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### AUTOMATED DATA ACQUISITION FOR ANALYSIS OF FLY ASH BY GRAPHITE FURNACE ATOMIC ABSORPTION SPECTROPHOTOMETRY

### A THESIS

### PRESENTED TO

The Faculty of the Department of Chemistry The College of William and Mary in Virginia

In Partial Fulfillment

of the Requirements for the Degree of

Master of Arts

Ъy

Eileen Dowds

### APPROVAL SHEET

This thesis is submitted in partial fulfillment of the requirement for the degree of

Master of Arts

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Approved, April 20, 1989

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Dedicated to

my mother, Kathleen Dowds

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### ABSTRACT

Fly ash deposits from a coal-fired electric generating plant were analyzed for seventeen elements present at trace levels. The amounts of these elements were determined using proton-induced X-ray emission (PIXE) and graphite furnace atomic absorption spectroscopy (GFAA). The effects of pH on leach rates from the fly ash for these elements were studied. Interfacing of the GFAA system to an Apple-Isaac computer data system was also developed. AUTOMATED DATA ACQUISITION FOR ANALYSIS OF FLY ASH BY GFAA

#### INTRODUCTION

### Environmental Significance of Fly Ash Disposal

the nation's seventh largest natural resource. Coal is Its combustion produces a tremendous amount of ash. In 1975, over 36 million tons of fly ash were produced in the U.S. (1,2). By the year 2000, particulate emissions may reach five million tons per Since the amount of fly ash produced is so great, many year. uses have been suggested for this material. These practical restoration projects, neutralization of acid mine include: wastes, highway construction, and fabrication concrete (3-6).

To safely utilize this material, it is important to understand mechanism of metal release from fly ash under natural the environmental conditions. The type of coal and area of origin effect levels of elements present. Coal has been into four groups based on appearance and physical categorized four groups are lignite, subbituminous, properties. The bituminous, and anthracite coal (7). The basis for this categorization is primarily carbon content and volatile matter. From these coal compositions, it is evident that combustion of certain coal types could release high levels of potentially elements. These elements include arsenic, selenium, toxic molybdenum, vanadium, nickel, boron, and lead. Most of cadmium,

these elements exist at higher concentrations in coal than in the earth's crust. These elements are further enriched, approximately tenfold, in fly ash due to the loss of organic matter during coal combustion. Levels of major elements (e.g., aluminum, calcium, iron, manganese, and silicon) are high in all coal types, as they are in the earth's crust.

Geographically, element profiles vary as the type of coal present in a region varies. Table 1 shows concentrations of major, minor, and trace elements by location. Generally, coals from the Western U.S. contain lower trace element and sulphur levels Supporting this, note that arsenic, cadmium, (7,9,10). and molybdenum concentrations are lowest in the Rocky mercury, Mountain coal shown in Table 1. In contrast, arsenic, cadmium, boron, lead, selenium, and molybdenum concentrations hiqh in the Interior Province coal. When coal are combustion takes place, elements incapable of escaping as a gas remain behind in various types of ash. The behavior of an dependent on both the temperature individual element is during coal combustion and the compound the element is reached present in. Most arsenic, cadmium, lead, and selenium would volatilize below 1550 degrees Celsius. species listed Under the right set of conditions in a furnace, these elements could escape as a gas and not be present in any residual ash. The presence of emission control devices in the furnace stack partly determines whether volatile elements are present in ash. Figure 1 illustrates the general configuration and flow in a coal-fired power plant.

Various stages of ash production are shown along with the stages of processing. Fly ash is a type of fine residual ash captured in emission control devices. As the name implies, this type of ash is airborne and can escape into the atmosphere if not caught by electrostatic precipitators. Bottom ash is another type of residual ash that remains in the furnace after coal combustion occurs. Bottom slag, or boiler slag, is produced when bottom ash is melted down. The amount of each ash produced is dependent on the type of furnace used.

Table 2 contrasts concentrations of some potentially toxic elements by type of ash and coal. Clearly, elements are enriched in all types of ash compared to coal, due to loss of organic matter and mass during the combustion process. The difference in concentration for each ash type can be explained by element differences in ash particle size and temperature/processing at various areas in the power plant. Fly ash is primarily composed of the smallest ash particles. Volatile elements condense on fly ash particles before they leave the furnace stack as a the data in Table 2, it appears that lead, qas. From molybdenum, and arsenic display this type of behavior to a great Elements not volatilizing under furnace temperatures degree. are found in uniform concentrations in both bottom slag and fly behavior are called matrix ash. Metals displaying this elements and include iron, aluminum, silicon, and calcium Aluminum has been used as a reference element to (13).for other elements since its compute enrichment factors concentration neither varies much within the fly ash particle nor

from fly ash to slag (12).

Release of trace elements from fly ash is more pronounced when in contact with fluids, where the metals are more mobile and Since fly ash is often disposed in pits open to soluble. environmental conditions, rainfall can start the leaching process. The solubility of each element is pH dependent. Table 3 contrasts metal concentrations in solutions leached from both soil. Note the dramatic increase in Mojave fly ash and solubility of certain fly ash elements as the pH was adjusted from 12.5 to 6.5. As these elements become soluble and escape from the original disposal site, groundwater quality can be adversely Runoff from the fly ash site could contaminate nearby effected. soil, well water, and vegetation.

### Figure 1 \*

Coal-fired power plant using a flue-gas desulfurization sludge system



Tab	1e	1 *
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## Average U.S. coal composition by location

Element	Appalachian Region	Interior Province	Gulf Province	Northern Great Plains Province	Roc <b>ky</b> Mountain Province	Alaska
Major elements						
(percent)						
Sulfur, total	2.3	3.9	1.9	1.2	0.6	0.2
Sulfur, sulfate	0.09	0.27	0.33	0.03	0.05	0.01
Sulfur, pyritic	1.56	2.37	0.59	0.76	0.19	0.07
Sulfur, organic	0.74	1.25	0.96	0.37	0.32	0.12
Nitrogen	1.3	1.2	0.4	0.9	1.2	0.7
Silicon	2.7	2	6.6	1.4	2.5	2.9
Aluminum	1.6	0.97	2.1	0.69	1.2	1.5
Calcium	0.12	1.2	1.2	0.97	0.59	1
Magnesium	0.068	0.089	0.291	0.255	0.104	2.5
Sodium	0.032	0.035	0.732	0.182	0.102	0.018
Potassium	0.23	0.16	0.3	0.04	0.076	0.12
Iron	1.9	3.3	2.2	0.75	0.45	0.38
Manganes <b>e</b>	0.062	0.014	0.024	0.0051	0.0036	0.0061
Titanium	0.09	0.052	0.16	0.042	0.061	0.077
Minor and trace						
elements (ppm)	1					
Antimony	1.2	1.7	0.9	0.6	0.4	2.7
Arsenic	27	21	6	3	2 `	3
Barium	100	70	200	500	200	700
Bervllium	2	3	2	0.5	0.7	0.7
Boron	30	100	100	70	70	70
Cadmium	0.7	7.1	1.3	0.2	<0.1	<0.2
Chromium	20	15	20	5	5	15
Cobalt	7	7	7	2	2	5
Copper	24	20.2	28	8.3	9.1	16.8
Fluorine	80	71	124	45	70	90
Gallium	7	5	10	3	3	5
Lead	15.3	55	20	5.3	5.5	5.9
Lithium	27.6	11	28	6.0	9.2	10.1
Mercury	0.24	0.14	0.18	0.09	0.06	4.4
Molybdenum	3	5	3	2	1.5	1.5
Nickel	15	30	20	3	30	10
Niobium	5	1.5	7	5	1	3
Scandium	5	3	7	2	2	5
Selenium	4.7	4.6	7	1.0	1.6	2
Strontium	100	50	200	150	100	100
Thallium	4.9	5.2	8.3	2.7	3.6	4.4
Uranium	1.4	3.3	3.2	0.9	1.6	1.2
Vanadium	20	20	50	10	15	30
Ytterbium	1	0.7	2	0.3	0.5	1
Yttrium	10	10	20	5	5	10
Zinc	20	373	40	25.6	9.9	24
						-

\* Source: ref. 7,8

## Table 2 \*

Progressive trace element enrichment in a coal-fired power plant

### (ppm)

Sample	Cu	Zn	As	Mo	SÞ	Pb	Se	Hg
Coal	9.6	7.3		0.99	-		1.9	0.070
Bottom ash	82	58	15	3.50	2.8	<5	7.7	0.140
Precipitator ash (inlet)	230	250	120	41.00	14.0	66	27	0.310
Precipitator ash (outlet)	320	370	150	60.00	18.0	130	62	-

## Table 3 \*

Comparison of element levels in saturation extracts of Mojave fly ash and 68 California soils

Element	Fly Ash (µg/ml) Water Soluble Before pH Adjust- ment (pH 12.5)	Water Soluble After pH Adjust- ment (pH 6.5)	Concentration in Soil Saturation Extracts (µg/ml) Mean Median		
Ca	476	38,234	128	60.0	
Mg	<1	8 <b>49</b>	38	12.4	
Na	287	900	524	45.0	
к	<100	<100	20	10.0	
Si	<0.6	<0.6	3.1	5.0	
в	<0.6	65	3.1	<0.1	
Ba	50	15	0.26	0.10	
Sr	61	333	0.93	0.18	
Al	< 2	<2	0.40	<0.01	
Cr	<1	<1	0.01	<0.01	
Fe	0.01	0.11	0.05	0.03	
Mn	<0.01	1.3	0.17	<0.01	
Cu	0.01	0.02	0.04	0.03	
2n	0.02	0.08	0.07	0.04	
Мо	0.12	1.11	0.73	<0.01	
Ni	0.01	0.13	0.02	<0.01	
Со	<0.01	0.09	0.06	<0.01	
v	<0.01	0.02	0.07	0.01	
Pb	<0.02	<0.02	0.05	<0.01	
Cđ	<0.01	0.03	-	-	
Ag	<0.001	<0.001	_	-	

\* Source: ref. 7,14,15

### CHAPTER I

### CHISMAN CREEK FLY ASH DISPOSAL SITE

The area examined during the current study was a fly ash disposal site located in Yorktown, Virginia. Fly ash was last deposited at this site in the mid-1970s by the Virginia Electric and Power Company (VEPCO). Figure 2 contains a map of the area under study.

Fly ash produced from 1957 to 1973 when the plant was stopped using coal and refinery coke and began using fuel oil. The refinery coke was obtained from the nearby AMOCO refinery. Refinery coke is derived from crude oil: therefore, certain elements, such as vanadium and nickel, present in especially high concentrations. When the VEPCO are plant was in full operation, one thousand tons of fly ash and two fifty tons of bottom ash were produced each week hundred This ash was deposited into three separate fly ash (16, 17). displays the location of the fly ash pits pits. Figure 3 in the study area.

fly ash had been deposited in the Chisman Creek Although watershed for several years, environmental effects went However, unnoticed during that time. in the early nearby residential wells were reported to 1980s, several green-colored water. The incident prompted the contain

State Board of Health and the State Water Control Board to undertake a testing program in the immediate area. Several residential wells were tested and found to contain high levels of vanadium. Vanadium and selenium were also present at higher than expected levels in Chisman Creek groundwater (18,19). The groundwater was probably contaminated by vertical and horizontal migration of fly ash. Later collaborative studies by the Virginia Institute of Marine Science and the Virginia Research Campus provided more detailed information Associated on the extent of trace metal contaminants. This work involved an extensive program of analyzing groundwater, stream water, fly ash, sediments, vegetation, and shellfish to understand the dynamics of trace metal mobility. The sampling program primarily involved pit C and the surrounding area, as shown in Figure 3.

geography and geology of the Chisman Creek watershed are The important in understanding the possible effects of drainage from the fly ash pits. The Chisman Creek watershed is located on the Virginia Peninsula, directly south of the VEPCO facility. The watershed encompasses about 4200 acres (16). The tidal creek is approximately 550 acres in area, 3.75 miles in length (in an east/west direction), .5 mile in width and 12 feet in average depth (16). Drainage from the tidal creek directed into Chesapeake Bay. The fly ash pits are located is on a 520 acre area of the watershed (16).

The geology of the area determines the flow of water as it emerges from the fly ash pits. The fly ash pits are located in

an area composed of two layers. The upper layer, known as the Tabb formation, is quite sandy and considered to be moderately permeable (16). This layer extends from the surface to an approximate depth of 20 feet (16).

The second layer, the Yorktown formation, is denser, less and composed of a fine clayish material (16). permeable, То excavate for the fly ash pits, material was only removed from formation. Data from previous studies suggested the Tabb when drainage reached the Tabb/Yorktown border, it would that flow laterally until reaching some surface stream (18,19). This phenomenon may limit contamination of areas beyond where the groundwater surfaced. The VIMS/VARC study indicated that only areas located downhill between an ash deposit and a major surface stream would be effected by this lateral flow.

information was determined from groundwater samples taken This wells in pit C during the VIMS/VARC through installed and wells previously installed by the the State Board of study Health and the State Water Control Board. Water samples contained many elements, including nickel and vanadium which were above background levels. The residential wells examined by the State Board of Health were located in an area subject to this drainage and were proven to be contaminated.

Another important mechanism for trace metal contamination was believed to be erosion and transport of fly ash into streams during storm activity (16). Since runoff from the fly ash pits during storm activity could be quite high, large amounts of particulate matter could be deposited in bottom

layers of nearby streams and made available to vegetation and other living organisms. Samples of stream water and surface particulates were taken and found to contain higher element levels during storm activity, which were correlated with more acidic conditions. The elements found at higher levels included nickel and vanadium. In addition, certain samples also contained higher levels of arsenic and manganese.

Surface sediments and sediment core samples were also analyzed. High levels of vanadium, nickel, arsenic, and copper were found in both samples types. The element levels in the surface samples to decrease with distance sediment were found fly ash pits. This phenomenon was downstream from the probably due to dilution, changes in sediment particle size, and particulate matter settling out (16). Oysters were also analyzed to determine the extent of element bioaccumulation. The samples were found to contain vanadium, up to 3.6 ppm (16). Since vanadium existed at relatively high levels in all samples analyzed in this study, a more in-depth future study of vanadium and its patterns seemed appropriate. Rather than taking samples from materials contaminated within Chisman Creek, fly ash would be taken back to the laboratory and subject to controlled conditions. The intent of the current study was to vanadium availability from fly ash under varying determine including changes in pH and ionic strength, which conditions, vanadium solubility. The profile of other would affect major, minor, and trace elements would be examined and compared to vanadium levels.

In order to improve analysis for vanadium during the course of the new study, existing instrumentation was modified and enhanced. A Perkin-Elmer graphite furnace atomic absorption spectrophotometer (Model 380) was interfaced with an Apple II+ microcomputer through an ISAAC (Instrumentation System for <u>A</u>cquisition <u>and C</u>ontrol) analog-to-digital converter. Control software for the Apple II+ computer was written in both Assembler and BASIC. This software controlled both the data acquisition and data reduction phase of vanadium analysis on the atomic absorption spectrophotometer.

Two modes of data acquisition were implemented. The first method involved sampling the analog signal from the recorder output. This method was a relatively slow form of data acquisition with the instrument providing a background-corrected signal. The second method involved making hardware modifications to the instrument so that components of signals available from method became accessible. The second method required a fast one algorithm for data acquisition; thus, Assembler routines became necessary. Since method two permitted the separation and display background and analyte element signals, chemical of both interferences became better recognized and understood.

## Figure 2\*

Chisman Creek study area and VEPCO facility in Grafton, Virginia



<sup>\*</sup>Source: ref. 16

## Figure 3 \*

Detailed map of Chisman Creek study area



\* Source: ref. 16

#### CHAPTER II

### EXPERIMENTAL METHODS

### Fly Ash Collection and Processing

During the current study, two sampling sites were selected from fly ash pit A (see Figure 3 for a map of the local area with locations of the three fly ash pits). Site 1 was located approximately one-quarter mile from Wolftrap Road adjacent to a pond and vegetation. Site 2 was located near a county landfill and probably received some landfill drainage due to the site's lower elevation.

At each sampling location, two sets of samples were taken. This included a shallow set from a depth of two to eleven inches, and a deeper sample from a depth of eleven to eighteen inches. Although procedures were used to limit contamination during sampling, the surface fly ash composition had been altered by environmental conditions since it was first deposited. Any sampling site could be composed of several types of ash and soil from the surrounding area. Consequently, metal concentrations varied considerably in this material.

The collected samples were stored in clean polyvinyl chloride (PVC) bags for transport back to the laboratory. All material from the same site was homogenized and sieved through acid-washed

polypropylene 2 mm screens. This removed most of the large material such as rocks, grass, and insects. The remaining material was placed on a double set of acid-washed polypropylene screens, 60 mesh on the upper tier and 120 mesh on the lower tier. This ash was then placed in an oven at 60 degrees Celsius to dry. The ash was gently sifted during the drying process so the sub-120 mesh particles could be collected in the lower tub. Although the sieved material reduced to a fine powder for both sites. the site 2 ash was noticeably darker and coarser than The shallow and deep samples were pooled from each site site 1. and set aside in a desiccator.

### Chemical Reagents and Buffers

Chloroacetic acid and sodium chloroacetate were chosen for the pH 3 buffer (pKa of chloroacetic acid = 2.85). Part a of Table 4 lists the concentrations of components used for the pH 3, pH buffers. The chloroacetic acid received was 5, and pH 8 analyzed by proton-induced X-ray emission (PIXE) and shown to contain lead, iron, nickel and other elements (see part b of for contaminants in each buffer component). Table 4 The chloroacetic acid was sublimed once prior to use in the pH 3 buffer. After each sublimation, the crystals were reenlisted and shown to be free from detectable contamination. by PIXE Once a 'clean' set of chloroacetic acid crystal was collected, it was used to produce sodium chloroacetate. The sodium required for this procedure was also analyzed and carbonate

found to be free of metal contamination. Sodium hydroxide was deliberately not used to produce sodium chloroacetate because of contamination with high concentrations of several elements.

the pH 5 buffer, acetic acid and sodium acetate were chosen For (pKa of acetic acid = 4.75). The acetic acid underwent two sub-boiling distillations prior to use. Both acetic acid and sodium acetate were analyzed PIXE and found to be by free from metal contamination. Several components were investigated for use in the pH 8 buffer. Among the most readily available and possessing a suitable pKb was tris(hydroxymethyl) aminomethane (THAM; pKb of THAM = 7.903). The protonated form of THAM easily synthesized by was combining concentrated nitric acid with THAM. THAM was also analyzed by PIXE and shown to be free from metal contamination. Any nitric acid used was distilled twice before use. The nitric acid was also used for cleaning laboratory glassware and utensils to prevent sample contamination.

The pH 3, pH 5, and pH 8 buffers were made up in the appropriate concentrations to result in a .01 ionic strength solution. The 5 buffer was also made up at .10 ionic strength to ЪЦ investigate the effect of differing ionic strengths on the dissolution of compounds in the fly ash. The site 2 fly ash required the higher ionic strength buffer to maintain the leachates ЪЦ 5. The lower ionic strength buffer was at sufficient to maintain the site 1 column effluents at pH 5. The columns were easy to maintain at pH 8 due to the natural Preparation of Leachates

Fly ash-filled chromatography columns were set up to simulate the natural leaching process occurring in the fly ash. A fly ash slurry was used to fill 50 ml glass burets equipped with Nucleopore polycarbonate filters. The filters (.40 µm pore size) prevented the escape of ash, while mobile phase flowed through the columns. The polycarbonate filters were held in place by a fine nylon mesh. Several types of mobile phases were chosen, based on properties such as pH and ionic strength. These properties had to be similar to those possessed by liquids presently leaching the fly ash in the disposal site. The columns were constantly replenished with these solutions, which were supplied as the columns drained at their own natural rate. The flow rate was determined by the origin of the ash (site 1 versus site 2), as well as the packing and settling of the columns. All site 2 columns drained more slowly than site 1, possibly due to the abundance of the dark coarse material which settled to the bottom of the Experiments were conducted to determine the best columns. ratio of ash to mobile phase that would give an acceptable flow rate and yield leachates with detectable metal concentrations. A ratio of 20 grams of fly ash to 20 mls of mobile phase was The fly ash slurry produced by this ratio could be selected. easily poured into the burets. Extra mobile phase was used to rinse all remaining ash into the burets and bring the final volume up to the same location in each buret. The column flow rates continued to vary and decrease until the ash reached a final stage of settling. This occurred about 120 hours after the columns were started. See Figure 4 for plots of flow rate versus volume eluted from each column.

A total of eight columns were leached to simulate a variety of conditions occurring in the fly ash disposal site. Three buffer solutions with pH 3, pH 5, and pH 8 were selected. The buffer solutions were produced with a low ionic strength so would not interfere with the natural dissociation they processes occurring in the ash. These buffers had to have a buffer capacity large enough to maintain the pH of the effluents to within a few tenths of the intended value column despite their ionic strengths. low Buffer components were chosen on the basis of their pKa/pKb values, freedom from metal contamination, availability, and required to maintain the desired pH. concentrations Where circumstances allowed, buffers components were synthesized from compounds already present in the laboratory. These compounds were analyzed and found to be free from metal contamination.

The eight columns will be referred to as column 31 (pH 3, site 1), column 32 (pH 3, site 2), column 510 (pH 5, site 1, ionic strength .01), column 511 (pH 5, site 1, ionic strength .10), column 520 (pH 5, site 2, ionic strength .01), column 521 (pH 5, site 2, ionic strength .10), column 82 (pH 8, site 2).

The buffers were made up in large quantities and supplied to the columns by gravity as the liquid levels dropped in the burets. This was accomplished by inverting a volumetric flask over the buret and allowing the buffer to flow from a connecting tube. The rate at which liquid was supplied was dependent on the flow of buffer solution through the ash in an individual rate column. The leaching of metals was dependent on the pH and flow rate of a column. The columns with lower flow rates could exhibit higher metal concentrations due to the extended period of time the ash remained in contact with the buffer. For most of the experiment, the leachates were collected every in 15 ml acid-washed preweighed eight hours polystyrene tubes. This procedure lasted approximately twelve weeks, until changes in metal concentrations levelled off for each column.

Almost all leachates collected during the first several days were analyzed by GFAA and PIXE. As changes in metal became more gradual, fewer samples had to be concentrations analyzed to obtain column profiles. By the time the columns were stopped, a total of 2010 mls had been collected from column 31 over 1710 hours, 2890 mls from column 32 over 1710 hours, from column 510 over 1750 hours, 2290 mls from 2040 mls 511 over 1750 hours, 1860 mls from column 520 over 1750 column 2560 mls from column 521 over 1990 hours, 1640 mls hours. from column 81 over 2310 hours, and 1170 mls had been collected from column 82 over 2230 hours.

Generally, at least 5 mls of leachate had to be available for

various analyses. For the leachates requiring filtration, due to presence of small fly ash particles, a larger volume had to be collected. Immediately after collection, the tubes were weighed. If filtration was required, it was performed before weighing. The samples were then preserved by addition of concentrated nitric acid until a pH of 2 or less was obtained.

All samples to undergo PIXE analysis were doped with Indium to produce a concentration of about 45 ppm of Indium. Samples to undergo atomic absorption analysis required no further preparation, other than dilution which varied from sample to sample.

### Table 4

Characteristics of pH 3, pH 5, and pH 8 buffers

a)	Composition of pH 3, pH 5, and pH 8 Solutions					
pH	Ionic Strength	Conjugate Base/Acid	Molarity	Acid/Base	Molarity	
pH 3	0.10	с1сн <sub>2</sub> 000 <sup>-</sup> Na <sup>+</sup>	0.0960	ClCH2000H	0.0680	
pH 5	0.01	CH3000-Na+	0.0096	сн <sub>3</sub> ссон	0.0054	
pH 5	0.10	СН <sub>3</sub> 000-Na <sup>+</sup>	0.0960	сн <sub>3</sub> ссон	0.0540	
pH 8	0.01	(CH2OH) 3 CNH2 (H)	+ 0.0096	(CH2OH) 3 CNH2	0.0120	

# b) Concentration of Trace Metal Contaminants in pH 3, pH 5 and pH 8 Buffers (concentration in ug/g) $% \left( \frac{1}{2} \right) = 0.0125$

	Fe	Ni	Zn	Pb	Ag	Sr
Concentrated pH 3 buffer $0.9600 \text{ M ClCH}_2 \infty \overline{\text{Na}^+}$ $0.6800 \text{ M ClCH}_2 \infty \text{OH}$	3.5	2.1	<1.1	52.0	2.2	5.1
Concentrated sodium carbonate solution 0.4850 M Na <sub>2</sub> CO <sub>3</sub>	<0.20	<0.08	<0.04	<0.07	<0.40	<0.04
Concentrated sodium acetate solution 0.8270 M CH <sub>3</sub> 000 Na <sup>+</sup>	<0.13	<0.05	<0.04	<0.07	<0.30	<0.07
Regular pH 8 buffer 0.0096 M ( $CH_2OH$ ) <sub>3</sub> $CNH_2(H)^+$ 0.0120 M ( $CH_2OH$ ) <sub>3</sub> $CNH_2$	<0.06	<0.30	0.027	<0.02	<0.06	0.016
THAM solid sample 0.00874 g (CH <sub>2</sub> OH) <sub>3</sub> CNH <sub>2</sub>	2.0	<0.40	<3.6	<0.30	<0.60	<0.13



Comparison of flow rates for pH 3, pH 5, and pH 8 columns



Total Volume Eluted versus Time for all Columns

### CHAPTER III

### ANALYTICAL METHODS

Both graphite furnace atomic absorption spectrophotometry (GFAA) and proton-induced X-ray emission (PIXE) were used for analysis of leachates obtained from all columns. Vanadium was determined by GFAA. All additional elements were determined by PIXE. The concentration of forty elements, from silicon to uranium, can be determined simultaneously with PIXE. In addition to chemical analyses performed on fly ash samples, the material was also analyzed via scanning electron microscopy (SEM). This analysis provided qualitative information on the fly ash, which was useful in comparing the current samples with materials reviewed in the literature.

### Scanning Electron Microscopy

Both site 1 and site 2 samples were analyzed by SEM at the Virginia Institute of Marine Science. The fly ash samples were mounted on aluminum stubs with carbon paint. The samples were then coated with a Au-Pd layer of 150 Å in thickness to enhance conductivity of the electron beam. The coating process occurred in a vacuum evaporator taken down to a pressure of 5 x  $10^{-7}$  torr. SEM micrographs were taken with an AMR 1000

scanning electron microscope operated at an accelerating voltage of 20 kV. Micrographs were taken at several magnifications from 6X to 5200X. The most useful micrographs produced were taken between 230X and 700X magnification. Two micrographs of the site 1 and site 2 fly ash are shown in Figure 9.

### Proton-Induced X-ray Emission

Samples were doped with a known amount of the internal standard indium. Indium was selected as the internal standard since it was unlikely to exist naturally in the samples and it did not interfere with the analysis of other elements. Liquid samples, like the leachates, were 'spotted' as a thin film  $(0.5 \text{ mg/cm}^2)$ on a graphite-impregnated polycarbonate target that is subject to a beam of high energy protons. Each element then produces a characteristic X-ray spectrum that contributes to the total X-ray spectrum emitted by the sample. There is an attenuation of X-rays that occurs for each element based on the amount of sample on the target. This prevents light elements (low atomic number) from being analyzed accurately. In order to separate each elements' spectrum to determine concentration, a least squares analysis is used to obtain a theoretical set of amplitudes that is compared to the spectrum of the composite sample (20). The isolated amplitudes for each element can be converted to a concentration for that element based on the concentration of the dopant, indium. Accuracy is checked by routinely analyzing check standards that contain a known amount of several elements.
#### Graphite Furnace Atomic Absorption Spectrophotometry

Native Perkin-Elmer GFAA Mode

Vanadium analysis was accomplished through graphite furnace (electrothermal) atomization of samples on the Perkin-Elmer 380 atomic absorption spectrophotometer (which will be referred to as the PE 380 system). The standard Perkin-Elmer system provided for deuterium background correction required on the samples, which contained a high concentration of salts due to presence of acids and buffers in the leachates. The HGA 400 furnace could be programmed for the temperature and times required in the dry, char, and atomize cycles. The system also possessed an AS 40 autosampler that allowed for sample injection replication and calibration/recalibration of standards as needed.

A recorder was present on the Perkin-Elmer system, which could only display corrected samples absorbances. Peak heights or peak areas could be seen on the instrument's LED digital display and reported on a PRS 10 printer. During native operation, the instrument microprocessor could perform internal calibration with up to three standards. However, the 380 microprocessor only used a single atomization transient for recalibration and the original data was erased. This may be acceptable for flame atomization but it is not compatible with the poorer precision often found with furnace atomization.

To gain flexibility and accuracy, data analysis was performed on a DEC PDP-11 computer after completion of a run. The PDP-11

software better compensated for stringent calibration more requirements, especially where more than three standards were required. The ability to use several standards was important in defining calibration curves that were frequently non-linear at higher concentrations. All sample and baseline readings, available on paper tapes generated by the atomic absorption spectrophotometer, were manually entered data tables on the PDP-11 computer. Average baseline into were then subtracted from all sample absorbances. The readings resultant absorbances were used for all further calculations. A summary report generated by the PDP-11 computer program data (21) is included in Table 5.

Each run included the analysis of several blocks of standards to obtain a calibration curve. Generally, five different concentrations of standard solutions were analyzed in triplicate. Replicate absorbances for each standard were averaged before entry into a second order non-linear regression algorithm. If the relative standard deviation for any standard's absorbances rose above about 3.0%, the average absorbance was not used to produce the calibration curve (see column 9 in Table 5). In Table 5, note that the standard labelled '31 77 5' failed this requirement. As a result, the notation '\*' appears in the right column, which means the results were not stored. Before installation of the data acquisition system, the RSDs had to be manually monitored, since an entire run could be lost if the standards had poor reproducibility. Normally, this information was not available until the final report was

generated on the PDP-11. The advantage of a real-time data acquisition system was that poor RSD values could be flagged immediately after a standard's final replicate was The PDP-11 summary report also included other quality analyzed. parameters, such as the RSD for computed standard control concentrations, as determined from the calibration curve. The computed standard concentrations are in column 6 and the RSD values are located in column 7 of Table 5. The ratios of computed standard concentration to solution concentration are located in column 11. These ratios ranged from about 95% to 105%. Values from 98% to 102% were considered acceptable.

# The Apple II+/ISAAC Data Acquisition Hardware

The major drawback of the PDP-11 was that it was an off-line system. It could not display real-time absorbances which occurred during the char and atomization cycles. For this reason, a data system was highly desirable. acquisition This acquisition system would be required to display both the element plus background and background peaks during atomization. These peaks would aid in showing the sample matrix subtraction process effect of this matrix on the graphite tube. The and the data acquisition system would still have to provide the analysis previously available with the PDP-11 software. capabilities The first step in selecting a data acquisition system was to pick an analog-to-digital converter capable of sampling instrument signals at a 60 Hz frequency.

Slower sampling rates, as those used in liquid chromatography automation, were not sufficient to measure changes that occurred during atomization since an entire peak could be generated in two seconds. The A/D converter had to operate with an Apple II+ microcomputer, which had already been purchased by the laboratory. The ability to read autosampler tray position and other logic signals was also a high priority.

A data acquisition system manufactured by Cyborg Corporation (55 Chapel St., Newton, MA.), the ISAAC Model 91A. was a 12-bit successive approximation selected. ISAAC used analog-to-digital conversion technique to produce the digitized The ISAAC system includes values. sixteen single-ended analog inputs (or eight differential analog inputs), sixteen binary inputs, four Schmitt triggers, a clock, a timer, a counter, and a buzzer/beeper. Software for data acquisition was provided in both BASIC and assembler. The BASIC software provided, LabSoft, loaded into a 12K memory extender card on the Apple II+. The A/D converter was connected to slot number three on the interface board of the Apple II+ computer Figure 5 for a block diagram of the (see configuration).

### Recorder Mode Data Acquisition

Initially, only the instrument recorder signal was interfaced to ISAAC. For each sample injected into the instrument, analog signals were collected for two time periods. The baseline

absorbance was measured at the end of the instrument char cycle. The background-corrected vanadium peak was sampled during the atomization cycle. This is illustrated by Table 6, which displays the programmed furnace temperatures for each injection. Steps 1-4 represent the dry, char, atomize, and clean-out cycles, respectively. The temperature row represents the maximum temperature reached during the corresponding cycle. The ramp is the number of seconds required to reach maximum time temperature while the hold time is the number of seconds the maximum temperature is to be maintained. The scorder was activated 24 seconds into the char cycle and remained on until the end of the atomize cycle. The 'read' row in Table 6 indicates the relative time each absorbance signal was Note the measured. that gas flow was reduced during the atomize cycle, as indicated by the mini-flow selection, to enhance the absorbance signal. The parameters on the right of Table 6 include the analyst ID, run number, sample analysis date, element, and absorbance versus description, concentration selection. The wavelength of the lamp output was reported with the filter position, slit width, deuterium also lamp selection (ON/OFF), lamp energy, peak versus area analysis selection, and sample injection volume. The remaining parameters were only required for special analysis types (e.g., standard additions).

The software that collected the recorder signals was written in BASIC and LabSoft. Recorder peaks were sampled at a rate determined by the execution time of the

LabSoft commands. Testing was performed to determine the length of time required to execute commands for reading a single analog input channel.

Under these conditions, the sampling rate was a limited 35 msecs per point because LabSoft commands had to be 'interpreted' first before execution. Commands preceded by an ampersand sign (&) directed program execution to Assembler routines located in the 12K memory extender card, where LabSoft resided. Considering the number of points obtained per injection, data from approximately 60 samples could be stored on a single side of a floppy disk. The principal benefits of the data acquisition system were the acquisition and storage of complete transients and the elimination of manual data entry. In addition to obtaining the recorder signal, the position of each sample was automatically read from the autosampler tray as a binary-coded decimal number. A total of six signals were read into the binary-coded decimal inputs of ISAAC and translated to tray positions one through 39.

Although the 'recorder mode' of data acquisition was easy to implement, the analytical capabilities of the Perkin-Elmer graphite furnace system were not affected. The principal benefits were the acquisition and storage of complete transients and the elimination of manual data entry.

# Enhanced Data Acquisition and Hardware Modifications

The primary purpose for implementing the more complex 'enhanced

mode' of data acquisition was the acquisition and display of the signal. background correction Direct measurement of the background correction signal was not possible with the instrument operating as received. However, display of background information is very useful in method development and detection of problem samples (high dissolved solids). Thus, the enhancements improve facilitate method accuracy and development. All hardware modifications were performed SO that original instrument performance was unaffected. In the event the computer or ISAAC malfunctioned, the instrument could still be used in its native mode.

The second method for acquiring instrument signals required making additional connections to several sample and hold amplifiers on the analog and background correction boards of the Perkin-Elmer 380 instrument. The amplifiers were interfaced with shielded cables and connected to the first four analog inputs on the ISAAC A/D converter. The signals to the deuterium and hollow cathode lamp signals corresponded from the sample and reference beams. These signals will be referred to as BG (background signal - sample beam), I (hollow cathode signal - sample beam ), BGO (background signal -(hollow cathode signal - reference reference beam), and **I**0 beam). Each set of signals was acquired every 16.67 msecs (60 times per second). ISAAC was programmed to wait for a 12 volt timing pulse before the four channels of data were acquired from the instrument (see Figure 6 for timing diagrams of these signals). The PE 380 timing diagram was measured with a

Tektronix 564 oscilloscope. The voltage corresponding to each signal was available to ISAAC at any time after the gating interval shown in the timing diagram.

for absorbance calculations and instrument Equations used electronic experiments are given in Tables 7 and 8. A correction was applied to the rapidly changing background signals since they were not sampled at the same time as the hollow cathode signals. To compensate for this time difference, contributions of signal values from the current and next sampling period had to be applied to the signal currently being collected. The time at the hollow cathode sample beam signal was collected was which selected to be 'time zero'. For example, the BG signal was actually obtained by ISAAC 13.77 msecs after the instrument had updated the hollow cathode sample beam signal (I) and 2.90 msecs before the hollow cathode signal would be updated again. In order to calculate a corrected BG value, 82.6% of the current signal value had to be added to 17.4% of the signal value from the next sampling period (details are given in Appendix A). This correction was only necessary for the BG signal. The reference beam signals, BGO and IO, remained at a nearly constant level during the char and atomize cycles.

The experiments used to determine appropriate equations for calculating absorbance are summarized in Table 8. These experiments involved using fine wire mesh screens (0.35 and 0.80 absorbance units) to block both copper lamp radiation and deuterium (background) radiation. During this process, a digital voltmeter was used to measure voltages at

the sample and hold amplifiers where the BG, I, four IO signals were available. The lamp current was BGO, and varied from nine ma through 12 ma to demonstrate compensation features of the instrument. In Table 8, note the instrument will always maintain the same voltage level for the hollow cathode reference beam signal (IO), even when the energy of the deuterium lamp changes (BGO). When the energy output of the hollow cathode lamp drifts, the auto gain circuit for the PMT will adjust to compensate for the drift. Thus, the hollow cathode signal in the reference beam is maintained at 2.48 volts, independent of the lamp current. Note that the reference sample signals not equal, even under balance and are conditions when there is no actual sample absorbance. Also note the voltages of BGO and IO remain at the same level (at constant hollow cathode lamp energy), although the conditions ranged absorbance (balance) to 0.80 absorbance. from zero This fact permits use of I' and BG' in the computation of hollow cathode and deuterium sample beam absorbances. The absorbance of radiation from the hollow cathode lamp can be defined for sample using expressions and reference signals that exist during the atomization process (see beam Log IO/I, A(t), represents the absorbance of Table 7). radiation from the hollow cathode lamp; IO and I are the lamp intensities after passing through the reference and sample respectively. Average values of IO and I, obtained and paths, computed during the char cycle, can be used to compute the hollow cathode lamp absorbance during atomization. A' is used to denote the absorbance under balance or 'no sample' conditions. The corresponding components are denoted as IO' and I'. Since instrument electronics force A' to a non-zero value, it must be subtracted from the total absorbance to obtain the true absorbance.

The absorbance expression in equation F simplifies to log I'/I, since IO' and IO should be equal. This principle applies to calculating the deuterium lamp absorbance, A(BG) in equation G. The final corrected absorbance, Acorr in equation I, is the difference between the hollow cathode, A, and deuterium absorbance, A(BG). A(t) - A' is the quantity of radiation the photomultiplier tube (PMT) that reaching has been atomic absorption (monochromatic absorption), reduced by molecular absorption, and scattering of light by particles. light reduced by other than monochromatic The amount of absorption can be determined and compensated for by quantifying the amount of deuterium lamp radiation absorbed.

### Atomic Absorption Application Software

Machine language software was required to obtain the BGO, IO, BG, and I signals from the instrument. Related software was available from Cyborg. However, to accommodate the speed at which data was updated on the instrument (every 16.67 msecs), additional machine language programming was necessary to overcome delays in BASIC and LabSoft. The BASIC section of the program was designed to call a machine

language subroutine when ISAAC received a 'start acquire' signal from the instrument. This signal was a 12 volt timing pulse present on the analog board of the instrument. As this signal appropriate 'high' state, which occurred every reached its 16.67 msecs. four analog signals were collected and The data was collected in the following order: digitized. BG, I. BGO, and IO. To preserve the data for later calculations, the individual digitized values were saved from each sampling period. Generally, for each sample injected, 300 sets of BGO, IO, BG, and I signals were collected over a five second period at the end of the char cycle. Acquisition during the char cycle was triggered by the recorder signal dropping to zero volts. Another 300 data sets were collected during the atomization cycle; that is, the collection was triggered by the atomize signal dropping from five to zero volts. The 600 sets of data were saved to disk between sample injections.

Within-run Calculations: Figure 7 is a flow chart illustrating the order of events during data acquisition and various calculations performed. Most of these computations were executed between sample injections, since there was approximately two minutes during this period. The actual computations performed in these equations are shown in Appendix A. Referring to the flow chart, the first set of signals to be collected was the hollow cathode/deuterium sample and reference beam signals occurring during the instrument char cycle. Next, these hollow cathode and deuterium sample beam signals were averaged and displayed on the monitor for the operator to view for each sample. Once the atomization cycle started, sets of these four signals were again collected. At this point, before any further calculations were performed, the operator was given the opportunity to ignore data for the current sample and go onto the next sample. If no intervention occurred, the timing corrections previously discussed were immediately applied to the deuterium sample beam signals collected during the atomize cycle.

All time-corrected signals were then used to compute the deuterium and hollow cathode lamp absorbance during the atomization cycle. The final corrected solution absorbance was then computed from these signals. Values for average deuterium and hollow cathode lamp drift were also computed background and corrected solution between samples. The absorbances were plotted on a monitor during the atomization cycle. This technique was very useful in predicting impending graphite tube failure. Frequently, the curve shapes plotted became erratic several injections before complete tube failure Figure 8 contains a real time plot of the deuterium occurred. background absorbance signal (highest signal) and the corrected hollow cathode absorbance signal. The peak height, peak area, and peak position were also reported for the corrected hollow cathode absorbance signal.

Data Storage: The flow chart in Figure 7 illustrates the parameters and values stored to disk for each run. These parameters include: tray position, peak crest, location of peak crest, each corrected absorbance value determined during the atomization cycle, each background absorbance value

computed from the first half of the atomization cycle and every fourth background absorbance value computed from the second half of the atomization cycle. The average hollow and deuterium lamp drift values were also stored. cathode If these values were less than .01 percent, zero was stored instead. Run parameters stored included the total number of injections per run, the element analyzed, and the number of points acquired per peak. The largest runs required three low density floppy disks for storage of both raw data and the computed values described above.

<u>Post-run Calculations</u>: The raw data values stored during the data acquisition were analyzed by secondary programs, which performed a number of calculations and sample labelling functions. A secondary file, or 'calculated' data file, was created by these programs after the raw data was analyzed. The main functions available in these programs are shown in Table B-1 in Appendix B.

The main goal after data acquisition was to compute concentrations for samples, standards, and QA samples. The next step in obtaining concentrations from raw data stored on the Apple II+ was to determine areas for each curve collected during data acquisition (item A for program 1 in Table B-1: area calculations). The difficulties encountered in this process are discussed in Appendix B along with special techniques used.

Once peak areas were computed, average peak height and peak area values had to be determined for each sample along with the relative standard deviation for peak height and area. The BCD

tray position previously stored with the raw data was used to determine replicates for each sample and was also later used in assigning sample identification information, dilution volumes, and adjusted sample weights. The sample labelling option (item C or D for program 1 in Table B-1) was generally selected next after area computation since any further data analysis required these labels. It was not necessary to enter adjusted weights if the sample was originally in a liquid state. sample Adjusted sample weights were only entered if samples were originally solid and underwent acid digestion. Other values entered upon selection of option C or D include label information ('STD' for standard, 'TEG' or 'EPA' for QA samples), dilution factors, and replicate number (the number of times an analysis was repeated on a sample). Data belonging to each sample could be reviewed to optionally flag outlying values if option B was selected in program 1.

After all options in program 1 were completed, the next task was to perform a second order non-linear regression on the data. This was accomplished by selecting options from program 2 listed in Table B-1. The regression was executed twice, once for peak height and once for peak area. Along with executing the non-linear regression, all other calculations necessary to determine final solution concentration, with the appropriate statistics, were also performed by program 2.

The regression was based on the first five standards analyzed during the run. Recalibration occurred every seven to eight samples to determine the direction and amount of any instrument

sensitivity change, which was compensated in calculation of solution concentration. Concentrations were computed for all solutions analyzed in the run, including standards, samples, Since there was no conversion back to a dry and OA samples. weight concentration for the fly ash leachates, the solution concentration was the final concentration reported for these A relative standard deviation was calculated for each samples. sample concentration and standard concentration computed from the Also, a mean relative standard deviation (MRSD) and regression. a relative mean standard deviation (RMSD) was computed for the first group of standards used in the regression. Finally, a third program was available to report the analysis results and raw data belonging to a run. The options for the report program are discussed in Appendix B.



Components of data acquisition system





PE 380 timing diagram (60 Hz)



# Figure 7

Data acquisition program flow chart

Apple II DAQ Program



Figure 8

# Transients produced during atomization as recorded by data acquisition system





# (a)

(b)

# Figure 9

Scanning electron micrographs of Site 1 (a) and Site 2 (b) fly ash

Trace element analysis results generated by PDP-11 computer

AA RE	ESULTS ANALY	FOF 7FD	ک ۱۳	410-5	174 Fi		ARSON		-	EU				
23	3-JUN-1	85		1.000	300100(	6 F 117 0	ACOUN	0	L CORNEL	TUR				
PO(P	ZERO)	P	1(P (	JNE) P	2 (P THO	) R	ASD(STAI	VIARI	DS)					
-	7780 <b>.7</b>		21	1582 -	.007706	9	,00501	35						
REGR	ESSION	EQ	JATI	IN, A/C	=P0+P1*	Atr2	*(A*72)							
1/	2	3	4	5	6	7	8	9	10	1	D	13	14	15
Ľ	SAMP	Ē	Ļ	DUTN	SOLTA C	0.1	ARCIDEAN			DPV 1		CONC		
TYPE	RKPG	אא.	REP	FACTOR	NEAN	IRSD	HEAN	28SD	UFIGHT	HEAN	LESP	1115.R.	LAR	
3:	1 77	2	0	.07611	.07582	2.4	457.2	2.3	0	,9961	2.4	.02373	STD	
3	1 77	3	0	.05742	.05788	1.0	380.9	,92	0	1.008	1.0	.01056	STD	
33	1 77	4	0	.03880	.03862	2.0	275.6	1.9	0	,9953	2.0	.01942	STD	
STOD	31 7	75	0	.01937	.01864	3.7	141.6	3.6	0	.01864	3.7	.000681	STDD	*
3	1 77	6	0	009850	.009857	2.3	76.10	2.2	0	1.001	2.3	.02281	STD	
HTX	15130	31	1	1.000	.000207	94	1.619	94	1.000	Less Tł	190	.000391	HTX	
TEGC	38904	1	1	9.000	.001509	21	11.76	21	0.	001509	21	.000312	TEGC	ŧ.
EPA1	22138	1	1	2,000	.03750	2.3	269.2	2.2	1.000	.07501	2.3	.001699	EPA1	
5102	1510	2	1	1.000	.019/1	2.0	347+6	1.4	1,000	.019/1	2.0	.000391	5102	
7100	1310		۰ ۸	1.000	070107	100	1+1/0	100	1.000	1 014	180 A A	.000077	0104 6TB	
3	1 77	5	ő	.01937	.01926	1.4	146.0	1.7	0	.9941	1.4	.01404	STD	
5105	1510	Š	1	1.000	.02399	2.0	174.0	1.1	1.000	.02399	2.0	.000476	5105	
5106	1510	6	1	1.000	00017	12	-1.333	.30	1.000	Less TI	nah	.000389	5106	
5108	1510	8	`1	1.000	.02665	2.0	191.9	•82	1.000	.02665	2.0	.000529	5108	
5101	1510	10	1	1.000	.01847	2.0	135.8	1.9	1.000	.01847	2.0	.000366	5101	
5107	1510	7	1	1.000	.02012	2.0	147.3	•45	1.000	.02012	2.0	.000399	5107	
5109	1510	9	1	1.000	.01758	3.2	129.5	3.2	1.000	.01758	3.2	.000564	5109	
3:	1 77	4	0	.03880	.03638	•89	261.8	•74	0	.9375	.89	.008342	STD	
	1 77	5	0	.01937	+01780	2.8	135.5	2.7	0	.9191	2.8	.02529	STD	
510	1510	11	1	1.000	.01844	2.0	123.5	.81	1,000	.01844	2.0	.000366	510	
510	1210	12	1	1.000	+01002	2.0	111.0	1.0	1.000	+01632	2.0	000328	510	
510	1510	14	1	1 000	+01070	2.0	01 14	110	1.000	01750	2.0	1000337	510	
510	1510	18	1	1.000	.01443	2.3	97.29	2.2	1.000	.01443	2.3	.000200	510	
510	1510	22	1	1.000	.01043	4.0	70.67	4.0	1.000	.01043	4.0	.000417	510	
3:	1 77	- 4	ō	.03880	.03239	.66	236.4	.42	0	.8349	.66	.005476	STD	
3	1 77	5	0	.01937	.01582	1.9	120.9	1.9	0	.8168	1.9	.01569	STD	
510	1510	26	1	1.000	.01276	4.7	75.76	4.7	1.000	.01276	4.7	.000600	510	
510	1510	27	1	1.000	.01012	2.5	60.24	2.2	1,000	.01012	2.5	.000250	510	
510	1510	32	1	1.000	.009552	2.6	56.90	2.1	1.000	009552	2.6	.000250	510	
510	1510	36	1	1.000	.008940	2.8	53.29	•27	1.000	008740	2.8	.000249	510	
510	1510	44	1	1.000	.00/405	3.3	44.19	.19	1,000	007405	3.3	.000247	510	
310	1210	21	1	1.000	+00/004	- <b>1.</b> V	41+81 204 4	4.0	1.000	2100	4.9	000281	010	
3. 21	1 77	7	۰ ۱	103000	+V2/07	+72	105 1	1 0		+/100	172	000007	עוכ מדס	
510	1510		1	1.000	.008004	3.4	41.74	1.3	1.000.	A00800	7.4	.000286	510	
510	1510	72	1	1.000	.006476	4.4	33.38	2.0	1.000	006476	4.4	.000285	510	
510	1510	81	1	1.000	.004532	6.3	23.38	2.8	1.000	004532	6.3	.000284	510	
510	1510	94	1	1.000	.005318	5.4	27.43	1.8	1.000	.005318	5.4	.000284	510	
510	1610	12	1	1.000	.002601	11	13.43	9.2	1,000	002601	11	.000283	510	
510	1610	39	1	1.000	.001282	36	6.619	36	1.000	.001282	36	.000463	510	
EFA1	22138	1	1	2.000	+03726	2.0	185.0	.85	1.000	.07452	2.0	.001480	EFA1	
3	1 77	4	0	.03880	+02418	1.4	181.0	1.4	0	+6231	1.4	1.008995	SID	
3	L //	5	Q	+01937	+01175	123	y <b>0.4</b> 3	,10	0	+6064	101	.003300	510	

 REPLICATES:
 BK
 PG
 NUH
 REP
 DRY
 HEAN
 S.D.
 REPS

 22
 138
 1
 .0747641
 .00170
 2

\* \* \*DATA STORED\* \* \*

# Table 6

Portion of PE 380 run sheet illustrating vanadium analysis conditions

	۱ · · · ·						
	STEP	1	2	3	4	5	Flameless AA Analyst DowDS
/	TEMP ( <sup>o</sup> C)	120	1600	2700	2800		Run $\frac{410}{5174}$ Sample FUXA64 LEACHATE Date 6/23/85
	RAMP TIME (s)	10	10	0	1		Element V ABS/CONC
	HOLD TIME (S)	60	20	2	3		Wavelength318.4nm Filter: IN/OUT
	RECORD		24				Slit(ALT).7 mm D <sub>2</sub> : ON OFF
	READ		24	-5			Lamp 15 ma MODE: TC INT PEAK
	BASELINE						Samp Volume 50 pl Method # /
	MINI FLOW			$\checkmark$			Alt volume STD ADD CONC
	ml/min						Alt soln
	STOP FLOW						
- 1							

Graphite Tube Number: 34 - 124 - 50

### Table 7

Equations derived for absorbance calculations

(' values indicate signals measured immediately before atomization)

- A) A = A(t) A'  $(A' = \log (Io'/I') \text{ and } A(t) = \log (Io/I))$
- B)  $A = \log (Io/I) \log (Io'/I')$
- C)  $A = \log (Io/I) + \log (I'/Io')$
- D)  $A = \log (Io/I * I'/Io')$
- E)  $A = \log (I'/I * Io/Io')$
- F)  $A = \log (I'/I)$  (since Io' = Io, autogain circuit)

Following this approach, equations for background absorbance and corrected absorbance were derived :

- G)  $A(BG) = \log (BG'/BG)$
- H) Acorr = A A(BG)

# Table 8

# Voltages measured at sample and hold amplifiers in the PE 380 system with corresponding computed absorbances

Copper lamp current/ absorbance conditions	BG0	BG	10	I	a <sub>A</sub>	<sup>b</sup> A(BG)
Cu lamp - 10ma						
Balance*	2.300	1.770	2.480	1.010		
0.35 absorbance 0.80 absorbance	2.300 2.270	0.800 0.277	2.480 2.480	0.458 0.160	0.343 0.800	0.345 0.805
Cu lamp - 12ma						
BGen 26.0**	1.695	1.308	2.480	1.017		
0.35 absorbance	1.693	0.582	2.480	0.455	0.349	0.351
0.80 absorbance	1.693	0.200	2.480	0.160	0.803	0.815
Cu lamp - 9ma						
BGen 44.5	2.590	1.990	2.480	1.013		
BGen 39.5	2.590	1.970	2.480	1.014		
0.35 absorbance	2.590	0.885	2.480	0.453	0.349	0.350
0.80 absorbance	2.590	0.305	2.480	0.161	0.800	0.812
$^{a}_{A} = \log (I'/I)$	I' n	epresents	balance c	xondition	s	
$^{b}A(BG) = \log (BG'/BG)$	BG' n	epresents	balance c	condition	S	
* PE 380 balan	ce conditio	ons from a	utogain c	ircuitry		
** deuterium la	mp energy					

#### CHAPTER IV

#### RESULTS AND DISCUSSION

# Chemical Availability from Chisman Creek Fly Ash

In this study, several sample types were analyzed for major, minor, and trace metal content by two different methods. Leachates of weathered fly ash were generated from chromatography columns operated at several pHs and ionic strengths. A profile of the leached elements was provided X-ray emission (PIXE) analysis. Since by proton-induced this allowed many elements to be analyzed technique complex profile of aluminum, silicon, simultaneously, a calcium, vanadium, manganese, iron, nickel, copper, gallium, germanium, arsenic, selenium, rubidium, strontium, yttrium, barium, and lead could be efficiently obtained. The leachates were also analyzed for vanadium by atomic absorption spectrophotometry. To provide an estimate of the total 'available' elemental content, other samples were produced by mixing 15.0 grams of fly ash with 20.0 mls of 5N nitric acid at room temperature. This technique was expected to elements available under strongly acidic estimate trace non-oxidizing conditions. These samples were also analyzed by proton-induced X-ray emission.

Table 9 shows the main elements released from a 5N nitric acid leach of both site 1 and site 2. As mentioned previously, the 5N leach more closely simulates natural conditions when evaluating total element availability than the hydrofluoric acid technique commonly used with geological samples. As trace with hydrofluoric acid, metals are released the silicate mineral's crystalline structure is actually dissolved. Although this method releases the greatest level of elements, it does not realistically model natural availability. The addition of fluoride has а further negative effect because of its subsequent PIXE analysis. interference with Another any alternate method, the aqua regia digest, also releases a larger portion of all elements than the 5N leach.

The ash analyzed in the current study had been exposed to the environment for a period of about 7 to 22 years. This weathering undoubtedly reduced the level of various elements found in the fly ash leachates compared to fresh fly ash. On the other hand, leachates analyzed in the present work provide an estimate of <u>future</u> environmental risk, since they measure trace elements currently available from the already weathered fly ash at the site.

In order to understand the trends in metal release from fly ash at various pHs, it is important to understand the structure and formation of fly ash particles. Coal is believed to form molten salt particles under the conditions and temperatures in a combustion furnace. As these particles condense, elements partition themselves into glass, quartz, and mullite phases (22). During this process, trace elements may be forced out of other phases and into the glass phase; the glass phase forms a 'cement' which holds the quartz and mullite phases As this partitioning occurs, certain elements become together. concentrated on the surface of fly ash particles Sulfur, arsenic, molybdenum, and other trace elements may be preferentially concentrated at the particle surface because they were last to condense on the fly ash particle core (22). The fly ash 'core' is composed of thermally stable minerals, called aluminosilicates (3,23,24,25). Some elements may undergo a 'diffusion' process from the inner particle to the surface (22). Regardless of what phenomenon is responsible for surface enrichment of elements, the elements in this surface layer should be available first from fresh ash when weathering occurs.

In the present work, the pattern of metal availability from weathered fly ash cannot be expected to follow the same trends displayed by fresh ash. Despite the age of the fly ash used in the current study, patterns for many elements are still evident.

Previous reports show that electron micrographs made of weathered fly ash are noticeably different than those made of Micrographs of relatively fresh ash contain fresh fly ash. spheres of actual fly ash particles that are often covered with a coating of salt crystals deposited on the surface (26,27). The particles are called plerospheres (28) because they contain smaller spheres inside, which is evident when the particle is broken. This structure has been proposed to

occur from the boiling and production of gas in the inner core as the particle enters the high temperature combustion zone (28). Any crystals present on the fly ash particle form slowly after the liquids on the surface crystallize (28). The larger crystals, 20 µm and larger, are composed chiefly of anhydrite gypsum (CaSO<sub>4</sub>  $\cdot$  2H<sub>2</sub>O) (29,28). (CaSO₄) and Calcium is probably chiefly present from intrusions of CaCO3. The larger crystals contain  $H_2SO_A$ , which contribute to leaching of metal oxides producing microcrystals of metal sulfates on the surface surface crystals are usually made of high (28). Other concentrations of sodium, calcium, sulfate, hydroxide, and boron. Micrographs of weathered ash show this surface coating to be flaky in appearance (26).

Electron micrographs of fly ash used in this study are shown in Figure 9. The 'plerosphere' structure discussed above is evident in the site 1 fly ash micrographs. Although several micrographs were taken of site 2 fly ash at the same magnification, this type of structure was not found. The micrograph included for the site 2 fly ash was the only picture taken that included larger The fly ash analyzed in the particles. current study contained high levels of calcium despite its weathered state, vet surface crystals were not present in any of the micrographs. Investigators have proposed that these elements are not just present as discrete salts but as surface adsorbed or admixed salts which cannot be dissolved in a short period of time (1,30).

The buffer compositions were chosen on the basis of recent

groundwater measurements in monitoring wells located in pit C (16). Wells unaffected by fly ash had a pH near 8. Wells within pit C had varying acidities and one well was measured at pH 3.5. Accordingly, pH 3.0 was chosen to represent the most strongly acidic, pH 5.0 represents mildly acidic conditions, and pH 8 represents the usual acidity of groundwater in this area.

Various plots were generated to show trends between concentration and volume eluted from each column under varying conditions, including pH, site, and ionic strength. The following paragraphs will compare and contrast element behavior under the varying conditions used in all columns. In all plots, 'open' symbols (such as circles, triangles, squares, etc.) are used to represent concentrations for the specific elements discussed. In cases where element concentration was less than instrument detection limits, the symbol for that particular concentration value is darkened.

Plots in Figure 10 display calcium, strontium, and barium concentrations during the leaching process for the pH 3, pH 5, and pH 8 columns. The exponential decrease in concentration exhibited in the plots could indicate these elements are present of salt which initially dissolves very quickly. as а type This is most evident for Ca in the 510, 511, and 521 columns, where the concentration dropped approximately 100 ppm during the first 200 mls of leaching. In whatever form Ca may have been present, its solubility is the greatest under pH 3 conditions as indicated by Ca concentrations remaining high throughout the leaching process. The same general trend exists

for Sr, although its soluble levels are about one order of magnitude lower than Ca. Note that there appears to be a pattern between the .01 and .1 ionic strength pH 5 columns. For both Ca and Sr, solubility seems to be hampered under the higher ionic strength conditions, as indicated by low Ca and Sr levels reached quickly. Trends displayed by Ba under pH 3 conditions, which were similar to Ca and Sr, disappeared in the pH 5 columns. In fact, trends for Ba in the 510, 511, and 520 columns show increasing concentration, although average Ba levels are not very different between the two pHs. No data was available for Ba at pH 8.

Sodium trends not investigated in any of the columns were because sodium was present in buffer components used for the pH 3 and pH 5 solutions. However, rubidium trends are shown in Figure 11 along with strontium and yttrium. In the pH 3 columns, both rubidium and yttrium display trends similar to strontium, as Strontium levels are consistently higher shown in Figure 10. than all other elements in any of the columns illustrated in Figure 11. This is consistent with the data in Table 9, showing soluble Sr levels to be about 15 to 40 times higher than either Rb or Y after a 5N nitric acid leach. Yttrium levels fell almost two orders of magnitude as the pH increased from 3 to 5, illustrating yttrium's greater solubility at pH 3. Rubidium levels also fell under pH 5 conditions, but not so dramatically. Fluctuations develop in both yttrium and rubidium patterns as their concentrations drop to near detection limits. No data was available for any of the elements under рH 8

# conditions.

Figure 12 includes plots of aluminum, calcium, iron, manganese, and silicon. Typically, concentrations of aluminum, silicon, and iron are high in fly ash, reflecting levels in the type of Silicon, aluminum, iron, and calcium are the coal used. third, fourth and fifth most abundant elements in the second. earth's crust, respectively. Manganese concentrations are lower in magnitude but it can still have an important effect on other elements. This phenomenon will be discussed in more detail in a later section. Note that detectable manganese levels in the pH 3 columns, being most soluble, are two to three orders of magnitude lower than the other elements.

Silicon concentrations remain at a fairly constant level within each column, supporting the theory that silicon is a matrix element. Many researchers believe that under conditions in a combustion furnace, the lighter elements in coal volatilize and eventually condense on the thermally stable aluminosilicate matrix (26,1,23). The matrix compounds include  $Al_2O_3$ , SiO<sub>2</sub>, CaO, FeO, and MnO (13).

Of the matrix compounds, both aluminum and silicon constitute approximately 84% of the fly ash content (31) yet these elements are eluted at low concentrations. About 91% of silicon and 63% of aluminum exist in aluminosilicate 'glass' (31). Elements locked in the aluminosilicate glass are tightly bound; that is, they are only soluble in HF.

Although silicon levels remain fairly steady within each column,

there is a small decrease in solubility when going from pH 3 columns to pH 5 and pH 8 columns. Iron levels fall drastically under the same changes in pH. Iron levels average about 50 ppm at pH 3 and only about .1 ppm at pH 5. This reflects iron's greater solubility at low pHs. Also, iron may be present in several forms of fly ash that explain its very different behavior at pH 5 and pH 8. About 68% of iron is believed to exist in the aluminosilicate 'glass' but it also exists in more soluble forms, such as magnetic spinels existing in fly ash and in sulfate, phosphate, and fluoride salts (31,22). Although aluminum levels were approximately the same as iron, silicon, and calcium in the pH 3 columns, its concentration was not well determined due to the inherent problems in its analysis with PIXE. Even aluminum levels of about 50 ppm were near the detection limits for this element.

The process that occurs when gases condense on the fly ash matrix causes a preferential concentration of certain volatile elements on the shell of the fly ash particle (26). These lighter elements include arsenic, cadmium, molybdenum, selenium, tungsten, cobalt, chromium, antimony, zinc, copper, and vanadium (12,32,33,34). This process becomes more important for smaller fly ash particles because of their larger surface area-to-mass ratio (13). The concentration of certain elements on the surface layer plays a critical role in the element's availability in the presence of leaching The elements comprising the outer core are usually agents. present as surface salts and should be available first when fresh

ash is leached. Eventually, the concentrations of these decrease elements should and the matrix elements (aluminum, iron, silicon, calcium, and manganese) should The pattern of metal release is more difficult to predominate. predict in older ash since the stage of particle dissolution is unknown.

# Environmental Significance of Trace Element Leachates

Trace elements frequently 'track' the matrix components silicon, aluminum, and manganese. Studies on leaching of matrix components have shown that there is an association between matrix components and certain trace elements (1,35,36). The experiments involved using extraction techniques with ammonium oxalate and hydroxylamine hydrochloride. Ammonium oxalate preferentially extracts aluminum and iron (1, 35)while hydroxylamine hydrochloride extracts manganese (1,36). Matrix compounds which contain aluminum, iron, and manganese attract trace metals due to the presence of active sites on their surface. Manganese has a sorptive capacity fifteen to twenty times higher than aluminum or iron (1, 37). compounds containing manganese are much more Consequently, those containing either iron active than or aluminum. manganese is often present in much lower However, since concentrations than aluminum or iron, its scavenging effect is usually not as great as these elements. Surface area and dielectric are the physical properties which cause trace element

attraction to iron, aluminum, and manganese compounds (1,29). The charged surfaces on these compounds change with pH. This phenomenon causes metallic cations, or any charged species, to be attracted to the compound's surface. Manganese may have a greater available surface area than either aluminum or iron, thus explaining its greater sorptive capacity (1,29). Studies of arsenic illustrate trace metal attraction to matrix compounds containing iron. Normally arsenic exists as oxyanions and oxyacids from pH 4 to pH 12 (30,38). The absence of arsenic oxyanions can be explained by an active iron surface scavenging this ion. Iron probably exists as amorphous iron oxyhydroxide, which is commonly present in fly ash (30). The iscelectric point of this compound occurs at high pHs so the surface retains a positive charge throughout most of the pH range (30,39). The charged surface probably attracts the negatively charged arsenic oxyanion, thus reducing its 'free' This process may also act on the uncharged concentration. arsenious acid.

To understand the association of matrix elements with other trace elements, it is important to understand both the mineralogy of coal and conditions present inside the coal furnace. Several studies have associated the matrix element iron with arsenic, chromium, cobalt, and zinc (29,1). Sulfides of iron and are volatile arsenic and may condense at the same temperature, further explaining the association of the two elements (1). Figure 13 contains plots of iron, arsenic, and The iron and arsenic data show many similarities in the copper.

pH 3 site 1 column. Trends are also evident in the pH 3 site 2 Some trends are present in the pH 5 data. columns, although both elements' concentrations are closer to detection limits which makes real similarities difficult to observe. Data was not available for either chromium or cobalt; therefore, these elements were not included in the plots. Zinc data was not included because it was not known what percentage of its concentration was due to contamination. Zinc is a common in the laboratory and, therefore, difficult to contaminant properly. Copper was substituted for zinc due to its analyze proximity to zinc on the periodic chart. The copper data followed a more exponential decrease in the pH 3 columns, which is quite different from iron's trend. Although no previous information was available about calcium and its behavior as a matrix 'scavenger', its pattern follows quite closely with copper (see Figure 14). The similarity may also be due to copper's, and other cationic metals, association with anionic salts, such as phosphate, and calcium's presence as apatite  $(Ca_{3}(PO_{4})_{2})$ This similarity is most noticeable in (31). the pH 3 experiments but also present at pH 8 and in some of the The concentrations are closer to detection limits pH 5 columns. in the pH 5 experiments as noted by the irregular patterns.

The matrix element manganese has been associated with cadmium, lead, and nickel (1). Previous studies have shown that about 24% of manganese is found on the surface of the fly ash particle, which is consistent with patterns for lead, cadmium, and nickel (1,40). Some researchers have proposed that these elements had prior geological association with manganese, which could explain an association between these elements. Cadmium was not included in any of the plots since it could not be detected in any of the samples. It is difficult to see any trends with manganese in the pH 3 data although the other elements are detected (see Figure 15). The pH 5 data for manganese, nickel, and lead are too close to detection limits to make any judgements. No data was available for any of these elements in the pH 8 experiments. Nickel concentrations are too close to detection limits in the remainder of the columns for any similarities to be seen.

It becomes more difficult to predict trends when attempting to leach trace elements from the matrix compounds. The processes controlling trace metal release are probably solubility dependent. The solubility of the matrix 'scavenger' may dictate the release of the associated trace metal (1). This process can be quite complex. Some trace elements display concentration patterns which cannot be explained by solubility of the matrix compound associated with it.

pH dependent, also play an which Other processes, are important role. Previous studies have determined the percentage several elements that should be leachable under strongly of acidic (pH = .5) or strongly basic (pH = 11.9) conditions (41). Data from this study illustrates that more than 30% of the arsenic, cadmium, fluorine, total amount of boron, and selenium should be available under acidic molybdenum, Between 10-30% of chromium and vanadium and between conditions.
1-10% of beryllium, copper, and zinc should be available under strongly acidic conditions. More than 30% of molybdenum should be available under strongly basic conditions. Between 1-10% of boron, fluorine, and selenium and between .1-1.0% of arsenic and chromium should be available under these conditions. Less than .1% of beryllium, cadmium, copper, vanadium, and zinc should be available at strongly basic pHs.

It is a well-known fact that many transition elements are more soluble at acidic pHs than neutral pHs (41,42). Transition metal behavior is unpredictable at high pHs due to the possible formation of anions. These anions are soluble at high pHs when the corresponding metallic cation would precipitate from solution (41,42). The formation of molybdate, borate, fluoride, selenate, arsenate, chromate, and vanadate are examples of this behavior (41).

Plots of arsenic shown in Figure 16 clearly display this dual trend at pH 3 and pH 8. Selenium displays greater solubility at pH 8 than at pH 3 in Figure 16, although the pH 3 site 1 levels are relatively high. Vanadium, on the other hand, clearly displays a higher solubility at pH 3. Refer to Table 9 for total 'available' levels of these elements as determined by the 5N nitric acid leach technique. Reliable data was not available for beryllium, zinc, fluorine, chromium, and cadmium to make any conclusions about their behavior.

# Figure 10a

Comparison of Ba, Ca, and Sr levels in pH 3, pH 5, and pH 8 columns







# Figure 10b

Comparison of Ba, Ca, and Sr levels in pH 3, pH 5, and pH 8 columns



1100

1000

900





BOC

Volume (ml)

700

....

-14

100

200

400

# Figure 10c

Comparison of Ba, Ca, and Sr levels in pH 3, pH 5, and pH 8 columns







## Figure 10d

Comparison of Ba, Ca, and Sr levels in pH 3, pH 5, and pH 8 columns



## Figure 11a

Comparison of Rb, Sr and Y levels in pH 3, and pH 5 columns



## Figure 11b

Comparison of Rb, Sr, and Y levels in pH 3, and pH 5 columns







# Figure llc

Comparison of Rb, Sr, and Y levels in pH 3, and pH 5 columns



Element Concentration versus Volume Eluted from Column 520

Element Concentration versus Volume Eluted from Column 521



Яb

&... Sr -0--- Y

KEY:

# Figure 12a

# Comparison of Al, Ca, Fe, Mn, and Si levels in pH 3, pH 5, and pH 8 columns



# Figure 12b

# Comparison of Al, Ca, Fe, Mn, and Si levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 510

KEY

Al Ca Fe

• Mn - S1



Comparison of Al, Ca, Fe, Mn, and Si levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 520

Element Concentration versus Volume Eluted from Column 521



KEY

Al Ca Fe Mn Si



Figure 12d

Volume (ml)

----[

1.00

# Figure 13a

Comparison of As, Cu, and Fe levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 31

# Figure 13b

Comparison of As, Cu, and Fe levels in pH 3, pH 5, and pH 8 columns



## Figure 13c

Comparison of As, Cu, and Fe levels in pH 3, pH 5, and pH 8 columns





Comparison of As, Cu and Fe levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 81

# Figure 14a

Comparison of Ca and Cu levels in pH 3 and pH 8 columns







# Figure 14b





Element Concentration versus Volume Eluted from Column 81



Element Concentration versus Volume Eluted from Column 82



# Figure 15

Comparison of Mn, Ni, and Pb levels in pH 3 columns







# Figure 16a

Comparison of As, Se, and V levels in pH 3, pH 5, and pH 8 columns







Element Concentration versus Volume Eluted from Column 32



# Figure 16b

Comparison of As, Se, and V levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 511



# Figure 16c

Comparison of As, Se, and V levels in pH 3, pH 5, and pH 8 columns





Element Concentration versus Volume Eluted from Column 521



# Figure 16d

Comparison of As, Se, and V levels in pH 3, pH 5, and pH 8 columns



Element Concentration versus Volume Eluted from Column 81



As

## Table 9

Concentration of elements in 5N nitric acid leachates of Site 1 and Site 2 fly ash (concentration in ug/g - '\*' denotes concentration in mg/g)

(' $\zeta$ ' indicates concentration below detection limits)

	Si	Ca	Mn	Fe	Ni	Al	
Site 1	37.0*	1480.0	<70.0	43.0*	53.0	33.0*	
Site 2	28.1*	1910.0	<60.0	48.8*	46.0	25.0*	
	As	v	Se	Cu	Pb	Rb	
	*						
Site 1	147.0	240.0	16.4	95 <b>.0</b>	61.0	87.0	
Site 2	244.0	250.0	10.1	97.0	67.0	76.0	
	Sr	Y	Ba				
		• • • • • • • • • • • • • • • • • • •					
Site 1	595.0	51.0	1520.0				
Site 2	680.0	49.0	1600.0				

### CHAPTER V

### SUMMARY AND CONCLUSIONS

The research presented in this paper has demonstrated the potential environmental hazards of weathered fly ash. Elements that pose these risks, such as vanadium, selenium, arsenic and nickel, were found at detectable levels after fly ash had undergone a long period of leaching in the laboratory. Vanadium was found in the leachates in the 1-10  $\mu$ g/ml range. Arsenic, selenium, and nickel were found to exist in the .1-1  $\mu$ g/ml range. Furthermore, the profile of these elements were still changing at the end of the study. In many cases, the concentrations were still above instrument detection limits.

In some cases, element levels varied significantly between site 1 and site 2. This illustrates the variability of fly ash within a single disposal site. Analysis of leachates from the .10 ionic strength pH 5 columns suggested possible suppression in solubility of certain elements due to the higher ionic strength. However, the reactions occurring between compounds in fly ash are complex and difficult to predict.

The ISAAC/Apple II+ system was an invaluable tool for automating the atomic absorption analysis of vanadium in fly ash leachates. The salt content of the buffer solutions accelerated graphite tube deterioration. Real-time plots of the data on the acquisition system showed when this process started to occur. The data acquisition system could be implemented in other difficult GFAA analyses to give similar benefits. The major drawback of the Apple II+ computer was the length of time required to complete an analysis of an entire run. In this situation, the PDP-11 computer was a far superior system. If the data acquisition system was implemented today, far more powerful microcomputers could be substituted for the Apple II+.

It is satisfying to report that conditions at the fly ash disposal site have improved since this study was undertaken. Money from the Environmental Protection Agency's super fund has been used to level and 'cap' the fly ash with a special clay, which should reduce element leaching and ash transport by wind.

### APPENDIX A

## Data Acquisition Software and Calculations

Table A-1 includes all computations performed during the enhanced data acquisition mode. The top portion represents computations performed during the instrument char cycle and the bottom portion contains computations performed during the instrument atomization cycle. Certain characters in the equations denote use of a real versus an integer variable. For example, the '%' is used to represent an integer value in Apple II+ BASIC. The other variable names without the '%' are used to represent real values, which require more storage space on the Apple II+. In many cases, it was only necessary to use integer variables to store many of the numbers obtained from the acquisition system since they were digitized voltages. In some equations with mixed variable types, the final result is stored in an integer variable with use of the 'INT' function. This function saves the integer portion of a real number for storage into an integer variable. The 'ABS' function shown in some equations is used to take the absolute value of a number. A listing of the enhanced data acquisition program is included in this appendix after Table A-1. At the start of the enhanced data acquisition program, the operator is asked to select the length of time, in seconds, for acquiring the atomization signal. The period selected is based

on the element about to be analyzed. For vanadium, this period was normally five seconds which translates to a total of 300 points collected using a 60 Hz acquisition frequency. During the instrument char cycle, the deuterium and hollow cathode sample and reference beam signals were acquired for the same length of time these signals would be acquired during the atomization The purpose of acquiring these signals prior to the cycle. actual sample atomization was to obtain 'baseline' values for each signal. These baseline values would be used for later computations to obtain the corrected hollow cathode absorbance. The average deuterium sample beam signal (SO) is computed in equation 1, the average hollow cathode sample beam signal (S1) is computed in equation 2, the average deuterium reference beam signal (S2) is computed in equation 3, and the average hollow cathode reference beam signal (S3) is computed in equation 4. The number of points sampled during the char cycle is represented by  $n_1$  in equations 1 - 4.

Equations 5 - 10 were only computed during the instrument atomization cycle. As previously mentioned, the deuterium sample beam signal underwent a correction to compensate for the difference in time between when the signal was sampled and when the signal was updated by the instrument. Equation 5 shows how this correction was applied to the deuterium sample beam signal, which was stored in  $S_{(0,ii)}$ . The corrected deuterium sample beam signal, stored in  $S_{(0,ii)}$ , was then used to compute the background absorbance, ABG\*(ii), at each point during the atomization cycle, as shown in equation 6. Note it is not necessary to use the deuterium reference beam signal to compute ABG%(ii) in equation 6. The average deuterium sample beam signal obtained during the char cycle, SO, was used instead. The total hollow cathode absorbance, AHC%(ii), is computed in equation 7 using the same logic. AHC% (ii) is computed for each point sampled during the atomization cycle using the hollow cathode sample beam signal, S%(1,ii), and the average hollow cathode sample beam signal obtained from the char cycle, S1. The corrected hollow cathode absorbance, ACRRT%(ii), is obtained in equation 8 by subtracting the background absorbance, ABG%(ii), from the total hollow cathode absorbance, AHC%(ii). ACRRT%(ii) represents only monochromatic absorbance with all other absorbance and scattering components subtracted out.

Finally, the average deuterium lamp drift (D2) and the average hollow cathode lamp drift (LD) values are computed in equations 9 and 10, respectively. The deuterium lamp drift is a measure of the average change in the deuterium reference beam signals acquired during the atomization cycle,  $S_{(2,ii)}$ , from the average deuterium reference beam signal obtained from the char cycle, S2. The variable  $n_2$  is the total number of points sampled during the atomization cycle. The hollow cathode lamp drift is a measure of this change for the hollow cathode lamp.

# Table A-1 Calculations performed during data acquisition

#### Char Cycle Computations

1)	S0 = 3	🗲 (S%(0,i))/n,*	average deuterium signal	-sample beam (BG)
2)	S1 =	$(S_{(1,i)})/n_1$	average hollow cathode signal	-sample beam (I)
3)	S2 = ;	$(S^{(2,i)})/n_1$	average deuterium signal	-reference beam (BGO)
4)	S3 =	$\hat{\Sigma}$ (S%(3,i))/n <sub>1</sub>	average hollow cathode signal	-reference beam (IO)

\*  $n_1$ ' represents total number of points collected during char cycle; 'i' represents individual signal values, which were summed over collection period. 's' sign on arrays indicates integer values are stored.

Atomize Cycle Computations

5)  $S(ii) = S_{1}^{*}(0,ii-1) \times .826 + S_{1}^{*}(0,ii) \times .174^{*}$ phase corrected deuterium signal-sample beam 6)  $ABG_{1}^{*}(ii) = INT(1000 \times (IOG_{10}(S0 / S(ii)) / 2.302))^{**}$ background absorbance 7)  $AHC_{1}^{*}(ii) = INT(1000 \times (IOG_{10}(S1 / S_{1}^{*}(1,ii)) / 2.302))$ total absorbance 8)  $ACRRT_{1}^{*}(ii) = AHC_{1}^{*}(ii) - ABG_{1}^{*}(ii)$ corrected absorbance (monochromatic) 9)  $D2 = (100 \times (ABS(S2 - S_{1}^{*}(2,ii)) / S_{2})) / n_{2}^{***}$ average deuterium lamp drift 10)  $ID = (100 \times (ABS(S3 - S_{1}^{*}(3,ii)) / S_{3})) / n_{2}$ average hollow cathode lamp drift

\* 'ii' represents individual signal values collected during atomize cycle.
\*\* 'INT' refers to taking the integer equivalent of a real number - decimal portions of the real number are truncated.

\*\*\* ' $n_2$ ' represents total number of points collected during atomize cycle. Individual deuterium lamp drift and hollow cathode lamp drift values were computed during atomize cycle; sum of these values were averaged at end of atomize cycle. 'ABS' refers to taking the absolute value of a number.

10 M = CHM (4) 12 GOTD 690 23 FOX (1 OR OK ) 3 THEN 6010 20 24 FOX (1 OR OK ) 3 THEN 6010 20 25 FOX (1 OR OK ) 3 THEN 6010 20 26 FOX (1 OR OK ) 3 THEN 6010 20 27 FOX = 1 THEN 6010 470 27 FOX = 1 THEN 6010 470 28 FOX = 1 THEN 6010 470 29 FOX = 1 THEN 6010 470 20 FOX = 1 THEN 6010 470 29 FOX = 1 THEN 6010 470 20 FOX = 1 THEN 6010 470 21 FOX = 3 THEN 400E : HTHB 1: VIB 23: PRIM \*(1) TO SAVE RAM DATA? HTHB 1: VIB 24: PRIM \*(2) TO SAVE RAM DATA? 20 FOX = 1 THEN 6010 20 20 FOX = 1 THEN 6010 40 21 FOX = 3 THEN 400E : HTHB 1: VIB 23: PRIM \*(1) TO SAVE RAM DATA? 20 FOX = 1 THEN 6010 20 20 FOX = 1 THEN 6010 20 20 FOX = 1 THEN 6010 40 21 FOX = 3 THEN 40E : HTHB 1: VIB 23: PRIM \*(1) TO SAVE RAM DATA? 21 FOX = 3 THEN 40E : HTHB 1: VIB 23: PRIM \*(1) TO SAVE RAM DATA? 22 FOX = 1 THEN 6010 40 23 FOX = 1 THEN 6010 40 24 FOX = 1 THEN 6010 40 25 FOX = 1 THEN 10 FOX = 10 FOX = 10 25 FOX = 1 TO FOX = 10 26 FOX = 10 FOX = 10 27 FOX = 10 FOX = 10 28 FOX = 10 FOX = 10 29 FOX = 10 FOX = 10 20 FO 1 METE 1
2 METE IF Y3 = 1 OR Y3 = 2 THEN 6 = 6 + 2: 6010 4004 61 NEXT I 71 NEXT K TEXT 0 **1**0

Enhanced Data Acquisition Program

IF ACK (I DR ACK) 2 THEN 6010 900 PRINT "ENTER ELEMENT NAME, & OF SECONDS FOR TRANSIENT (SEPARATE RESPONSES BY COMMO AND ONLY INTEGERS FOR TIME": INPUT SYS,N:LN = N:\* 60:L = 300 - 8:IL = L + 1 + N:F = IL:ED = L + 1 ": NDRMAL : BOSUB 100 HTAB 1: VTAB 23: PRINT "(SSC) TO PREVENT DATA STORGE": HTAB 1: VTAB 24: PRINT "(RTN) TO MORK RECORD FOR LANSING DATA": HTAB 1: VTAB 21: PRINT "COMPUTING 1250 ADERTY(M) = AMCK(M) - ABES(M) 1253 REURINT "S(M) = AMCK(M) - ABES(M) 1270 IF K = FTHEN GOTO 1280 1275 M = MN + 1 1280 MEXT K 1295 GESLB 300; REM - AREA SURBOUTINE 1175 REM RINT "ANG: SO"; SO; SI"; SI; S2"; S2"; S3"; S3 1230 PHCK(NN) = INT (T + ( LOG (SI / SX(1, K)) / X)) 1220 REEX(NN) = INT (T + ( LOG (SO / S(NN)) / X)) IF K = F THEN S(NN) = SX(0,K); 60T0 1220 1210 S(NN) = 5x(0,K) + 2 + 5x(0,K + 1) + Y 1190 FOR K = ED TO F 1180 NN = 080

CdL - 162 ADC 7(0, CUL - 132) ADC 101, CUL - 132) ADC 110, CUL - 132 ADC 110, CUL - 142 ADC 000, CDC 6(1,0) ADC 6(1,0) ADC 6(2,02); FUE ADC 6(1,0) ADC 6(1,0) ADC 6(1,0) ADC 6(1,0) ADC 6(1,0) ADC 7(0, CUL - 142) ADD 100, CDC 7(10, CUL - 142) ADC 7(1,0) ADD 100, ADD 100

2016 1 BUZZ STUP : JF HES = 2 GOSUB 20 2018 PRIMT "HIT ANY KEY AFTER NEW DISK IS INSERTED": INPUT ";KEY3; HOME 4000 JF Y3 = 177 DR Y3 = 178 THEN 6 = 6 + 2 4004 GUTD 55 4770 PRIMT DS; "DDGN";295;",L", P1;N; ",D';R2X 4770 PRIMT DS; "DDGN";295;",L", PRIMT 6: PRIMT SY1; PRIMT N: PRIMT RI; PRIMT RP 4774 PRIMT DS; "CLOSE";294

## APPENDIX B

## Data Analysis Software and Calculations

The details of the post-acquisition calculations are discussed in this section. A list of important equations used by the two primary analysis programs are included in Table B-2. Computer listings of the analysis programs as well as a reporting program are included in this section.

A summary of the post-acquisition calculations would not be complete without a discussion of difficulties encountered in vanadium peaks collected during data acquisition. integrating One of the more challenging tasks was to determine an algorithm that could reproducibly determine peak area for very small vanadium peaks well as the larger peaks produced by as The vanadium peaks were standards. subject to spiking at peak maximum (due to graphite tube deterioration), edges (due to processes involved prolonged trailing in of atoms after atomization), and unpredictable the decay peak fronts (due to matrix interferences). The slope at the beginning of most vanadium peaks was very gradual. This noise discrimination difficult. The final made peak versus algorithm included three tests to determine if a peak was 'eligible' for integration. The first test, to mark 'start of required an increasing difference between five peak',

successive data points. The average difference between each of the five points was saved and used for later tests. The required that the difference in three successive second test data points, located after the peak maximum, be less than the average difference determined from the first test. Satisfying the second requirement guaranteed that a decreasing trend occurred just after the peak maximum. The third test required that the average of three successive points, located after the points that satisfied the second test, be less than 4% of the peak maximum. If the third test was satisfied, it indicated the data points had returned to some baseline value. If all three tests were satisfied, area integration began at the first data point comprising the set of five 'beginning' points. Area integration stopped at the first data point comprising the of three 'ending' points. If the 'peak start' was not set located before the peak maximum, area calculations were aborted. The peak was also flagged and not used as a replicate in the calculation of average peak area. If this all replicates of a sample or standard, the occurred on average area computation was aborted and flagged for future reference.

After area computations and the sample labelling step were completed, the next task was to perform a second order non-linear regression on the data. The non-linear regression equation is shown on line A in Table B-2. The equation is also shown below:
$$A/C = PO + P1 \times A + P2 \times A^2$$
 (Routh, 1981)

Concentrations were determined for all solutions analyzed in including standards. Equation B in Table B-2 the run, shows the computation used to obtain the solution concentration, which yet corrected for instrument sensitivity changes. was not every Recalibration occurred seven to eight samples to determine the direction and amount of the instrument sensitivity change. A correction factor, CORR(i), was computed to compensate for this sensitivity change (see equation C in Table B-2). When CORR(i) values were greater than one, this indicated the sensitivity of the instrument had increased during the run. For groups or blocks of standards run together, an average correction factor was computed as AVGCORR(i) (see equation D in Table B-2). Finally, an average AVGCORR(i) value, represented by AVGCORR2(i), was computed using the two sets of standards which surrounded a group of samples (see equation E in Table B-2). bracketed by two sets of standards had an average Samples sensitivity correction factor applied to their

concentration values (see equation F in Table B-2). The adjusted sample weight, ADJWT(i), was folded into the final equation for corrected solution concentration in equation G. The mean relative standard deviation, MRSD, and relative mean standard deviation, RMSD, were computed for the first group of standards used in the regression, as shown in equations H and I in Table B-2. A solution concentration error, SCERR(i), was also calculated for each sample analyzed in the run (see equation J in Table B-2).

### Table B-1

Functions available from post acquisition processing programs

Main Menu Available from Data Analysis Program

A) Area calculations
B) Second pass through data
C) Make labels
D) Edit labels
E) Transfer weight/tray position to 'C' file
F) Exit

Main Menu Available from Nonlinear Regression Program

- A) Peak Height / Area Calibration and Regression
- B) Peak Height Calibration and Regression
- C) Peak Area Calibration and Regression
- D) Calibration Plot/Hardcopy
- E) Exit

# Table B-2

Equations for non-linear regression and corrected solution concentration

- A)  $A/C = PO + P1 \times A + P2 \times A^{2*}$ general form of non-linear regression equation.
- B)  $C(i) = A(i) / (P0 + P1 \times A(i) + P2 \times A(i)^2)$ concentration not corrected for instrument sensitivity changes.
- C) CORR(i) = C(i) / SOLNC(i)\*\* instrument sensitivity correction factor.
- D)  $AVGCORR(i) = \mathcal{L}(CORR(i)) / NSTDS^{***}$ average instrument sensitivity correction factor.
- E) AVGCORR2(i) = (AVGCORR(i-1) + AVGCORR(i)) / 2.0<sup>\*\*\*\*</sup> block averaged instrument sensitivity correction factor.
- F) CONC(i) = (A(i) / AVGCORR2(j)) / DIL(i) sample concentration corrected for average instrument sensitivity change and dilution factor (DIL(i)).
- G) CONC(i) = ADJWT(i) x CONC(i) sample concentration corrected for adjusted sample weight (ADJWT(i))
- H) MRSD =  $(\int (\mathcal{L}(SD(i) / A(i))^2) / N) \times 100.0^{*****}$ mean relative standard deviation.
- I) RMSD =  $\int (f(1 CORR(i))^2) / N^{*****}$ relative mean standard deviation.
- J) SCERR(i) =  $\int (((SD(i)^2) / (PO+P1xA(i)+P2xA(i)^2)) + ((RMSDxC(i))^2))$ sample solution concentration error.

\* in the regression equation, X is absorbance (A) and Y is absorbance divided by solution concentration (C); P0, P1, and P2 are regression coefficients.

\*\* 'SOLNC(i)' is solution or 'known' concentration; equation C only applied to standards.

\*\*\* 'NSTDS' is number of standards analyzed together in a set; equation D only applied to standards.

 $^{\star\star\star\star}$  equation E applied to pairs of standard sets surrounding a group of samples.

\*\*\*\*\* equation H computed with first group of standards analyzed in run; 'N' is number of standards in the first group. 'SD(i)' is relative standard deviation (computed with solution concentration, SOLNC(i), and concentration obtained from regression, C(i)).

\*\*\*\*\*\* equation I computed with first group of standards analyzed in run.

11. The fore control cont

## Data Analysis Program

IF RIGHTS (294,1) = "C" THEN PRINT DS; CLOSE"; 234: PRINT "INCORRECT NOME FORMET FOR REQUING RAW DATA FILE- DO NOT APPEND NAMES WITH A "C" ; 6010 500 IF D ) O THEN PRINT "AREA ALAREADY CALCULATED" DO YOU MISH TO REDOT": INOUT DOS: IF LET'S (DOS,1) ( ) "Y" GOTO 1490 PRINT DS;"READ BP ";NUMRS(J);",R";JK: INDUT UIS; INDUT U2; INDUT U3; INDUT U4; INDUT U5; INDUT U65 PEEK ( - 18962) ( 2 THEN PRINT DS;"CLOSE"; 294: PRINT "FILE HAS NO ROW DATA": GOTO 500

NEW THE ROOKE LOOP HAS POINTS INITIALLY THAT ALL DOM'E SCEEDTHRESHOLD SO RE COURSE 11 MUST FIND 3 CONSECUTIVE POINTS THAT EXCEED THATE SACE AND AGAIN SO THAT REAL LOOP CAN BE EMERCE 1160 NEXT K 1190 REM 1200 REM THE ABOVE LOOD HAS POINTS INITH 1210 RA POMAIS(R, PR) • . . OM 1220 FOR H = K - Z TO N. REM PRIN"H=";H 1220 HI = HI + I

IF (ACRATA(I + 1) - ACRATA(I)) ) = AR THEN CTR1 = CTR1 + 1:SUM = SUM + (ACRATA(I + 1) - ACRATA(I)); GOTO 1040 IF P ( ) TP THEN KEIK(N) = PR \_ 1:PR = 1 + 1:PR = 1 + 1:PLN PMINI DF TLUES 7:294: 6010 1410
IF P ( ) TP THEN KEIK(N) = PR \_ 1:PR = 1
IF R ) S AND XS ) O THEN SIX = SIX / XSX:EIX = 0
PRINT DF \*E&D\*:294; P\*:H(\*):B1: IAPUT P: INPUT PMEX(R,PR): INPUT LUC:PMAX4(R,PR) = STR4 (PMEX(R,PR))
PRINT DF \*E&D\*:294; P\*:H(\*):B2:
PRINT DF \*E OF 10 \*
INPUT PC TO N
INPUT PC TO N
INPUT PC TO N PRINT DA; FREAD; 1394; , RO, BO': I APUT 6: INPUT 5Y: INPUT N: INPUT N: INPUT RI: INPUT R2: PRINT D4; °CLIDE; 1295 DIM ACTRATIKIN; PMATXIG, 9), 44(6), RE1X(6), REAS(6, 9), A45(6), Y1YX(6), ALMI(6), SID(6), S2D(6), PMATS(6, 9), P45(6) PRINT D5; "OPDN"; 2394; "L"; R1 PRINT "HUM MONY SIDES OF DISKS USED FOR DATA": INPUT 1YS IF W = 1 THEN TP = P:R = 1:PR = 1 PRINT D8;"READ"; [395;", R"; N;", BIESA": INPUT FLZ5: IF FLZ5 = "" THEN PRINT D5;"CLOSE"; [395: 60T0 1410 IF W = (ICZ#(IY) + 1) THEN PRINT "TURN OVER DISK AND HIT ANY KEY AND (RTN)"; INDUT 225 IF (ACARTA(I + 1) - ACRATA(I)) ) 0 THEN RR = (ACRATA(I + 1) - ACRATA(I)):CTR1 = 1 If 1 = N RHUD PMAXX(R, PR.) 40 THEN AREAS = "0": GUTD 1320 If 1 = N THEN PRINT "ND 5 POINTS THAT EXCEED THRESHOLD": AREA = 0.0; BOTD 1300 If ACRRTS(1) ( 0 BOTD 1050 IF REG (RCRRTX(K) - ACRRTX(K + 1)) ) = RR THEN R1 = 0: 6010 1160 750 PRIME DAY, TEADY, TAN, DOY, THOUT GE, THOUT SYE, THOUT NE, TEADY, TSYE, LY, FRIST, CAR, SUES, DE DAMARD, THOUT YAS, SUES, S

NEW THE ROOF LOOP HAS ALL POINTS ENTERING MICH EXCED THE THRESHOLD ON THE SLOPE DOMEWING A MAST COURT 3 CONSECUTIVE POINTS THAT DOME EXCED TH ORDER TO STOP PREA DALOTIONS 5 /\* \*\*\*\* 1.36 // million 5 /\* for the first interpretation 5 /\* for the first

REW 60701990 

DIK SSK(6), SK(6), RE1X(6), PROX8(G, 9), REA8(G, 9), RA48(G), LBL18(G), NRD18(G), RED18(G), RU11(6(), SU10(G), SU10(G), V1YX(G), PA8(G), SHK(G) 238 F (2) 0 The File = 1: 800 cs0 200 f File (3), (10, (30, 10) = 11: 800 cs0 200 f File (3), (10, (30, 10) = 11: 800 cs0, 100 f File (3), (100 f File (3) 50 50 11 = 1 0 LGN 101 51 = 1 0 LGN

IF (LETTS (YS,1)) = "Y" THEN PRINT "DRIVE \$?"; INDUT R2X = 2 THEN PRINT DS,"COTALOG,D";R2X PRINT "ENTER FILENME AND DRIVE & (SEPARATE RESPONDED WITH A 'C' ": INDUT 298, R2X IF W ) O OR JS = 4 THEN PRINT DS; "MRITE':295," RO, BW": FRINT Q2: PRINT DS; "MRITE':295,", RO, BD": PRINT G IF GBS = 3 OR GBS = 4 THEN PRINT DS; "WRITE':295,", RO, BD": PRINT X PRINT DS; "MRITE':295," RO, BION: PRINT 1 IF FD = 5 THEN PRINT DS; "WRITE':295,", RO, BD': FRINT O: PRINT DS; "WRITE',295,", RO, BIS': PRINT '\*' PRINT DS; "THEN PRINT DS; "WRITE';295,", RO, BIS': PRINT '' PRINT DS; "UGE':295 PRINT DS; "UGE':295 PRINT D&:"WRITE",[294,",R",N,",ESC": PRINT LETTS ("1.0000000",5): 6010 3740 PRINT D\$:"WRITE",[294,",R",N,",B40": PRINT LETTS (TLCS(N),10): PRINT LETTS (T2CS(N),10) PRINT D\$:"WRITE",[294,",R",N,",B42": PRINT LETTS ( 5TRS (ALMI(N),18) PRINT D\$;"WRITE"; 194;", R";N;", BC54": PRINT LEFTS (( STRS (SCD(N))), 4) PRINT D5;"WRITE";295;",R";N;",B140": PRINT LEFT\$ (( STR\$ (SID(N))),4) IF RSC ( LEFTs (294,11) ( 65 OR RSC ( LEFTs (294,11) ) 30 6010 3940 IF 0245 = 2 07 02454 = 3 07 02545 = 4 07 02545 = 5 THEN GOSUB 2280 IF 0255 = 5 60710 4020 PRINT DS;"4RITE";295;",R",N;",BT7"; PRINT REIX(N) PRINT DS;"4RITE";295;",R";N;",B80"; PRINT LEFTS (P46(N),6) PRINT DS;"4RITE";295;",R";N;",B80"; TEXT : HONE : PRINT "MENU OPTIONS": CLEAR : DS = CHR\$ (4) PRINT "(2) END PASS THROUGH DATA-EDIT DATA/LABELS" PRINT "CATALOG TO LOCATE FILE? (Y/N)": INPUT YS print "(4) Edit Labels" Print "(5) Indut Sample Mis, /Labels From Disk" Print "(6) Guit" HTPB 1: VTPB 5: PRINT "(1) AREA CALCULATIONS" PRINT D\$;"DPEN";29\$;",L350";",D";R2% IF Q8% = 1 THEN PR = 1: GOSUB 720 PRINT D\$;"HRITE"; 29\$;", R";N;", B52" IF REX ( 1 DR REM ) 2 60T0 3940 IF GBX = 3 DR GBX = 4 6010 3740 IF DAX ( 1 DR DAX ) 5 6010 3890 IF FD ( 2 THEN BOTO 3740 PRINT "(3) NAKE LABELS" FOR II = 1 TO REIS(N) FOR II = 1 TO YIYX (N) IF 08% = 6 6010 4020 IF J% ( 4 60T0 3740 IF 02 = 0 60T0 3740 PRINT PNRXS (N, 11) PRINT AEAS (N, 11) PRINT RUN1 (N) INPUT 0.8X 6070 3820 NEXT II NEXT 11 NEXT N RETURN 8 25250 25500 25500 25500 25500 25500 25500 25500 25500 25500 25500 255000 Non-linear Regression Program continue continu 470 FOR I = 2 TO J; REN THIS LOOP TO BET ANERGES FOR BLOCKS OF STDS SURROLMOING A GROUP OF SUPPLES 475 IF KM = 1 AND LBLIS(1) ( ) "STD" THEN INCRFT(KM) = INFT(1); KM = KM + 1 480 INCRFT(KM) = (INFT(1 - 1) + INFT(1)) / 2.0; PRINT "INCRFT(", I - 1,")=","INCRFT(I - 1) 480 INCRFT IS THE ANERGE OF INCRFTS SURROLNDING A BLOCK OF DATA 1 + MX = MX 061 1 = 101 594

 

 05
 Fill

 05
 Fi PRINT DF\*UNLOX\*;295;\*1.D\*;R24 PRINT DF\*OPEN\*;295;\*1.D\*;R24 PRINT DF\*OPEN\*;295;\*1.BVUT R55; JF R58 = \*1\* GOTO 6.65 PRINT DF\*CORE\*;295;\*1.BVLT \*REGRESSION NOT PREVIOUSLY DOME\*; FOR NB = 1 TO 1000; NEXT NB; GOTO 2660 PRINT DF\*RED\*;295;\*1.BNLP1; I: HOUT P2 PRINT DF\*RED\*;295;\*1.BNLP1; PRINT DF\*LOOX\*;295 HOLOT 20,0 TO 20,137: HOLOT 20,137 TO 279,137: HOLOT 279,137 TO 279,0: HOLOT 279,0 TO 20,0 POKE - 16304,0: POKE - 16297,0: POKE - 16299,0: POKE - 16302,0 PRINT D\$;"WRITE";29\$;",R0,B18": PRINT '1" PRINT D\$;"CLOSE";29\$ PRINT D\$;"LOCY";29\$ 680 PRINT DA; LOCK"; 29 700 POKE - 16304,0: POKE - 16297 800 RDT= 0 808 X = 20:Y = 134.K = 17:P = 17 FOR I = 140 TO 0 STEP - 20 IF R = 2 THEN 60T0 920 IF I ( 10 THEN 60T0 920 DRAM IS AT 6, I IF P ) 26 THEN P = 17 DROW P RT 11,1 JF R = 2 THEN X = 274 PRINT DS; NON, C, I, 0 HCOLOR= 3: SCALE= 1 FOR IL = 1 TO 2 FOR R = 1 TO 1 DRAW 14 RT X, I Speed R 601 700 0 #Hd HGR2 816 10 8 3 333333566 ŝ 23 8 8 92 **86**≜ a a a a a ŝ ğ 8

1355 MEXT I 1360 POS = STRS (P0):PIS = STRS (P1):P2S = STRS (P2):XE = 10 2580 FDR I = 1 TD 13 2590 A(1) = 0.0: NEXT I

IF 02 (1 THEN PRINT DS;"CLOSE";29%. PRINT THUS FILE MOS NO DATA RECORDS YET"; 6010 25 IF 01 (1 THEN PRINT DS;"CLOSE";29%. PRINT "NOT ENOUGH LABELS IN RECORDS TO CONTINUE WITH ROUTINE"; 6010 25 IF 01 ( ) 02 THEN PRINT DS;"CLOSE";29%. PRINT "N DATA RECORDS NOT = 0 OF LABELS IN FILE"; 6010 25 RED F ( PEEX (106) + 226 + PEEX (105)) = 16330 THEN PRINT D8;"CLOSE";29%.LUM = 16390: 6010 2210; REN LUMEN A16.041N MAS BEEN USED FROM EARLIEN SJB PRINT "ENTER NOME OF FILE AND DRIVE & (FILENOME MUST BE RAW DATA FILENOME APPENDED WITH A "C" FOR THIS CALCULATED DATA FILENOT 294,R2X DJM LBL15(01), WB15(01), 01(L15(01), A45(01), A246(01), 9), PMR75(01, 9), 50(2, 01), 945(01), 520(01), 720(01), 7175(01), 7175(01), 7215(01) REM RINT D5;"BLOOD PIC22,AI6384";",D1" PRINT D5;"BLOOD PIC22,AI6384";",D1"; PORE 23, INT (36847 / 256); PORE 22,((36847 / 256) - INT (36847 / 256)) + 256; MINDN: 36846 PRINT D5; RE00"; Z95; ", R0, 80": INDUT Q1: PRINT D5; RE40"; Z95; ", R0, 84": INDUT G2: PRINT D5; "RE40"; Z95; ", R0, 84": INDUT G IF PEEK ( - 18962) ( 2 THEN PRINT DS, "CLOSE", 296; PRINT "THIS FILE NOME HAS NO LABEL/DATH RECORDS"; 6070 25 PRINT DAT "BLOOD EPSON, HIRES. OBJ, PRE000": LONEN: 25400: REM LOADED AT 24576 FOR LENGTH OF APP. 3 SECTORS print ds;"red";z9s;",R";11;",D140": INPUT 50(1,11) print ds;"redu";z9s;",R";11;",B140": INPUT Y1Y5(11); Print ds;"redu";z9s;",R";11;",B180": INPUT A45(11) print ds;"redu";z9s;",R";11;",B190" 2630 NEXTJ 2640 NEXT1 2650 J = 0:K = 0:31M = 0.0:CTR = 0:5 = 0.0:KK = 0:MRSD = 0.0:MMSD = 0.0:MM = 0:XX = 0:X2 = 0:XE = 0 PRINT "ENTER DISK WITH SHOPE TABLE/EPSON, HIRES, OBJ IN DRIVE I AND HIT ANY KEY AND (RTN) TO CONTIME" IF LEFTS (YS,1) = "Y" THEN PRINT "DRIVED ": INPUT REAL IF REA ( 1 OR REAL) 2 THEN 6010 2830 IF ASC (LEFTs (294,1)) ( 65 0R ASC (LEFTs (294,1)) ) 90 6010 2630 IF RIGHTS (795,1) ( ) "C" 6010 2630 HTAB 1: VTAB 3: PRINT "(1) HT/AREA CALIBRATION & REGRESSION HTAB 1: VTAB 4: PRINT "(2) HT CM. LBARTION 4 REGRESSION" HTAB 1: VTAB 5: PRINT "(3) AREA CM. LBARTION 4 REGRESSION" HTAB 1: VTAB 6: PRINT "(4) CM. LBARTION A.OT/HAROCOPY" PRINT "CATALOG TO LOCATE FILENDUE NEEDED?": INDUT YS CLEAR ; HONE ; HTAB 1: VTAB 2: PRINT "NENU OPTIONS" PRINT D8;"READ";294;","R";11;", 8254": INPUT 5D(2,11) NEXT 11: PRINT D5;"CLOSE";294: PRINT D5;"LOCK";294 PRINT DS; READ"; 294; ", R"; II; ", BCC": INPUT RUNI (II) PRINT D5;"READ";295;", R";II;", B77": INPUT REIX(II) PRINT D5;"READ"; 295;", R"; 11;", 880"; INPUT P45(11) REN DNE 10, 76: POKE 11, 00: POKE 12, 96: IF CARK ( I DR CARK ) 5 THEN 6010 2740 

 2550
 RCUIR
 RETURN

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 REUN
 SEM

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 REM
 RM

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 REM
 RM

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 REM

 2740
 LEM
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 2750
 REM
 REM

 2750</td 2652 IF DAX ( ) 4 6010 2660 2620 R(I, J) = 0.0:T(J) = 0.0FOR 1 = 1 TO 7 FOR J = 1 TO 8 82 0 87 0

3180 DIM MCM (22), (130), (RFT (33), MFT (115), MCRFT (115), A(13), A(13), A(13), A(10), SCER (01), DX (01) 3190 LON(1) = 140:LON(2) = 254:LOX(1) = 145:LOX(2) = 260:L4X(1) = 276:L4X(2) = 282:L6X(1) = 306:L6X(2) = 324:L6X(1) = 201:BX(2) = 200 200 F Cox = 1 MEN GSUB 650 200 F Cox = 1 OR Cox = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR Cox = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR (1) 200 E Cox = 1 OR Cox = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR Cox = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR Cox = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 200 E Cox = 1 OR COX = 2 OR COX = 3 THEN GSUB 65 200 G17 240 20

PRINT D&; "READ"; 294; "AN BIO": 140-T FLAS IF FLAGE = \*\* THEM PRINT DA; "CLOSE"; 294; PRINT THIS IS RAW DATA FILE - ENTER CORRECT NOVE \*. 60TO 3000 PRINT DA; "READ"; 294; "AN, BO": 140-UT DA; "CLOSE"; 294; "AN, BM": 140-UT 6; PRINT DA; "READ"; 294; ", AN, BIS": 140-UT ALMS: PRINT DS; "CLOSE"; 294; "AN, BM": 140-UT ALMS: PRINT DS; "READ"; 294; "AN, BM": 140-UT ALMS: PRINT DS; "PRINT DS; "PR DIA SSK(6), SK(6), RE1X(6), PPAKK(G, 9), AEA6(G, 9), AA6(G), LBL16(6), AB016(6), RE14(6), AD01(6), 220(6), 220(6), 2114(6), P46(6), 262(6), 22(6), 22(6), 220(6), 22(6), 2 PRINT DS:\*660\*,129;",R",M",BIAS': INDUT NONCIN): PRINT DS:"READ':239;",R",M", R276": INDUT SCER(N): PRINT D5;"READ':294;",R",M", B320\*: INDUT D2 (N) PRINT DS:\*660\*,1294;",R",M", B260\*: INDUT NONC(N): PRINT D5;"READ':294;",R",M,",B222\*: INDUT CCCER(N): PRINT D5 PRINT D5;"READ':1294;",R",M",B177\*: INDUT V174(N) "C" THEN PRINT DS: "COSE": 295: PRINT "INCORRECT NOVE FORMAT FOR READING CALCULATED DATA FILE": SUTO 3000 IF RIGHTS (294,1) ( ) "C" THEN PRINT DS;"CLOEF", 259: PRINT "INCORRECT NAME FORMAT FOR READIN IF PEEK ( - 18962) ( 2 THEN PRINT DS;"CLOEF", 239: PRINT "FILE HAS NO INFORMATION": 8070 3000 IF IB = (ISIDEX(IU) + 1) THEN PRINT "TURN DVER DISK AND HIT ANY KEY AND (ATN)": INDUT TES PRINT D\$;\*READ^;25\$;\*,R0,B20\*; INOUT P0; INOUT P1; INOUT P2 PRINT D\$;\*READ^;25\$;\*,R0,B200\*; INOUT P3; INOUT P4; INOUT P5 PRINT D\$;\*READ^;25\$;\*,R0,B60\*; INOUT T0; PRINT D\$;\*READ\*;25\$;\*,R0,B240\*; INOUT T3 INPUT NUBIS(N): INPUT LULIS(N): INPUT DILLIS(N): INPUT REPIS(N) PRINT D\$;"READ";129\$;",R0,B76": INDUT U0 PRINT D\$;"READ";129\$;",R0,B256": INDUT U3;MX = 102 PRINT DS; "READ"; 295; ", R"; N; ", B140": INPUT SID(N) PRINT D\$; READ ; 295; ", R"; N; ", B80": INPUT P45(N) IF Q8% ( ) 4 AND Q8% ( ) 5 THEN 60TD 870 PRINT D&: "DPD"; 295; ", L"; RI PRINT D\$; "MRITE"; 295; ", L"; RI PRINT "0" PRINT D\$; "CLOSE"; 295 PRINT DS; "OPEN"; 294; "LISO"; "D"; REX PRINT DS; OPEN"; 295; , L350, D"; R25 PRINT DS; READ ; 295; , R ;N; , 852 PRINT D\$;"READ"; 795;", R";N;", B77" PRINT D5; READ ; 295; , R ;N; , B08 PRINT DS; READ ; 295; , R ;N; , B2 PRINT DS; "READ"; 295; ", RO, B"; MKX PRINT D6; "READ" (295; ", RO, B"; NKX TEXT : D5 = CHR5 (4): 6070 3820 INPUT FO(IY) : MKK = MKK + 4 INPUT F3(IY) INKX = NKX + 4 IF 081 ( ) 5 60T0 970 IF 084 ( ) 5 60T0 848 FOR IK = 1 TO RE15(N) FOR IB = 1 TO 6 FOR IU = 1 TO ANX NEXT IY: MKK = 280 INPUT PNAXS (N, IK) FOR IY = 1 TO UO FOR IV = 1 TO U3 FOR N = 1 TO 02 INPUT RUNI (N) INPUT RELA(N) NEXT IU NEXT IY NEXT 18 NEXT IK Non-Lay õ ---

PRINT \* MB16 (N ) H5 (-BL16 (N ) H5 PRINT H5; WBI5(N); H5; LBL15(N); H5; DIL15(N); H5; REP15(N); H5; AJN1(N); H5; A5(N); H5; S1D(N); H5; S2D(N); H5; S CHIRS (27)\*0\* CHIRS (1) CHIRS (11) CHIRS (21) CHIRS (36) CHIRS (46) CHIRS (56) CHIRS (66) CHIRS (66) CHIRS (101) PRINT D\$;"READ"; 295;", R";N;", BI80": INDUT A45(N) PRINT D\$;"READ"; 295;", R";N;", BI90" \$ 8

IF KIK = (15106%(1U) + 1) THEN PRINT "TURN OVER DISK AND HIT ANY KEY AND (RTN)": INPUT TTS: 6010 2244 IF KI = (ISIDEX(II) + 1) THEN PRINT "TUBN DVER DISK AND HIT ANY KEY AND (RTW)": INDUT YTS: BOTO 2740 PRINT TRB( 2); P(IRECS(IIK)); TRB( 20); PMAXK(IRECS(IIK)); TRB( 35); LOC(IRECS(IIK)) IF RTX = 2 And IC = 1 6070 2763 Print "Enter 4 of records you world line to onerrate for option";1c; 1.NPUT naunx IF IRECY(1.1) ( 1 OR IRECX(1.1) ) 6 THEN 6010 2650 IF IC = 1 THEN PRINT "ENTER NEW VALUE FOR TRAY POSITION": INPUT PUTRECX(1.1)) PRINT TRB( 2);"TRAY ID"; TAB( 20);"PEAK MAX."; TAB( 35);"PEAK LOCATION" PRINT D5;"00EN";295;",L";RI JF IC = 2 THEN PRINT D5;"MRITE";295;",R";IRECX(JJ);",D1554"; PRINT \*\*\* INDUT ISIDEX(U); IF ISIDEX(U) ( 1 OR ISIDEX(U) ) 6 THEN 60TO 2221 PRINT "HOULD YOU LIKE TO OVERWRITE ANY OF THE VALUES?"; INPUT RTS PRINT "DO YOU WISH TO GET A REPORT OF THESE VALLES?": INDUT YYRS PRINT "HOW NEWY SUDES OF DISK ARE USED FOR DATA?": INPUT ANY DIM ACCRTX (N), PMAXX (6), RE14 (6), Y1YX (6), LOC (6), P (6), IRECX (6) PRINT "HOW NEW RECORDS NEED TO BE RECOLLED?": INPUT NUM PRINT "RECORD #"; LIN: INPUT IRECK(IIK) IF IRECK(IIN) ( 1 OR IRECK(IIK) ) 6 THEN 60T0 2231 IF RIGHTS (YYRS, 1) ( ) "Y" THEN 60TO 2600 IF RIGHTS (RTS, 1) ( ) "Y" THEN GOTO 3000 PRINT "ENTER LAST TRANSIENT # ON SIDE"; IU PRINT "ENTER RECORD . ; IJ: INPUT INECX(IJ) IF KNUMS ( 1. OR KNUMS ) 6 THEN 60TO 2620 IF NUMS ( 0 OR NUMS ) & THEN 6010 2227 IF KIK ( ) INEEX (IIK) THEN 6010 2256 IF RTA ( 1 OR RTA ) 3 THEN 60T0 2605 PRINT "ENTER TYPE OF CHANGE DESTRED" NEXT IU: IF 08% = 2 THEN 60TO 2225 POKE 33, 33: PR# 1: PRINT CHR\$ (15) IF ANY ( I DR ANY ) 6 6010 2218 IF RTX = 1 PMD IC = 2 8010 2763 IF KI ( ) IREC×(1J) 60T0 2760 PRINT \* (2) DISCLUDE RECORD\* PRINT \* (3) BOTH\*: INPUT RT# PRINT DS; "DPEN"; 295; ", L"; RI PRINT \* (1) TRAY POSITION\* 605UB 200; 60TO 3000 PRINT DS;"MON, C, I, 0" FOR IU = 1 TO PNK FOR IIK = 1 TO NUMY FOR IIX = 1 TO NUMS FOR 11X = 1 TO NUMX FOR 1.J = 1 TO KNUM FOR 1.J = 1 TO KNUMS FOR KI = 1 TO 6 FOR II = 1 TO ANY FOR IU = 1 TO ANY FOR KIN = 1 TO 6 FOR IC = 1 TO 2 NEXT LIK NEXT KIK NEXT 11K NEXT 11K NEXT IU NEXT II NEXT 1J 0 884 õ 

2'4 If L = 1 Pel RENT De, REUP, 294, PF (RECKLU), 1, 91'F TOUT PARI, HOUT PARI, REUKLUD): REUM LOCHREG(LU);
2'5 IF LC = 1 THEN REUM PS, REUP, 294, PF, (RECKLU), 1, 91'F TOUT PARI, REUKLUD): REUM LOCHREG(LU);
2'5 MCT LU
2'6 MCT LU
2'7 MCT RU
2'7 MCT RU
2'8 MCT LU
2'9 MCT LU
2'10 MCT RU
2'10 MC

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