.00-144.40	4.40	SS	0.03*	0.94	0.64	0:30		7.5	143	1.67	1.08
1	144.40-145.60	1.20	SS	0.26*	8.13	-6.31	14.44		5.0	77.6	11.8	13.5
48	145.60-151.25	5.65	SS	0.07	2.19	3.67		1.48	7,3	364	2.62	3.26
9	161.25-164.00	3.65	55	0.13	904	9.95		6.79	7.8	744	3.24	4.60
8	154,90-158,30	3.40	SS	0.08	2.50	4.30		1.80	2.9	8	2.61	270
15	158.30-158.90	0.60	SS	0.13*	4.06	1.77	229		7.4	1270	3.42	4.84
52	158.90-159.25	0.35	SS	0.52*	16.25	10.87	5.38		7.3	5290	8.71	11.6
3	159.25-161.00	1.75	WS	-20.0	0.63	1.39		0.76	8.3	223	Q	g
3	161.00-162.40	1.40	WS	0.08	2.50	10.74		8.24	8.4	28.0	Q	Q
88	162.40-164.30	1.90	SS	.00	1.56	3.92		2.36	8.3	1330	0.45	1.30
8	164.30-164.80	0.50	WS	.90.0	1.88	-0.25	2.13		8.1	141	Q	Q
15	164.80-167.00	220	HS	0.08	2.50	10.11		7.61	8.3	1300	0.38	2.99
38	167.00-167.20	0.20	HS	0.15*	4.69	0.01	4.68		1.7	111	Q	Q
8	00 121 VC 201	410	HS	010	3.13	7.08		3.96	8.3	745	0.32	2.50

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- pH 3 Sulfuric Acid Manganese Extraction

** - 1M Ammonium Acetate(0.2% Hydroquinone Manganese Extractant

REIC LABORATORY

Acid-Base Account ~

AL CORPORATION	
V COA	-21
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pany:	с С С
Con	Site:

ate: C	Date: October 17, 1995				Calcium Carbo	Calcium Carbonate Equivalent in Tons/1000Tons of Material	Tons/1000Tons	of Material			By/Suu	
EIC J	REIC Job #: 0895-36981 and 0995-37716	and 0995-	37716									
SWPLE	SAMPLE	THOORESS	ROCK	% SULFUR	POTENTIAL	NEUTRALZATION	DEPICIENCY EXC	EXCESS	PASTE No	MANDANESE	NCID NCID	REDUCIBLE MN -
8	171.30-171.55	0.25	SHIC	0.57	17.81	1.64	19.45		7.0	13.4	0.32	4.14
15	171.55-172.70	1.15	SH	.90.0	1.56	0.64	0.92		5.9	76.1	QN	QN
82	172.70-172.90	0.20	COAL	0.70*	21.83	8.46	13.52		6.8	36.2	1.64	6.26
8	172.90-173.10	0.20	HS	0.04*	1.25	2.66		1.41	8.1	11.7	Q	4.23
3	173.10-174,50	1.40	WS	0.01*	0.31	1.01		0.70	8.3	3.25	Q	Q
8	174.50-175.60	1.10	SH	0.01	0.31	0.38		0.07	8.4	3.40	QN	QN
89	175.60-175.75	0.15	SH	0.03*	0.94	0.64	0:30		7.9	3.50	Q	Q
	175.75-178.00	2.25	COAL									
19	178.00-178.70	0.70	SH	60.0	2.81	3.92		1.11	8.3	156	Q	Q
8	178.70-180.45	1.75	SH	60'0	2.81	6.07		3.25	8.2	168	Q	Q
8	180.45-185.55	5.10	SS	0.04	1.25	13.01		11.76	8.3	431	1.04	6.37
2	185.55-190.15	4.60	SS	90:0	1.88	14.65		12.78	8.3	510	1.02	6.41
1	190.15-191.80	1.65	R	0.13	4.06	9.22		5.16	8.2	875	0.62	4.69
2	191.80-193.30	1.50	ss	0.13	4.06	8.59		4.53	7.4	623	3.30	6.03
2	193.30-194.35	1.05	HS	60.09	2.81	6.95		4.14	8.1	457	0.55	3.71
74	194.35-194.50	0.15	HS	90.06	1,88	6.57		4,69	8.2	407	0.32	3.11
	194.50-204.45	9.95	COAL									
75	204,45-204,55	0.10	SH	0.48*	15.00	-5.80	20.80		6.8	57.8	0.42	1.59
82	204.55-208.05	3.50	R	0.04*	1.25	1.90		0.65	8.1	131	g	1.36
4	208.05-213.05	200	ŝ	90,06	1.88	4,55		2.68	8.1	81.7	1.28	2.76

- pH 3 Sulfuric Acid Manganese Extraction
 - 1M Ammonium Acetate0.2% Hydroquinone Manganese Extractant

P.O. Box 286 Beaver, WV 25813 Phone: 1-800-999-0105 / (304) 255-2500 / FAX: (304) 255-2572

Research, Environmental and Industrial Consultants INC.

	_	-	-	-	_				-					 		 				
	EASLY	REDUCIBLE MN		0.51	0.68	0.57						•								
вувш	ACID	LEACHABLE MN*		QN	QN	QN													1c	Cuons
	TOTAL	MANGANESE		4.90	29.2	133											fleenalle	illesons,	55-2572 DI -11-	1 1 11
	PASTE	Ŧ		9.0	8.4	8.4											a base and a set of the	verburden and w	255-2572	APPROVE
/ Material	ALCERS	EXCESS			0.58												C et	Applicable to United in Un	500 / FAX: (304)	
ons/1000Tons o	NET NEUTRALIZERS	DEFICIENCY		0.05		1.50												xoratory methous Mai Consultants	05 / (304) 255-25	
Calcium Carbonate Equivalent in Tons/1000Tons of Material	NEUTRALIZATION	POTENTIAL		69'0	0.69	0.38												EPA manual: EPA-600/2-78-054; Field and Laboratory Methods Applicable to Overburgen and Minesons. Research Environmental and Industrial Consultants INC.	266 Beaver, WV 25913 Phone: 1-800-999-0105 / (304) 255-2500 / FAX: (304) 255-2572	
Calcium Carbon	POTENTIAL	ACIDITY		0.94	0.31	1.88												EPA-600/2-/16-	N 25813 Phot	
	% SULFUR	- Pyrtik		0.03*	0.01	-90'0												PA manual:	86 Beaver, W	
37716	ROCK	TYPE	COAL	ΒH	RH	SH												rence in E	P.O. Box 2	
and 0995-	THICKNESS		3,85	0.40	2.35	2.30												~As Reference in	a.	
Date: October 17, 1995 REIC Job #: 0895-36981 and 0995-37716	SAMPLE	INTERVAL	213.05-216.90	216.90-217.30	217.30-219.65	219.65-221.95														
Date: O	SAMPLE	NUMBER		78	79	8	+	-	1		1		T	t	-					

** - 1M Ammonium Acetate/0.2% Hydroquinone Manganese Extractant

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Acid-Base Account ~

REIC LABORATORY

Page 2 Pen Coal Corporation Job #: 0496-41821

CORE #: 95-21

Pen Coal Sample #	pН	Aluminum	Iron	Manganese
#2 pH 2.0	2.29	36.05	48.0	47.95
#2 pH 3.0	3.80	0.50	2.80	5.10
#2 pH 4.0	5.34	ND	ND	0.35
#2 pH 5.0	5.54	ND	ND	ND
#2 pH 6.0	5.70	ND	ND	0.30
#2 pH 7.0	5.82	0.80	ND	ND
#2 pH 8.0	5.78	1.10	ND	ND
#10 pH 2.0	2.09	85.0	6150.0	2.45
#10 pH 3.0	2.58	86.0	6800.0	2.35
#10 pH 4.0	2.69	74.0	6150.0	2.00
#10 pH 5.0	2.72	70.0	5400.0	2.10
#10 pH 6.0	2.68	83.0	6450.0	2.35
#10 pH 7.0	2.67	91.0	6400.0	2.70
#10 pH 8.0	2.68	86.0	860.0	2.45
#20 pH 2.0	2.14	33.80	855.0	20.15
#20 pH 3.0	3.16	17.60	142.5	17.30
#20 pH 4.0	3.70	13.15	96.5	16.25
#20 pH 5.0	3.79	13.05	109.0	16.10
#20 pH 6.0	3.92	11.90	109.0	16.60
#20 pH 7.0	3.83	12.15	104.0	16.85
#20 pH 8.0	3.82	11.65	117.0	14.35
#24 pH 2.0	2.17	20.45	113.5	35.95
#24 pH 3.0	3.98	ND	14.20	8.45
#24 pH 4.0	6.50	ND	ND	0.25
#24 pH 5.0	6.63	ND	ND	0.35
#24 pH 6.0	6.72	ND	ND	0.35
#24 pH 7.0	6.82	ND	ND	ND
#24 pH 8.0	6.90	ND	ND	ND

ND - None Detected at MQL

MOL - Minimum Quantifying Level

Page 3 Pen Coal Corporation Job #: 0496-41821

CORE #: 95-21

Pen Coal Sample #	pН	Aluminum	Iron	Manganese
#44 pH 2.0	2.13	12.40	234.0	2.00
#44 pH 3.0	4.45	ND	73.5	0.90
#44 pH 4.0	6.34	ND	ND	ND
#44 pH 5.0	6.60	ND	ND	ND
#44 pH 6.0	6.83	ND	0.90	4.55
#44 pH 7.0	6.84	0.55	ND	ND
#44 pH 8.0	6.96	17.30	4.05	ND
#47 pH 2.0	2.14	36.1	545.0	16.65
#47 pH 3.0	3.02	31.4	413.50	15.90
#47 pH 4.0	3.37	28.95	361.50	15.95
#47 pH 5.0	3.44	27.20	349.0	16.0
#47 pH 6.0	3.31	28.95	316.5	15.60
#47 pH 7.0	3.40	28.60	347.5	16.70
#47 pH 8.0	3.30	28.50	332.5	16.55
#70 pH 2.0	2.73	21.60	454.5	13.35
#70 pH 3.0	6.61	ND	ND	0.80
#70 pH 4.0	7.82	ND	ND	ND
#70 pH 5.0	8.05	ND	ND	ND
#70 pH 6.0	8.10	ND	ND	ND
#70 pH 7.0	8.04	ND	ND	ND
#70 pH 8.0	8.25	ND	ND	ND

ND - None Detected at MQL

MQL - Minimum Quantifying Level

DATE: 3/3/97

APPROVED: PA

Phil Evans

Page 3 Pen Coal Corporation Job #: 0496-41821

CORE #: 95-21

Pen Coal Sample #	pН	Aluminum	Iron	Manganese
#44 pH 2.0	2.13	12.40	234.0	2.00
#44 pH 3.0	4.45	ND	73.5	0.90
#44 pH 4.0	6.34	ND	ND	ND
#44 pH 5.0	6.60	ND	ND	ND
#44 pH 6.0	6.83	ND	0.90	4.55
#44 pH 7.0	6.84	0.55	ND	ND
#44 pH 8.0	6.96	17.30	4.05	ND
#47 pH 2.0	2.14	36.1	545.0	16.65
#47 pH 3.0	3.02	31.4	413.50	15.90
#47 pH 4.0	3.37	28.95	361.50	15.95
#47 pH 5.0	3.44	27.20	349.0	16.0
#47 pH 6.0	3.31	28.95	316.5	15.60
#47 pH 7.0	3.40	28.60	347.5	16.70
#47 pH 8.0	3.30	28.50	332.5	16.55
#70 pH 2.0	2.73	21.60	454.5	13.35
#70 pH 3.0	6.61	ND	ND	0.80
#70 pH 4.0	7.82	ND	ND	ND
#70 pH 5.0	8.05	ND	ND	ND
#70 pH 6.0	8.10	ND	ND	ND
#70 pH 7.0	8.04	ND	ND	ND
#70 pH 8.0	8.25	ND	ND	ND

ND - None Detected at MQL

MQL - Minimum Quantifying Level

DATE: 3/3/97

APPROVED: PA

Phil Evans

REIC LABORATORY

Acid-Base Account ~

Company. PEN COAL SERVICES Site: CORE 94-7

		DODD DIED D	20120									
SAMPLE	SAMPLE	THICKNESS	ROCK	% SULFUR	POTENTIAL	NEUTRALIZATION	NET NEUT	NET NEUTRALIZERS	PASTE	TOTAL	Ħ	EASILY
NUNBER	INTERVAL		BdvL	- Pyrtic	ACIDITY	POTENTIN	DEFICIENCY	EXCESS	ł	WANGANESE	Adjusted Mn	REDUCIBLE MN **
58A	175.31-175.51	0.20	BONE	0.34	10.63	5.18	5.45		2.7	871	7.05	46.0
568	175.51-175.99	0.48	COAL	0.03*	0.94	1.50		0.56	7.8	4,95	Q	1.35
580	175 99-176 60	0.61	COAL	0.02*	0.63	0.75	and the second second	0.12	7.6	2.80	QN	QN
580	176.60-177.01	0.41	BONE	0.02*	0.63	2.95		2.32	7.8	5.90	QN	0.64
SBE	177.01-177.83	0.82	COAL	0.01	0.31	0.88		0.57	2.7	16.2	09:0	4.22
S8F	177.83-176.27	0.44	SH	0 10*	3.13	3.45		0.32	7.8	12.7	0.25	2.47
58G	178.27-179.58	1.31	COAL	0.12*	3.75	1.50	2.25		7.7	35.5	09:0	6.02
HBS	179.58-179.99	0.41	SH	0.01	0.31	3.95		3.64	7.5	12.3	QN	6.25
581	179.99-180 72	0.73	COAL	0.08	2.50	1.38	112		7.0	9.40	0:30	1,44
58J	180.72-180.83	0.11	SS	0.18	5.63	6.40		0.77	7.0	16400	0.60	60.1
SBK	180.83-182.21	1.38	COAL	*10.0×	0.31	1.00		0.69	7.2	21.7	0.40	2.92
SSL	182 21-182 83	0.62	5	0.03-	0.94	2.23		1.29	7.2	17.3	0.30	8

P.O. Box 286 Beaver, WV 25813 Phone: 1-800-999-0105 / (304) 255-2500 / FAX: (304) 255-2572

Research, Erwironmental and Industrial Consultants INC.

** - 1M Ammonium Acetate/0 2% Hydroquinone Manganese Extractant pH 3 Sulfuric Acid Manganese Extraction

APPROVED And Elone

Phil Evans