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# Predicting Potential Aluminum Contamination of Surface and Ground Waters from Acid Sulfate Enriched Drainages Emanating from Low Neutralization Capacity Watersheds

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### PREDICTING POTENTIAL ALUMINUM CONTAMINATION OF SURFACE AND GROUND WATERS FROM ACID SULFATE ENRICHED DRAINAGES EMANATING FROM LOW NEUTRALIZATION CAPACITY WATERSHEDS

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1988

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United States Department of the Interior Agreement Number(s): 14-08-0001-G1227 (FY 1986) 14-08-0001-G1424 (FY 1987)

P.L. 98-242

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Research Report No. 170

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Project Numbers: G 1227-04, G 1424-04\*, (A-109-KY)

Agreement Numbers: 14-08-0001-G1227 (FY 1986) 14-08-0001-G1424 (FY 1987)

Period of Project: July 1986 - June 1988

Water Resources Research Institute University of Kentucky Lexington, Kentucky

The work upon which this report is based was supported in part by funds provided by the United States Department of the Interior, Washington, D.C., as authorized by the Water Resources Research Act of 1984. Public Law 98-242.

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#### ABSTRACT

The composition of soil solutions and surface waters emanating from unreclaimed, or partially reclaimed, strip-mined watersheds with low buffering capacity in Kentucky were compared with soil solution compositions of unaffected strata in the watershed. The data suggest that almost 20 years after mining, most soil solutions and surface waters of the disturbed areas still contain high levels of dissolved Al, controlled primarily by the solubilities of a jurbanite-like mineral (upper limit) and alunite (lower limit). Soluble Al in solutions of undisturbed areas was consistent with the solubility of kaolinite or gibbsite. The absence of jurbanite x-ray diffraction peaks suggested the presence of an amorphous mineral or one stoichiometrically similar to jurbanite. Despite greater residence times for soil solutions as compared to surface waters, their compositional differences were insignificant. The control of soluble Al by basic aluminum sulfate minerals was not affected by the variable mineralogical and textural composition of soil and geologic strata in the watershed. Apparently, this is the result of low buffering capacity. At pH < 4, pH and sulfate activities can be used to accurately predict the levels of soluble  $Al^{3+}$  in surface and ground waters of the watersheds. Similar predictions from pH and  $\mathrm{SO}_4^{2-}$ activities can also be made for dissolved  $Fe^{3+}$  levels, supporting the stoichiometry but a much higher solubility than that of jarosite.

Column leaching experiments of selected spoil, soil and sediment samples contaminated by acid mine drainage over a period of five months also confirmed that pH,  $SO_4^{2^-}$ , Al and Fe are the major components controlling the behavior of these systems. Alternating wetting and drying cycles appeared to cause more drastic changes in the above components than different flow rates simulating rain events of different intensity. Kinetic profiles of Al and Fe released from the studied materials suggested that at different stages of leaching these materials may be releasing different levels of Al and Fe controlled by the solubility of characteristic sequences of basic Al-sulfate, Al-hydroxide and aluminosilicate minerals or Fe-sulfate, Fe-hydroxide minerals, which are watershed specific. The jurbanite-jurbanite, jurbanite-alunite and the jurbanite-alunite-microcrystalline gibbsite appeared to be the dominant sequences controlling the solubility of Al. Similarly, the jarosite-amorphous Fe (OH<sub>3</sub>) or amorphous Fe(OH<sub>3</sub>)- goethite sequence appeared to control

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the solubility of Fe. These findings suggest that equilibrium relationships, although quite useful, may not be sufficient for predicting the dynamic behavior of systems such as those included in this study. Kinetic relationships which reflect Al and Fe released as a function of time and the sequence of mineral phases controlling Al and Fe solubility are also necessary in order to obtain more realistic models of Al and Fe release into aquifers of such systems.

Descriptors:

Dissolved Aluminum, Aluminum Sulfates, Water Flow Rates, Mineral Equilibria, Acid Soils.

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# PART I

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# ALUMINUM AND IRON EQUILIBRIA IN SOIL SOLUTIONS AND SURFACE WATERS

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#### PART I

#### INTRODUCTION

High concentrations of dissolved Al are typical in acid drainages running off or percolating through soil and geological strata of surface mined watersheds (Bieseker and George, 1966; Van Breemen, 1973; Rogowski et al., 1977; Caruccio and Geidel, 1978; Nordstrom, 1982;). There is insufficient evidence that Al is toxic to fauna or humans, but high Al levels in acidified lakes cause reduced fish populations (Baker and Schofield, 1982).

In most naturally acidified terrestrial or aquatic environments soluble Al is assumed to be controlled by the solubility of gibbsite or kaolinite. However, most acid mine drainages are sulfate-rich solutions acidified by pyrite oxidation. The aqueous geochemistry of Al is modified considerably in such sulfate systems (Nordstrom, 1982). Gibbsite and kaolinite no longer dictate Al levels in solution; rather Al levels are controlled by newly precipitating basic Al-sulfate minerals (Van Breemen, 1973; Nordstrom, 1982; Anthony and McLean, 1976; Evangelou and Karathanasis, 1984). The nature and solubility of basic aluminum sulfates vary with soil pH and the physicochemical and mineralogical characteristics of soil and geologic strata in the watershed (Singh and Brydon, 1969; Frink, 1973; Adams and Hajek, 1978; SSSA, 1982; Driscoll et al., 1984; Rodriguez and Hidalgo, 1985; Nordstrom and Ball, 1986). For pH values < 4.5, surface mine drainages frequently appear to be undersaturated with respect to gibbsite and kaolinite, and their Al concentrations appear to be related to the solubilities of the aluminum sulfate minerals alunite, jurbanite and basaluminte (Nordstrom, 1982). These minerals are less soluble than gibbsite or kaolinite and can dramatically modify the upper limits of dissolved Al in surface and ground waters of acid mine drainage inflicted watersheds (Van Breemen, 1973; Nordstrom, 1982; Evangelou and Karathanasis, 1984). Aluminum concentrations in solutions reaching the aquifer are expected to be especially high in watersheds dominated by sandstone, shale and siltstone geologic strata which lack the neutralization capacity of limestone beds (Biesecker and George, 1966).

The equilibrium of Fe is also significantly modified in an acid sulfate system (Van Breemen and Harmsen, 1975; Nordstrom et al., 1979, SSSA, 1982). While goethite or amorphous  $Fe(OH)_3$  control Fe levels in most natural aquatic

systems (Stumm and Morgan, 1981), dissolved Fe in acid sulfate-rich solutions appears to be more consistent with the solubility of the basic iron sulfate mineral jarosite. This mineral, however, is metastable and gradually transforms under field conditions to the more stable goethite or hematite (Van Breemen and Harmsen, 1975).

The research objectives of this study were:

- 1. To identify and characterize indigenous and newly precipitated stable Al and Fe minerals, forming through acid sulfate drainage - soil interactions, likely to control the solubility of Al and Fe in effluents of runoff and infiltration waters of watersheds with diverse mineralogical composition.
- 2. To determine the role of clay-colloids present in specific low neutralization capacity soil or geologic systems in modifying the composition of water solutions emanating from acid sulfate drainage inflicted watersheds.
- 3. To establish empirical relationships defining and predicting potential levels of Al and Fe released in water reservoirs by acid sulfate drainage contaminated soils and geologic strata of specific geographic regions of Kentucky.

Although the solubility product-equilibrium approach used in these watersheds presents several inherent limitations (the systems are heterogenous, non-equilibrium, and open), comparisons between predicted equilibrium compositions and actual solution compositions should be useful in better understanding the Al and Fe physicochemical behavior of the watersheds.

#### MATERIALS AND METHODS

The study sites designated as OH, BD, and CL were located in Ohio, Boyd, and Clinton Counties of Kentucky, respectively. The watersheds were chosen to represent different geological strata, major mineable coal seams, and soils with low neutralization capacity (absence of carbonates) in western (OH), eastern (BD) and southern (CL) areas of the Kentucky coal fields. Site characteristics and descriptions of the collected samples are given in Tables 1 and 2. All watersheds were mined prior to 1973 and were subjected to little or no reclamation. Twelve undisturbed soil and sediment samples contaminated by acid mine drainage, 10 spoil samples, and 8 undisturbed, uncontaminated soil and sediment samples were collected in October of 1986. Eleven surface water samples from seeps, sedimentation ponds and streams were also collected from the same watersheds. Soil and sediment samples were collected with their natural moisture and sealed in polyethylene bags. Thereafter, the samples were uniformly saturated with sufficient deionized water to approach their water holding capacities and left undisturbed in a controlled temperature  $(25^{\circ}C)$  storage chamber for 90 days. Following the equilibration period, interstitial soil solutions were extracted by centrifugation and filtered through a 0.45  $\mu$ m membrane filter. Extracts were analyzed in duplicate for pH, A1, Fe, Ca, Mg, K, Na, Mn, Si, S, and Cl. Surface water samples were analyzed similarly within a week after sampling. Iron, Ca, Mg, K, Na and Mn were determined by atomic absorption using an Instrumentation Laboratory S-11 spectrometer. Aluminum and Si were determined colorimetrically using the eriochrome cyanine- $R_2$ , and the amino acid-reduced molybdosilicic blue method (APHA, 1976), respectively; sulfate - S was measured turbidimetrically, and Cl titrimetrically, using standard methods (American Public Health Association, 1976).

Following extraction, soil, sediment, and spoil subsamples were airdried, gently crushed to pass a 2 mm sieve, and analyzed for particle size distribution, pH, extractable bases, CEC (SCS, 1982) and mineralogy of the whole-soil and the < 2  $\mu$ m clay fraction (Karathanasis and Hajek, 1982). Whole-soil mineralogical analysis was performed on powder samples by x-ray diffraction (XRD). Mineral quantities were estimated from peak areas of XRD and thermogravimetry (TG) using an 1840 Philips Cu-K $\alpha$  diffractometer and a Dupont 1090 Thermal Analyzer, respectively (Karathanasis and Hajek, 1982).

Solution ionic activities were calculated by using the extended Debye-Huckel equation and a method of successive approximations (Adams, 1971), which included the following ion pairs:  $CaSO_4^{0}$ ,  $MgSO_4^{0}$ ,  $KSO_4^{-}$ ,  $NaSO_4^{-}$ ,  $CaHPO_4^{0}$ ,  $MgHPO_4^{0}$ ,  $AISO_4^{+}$ , and  $FeSO_4^{+}$ . Solution Al was distributed among the following species: Al(OH)<sup>2+</sup>, Al(OH)<sub>2</sub><sup>+</sup>, Al(OH)<sub>3</sub><sup>0</sup>, and AlSO<sub>4</sub><sup>+</sup>, using dissociation constants from Adams (1971), and May et al. (1979) (Table 3). All solutions had negligible or non-detectable levels of P, F and dissolved organic carbon. Iron (Fe<sup>3+</sup>) activity in solution was calculated from total Fe distribution into ionic species similar to those used for Al speciation, plus Fe<sup>2+</sup>, which was determined by the orthophenanthroline method (APHA 1976). The dissociation constants used for Fe speciation were taken from Lindsay (1979) (Table 3).

#### RESULTS AND DISCUSSION

#### Physicochemical and Mineralogical Characteristics

Particle size distributions, soil pH, extractable bases and cation exchange capacity (CEC) of the samples studied are reported in Table 4. The samples covered a wide textural range representing loamy sand, sandy loam, loam, silt loam, silty clay loam, clay loam, sandy clay loam, silty clay and clay textures. The pH of the contaminated soil/spoil/sediment samples ranged from 3.20 to 4.27 with an average value of about 3.7. The uncontaminated samples had a pH range of 4.37 to 6.06 with the highest pH's (CL-1A and CL-1B samples) being the result of the buffering capacity of underlying limestone beds at the base of the watershed. The extractable base distributions in most contaminated samples favored Mg over Ca. Uncontaminated samples showed the opposite trend. Based on comparisons between CEC and total extractable bases, Al appeared to be the dominant saturating cation of the exchange phase in most samples. The relatively high CEC's of samples OH-1B, OH-1D, and OH-1E are the result of high clay content or presence of montmorillonite in the clay fraction. These samples were the only ones to contain considerable amounts of minerals with high buffering capacity (montmorillonite, vermiculite).

Mineralogical compositions of whole soil (< 2 mm) and clay (< 2  $\mu$ m) fractions are reported in Tables 5 and 6, respectively. Of the aluminosilicate minerals, kaolinite was the dominant mineral in the clay fraction of all but one sample (BD-3A), in which mica predominated. Kaolinite was also the second most dominant mineral after quartz in most whole soil samples, except those high in jarosite. Mica was in general the next most abundant aluminosilicate of whole soil samples followed closely by feldspars. Vermiculite (OH1, OH2, BD) and 1.0/1.4 nm interstratified minerals (OH1, OH2) were also found in minor quantities. In addition to the OH-1B, OH-1D, and OH-IE samples, three other samples from the OH-1, OH-2 watersheds, two from the BD and four from the CL watershed also contained some smectite or vermiculite in their clay fraction. Clay size mica was within the same range as in whole soil samples. Other silicate minerals in small quantities included HIV (high in CL-1A and CL-1B), 1.0/1.4 nm interstratified, guartz, and feldspars. Goethite was common in most samples while gibbsite was detected only in samples OH-2E and CL-1A.

Aluminum sulfate minerals were detected in small to moderate quantities primarily in the whole soil fraction of most of the contaminated samples. Alunite and aluminite were most common with sporadic appearances by basaluminite and hydrobasaluminite. Of the other sulfate minerals, jarosite was the most frequent encounter with sporadic appearances by gypsum and anhydrite. The high concentrations of jarosite and alunite in some clay fractions are indicative of fairly good crystallinity. On the other hand, the absence of jurbanite XRD peaks in all samples, in spite of support from solution compositions, suggests a very poorly crystalline component. The latter discrepancy, which appears to be common in acid mine drainage environments, prompted Van Breemen (1973) to call jurbanite a "mystery mineral".

#### Soil Solution and Surface Water Composition

Elemental and ionic composition of soil solutions and surface waters are listed in Tables 7, 7a, 8 and 8a, respectively. Solution and water pH values were generally considerably lower than soil pH's, ranging from 1.96 to 3.96 in contaminated samples. Solution pH values of uncontaminated samples, although lower, were closer to their respective soil pH. The pH of surface waters agreed fairly well (some waters had slightly higher pH) with the pH of interstitial soil solutions percolating through the spoil, sediment or soil strata of the watershed. There was no consistent relationship between amount of clay or type of mineralogy and the pH of soil solutions or surface waters in the watershed. This indicates that the low buffering capacity of these materials has already been consumed by the chemistry of the acid mine drainage waters.

The highest Al concentrations were found in the OH1 and OH2 watersheds. The water sample collected from the flume of the CL watershed had no detectable Al. Generally, surface waters contained lower Al concentrations than percolating soil solutions and the Al concentrations decreased with distance from the disturbed area. This trend is best illustrated in the BD watershed where the flume water sample (BD-7) at the base of the watershed contains only 0.23 mM Al while samples BD-8 and BD-9 representing drainages of increasing proximity to the disturbed area contain 0.66 and 0.71 mM Al, respectively. Similar trends were observed in all watersheds. Again, as with

the pH, there was no evidence of buffering of the amounts of Al released in the watersheds by increases in clay content or presence of high charge minerals. The reduced Al concentrations found in surface waters, versus those found in percolating waters of comparable pH levels are attributed to dilution effects and precipitation of aluminum sulfate minerals, which remove considerable amounts of Al from the drainage water. Aluminum sulfate precipitation is supported by mineralogical and solution composition data as well as by the good correlation (r = 0.92) between Al and SO<sub>4</sub> concentrations. The high Si concentrations of most solutions and surface waters (compared to the solubility of quartz) verify that their high Al level is the product of extensive destruction of aluminosilicate minerals from the interaction between acid mine drainage and spoil/soil/sediment strata. The relatively good agreement between total Fe and  $SO_4$  concentrations also suggested a possible control of soluble Fe by iron sulfate minerals especially in solutions with high iron concentrations. This was supported by mineralogical data indicating small to moderate amounts of jarosite in several contaminated samples. The low Fe and SO<sub>4</sub> content of uncontaminated soil solutions and some surface waters suggests Fe-control by a sulfate-free Fe mineral (Fe-hydroxide or Fehydroxyoxide).

From the other solution components the higher Na levels over those of K are attributed to the abundance of albite versus orthoclase (confirmed by XRD) and the lower interaction of Na as compared to that of K with the solid phase. Calcium, Mg, and sulfate concentrations, although high, are within the expected range for acid sulfate weathering materials. Calcium concentrations were generally close to the range supported by the solubility of gypsum, especially in the OH watersheds.

#### Mineral Phase Regulation of Aluminum Levels

Solution ionic activities were used to assess the potential equilibrium of Al<sup>3+</sup> in soil solutions and surface waters with mineral phases. Mineral phases that were evaluated included gibbsite (poorly and well crystalline with pK's of 32.4 and 33.9, respectively), kaolinite (pK = 76.4), alunite (KAl<sub>3</sub>(SO<sub>4</sub>)<sub>2</sub>(OH)<sub>6</sub>; pK = 85.4), jurbanite (Al(SO<sub>4</sub>)(OH).5H<sub>2</sub>O; pK = 3.8) and basaluminite (Al<sub>4</sub>(SO<sub>4</sub>)(OH)<sub>10</sub>.5H<sub>2</sub>O: pK = 117.6).

A plot of pH versus the negative logarithm of the activity of  $Al^{3+}$  in Fig. 1 suggests at least two possible controls of Al equilibrium in solutions with pH < 4.2 and those with pH > 4.2. Although the break in slope may cover the pH range from 4.2 to 5.0, it is apparent that the activity of  $Al^{3+}$  in solutions and surface waters of low pH (< 4.2) is consistent with the stoichiometry of jurbanite and those of pH > 4.2 with the solubility of kaolinite or gibbsite. These trends are in agreement with data reported by Nordstrom and Ball (1986), indicating a slope break in the pH range of 4.5 to 5.0 as the separation boundary for  $Al^{3+}$  control between basic aluminum sulfates and aluminosilicates. This suggests that  $Al^{3+}$  levels in all solutions in contact with contaminated samples and surface waters are controlled primarily by the solubility of basic aluminum sulfate minerals since their pH is < 4.2. On the other hand,  $Al^{3+}$  in solutions of uncontaminated soils is controlled by aluminosilicate minerals.

Although surface waters, because of their shorter residence time, should exhibit increasing tendencies toward mineral saturation with aluminosilicates rather than basic aluminum sulfate minerals, no such trend was detected in the watersheds studied. The only exception was the water sample collected from the flume of the CL watershed located below Mississippian limestone beds. This sample had undetectable amounts of Al and therefore it is not plotted in Fig. 1.

The control of  $Al^{3+}$  activities in soil solutions and surface waters of the contaminated watersheds by basic aluminum sulfate minerals is also illustrated in Fig. 2. The positive slope of this regression line (1.3) agrees very well with that reported by Nordstrom and Ball (1986) for acid mine waters of California and is explained by the conservative  $S0_4^{2-}$  behavior attributed to dilution factors. A plot of  $Al^{3+}$  activities versus pH + pSO<sub>4</sub> in Fig. 3 leaves no doubt that solution  $Al^{3+}$  is controlled by the solubility of an aluminum sulfate mineral with 1:1:1 ( $Al:SO_4:OH$ ) stoichiometry. The excellent fit of the data to this regression line suggests that pH and  $S0_4^{2-}$ activities can predict with 85% accuracy  $Al^{3+}$  levels in soil solutions and acid mine waters. The stoichiometry of this empirical relationship, which is consistent with the solubility of jurbanite, appears to be independent of the non-carbonate mineralogical composition or texture of the spoil, soil, or geologic strata in the watershed.

Figure 4 depicts a solubility diagram similar to that used by Van Breemen (1973) and Nordstrom (1982) to demonstrate potential  $Al^{3+}$  control by aluminosilicate and basic aluminum sulfate minerals in acid sulfate waters. Soil solution and surface water composition points of the contaminated samples congregate rather well about the jurbanite solubility line at low  $2pH + pSO_{d}$ values, whereas at values > 11, representing the uncontaminated samples, there is a tendency for undersaturation with respect to jurbanite and confinement to the region bounded by the gibbsite and kaolinite saturation lines. Only four of the 22 contaminated soil/spoil samples and about half of the surface waters were slightly undersaturated with jurbanite. These undersaturated solutions and waters, nevertheless, were still confined to the region bounded by the jurbanite and alunite solubility lines. The data suggest that precipitation of jurbanite dictates the upper limit of dissolved  $Al^{3+}$  in most soil solutions and some surface waters of contaminated areas in the watershed while alunite provides a lower solubility limit. The absence of jurbanite in the mineralogical composition of the samples studied supports Van Breemen's (1973) speculation that this mineral may be amorphous or present in quantities not detectable by XRD. Regardless of its presence or crystallinity jurbanite is only an intermediate metastable phase controlling  $Al^{3+}$  in solution, with an eventual control transfer to the more stable minerals of the system, alunite and kaolinite. Equilibrium with alunite, however, may rarely be reached because of slow nucleation and precipitation kinetics (Nordstrom, 1982). Kaolinite appears to be the most stable mineral controlling  $A1^{3+}$  in uncontaminated samples (Fig. 4). Gibbsite, K-aluminite, and basaluminite do not appear to influence dissolved  $A1^{3+}$ . These interpretations are supported by comparisons of solution ion activity products and solubility product constants of the above minerals (Tables 9, 10 and 11).

Control of  $Fe^{3+}$  in Solution

Equilibrium evaluations similar to  $Al^{3+}$  were made for soluble  $Fe^{3+}$  using amorphous  $Fe(OH)_3$  (pK = -3.54), goethite (FeOOH; pK = 0.02), and jarosite (KFe<sub>3</sub>(SO<sub>4</sub>)<sub>2</sub>(OH)<sub>6</sub>; pK = 12.5) as potential mineral phases controlling the activity of Fe<sup>3+</sup> in solution. Figure 5 depicts the relationship between pH and the negative logarithm of Fe<sup>3+</sup> activity in the studied solutions and surface waters in reference to the solubility of the above minerals. The considerable scatter of the points in the diagram suggests that the solutions do not appear to be in equilibrium with a specific Fe mineral. Although all solutions are supersaturated with goethite, their Fe<sup>3+</sup> activities are considerably higher than those supported by the solubility of this mineral. Surprisingly, some of the most supersaturated solutions were those of surface waters which along with a few soil solutions supported the metastability of amorphous Fe(OH)<sub>3</sub> at low pH levels (2.75 to 4.00). The remaining contaminated solutions were mostly confined in the metastability region bound by the amorphous Fe(OH)<sub>3</sub> and the jarosite saturation lines.

A plot of the solution points, in a pH + pSO<sub>4</sub> versus pFe<sup>3+</sup> system (Fig. 6), suggests that acid mine drainage solutions in the OH1 and OH2 watersheds obey a different regression relationship than those of BD and CL watersheds. Although the two regression lines are almost parallel to each other (slope - 1.6) the same level of pH + pSO<sub>4</sub> predicts higher Fe<sup>3+</sup> activities in the BD and CL than in the OH watersheds. The above relationships which also imply conservative behavior are attributed to dilution effects and possible redox potential variations in the studied watersheds.

#### CONCLUSIONS

Dissolved Al released in ground and surface waters of acid mine drainage contaminated watersheds with low buffering capacity appeared to be controlled by the solubility of the basic aluminum sulfate minerals jurbanite and alunite. The detection of only alunite by XRD suggests that jurbanite may be amorphous or present in very small quantities. The control of soluble Al by basic aluminum sulfate minerals did not appear to be affected by the diverse mineralogical and textural compositions of the soil and geologic strata in the watershed, which may have been limited by the buffering of the pH of the systems. Soil solutions in contact with uncontaminated soil and geologic strata in the watersheds supported soluble Al levels consistent with the solubility of kaolinite. The data indicate that in solutions with pH < 4 the pH plus activity of SO<sub>4</sub><sup>2-</sup> can accurately predict the levels of soluble Al<sup>3+</sup> in surface and ground waters of the watersheds. Similar predictions can be made for soluble Fe<sup>3+</sup> levels but the pH - SO<sub>4</sub><sup>2-</sup> - Fe<sup>3+</sup> relationship appears to be watershed specific.

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Table 11. Ion activity products (pQ's) for common sulfate and aluminosilicate minererals in water samples.

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Vatershed	Location	Prevalent Rock Types	Prevalent Soil Types	Area ha	Approx. Mining Date	X Disturbed
IHO	Tributary to Muddy Creek, near Horton, Ohio County	Sandstone, siltstone and shale, Lower and Middle Pennsylvanian, Tradewater and Caseyville Formations, Mining Citv and Elm Lick coal beds.	Bethesda (loamy-skeletal, mixed, acid, mesic Typic Udorthents); Fairpoint (loamy- skeletal, mixed, nonacid, mesic Tvoic Udorthents): Zanesville	80	1970	70
042	Tributary to Bens Lick Creek, near McHenry, Ohio County		(fine-silty, mixed, mesic Typic Fragludalfs).	20	1970	60
BD	Tributary to Fourmile Creek near Garner, Boyd County	Sandstone interbedded with shale, Middle Pennsylvanian, Breathit Formation, Princess No. 7 coal bed.	Latham (clayey, mixed, mesic Aquic Hapludults); Shelocta (fine-loamy, mixed, mesic Typic Hapludults).	4 N	1960	18
ป	Tributary to an unnamed depression near Nora, Clinton County	Sandstone interbedded with shale, Middle and Lower Pennsylvanian, Breathit Formation underlain by Mississippian lime- stone, unnamed coal bed.	Shelocta (fine-loamy, mixed, mesic Typic Hapludults).	23	1973	16

Table 1. Watershed Characteristics.

Vetershed	•	Samp 1e	Characteristica
	VI-H0	a bot	Upper 10 cm of graded spoil-soil overburden material
	0H-18	and iment	
	04-IC	sediment	Clayey and shely undistrubed beda below the mined coal seems percolated by acid
	01-10	sediment	stine drainage
	0H-1E	sectment	
	1- <del>1</del> 5	water	Ground water seeping from spoil material
	2-HQ	water	Surface stream watar draining the watershed
570	40-10	[]	South metanets is a second southed with and twented for another the
ž			opoit motoling in peacons vortatt with addimentation party moto-
		1110001 DBD	urange mineral presipitates on pond benk this alarest sectorization as and hist
	3-5		annia minaral pracipitatas an puma sama Amadrian-brad and matamini in mininal anatas utab and manakatian
	26-140		versigner op oor menster at in poordiel contact with povimentation and weber (AM-90-4 a verse B tarters AM-95 - Jacks B a C burkes)
		100	pord weter (onteder a upder a societai, unted = lower a v darieun) societ kock (o do control) accelte disk andioceasian acceleration
	¥-45	i Loda	Spoil Dank in in gestons i contact with sequenciation powermater 
	9 : 5		HIG-slope of spot Dank
	5-40	apot 1	Creat of spoil bank
	8-8	spotl	Summit of spoil bank
	5-5	1 t oqt	Exposed floor of old sedimentation pond
	6-3	water	Sedimentation pond water
	7-10	mater	Sedimentation pond water
	0#-5	water	Depression in an old sedimentation pond floor
	0H-6	water	Sedimentation pond water
00	80-1V	sed iment	Undisturbed shaly sandstone beds over the mined coal seem
	80-1B	1100	Depressional area and receiving acid mine drainage waters from adjacent spoil
	90-IC	1108	bank (80-18 = upper slope; 80-1C = lower slope)
	80-2A	ao 1 1 f	Uncontaminated sandstone bed, parant material of Shalocta soil (C horizon)
	80-28	teo i 14	Uncontaminated A and BA horizon of Shelocta soll
	B0-2C	aoi 11	Unconteminated Bt horizon of Shelocta soil
	80-3A	Epol ]	Highly oxidized spoil bank
	80-38	1 tode	Spoil meterials less oxidized than 80-3A
	BD-3C	1101	Subsoil meterial below the graded spoil bank
	80-7	water	Flume in stream draining the strip-mined watershed
	80-8	water	Intermittent surface water draining into the stream (60 m above flume)
	6-09	tater	Spoil seep draining into the stream (100 m above fiume)
ರ	CL-1A	ao11†	Top soil (A + BA Norizons) along stream bank (base of wetershed, near flume)
	CL-18	#01]†	Subsol) (Bt + BC horizons) ajong stream benk (base of watershed, near flume)
	CL-2A	ao! 1f	Surface soil (A horizon) of noncontaminated Shelocta soil
	CL-28	111 so 117	Subsoil (Bt horizon) of noncontaminated Shelocta soil
	CL-2C	sed iment f	Sandstone-shale interbedded C horizon of uncontaminated Shelocta spil
	CL-20	sed iment	Shaly undisturbed layer above the seam seeping acid mine drainage
	01-10	weter	flume in stream draining the strip-mined watershed

t Uncontaminated soil and sediment samples.

Table 2. Sample description.

Equation	Equilibrium Constant	Reference
$Al^{3+} + H_{2}0 = Al(OH)^{2+} + H^{+}$	pK = 4.99	Adams, 1971
$Al^{3+} + 2H_2^0 = Al(OH)_2^+ + 2H^+$	pK = 10.13	May et al., 1979
$Al^{3+} + 3H_20 = Al(OH)_3^0 + 3H^+$	pK = 15.2	Adams, 1971
$Al^{3+} + SO_4^{2-} = AlSO_4^+$	pK ≖ 3.2	Adams, 1971
$Fe^{3+} + H_{2}0 = Fe(OH)^{2+} + H^{+}$	pK = 2.19	Lindsay, 1979
$Fe^{3+} + 2H_20 = Fe(OH)_2^+ + 2H^+$	pK = 5.69	Lindsay, 1979
$Fe^{3+} + 3H_20 = Fe(OH)_3^0 + 3H^+$	pK = 13.09	Lindsay, 1979
$Fe^{3+} + SO_4^2 = FeSO_4^+$	pK = 4.15	Lindsay, 1979
$Al(OH)_3 = Al^{3+} + 3OH^{-}$	pK <sub>sp</sub> = 32.65	Johnson et al., 1981
microcrystalline gibbsite		
AL(OH)3 = AL <sup>3+</sup> + 30H	рК <sub>sp</sub> = 33.9	Johnson et al., 1981
well crystalline gibbsite		
$KAl_3(SO_4)_2(OH)_6 = K^+ + 3Al^{3+} + 2SO_4^{2-} + 6OH^-$ alunite	рК <sub>sp</sub> = 85.4	Nordstrom, 1982
$Al(so_4)(OH)_6 + H^+ = Al^{3+} + so_4^{2-} + H_2^{0}$ jurbanite	рК <sub>sp</sub> = 3.8	Nordstrom, 1982
$Al_4(SO_4)(OH)_{10} = 4Al^{3+} + SO_4^{2-} + 100H^{-}$ basaluminite	рК <sub>sp</sub> = 117.6	Wordstrom, 1982
$Al_2Si_2O_5(OH)_4 + 5H_2O = 2Al^{3+} + 2H_4SiO_4 + 6OH$ kaolinite	pK <sub>sp</sub> ≖ 76.4	Helqeson et al., 1969
$Fe(OH)_3 + 3H^+ = Fe^{3+} + 3H_2O$ amorphous	рК <sub>sp</sub> = -3.54	Lindsay, 1979
KFe <sub>3</sub> (SO <sub>4</sub> ) <sub>2</sub> (OH) <sub>6</sub> +6H <sup>+</sup> = K <sup>+</sup> + 3Fe <sup>3+</sup> + 2SO <sub>4</sub> <sup>2-</sup> + jarosite	6H <sub>2</sub> 0 pK <sub>sp</sub> = 12.5	Lindsay, 1979
FeOOH + $3H^+$ = Fe <sup>3+</sup> + $2H_2O$ goethite	pK <sub>sp</sub> = 0.02	Lindsay, 1979

Table 3. Equilibrium relationships used in this study.

Table 4.	Physical	and che	mical	properties of so	soil, spoil	and sediment	t samples.			
	Par	Particle Si	Ze .	TT			Extractabl	ble Bases		
Sample	Sand	Silt	Clay	Class	PH†	Ca	ĝ	¥	Na	CEC†
		<b>%</b>	1 1 1 1 1 1					cmol(+)kg <sup>-1</sup>		
AL-HO	54.0		•	_	1	•	ဖ	· ?	Γ.	-
0H-18	<b>*</b> 0	50.8	36.7		3.52	1.28	2.30	-	0.23	6.0
			٠	510		•		-; c	Ņ-	
OH-1E	17.4	24.2	58.4	1	, m	2.32	5.43	0.24	<u>- </u>	28.27
OH-2A				-		9				0 7
OH-2D				: 11s	• •	? ~!	• •	• •	<u>;</u>	, o
OH-2E						1		•	<u>ب</u>	8
OH-4A				Ţ	•	Γ.	•	٠	ų,	1.2
0H-4B	•			sic/sicl	•	9		•	2	4.7
OH-4C		٠		<u>2</u>		- '	•		2	9
	80. 4 60. 7	51.8	38.4	sicl	3.21	7.89	2.92	0.07	0.31	17.62
	•			_	•	•		•	N	0. V
80-1A	_		e.	51]		1	۰.	•	~	<u>.</u>
80-18	<b>60</b>	•	•	[].	•		αļ	•	~ יי	ς,
80-1C	s e	٠			•	Ξ, c		٠	ŊĊ	Г
BU-24 BD-284	• •		é e	- 5 - 1 - 5	•	5	-		Ĵ.	
BD-2C1	34.1	39.6	26.3	]/c]	4.46	0.56	0.83	0.11	0.23	7.92
BD-3A	· LO		~	~	•	ŝ	σ,		۳.	Ľ.
BD-3B	+	•	ġ.	$\sim$	•	<b>`</b> .'	٠.		-	ŝ
BD-3C	∞ –		÷.	l/scl	<b>4</b> .03	Ú.	~	٠	Ξ.	Ϋ́.
CL-1A1		-	•	51]	<u></u>		<u>م</u>	଼	•	•
CL-18‡	æ.	ص	5		0	1	~	. ا	٠	÷
CL-2A‡	പ്	<u>ن</u> م	<del>.</del>	s]	<u> </u>	ģ				•
CL-28 <b>‡</b>	ດີ	<u>.</u>	i.	ບ່	ې	<u> </u>	ς, i	<b>-</b>	•	<u>.</u> ,
cL-2C cL-2D	28.6 63.2	34.3 25.9	37.1 10.9	5 [J	4.44 4.06	0.61	0.41 1.18	0.15	0.20 0.18	14.30 6.55
		water; CE	C - NH <sub>a</sub> oac	IC pH 7						
‡ Unconta	Úncontaminated :	samples	•							

	A 111 / A	111	Ē	۷	7	3	-		E.	50	200	JAK	AL
						6	kg <sup>-1</sup>	X 10 <sup>-</sup>	1				
H-1A	1		17	61	36		Ş					I	
	¥	6	2	4 5		4	ŝ	1 1	•	• •	4 1	ŝ	-
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u-10	•	•	20	16	45	9	æ	( 1	2	•	•	~	י י
H-10	1	•	2	15	29	16	თ	•	ŝ	, ,	•		α
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1-2A,	ú	¥0	12	10	24	4	1	1					
OH-2B	•	•	3	2	•	r	;	4 9	n	) 1	1	81	5
t-2C₹													
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1-2E	m	( 1	ŝ	9	67	• •	• •	46	~	<b>~</b>		ł	n
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Vermiculite; INT - Interstratified; JAR - Jarosite; K - Kaolinite; MI - Mica; Q - Quartz; SM - Smectite; V - Vermiculite. amorphous precipitates Uncontaminated samples

++- von

Sample	WS	>	ЧIV	INT	ĨW	¥	ð	9	LL.	M	BA	JAR	AL	61
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Sample	Æ	AI	fe§	Na	¥	Ca	Mg	Mn	Şł	50 <b>4</b>	cı
			8 8 8 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9	1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1							
0H-1A	2.78	2.66	.13 (	0.37	•	•	•	0.27	3.39		0.03
OH-18	3.20	10.33	.05 (0)	2.20		•	_	1.53	3.16		0.23
OH-1C	2.56	8.52	ė	1.13	0.20	8.01	18.58	2.18	0.83		0.17
OH-10	2.63	5.00	.14 (0.0	0.76		•	0	2.74	1.57	• •	0.01
<b>31-10</b>	2.71	4.93	.47 (0.1	0.81			÷	3.20	2.16	25.93	0.27
0H-2A	3.61	1.93	.37×10 <sup>-2</sup> (	0				19 0	1 67	16 07	13
0H-20	306	74	16v10	; - ~~		•		- 0. - 10		10.01	10.1
	17 6			- r				2.6/	1.05	18.21	0.32
		1.00 7			•			3.92	0.17	33.75	0.24
	28.2	0.35			•			6.39	1.25	27.50	0.02
84-HO	3.8/	0.14	.07×10 <sup>-</sup> 5 (	o O				0.03	0.43	1.44	0.00
0H-H0	3.86	0.51	<u> </u>	) 0.67	0.22	7.50		0.02	0.72	9.53	0.12
01-HO	2.73	30.30		•	•			15.66	3.07	73.59	0,10
0H-5A	1.96	195.93	.03 (15.	•		10.10	162.08	9.38	+QN	QN	9
80-1A	3.86	0.14		~	0.43		9F 0		0 60	1 13	
BD-18			10/ 10	ic		•	52.0			1.16	
BD-1C	3.43		C	•			1.07				
BD-24	4.40		0.01×10 <sup>2</sup> 2	; c		•		0.07~10	-1 0 20	20.0	
BD-2B		0.02210-1	0.07×10-2	; c		•			00	C7.0	0.0
RD-2C	1.15		37410-2	; c	2	•		23	17.0	29	22
RD- 34	30.6	101 0	201012	5 u					17.0		
	2.6	10.0	17410-2	, c	1.02	•	0/.0	01.0	2.1.2		0.10
	2.00 20	10-0	0 07/10-2 (0)			2.0			11.0	200.0	<b>50°0</b>
200	10.0	00.0	01210.	5	00		<b>*</b> 1.0	10.0	01.1	70.0	00.0
CI - 1A‡	5 71	0 01×10-1	0.07×10-2 /	0	0 07		0 51	07~10	-2 0 15		
C1 - 184	5	0.02210-2	0 12×10		0.05410-1	•	5.0	2		•	
CL - 24	4 78	0.05		; c	0.00140		1 2 1	20.0		•	01.0
CL - 284	4.52	.07×10	1 0.03v10 <sup>-2</sup> /	0.06	0.08410-1	0.05		0.05~10	-1 0 15	0.03	
ci - 204	4.32	0.01	0 12×10-2 (0			٠	0.07v10	01/20			8.0 0
CL - 20	3.24	3.66	0.23 (0)	50		•		46 46	201.0		
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	NU = not determined due Unrontaminated samples	ined due to	Insuffictent	amount o	of solution.	·					
		anpres konic nonko:	+ c.2+			L		****			
	In parentnesis	nesis represen		CONCENTRALIONS	ns measured	oy the	ortnopnenan	thraine	method	(ALPHA, 19/6)	

Table 7a. Negative logarithms of ion activities in interstitial soil solutions.

Na <sup>+</sup> pK <sup>+</sup> pCa <sup>2+</sup> pMg <sup>2+</sup> .55 4.72 2.45 3.00 .79 3.84 2.95 2.52
3.09 3.89 2.73 2.32 3   3.24 4.66 2.85 2.48 3   3.21 5.09 6.48 2.40 3
3.12 4.15 2.71 2.52   2.95 3.28 2.56 2.57   2.64 3.44 2.56 2.57   3.16 3.36 2.56 2.27   3.16 3.36 2.56 2.31   3.70 3.66 3.46 3.51   3.57 4.19 2.52 3.31   3.82 4.77 2.77 2.49
2.65 3.41 3.66 3.58 4   3.33 3.94 3.24 3.07 4   3.27 3.94 3.15 3.01 3   3.27 3.94 3.15 3.01 3   4.15 3.90 4.30 4.15 5
2.38 2.90 2.97 2.64   4.15 3.93 3.68 3.89   4.10 4.12 3.64 4.00
3.72 4.19 3.37 3.11   4.40 5.22 4.00 5.40   3.49 3.11 3.23 2.96   4.22 4.11 4.30 4.70   3.42 4.11 4.30 4.70   3.42 4.04 4.10 4.30   3.25 4.21 2.93 2.53

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Total elemental concentrations of surface water samples. Table 8.

Sample	Ħ	Al	Fet	Na	Mg	Ca	×	Si	so4	CI
1-H0	2.81	4.29	1.21 (0.	1.21 (0.09) 1.13	16.27	- 1811	0.19	1.16	4.25	39.81
0H-2	3.08	0.64	1.05 (0.1	.04) 1.05	7.21	3.51	0.28	0.54	12.25	0.02
0H-3	3.25	0.86	0.90 (0.0	0.02) 0.90	9.33	5.70	0.12	0.76	16.72	0.00
0H-4	2.80	1.94	1.02 (0.0	04) 1.02	1.84	10.40	0.40	1.18	32.06	0.24
0H-5	2.72	26.00	3.59 (0.3	39) 3.59	84.58	11.28	0.08x10 <sup>-1</sup>		135.47	0.21
0H-6	3.36	0.18	0.42 (0)	0.42	7.10	6.0	0.26		13.62	0.00
BD-7	3.30	0.23	0.93 (0)	0.93	0.68	0.83	0.13	0.55	2.34	0.05
BD-8	3.72	0.66	1.92 (0)	1.92	1.59	1.31	0.11	0.74	4.84	0.10
BD-9	4.03	0.71	2.06 (0)	2.06	1.36	1.23	0.09	0.65	4.65	0.13
CL-10	6.30	0.00	0.20 (0)	0.20	0.36	0.69	0.79	0.05	1.63	0.00
CL-11	3.79	2.70	0.07 (0)	0.07	3.83	1.85	0.06	1.49	9.72	0.05
+			-6							

I Values in parenthesis represent Fe<sup>2+</sup> concentrations measured by the orthophenanthroline method (APHA, 1976).

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Table 8a.	
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Sample	Hd	pNa <sup>+</sup>	pĶ <sup>+</sup>	pCa <sup>2+</sup>	pMg <sup>2+</sup>	pMn <sup>2+</sup>	pSŐ <u>4</u>	pC1 <sup>-</sup>	pAl <sup>3+</sup>	pFe <sup>3+</sup>
0H-1	2.81	3.05	3.83	2.57	2.15	0	3.00	1.40	3.18	4.24
0H-2	3.08	3.07	3.66	2.89	2.55	0	2.35	4.77	4.22	4.84
0H-3	3.25	3.15	4.02	2.73	2.48	0	2.27	5.19	4.16	5.03
0H-4	2.80	3.12	3.54	2.57	2.28	0	2.10	3.62	3.96	5.07
0H-5	2.72	2.66	5.31	2.80	1.87	0	1.70	3.68	3.19	4.90
0H-6	3.36	3.47	3.69	2.67	2.57	0	2.33	5.41	4.79	5.35
BD-7	3.30	3.08	3.92	3.31	3.39	0	2.84	4.29	4.29	4.67
BD-8	3.72	2.78	4.01	3.19	3.10	0	2.59	4.00	4.00	4.82
BD-9	4.03	2.75	4.11	3.22	3.15	0	2.60	3.90	3.98	5.17
CL-10	6.30	3.75	3.14	3.35	3.64	0	2.96	0	8.45	13.89
CL-11	3.79	4.27	4.31	3.14	2.80	0	2.36	4.30	3.58	6.40

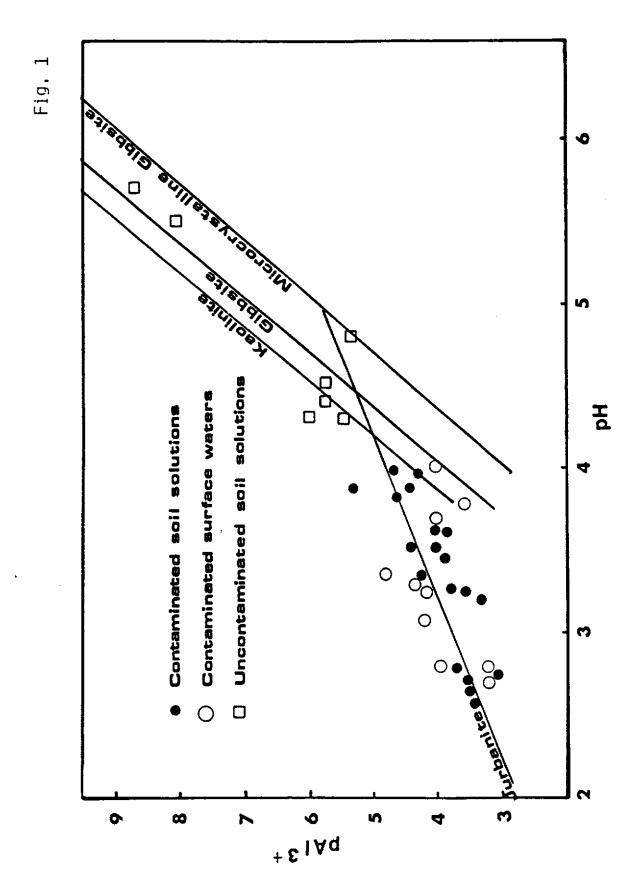
Samples		Gibbsite	Kaolinite	Alunite	Jurbanite	Basaluminite	Jarosite
Uncontaminated soil solutions	s0 s0	33.6 ±0.9	74.7 ±1.7	85.1 ±2.3	5.1 ±0.9	120.0 ±2.7	13.9 ±1.2
Contaminated soil solutions	sot	34.3 <u>+</u> 7.5	77.7 <u>+</u> 1.9	84.6 <u>+</u> 2.1	3.0 ±0.7	124.8 <u>+</u> 3.1	9.5 <u>+</u> 3.1
Surface waters	sot	35.6 <u>+</u> 1.8	77.9 ±3.3	84.5 <u>+</u> 2.6	3.2 ±0.8	124.2 ±5.3	5.1 <u>+</u> 3.2
Reference minerals	₽K∱	33.9	76.4	85.4	3.8	117.6	12.5
tpq = pal <sup>3+</sup> +	3pOH <sup>-</sup> 1	pAl <sup>3+</sup> + 3pOH <sup>-</sup> for gibbsite;	te; 2pAl <sup>3+</sup> + 2pH <sub>4</sub> SiO <sub>4</sub> + 6pOH <sup>-</sup> for kaolinite;	Si04 + 6pOH <sup>-</sup>	for kaolinite		
pK <sup>+</sup> + 3p	A1 <sup>3+</sup> +	pK <sup>+</sup> + 3pAl <sup>3+</sup> + 2pSO <sub>4</sub> <sup>2</sup> + 6pO	6pOH <sup>-</sup> for alunite; pAl <sup>3+</sup> + pSO <sub>4</sub> <sup>2-</sup> - pH for jurbanite;	: pAl <sup>3+</sup> + pS	0 <mark>4</mark> - pH for j	jurbanite;	
4pal <sup>3+</sup> +	. pS04	4pA1 <sup>3+</sup> + pSO <sub>4</sub> <sup>2-</sup> + 10pOH <sup>-</sup> fo	ır basaluminite	; and pK <sup>+</sup> +	3pFe <sup>3+</sup> + 2pS04	for basaluminite; and $pK^+$ + $3pFe^{3+}$ + $2pS0_4^{2-}$ · 6pH for jarosite.	ite.
† solubili	ity proc	solubility product constant.	.:				

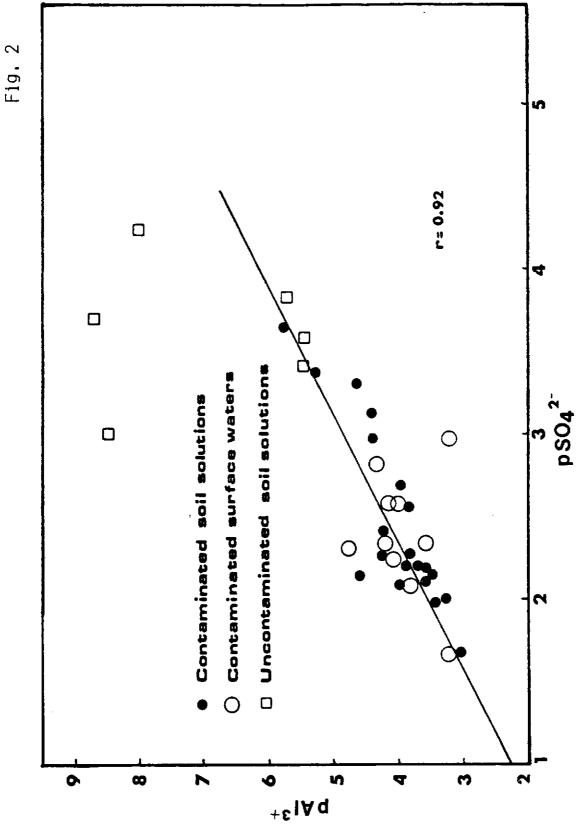
Table 10.	Ion activity products	products (pQ's)	for	common sulfate and alu	aluminosilicate	minerals in soil	l solutions.
Sample	Gypsum	Gibbsite	Alunite	Kaolinite	Jarosite	Jurbanite	Basaluminite
OH-1A	}		1	<u>ب</u>	-		29.2
. 1	•	~ 12	יי כ		; ~ ~	1	3.1
· •	4.73	) r~	86.70	81.59	8.61	2.83	129.97
OL-HO	• •	. ~	. o	0		0	29.8
0H-1E		37.38	9.	-	6.5	<u>о</u> .	29.1
	A 07		A C	Ľ	-	4	21.4
04-20	ŗα	* =	80.79	74.70	8.88	2.55	119.66
01-55 0H-25	y c	- 10	. с . –			7	21.9
OH-4A	2	ັດ	2.7	2	·	്	22.5
0H-4B	. T	0 10	2.7	က	3.7	8	25.4
0H-4C	: <b>•</b>	6 60	2. 8	ထ	4.5	<del>ر</del>	26.1
0H-4D	4.52	36.87	5.0		ຸດ	0	26.6
0H-5A							
	6 78	V		6.3			22.0
	5.00 7	, rur	•	2 2 2 2	•	• •	23.4
	5.74	י ש ל	• •	20		3.00	123.70
BD-2A	7.95	34.56	86.15	76.19	15.32	•	22.7
BD-2B	1						
BD-2C					1	•	
BD-3A	5.17	35.94	82.98	77.00	10.77	2.69	124.50
BD-3B	•	5.7	•	4.6	1.5	Σ,	20.0
BD-3C	٠	4.7	•	5. 3	2.5	<u>б</u> .	22.1
C! - 14	C	~	~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~	4.7		9.	1.3
- 1	? <b>`</b>		1	4.7	14.44	<b>~</b> .	1.2
	ייב		: en	3.0		<u> </u>	2.0
- 1	2		0	6	4	9	1.6
1	:	- 4	85.80	76.48	12.21	4.79	122.23
CL-20	5.13	35.85	ნ.	6.7	9	പ്	4.1
Reference	pK's 4.6	33.9	85.4	76.4	12.5	3.8	117.6

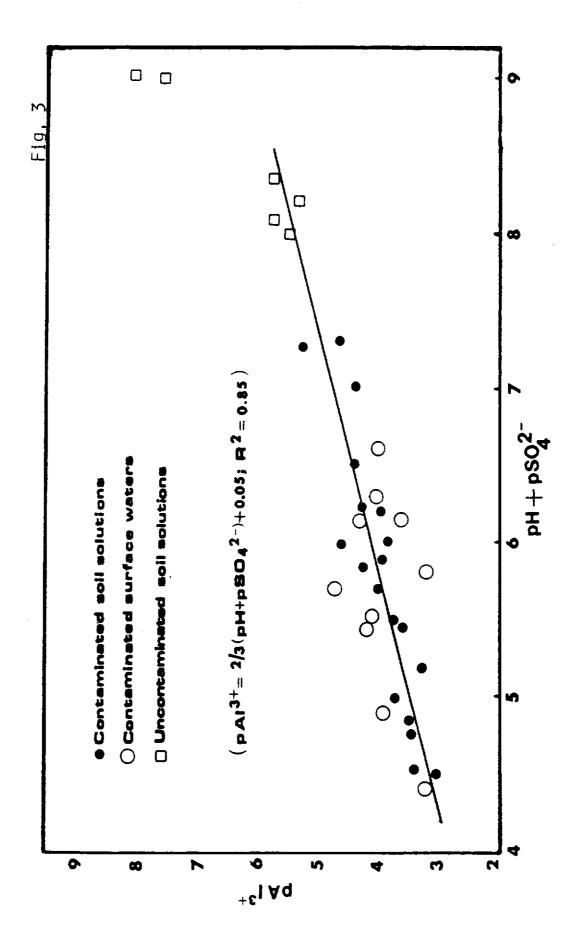
Table 11.	Table 11. Ion activity products (pQ's) for common sulfate and aluminosilicate minerals in water samples.	oducts (pQ's)	for common	sulfate and alu	minosilicate	minerals in v	vater samples.
Sample	Gypsum	Gibbsite	Alunite	Kaolinite	Jarosite	Jurbanite	Basaluminite
0H-1	5.57	36.75	86.51	79.37	5.69	3.37	127.62
0H-2	5.23	36.97	86.54	80.50	4.40	3.49	128.42
0H-3	4.99	36.41	85.54	79.07	4.15	3.18	126.42
0H-4	4.66	37.56	86.82	80.98	6.15	3.26	129.93
0H-5	4.49	37.03	85.96	81.69	7.09	2.17	127.25
0H-6	5.00	36.70	86.56	80.13	4.24	3.76	127.87
BD-7	6.14	36.38	86.67	79.29	3.81	3.83	126.99
BD-8	5.78	34.84	82.87	75.95	1.33	2.87	121.40
80- <b>9</b>	5.81	33.89	81.07	74.16	0.64	2.55	118.23
CL-10	6.31	31.55	80.61	71.72	12.93	5.11	113.76
CL-11	5.50	34.20	81.03	74.06	5.49	2.15	118.76
Reference pK's	pK's 4.6	33.9	85.4	76.4	12.5	3.8	117.6

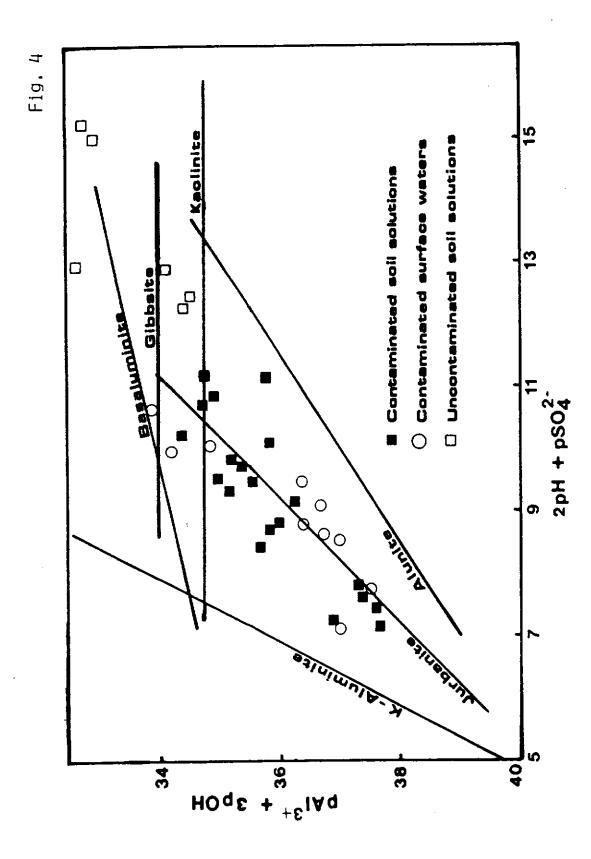
## FIGURE LEGEND

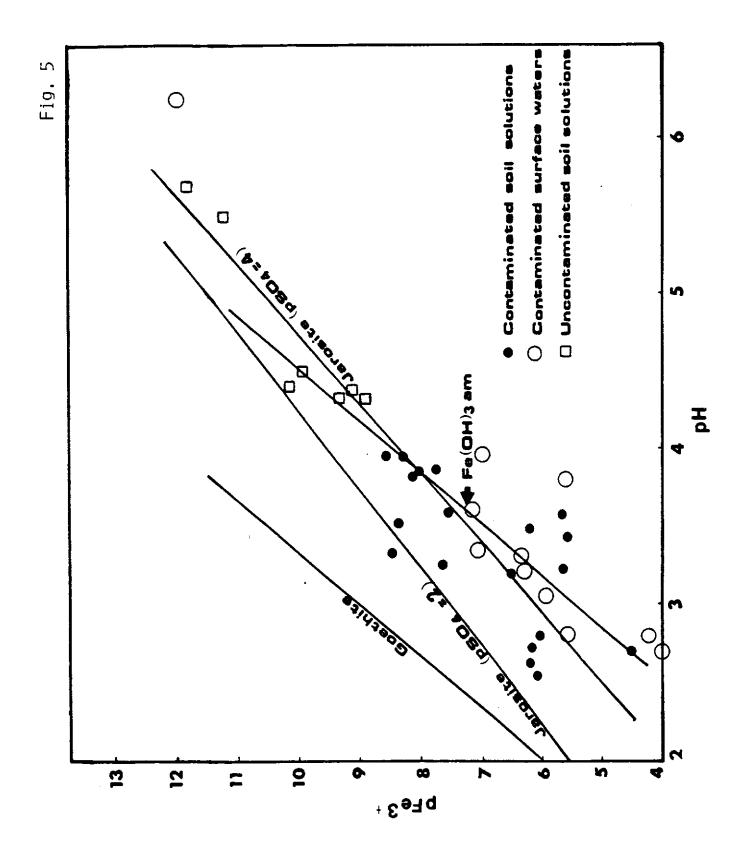
- Fig. 1. Values of pH and pAl<sup>3+</sup> for soil/spoil/sediment solutions and surface waters in reference to jurbanite, kaolinite and gibbsite solubility lines (pH<sub>4</sub>SiO<sub>4</sub> = 3.5;  $pSO_4^{2^-}$  = 3.0).
- Fig. 2. Relationship between  $pAl^{3+}$  and  $pSO_4^{2-}$  in soil/spoil/sediment solutions and surface waters.
- Fig. 3. Relationship between  $pAl^{3+}$  versus  $2pH + pSO_4^{2-}$  in soil/spoil/sediment solutions and surface waters.
- Fig. 4. Solubility diagram for potential aluminosilicate and basic aluminum sulfate minerals controlling dissolved  $Al^{3+}$  in the pAl<sup>3+</sup> versus 2pH + pSO<sub>4</sub><sup>2-</sup> system.
- Fig. 5. Values of pH and pFe<sup>3+</sup> for soil/spoil/sediment solutions and surface waters in reference of goethite, amorphous  $Fe(OH)_3$ , and jarosite solubility lines.
- Fig. 6. Relationship between  $pFe^{3+}$  and  $pH + pSO_4^{2-}$  in soil/spoil/sediment solutions and surface waters of the studied watersheds.

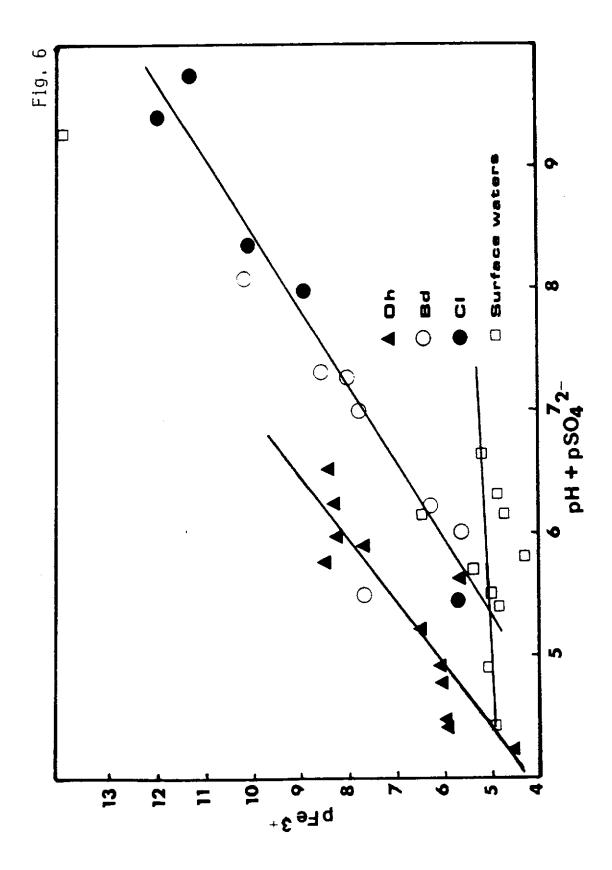












# PART II

# KINETICS OF AL AND FE RELEASE

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## PART II

#### INTRODUCTION

When sulfate enriched acid mine drainages interact with soil and geological materials they form soluble Al and Fe sulfate compounds capable of releasing toxic levels of Al and Fe into adjacent water reservoirs (Caruccio and Geidel, 1978; Driscoll et al., 1984; Evangelou and Karathanasis, 1984; Nordstrom et al., 1979; Van Breeman, 1973; Harmsen and Van Breemen, 1975). Using equilibrium models researchers have suggested that in addition to gibbsite and kaolinite, aluminum sulfate minerals, such as jurbanite, alunite, basaluminite, and aluminite may control the solubility of Al in solutions percolating through these systems (Nordstrom 1982; Nordstrom and Ball, 1986; Anthony and McLean, 1976; Rodriguez-(Clemente and Hidalgo-Lopez, 1985; Karathanasis et al., 1988; Van Breemen, 1973; Johnson et al., 1981). Similarly, in addition to the various Fe-hydroxides (goethite, amorphorus Fe(OH)<sub>3</sub>), jarosite (K Fe<sub>3</sub>(SO<sub>4</sub>)<sub>2</sub>(OH)<sub>6</sub>) may control the solubility of Fe<sup>3+</sup> in effluents of such systems (Harmsen and Van Breemen, 1975; Nordstrom et al., 1979; Singer and Stumm, 1968; Karathanasis et al., 1988). Thermodynamic equilibrium models utilizing the solubility product principle, although useful, present several inherent limitations in accurately predicting Al and Fe behavior in acid mine drainage contaminated watersheds because they are open, heterogeneous and non-equilibrium systems. Furthermore, these models do not take into account kinetic effects that can drastically change the composition of the system over time. This is because the amount and rate of Al and Fe released from the contaminated watershed depends not only on the solubility of the mineral controlling Al and Fe in the effluent but also on physicochemical, mineralogical and hydrological factors controlling the rate of water flow through the soil or geological materials (Rogowski et al., 1977; Lewis and Grant, 1979; Singer and Stumm, 1968; Lowry and Finney, 1962; Longbein and Dowdy, 1964). Significant variations in the chemistry of stream waters draining acid mine contaminated watersheds have been observed as a function of discharge including dilution, concentration or no change by increasing discharge (Johnson et al., 1969). These chemical changes can be explained variously in terms of reaction kinetics (Longbein and Dowdy, 1964) or on the basis of simple mixing models (Johnson et al., 1969; Hall, 1970).

Experimental leaching studies of pyritic coal mine spoil materials designed to predict long term potential acidity also have suggested drastic changes both in dominant chemical species and in concentration over a long time period. In a system with moderate pyrite and low carbonate content, the evolution of drainage water chemistry may progress from an effluent with a high  $SO_4$ -moderate pH, upon exhaustion of the available carbonates, to that with a low pH-high  $SO_4$ , and then to one with a low  $SO_4$ -moderate pH upon exhaustion of the available pyrite in the oxidation zone (Vimmerstedt and Struthers, 1968). These chemical changes over time suggest that the solubility control of Al and Fe in effluents of these systems may be transferred to various minerals, following a mineral sequence pattern which is site and material specific.

The objective of this study was to determine the effect of water percolating at different flow rates through acid sulfate drainage inflicted spoil, soil, and geological materials on the amount of Al and Fe released in a given aquifer.

#### MATERIALS AND METHODS

Two pyritic spoil materials represented by samples OH-4D and BD-3A, two contaminated sediment samples represented by samples OH-1E and CL-2D, and two soil samples (one contaminated by acid mine drainage, BD-1C; the other almost uncontaminated, CL-1A) were selected for the leaching column experiments. The pysicochemical and mineralogical properties of these materials and of the watersheds from which they were sampled are listed in Tables 1, 2, 4, 5 and 6 of part I. Compositions of in situ solutions contacting these material are listed in Tables 7, 7a, 8 and 8a of part I.

The samples were air-dried, ground and sieved through a 2-mm sieve and packed into 50 cm long plexiglass columns of 6 cm diameter. The height of the material in the column was approximately 40 cm. The lower end of each column was attached to a funnel with a 0.45  $\nu$ m pore size filter. A drainage tube was attached to the funnel draining the effluent into a collection flask. The top of each column was closed with a rubber stop with two openings. One of the openings was used as inlet for the leaching solution and the other for aeration. The inlet tubes of each column were connected to distilled water reservoirs, which provided water flow rates of 1 and 3 ml/hr to two sets of duplicated columns. The water flow rates were adjusted and regulated by a persistaltic pump. Decreasing hydraulic conductivity problems occurred in several of the columns (especially with the 3 ml/hr flow rate set) in 3 to 4 instances during the experiment. At that time the water flow was stopped in all columns to prevent overflow and restarted after 5 days to allow stabilization of particle distribution. These stabilization periods also served to study the effect of simulated alternating wetting and drying cycles on effluent compositions. Effluent samples were collected at certain time intervals and at the beginning and end of each wetting cycle over a period of 120-150 days. The effluent samples were analyzed in duplicate for pH, Al, Fe, Ca, Mg, K, Na, Mn, Si, S and Cl according to methods listed in part I of this report. Solution ionic activities were calculated by using the extended Debye-Huckel equation and a method of successive approximations (Adams, 1971), and are reported in Tables 1 through 6a. Ionic speciations were based on dissociation and solubility product constants listed in Table 3.

## **RESULTS AND DISCUSSION**

#### Changes in Effluent Solution Composition with Time

Figures 1, 2, 3, 4, 5, and 6 show effluent composition changes in pH,  $SO_4^{2-}$ , Al, Ca and Mg, respectively, during the 5 months of the experiment. Some major spikes in the distribution of elements are associated with the beginning of a new wetting cycle. The smaller deviations (smaller peaks) in the distribution curves of each element are the result of fluctuations in the hydraulic conductivity of the materials during the course of the experiment.

The composition point distribution in Figure 1 indicates that while there was a very small rise in the pH of effluents BD-3A, OH-4D, CL-2D, and OH-1E during the 120 day period, sharp pH increases occurred in the effluents of CL-1A (4.8 to 8.3) and BD-1C (3.0 to 5.5) samples after 20 and 38 days of leaching, respectively. The fact that the pH of these two samples remained relatively constant for the rest of the experiment suggests that a new pH steady state was established after the specified critical time period. The only other sample showing a moderate but gradual pH increase (2.25 to 3.0) was the OH-4D effluent.

The sharp pH increase in the CL-1A and BD-3A effluents correlates quite well with the drastic decrease of  $SO_4^{2-}$  concentrations from about 14.0 to almost 0.0 mM after 20 days (CL-1A) and from about 14.0 to about 1.0 mM after 40 days (BD-3A) of leaching (Fig. 2). Apparently, after the dissolution and removal of sulfate salts another mechanism prevails and assumes control of the pH in the effluents of these samples. Although significant decreases in  $SO_4^{2-}$  concentrations also occurred in all effluents (Fig. 2), the reduction was not adequate to change the existing pH control mechanism.

The pH or  $SO_4^{2^-}$  changes also had drastic effects on the Al concentration of most effluents (Fig. 3). Aluminum concentrations were 2 to 3 times lower after 120 leaching days. For most of the samples this reduction occurred between the 20 and 40 day period, except for OH-4D, which showed a gradual decrease throughout the experiment. These decreases were even more spectacular for the activity of Al<sup>3+</sup>, dropping 10<sup>8</sup> times in effluent CL-1A,  $10^4$  times in BD-1C,  $10^2$  times in OH-4D, CL-2D, and OH-1E and about 5 times in the BD-3A effluent (Fig. 4).

The fact that the sum of Ca+Mg (mM) leached from the CL-1A and BD-1C samples is about equal to the total mM of  $SO_4^2$  suggests that the complete dissolution and removal of Ca and Mg sulfate salts may be responsible for the dramatic pH,  $SO_4^2$  and Al<sup>3+</sup> changes in these particular samples (Fig. 5 and 6). The constant Ca concentration in the OH-4D effluent during the experiment and the presence of abundant  $SO_4^2$  suggests equilibrium with gypsum (Fig. 5).

These elemental distributions were consistent in both 1 and 3 ml/hr flow rates. Greater consistency, however, was observed with the lml/hr rate because they posed fewer problems in terms of uniformity of hydraulic conductivity and regulation of the water flow.

Effect of Kinetics on Al Solubility Control

Effluents from the two spoil materials OH-4D and BD-3A remained saturated with respect to jurbanite during the entire length of the leaching experiment (Fig. 7). This is also illustrated from the clustering of the effluent composition points around the jurbanite solubility line in Fig. 8 and 9, indicating a near equilibrium condition of these solutions with the above mineral. The uniform distribution pattern of both 1 and 3 ml/hr effluents suggests no apparent effect of the flow rate on soluble Al levels released by these materials. It should be pointed out, however, that the majority of the BD-3A effluent points are also confined in the region bounded by the kaolinite and gibbsite solubility lines, suggesting possible control of Al in these samples also by the above minerals. In contrast, soluble Al levels in effluents of CL-1A, BD-1C, and CL-2D dropped below the jurbanite solubility limit within two weeks of leaching, while the OH-1E effluent supported Al concentrations within the range of jurbanite solubility for the first 45 days and gradually became undersaturated thereafter (Fig. 7).

It is also evident from Fig. 10 that Al levels in the CL-1A effluents after a few days of leaching, became consistent with the solubility of microcrystalline gibbsite, when the few sulfates present in the original material were completely dissolved. A similar trend is also observed in the BD-1C effluents (Fig. 11), where Al concentrations were confined in the region bounded by the jurbanite-alunite saturation lines during most of the experiment, but moving gradually towards the gibbsite and microcrystalline gibbsite solubility regions, where they concentrate during the final stages of

leaching. Again as was the case with the other materials there is no apparent effect of the different flow rates.

A somewhat different pattern of Al solubility control was observed in the CL-2D and OH-1E samples. Aluminum levels in effluents of both of these samples, after an original saturation with jurbanite, appeared to move away from jurbanite and eventually became consistent with the solubility of alunite (Fig. 12 and 13). The Al control transfer from jurbanite to alunite appears to occur faster in the CL-2D material and much slower in the OH-1E material. This is probably the result of the higher buffering capacity of the OH-1E (higher clay content and CEC, presence of expanding minerals) as compared to the CL-2D material (Tables 4, 5, and 6 of Part I).

## Kinetics of Fe Solubility

Total Fe concentrations in the effluents of CL-1A and BD-1C decreased sharply the 8th and 14th day of leaching, respectively, probably as a result of prevailing reducing conditions in the column (Fig. 14). These concentrations dropped below the original Fe levels after the 20th day of the experiment as a result of dramatic pH increases. Abrupt Fe fluctuations at certain time periods in the effluents of OH-4D and BD-3A are attributed to alternating reducing and oxidizing environments, following the pattern of alternating wetting and drying cycles. The above interpretations were supported by the generally poor correlations of Fe with pH and  $SO_4^{2-}$  observed for OH-4D and CL-2D samples.

Iron (Fe<sup>3+</sup>) solubility in the effluents of all but the two soil materials (BD-1C and CL-1A) fluctuated between that of jarosite and amorphous Fe(OH)<sub>3</sub> (Fig. 15). There was a cycling pattern between these two minerals for Fe<sup>3+</sup> solubility control during which partial precipitation of one or the other may have occurred. The clustering of many points around the goethite saturation line suggests that this mineral may have also been involved in the control cycle.

Iron (Fe<sup>3+</sup>) activities in the BD-1C effluent, after an initial undersaturation period consistent with the solubility of jarosite, approached gradually the amorphous Fe (OH)<sub>3</sub> solubility line along which they clustered during the rest of the experiment. The solubility of amorphous Fe (OH)<sub>3</sub> appeared to also control Fe<sup>3+</sup> activities in the CL-1A effluents throughout the

experiment, with only some points, representing the final leaching stages, moving towards goethite solubility control (Fig. 15). As it was the case with Al the kinetics of Fe solubility did not appear to be influenced by the different leaching rates.

#### CONCLUSIONS

The column leaching experiments also confirmed that the major components controlling the behavior of these systems over time are primarily pH,  $SO_4^{2-}$ , Al and Fe. Alternating wetting and drying cycles appeared to cause more drastic changes in the above components than different flow rates. Two of the studied materials exhibited pH increases from 5.0 to 8.0 and from 3.0 to 5.5, respectively, within the first 30 days of leaching. These changes caused dramatic decreases in soluble Al levels. In most cases, however, these Al levels were maintained under the solubility control of Al-sulfate minerals. Two materials remained within the solubility range of jurbanite, two shifted from the jurbanite to the alunite range, and two from jurbanite to that of microcrystalline gibbsite. Iron release appears to follow a cycling pattern during which the solubility control of Fe in the effluent is alternating between jarosite, amorphous Fe (OH)<sub>3</sub> and goethite. These findings suggest that equilibrium relationships, although quite useful, may not be sufficient for predicting the dynamic behavior of systems such as those included in this study. Kinetic relationships reflecting Al and Fe released as a function of time and the sequence of mineral phases controlling Al and Fe solubility are also necessary in order to obtain more realistic models of Al and Fe release into aquifers of such systems.

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Ion activities (negative logs) in effluent solutions of the CL-1A soil material leached at ml/hr rate as a function of time. Table 6a.

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Ion activities (negative logs) in effluent solutions, of the BD-3A spoil material as a function of time. Table 1:

pFe <sup>3+</sup>	•	•	٠		6.07		•	•	•		•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•
pMn <sup>2+</sup>	2	9	æ.	<u>م</u>	5.20	4	r.	<u>و</u>	æ.	æ	ŝ		4.	പ്	പ്	e,	9.	9	5	ő	9	ŋ	9	9	ş	6.02	æ.
pA1 <sup>3+</sup>	ω,	0.	-		4.00	2	ഹ	.9	Γ.		4	9	Γ.	<del>م</del> .	σ.	0.	σ.	<u> </u>	6	<u>б</u>	ი.	<del>ە</del> .	ი.	σ.	σ.	4.10	ຕ.
pNa <sup>+</sup>	4	<u>б</u> .	0	2	3.51	9.	9	9	2	æ.	0	9	5	5	<b>~</b>	5	5	σ.	0	0	0.	0.			Γ.	9	0
pK <sup>+</sup>	<u>م</u>	σ.	0	0	3.16	~	2	~	~	e.	ີ.	ີ.	?	ς.		<del>ر</del> .	4.	4.	പ	പ	ŝ	പ	<u>د</u>	<u>د</u>	<u>و</u>	<u>د</u>	<u>د</u>
pMg <sup>2+</sup>	Ŀ.		ື	2	3.72	Γ.	0	г.	9.	5	۳.	Γ.	ς.	2	ς.	4	2	പ	۲.	σ,	σ.		ς.	б.	8.	5.59	æ.
pCa <sup>2+</sup>	8	.1	<u>.</u>	4	3.70	æ.	æ,	ω,	ထု	0	<u> </u>	0	Γ.	Γ.	Ξ.	~	<u>.</u>	e,	ц,	9	œ.		-	ſ.	5	ຕຸ	9.
pS04-	2	2	9.	1	2.97	-	Γ.	2	2	2	e,	2	2	Γ.	٣.	2	2	۳.	<del>ر</del>		е.	٣.	e.	e,	4.	4.	. 2
Hd	•		•	•	3.75	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	•	5	5	3.78	6.
Rate(m])	I	1	-		1	1	1	-1			1	-1	1	-	-		1	Π		4	-1	1	1	4	r4	Ч	I
Date	6-25	7-2	7-8	7-17	7-27		1	-	2	2	2	2			6	7	7	7	10-22	$\sim$	2	ΞĒ.	11-5	11-9	11-13	11-17	11-23

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Ion activities (negative logs) in effluent solutions of the BD-3A spoil material as a function of time. Table la:

pFe <sup>3+</sup>	ပ္	2	~	ę.	ب	<u> </u>	ς.	Γ.	ŝ	æ,	က္	~	9	4	æ.	~	ц,	œ.	e.	ς.	Γ.	σ.	æ.	7.99	e.	പ്	ຕຸ	5.	Γ.
pMn <sup>2+</sup>	. с.	5	4	.1	.1	5.26	4	σ.	<u>.</u>	4.		S.	9	<del>م</del>	Γ.	Γ.		Γ.	0.	0.	0	<u>.</u>	<u>.</u>	6.08	۰.	0		e.	<u>ب</u>
pA1 <sup>3+</sup>	8	<u>.</u>	2	0.	Γ.	Г	0	ω.	æ.	4.	2	2	0	Γ.	0	0		-	~		-			4.14	Γ.	$\sim$	$\sim$	2	4.22
pNa <sup>+</sup>	.6		ő	2	9	4.10	Γ.	Γ.	~	Γ.	5	r.	0	Γ.	2	<del>ب</del>	ņ	ņ	<u>ب</u>	4	<u>.</u>	4	4	4.48	4	4	<b>T</b>	4	4.54
pK <sup>+</sup>	σ,	0	~	ч.	4	4.	പ	ŝ	e.	4	4	4	<u>с</u>	ц С	φ.	۲.	5	5	5	7.		L.	٢.	3.76	٢.	5	<b>`</b> .	Γ.	æ.
pMg <sup>2+</sup>	<u></u> .	~	0	æ,	2	9	æ,	~	4	<u>с</u>	ŝ	5		5	<u>ب</u>	2	2	<u>.</u>	6.24	4.	~	-	r,			<u> </u>	1.	5.96	æ.
pCa <sup>2+</sup>	6.	2	9.	<u>.</u>	0.	4.31	4	.6	ω.	<u>ں</u>	9.	9.	2	8	0	0.	0			ς.	œ.	2	ς.	5.38	~	Γ.	5	4.	4.40
pS0 <sup>2</sup> -	ന	e.	Γ.	2	<b></b>	ς.	<u>с</u>	~	2	4.	3	ŗ.	<i>с</i> .	4	4	ŝ	<u>د</u>	ŝ	9	9	2	<u>ە</u>	و	3.60	പ	9	ပ	പ്	9
Hq	ഹ	<b>~</b>	ω.	ω.	5	5	5	ŝ	œ.	Γ.	<u>.</u>	<u>.</u>	<u>б</u>	œ.	r.	ω.	ω.	ω.	7	ω.	ω.	L.	L.	3.75	5	5	ი.	ω.	ω.
Rate(ml)	ę	ო	ო	ო	ო	ო	ო	ო	ო	m	ო	ო	ო	m	m	e	ო	m	ę	ę	m	m	ო	m	с	ę	m	e	ę
Date	ہے۔ ا	1	Ϋ́	1	- í	-		-	7	7		-	2	$\sim$	6	6	6	1-0		0-1	0-2	0-2	0-2	1-2	1	- 1	-	•	-2

Ion activities (negative logs) in effluent solutions of the OH-4D spoil material as a function of time. Table 2:

pFe <sup>3+</sup>	ິ.	5	<b>~</b> .	2	<u>ب</u>	~	?	4.91	0	<u> </u>	<b>·</b> ·	<u> </u>	ς.	ŝ	0	<u>ى</u>	9	9	0.	<u>م</u>	<u>с</u> .	æ.	5	Ģ	و	ς.		<del>ب</del>
pMn <sup>2+</sup>	•	•	4.69	•	•	٠		ч.	ς.	2	5.	ŝ	5	æ,	0	0	0	<b>.</b>	<del>ر</del>	?	<b>.</b>	۳.	<u> </u>	ີ.	e.	ę.	6.28	<del>د</del> .
pA1 <sup>3+</sup>	0.	4	ŝ	9	5	5	9	3.71	5	Ŀ.	ω.	φ	6	φ.	ი	б.	ς.	б.	ς.	Γ.	0	0.	0	਼	0.	Γ.	œ.	
pNa <sup>+</sup>	4.13	4.	4.	<del>د</del> .	4.	4.	Γ.	4.22	e.	4.	4.	ഹ	ĥ	പ	5	φ.	?	9.	പ	Ģ	9.	പ	4	ო	റ	ŝ	ŝ	9.
pK <sup>+</sup>	Ŀ.	9.	9	ς.	8.	9	e.	4.47	ŝ	ġ.	æ.	ŝ	•	9	5	æ,	4.	<b>·</b> ·	5	ω.	5	9.	പ	4.	9		5	σ.
pMg <sup>2+</sup>	~	æ	0	<u>°</u>		۳.	e.	3.42	4	°.	8.	<del>م</del>	~	<u> </u>	ഹ	r,	?	ŝ	9.	æ,	ອຸ	0	æ.	œ.	æ.	σ,	<u> </u>	~
pCa <sup>2+</sup>	<u>د</u>	ς.	<b>m</b>	ς.	'n.		2	2.30	~	S.	۳.	<del>ر</del> .	ς.	ę.	ຕ.	ς.	2	ņ	ς.	2	'n	<b>m</b>	2	~	<del>ب</del>	2	·Ω	2
ps0 <b></b> <sup>2</sup> -	1.72	0	-	Ë,	Ē,	Ξ.	۳.	2.16	Ē.	Ċ,	2	~	~	2	~	2	~	2	2	2	~	2	~	~	2	~	ഴ	2
Hq	~	4	œ.	<u>د</u>	<u>د</u>	<u>۲</u>	9.	2.76	٢.	æ.	8.	S.	ົ	<del>م</del>	σ.	ς.	σ.	<u> </u>	ი.	σ.	a,	ົ	ດຸ	റ	<u> </u>	<u> </u>	ດ	6.
Rate(ml)	<b></b> 4	-1		-1		Ħ	-	1	-	-		-		-1	-1	-4	1		-			-	-1	1	4		-4	1
Date	1	1	E	- I -	2	ဗို	-	8-14	2	2	-	2	-2	2	- I -	1		-	7	-	- I	2	2	11-2	11-9		11-17	11-23

Ion activities (negative logs) in effluent solutions of the OH-4D spoil material as a function of time. Table 2a:

													•
•	pFe <sup>3+</sup>	5.37	5.78	6.17	6.39	6.52	6.85	6.50	6.70	4.11	4.52	4.94	5.33
¢	pMn <sup>Z+</sup>	3.93	4.48	4.92	5.34	5.47	5.58	5.65	5.75		5.72	5.65	5.73
•	pAl <sup>3+</sup>	3.01	3.41	3.76	3.82	3.90	3.97	4.04	4.09	4.31	4.61	4.40	4.45
	pNa <sup>+</sup>	4.13	4.52	4.81	4.79	4.98	4.97	5.04	5.13	3.72	4.21	3.93	4.30
	⁺ ¥d	5.79	5.69	5.12	5.27	5.54	6.84	6.84	6.09	4.10	4.19	4.13	4.26
ć	pMg <sup>Z+</sup>	2.52	2.75	3.17	3.45	3.69	3.76	3.64	3.72	3.91	3.64	3.96	4.02
Ċ	pCa <sup>2+</sup>	2.51	2.34	2.17	2.30	2.29	2.28	2.29	2.26	2.24	2.25	2.29	2.28
c	pS04-	1.75	2.00	2.07	2.20	2.21	2.22	2.24	2.23	2.23	2.24	2.28	2.28
	Hd	2.28	2.45	2.54	2.76	2.79	2.85	2.78	2.84	3.22	3.44	3.35	3.37
	Rate(ml)	ო	ო	ო	ო	ო	m	ო	m	m	m	Ś	ო
	Date	6-25	6-26	7-2	7-8	7-10	7-13	7-15	7-17	8-12	8-14	9-21	9-25

Ion activities (negative logs) in effluent solutions of the OH-IE sediment material as a function of time. Table 3:

pFe <sup>3+</sup>	5.58	4.97	5.45	5.11	5.33	5.39	5.56	5.86	5.49	5.65	5.36	5.66	5.75	5.86	5.96	6.03	6.01	6.08	6.15	6.15	6.18	6.24	6.22	6.26	6.12
pMn <sup>2+</sup>																4.16									
pAl <sup>3+</sup>	4	L.	3.99	•	2	4.29	<b>n</b>	0	m.	Ģ	Ģ	ŝ	E C	-	Ψ.	4.72	<b>·</b>	5		~	~		-	Ч.	Ч.
pNa <sup>+</sup>	ထ္	4	°.	5	4.09	4	<u>с</u>	9.	ς.	5	Ψ.	5	9	٦.	Ψ,	4.60	5	-	5	4	17	٦.	9	4.61	9.
pK <sup>+</sup>	က	<u>ں</u>	ဖ	<b>O</b>	<u>ں</u>	2		~	r.	r.	<b>~</b>	Γ.	Ξ.	ω.	Ξ.	3.83	ŝ	Ξ.	ω,	Ξ.	Ξ.	α,	σ.	5	•
pMg <sup>2+</sup>	ົດ	ഹ	ഹ	œ	8	4			ന	2	-	l N	2		က	3.36	<u>ح</u>	5	Γ,	α				<u>ц</u> ,	
pCa <sup>2+</sup>	2.63	3.06	ο Ο	0	3.26	4	3.42	4	3.52	ഹ	ġ (C	j O	e.	р С		3.76	5	5	5		÷Ψ				<b>.</b>
pS02-	1.79																								3.26
łd																3.03						1	•	. –	3.06
Rate(m])	<b></b>		•	4	4	۱	•	4	<b>،</b>		4 +	•	4	•	4	•		•	•	•	4	•	4	4	44
Date		7-2	7-8	) –	10	1 (*	· .	i٩	' c	ı ç	10	5.	2	0 - 0 - 0	$\sim$	-	10-19		10	10	10		0-11	<b>ا</b> ا	11-23

time.
function of tir

pFe <sup>3+</sup>	8 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5
pMn <sup>2+</sup>	2.71 3.72 3.55 3.55 3.55 3.55 3.55 3.55 3.55 3.5
pAl <sup>3+</sup>	3.38 3.69 4.20 4.20 4.50 4.50 50 4.50 50 50 50 50 50 50 50 50 50 50 50 50 5
pNa <sup>+</sup>	3.00 3.38 3.38 3.38 4.21 4.56 55 55 55 55 55 55 55 55 55 55 55 55 5
pK <sup>+</sup>	3.40 3.71 3.71 3.73 3.73 3.73 3.73 3.73 3.73
pMg <sup>2+</sup>	2.46 2.73 2.94 2.94 2.99 2.99 2.99 2.99 2.99 2.99
pCa <sup>2+</sup>	2.48 2.48 3.26 3.26 3.34 3.43 3.26 3.43 3.26 3.26 3.26 3.26 3.26 3.26 3.26 3.2
pS0 <b>2</b> -	2.83 2.83 2.83 2.83 2.83 2.83 2.83 2.83
Hq	2.60 2.75 2.95 2.95 3.01 3.01 3.01 3.23
Rate(m])	ო ო ო ო ო ო ო ო ო ო ო ო
Date	6-25 6-25 6-30 7-6 7-13 8-17 8-17 8-17 8-17 8-17 9-21

Ion activities (negative logs) in effluent solutions of the CL-2D sediment material as a function of time. Table 4:

pFe <sup>3+</sup>	6.37	പ	Γ.	2	4	ų.	e.	ς.	œ.	2	ຕ.	°.	°.	<u>و</u>	~	Ľ,	ŝ	٩.	ŝ	Υ.	e.	٩,	ີ.	ግ	Ş		~	ষ্	с;
pMn <sup>2+</sup>	3.55	പ	æ.	Ŀ.	9	9	0	2	ີ.		ę.	e.	N,	2	4	٩,	শ্	٩.	٦,	শ	٩,	٩.	٩.	٦,	٦.	٩,	÷	ę	<b>`</b>
pA1 <sup>3+</sup>	3.72	0	ņ			4	4.36	4	ດ	ŝ	ŝ	റ	9.	ဗ	œ.	<b></b>	<b></b>	Γ.	<u></u>	ę	e.	ę	ę	φ.	9.	ς.	ω,	α	αļ
pNa <sup>+</sup>	ω.	ŝ	ŝ	4	ŝ	4	4.32	4	4	4	4	4	4	പ	e.	5	2	5	Ľ,	5	5	7	Υ.	٩.	<u>ج</u>	٦.			÷
pK <sup>+</sup>	<u>б</u> .	<del>م</del> .	Ξ.	σ.	0	ц С	3.53	9.	œ.	L.	r.	r.	r.	r.	ω.	æ.	<u> </u>	æ.	L.	5	ς.	C.	5	5	Γ,	Γ.	5	5	<b>·</b>
pMg <sup>2+</sup>	ပ	œ	ထ္	ື	~	റ	3.29	<u>.</u>	œ.	°.	4	4	°.	4	ି	ц С	പ		2	÷	ω.	Ξ.	Γ,	Γ.	Ψ.	Ψ.	Γ.	<b>_</b>	Ψ,
pCa <sup>2+</sup>	୍		က္	ц С	<b>_</b>	<u> </u>	3.66	ġ	<u></u>	æ	<u> </u>	Ξ.	α	<u> </u>	<u> </u>	α	S,	5	ယ္	2	ų	ယ္	Ψ.	ι Ψ	<u> </u>	9	ω,		-
pS04 <sup>-</sup>	ີ.	ц С	9.	5	ດຸ	0	2.98	0	0	2	Γ.		~	~	4	2	2	2		<u>ج</u>	<u>ج</u>	က	<u>ر</u>	<u>ີ</u>	പ	പ	പ	7	Υ.
Hq	ω.	5	0	ω.	ົ	σ.		-		~		Ξ.	Ч.	~	2	2				-	2		-						3.17
Rate(m])	-	1	1		-1	, ,		-					-	. –	1	. –		. –	( <b></b>	▎┍╍┙	. –		-		) <del></del>	1	ı —		-
Date					2	က		-	l N	2	-	l N	9-2		-			•		-	i N			۱ م	- 11 - 12	6-11	11-13	11-17	11-23

	pFe <sup>3+</sup>	6.19	5.83	6.47	6.23	5.28	5.47	5.55	5.59	5.73	4.54	5.10	4.63
	pMn <sup>2+</sup>	3.56	3.90	4.38	4.68	4.63	4.69	4.80	4.80	4.79	4.04	4.43	4.27
	pAl <sup>3+</sup>	3.73	4.17	4.67	4.97	4.88	5.07	4.97	4.93	4.92	4.74	5.02	4.76
	pNa <sup>+</sup>	4.00	4.70	5.19	5.12	4.68	4.79	4.79	4.79	4.79	4.03	4.27	4.14
	pK <sup>+</sup>	3.94	4.00	4.17	4.16	3.94	3.94	3.98	3.94	3.94	3.40	3.53	3.39
	pMg <sup>2+</sup>	2.72	3.28	3.43	3.75	3.67	3.73	3.75	3.75	3.77	3.55	4.03	3.59
	pCa <sup>2+</sup>	2.99	3.37	3.52	4.02	3.99	4.02	4.06	4.11	4.18	3.83	3.83	3.84
	ps0 <sup>2</sup> -	2.34						3.45					
function of time.	На	2.78											3.29
function	Rate(ml)	ę	ო	ო	m	m	ო	с	ო	ო	ო	ო	m
	Date	6-25	6-26	6-30	7-2	7-8	7-10	7-13	7-15	7-17	8-12	8-14	9-18

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Table 4a: Ic	fi
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	-	-			_							
pFe <sup>3+</sup>	5.60	4.49	4.32	4.48	5.19	7.24	8.07		8.47	8.35	7.93	90.06
pMn <sup>2+</sup>	3.53	3.58	3.75	3.98	4.24	4.32	4.33		4.36	4.34	4.61	4.76
pAl <sup>3+</sup>	3.59	3.88	4.46	4.52	4.49	4.57	6.80		6.60	7.49	6.81	7.70
pNa <sup>+</sup>	3.26	3.95	4.30	4.55	4.91	4.90	4.81		5.02	4.94	5.03	4.93
pK <sup>+</sup>	3.50	3.57	3.62	3.71	3.80	3.84	4.08		4.13	4.18	4.30	4.26
pMg <sup>2+</sup>	2.99	3.04	3.02	3.31	3.49	3.79	3.78		3.74	4.04	3.88	4.11
pCa <sup>2+</sup>	2.94	3.08	3.25	3.54	3.80	3.80	3.81		3.99	3.88	4.04	4.22
ps0 <mark>4</mark> -						3.48				•	3.57	
Hd						3.52						
Rate(ml)	1	-	-	-		1	٦	1	П	п	П	I
Date	6-25	7-2	7-8	7-17	7-27	7-31	8-14	8-20	8-27	9-18		<b>57-6</b> 59

Ion activities (negative logs) in effluent solutions of the BD-IC soil material as a function of time. Table 5a:

pFe <sup>3+</sup>	5.20 4.80 5.24 8.18 8.18 8.18 7.59 7.59 6.99
pMn <sup>2+</sup>	3.55 3.55 3.92 3.92 4.54 4.51 4.54 4.50 4.56 4.50 4.50 4.50 4.50 54 4.50
pA1 <sup>3+</sup>	3.73 5.10 5.99 5.90
pNa+	44 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5
pK <sup>+</sup>	4 4 4 4 4 4 4 3 3 3 3 4 5 4 5 4 5 3 3 3 4 5 5 5 5
pMg <sup>2+</sup>	3.05 3.05 3.91 3.91 4.17 4.17 4.19 19 19 19 19 19 19 19 19 10 10 10 10 10 10 10 10 10 10 10 10 10
pCa <sup>2+</sup>	2.98 3.385 3.95 3.95 4.01 4.01 4.02 4.01 78 7.03 4.01 78 7.03 78 78 7.03 78 78 78 78 78 78 78 78 78 78 78 78 78
pS04-	2.43 3.17 3.55 3.38 3.38 3.34 3.34 3.34 3.34 3.34 3.34
Hd	2.84 2.25 2.25 2.25 2.25 2.25 2.25 2.25 2.2
Rate(m])	ო ო ო ო ო ო ო ო ო ო ო ო
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Table 6:

pFe <sup>3+</sup>	8.30	5.33	14.04	16.95	16.41	15.74	15.80	15.97	15.64	15.44	15.18	14.24	17.24
pMn <sup>2+</sup>	3.99	3.68	4.19	4.65	4.64	4.48	4.60	4.51	4.14	4.99	5.60	4.77	7.52
pAl <sup>3+</sup>	6.15	6.24	13.16	15.70	14.64	13.80	14.51	14.27	13.86	14.23	14.66	13.79	16.34
pNa <sup>+</sup>	3.60	3.80	3.95	4.16	4.23	4.32	4.36	4.06	4.03	3.82	4.03	4.03	4.19
pK <sup>+</sup>	2.95	3.24	3.39	3.57	3.62	3.55	3.58	3.40	3.42	3.35	3.36	3.44	3.42
pMg <sup>2+</sup>	2.85	3.19	3.55	4.10	4.16	4.03	4.02	3.75	3.63	3.56	3.58	3.85	4.12
pCa <sup>2+</sup>	2.09	2.34	2.88	3.49	3.48	3.32	3.34	3.02	2.87	2.93	3.19	3.19	3.86
ps0 <mark>4</mark> -	2.06	2.28	3.27	5.12	5.72	6.18	6.84	5.51	6.26	5.83	6.05	5.69	7.33
Hd	4.82	3.91	7.74	8.41	8.15	7.91	8.05	7.98	7.85	7.97	7.89	7.97	8.50
Rate(m])	-	-	٦	1	1	-1	-	1	1	1	1	1	1
Date	6-25	7-2	7-8	7-13	7-17	7-27	7-31	8-12	8-14	8-20	8-27	8-31	9-25

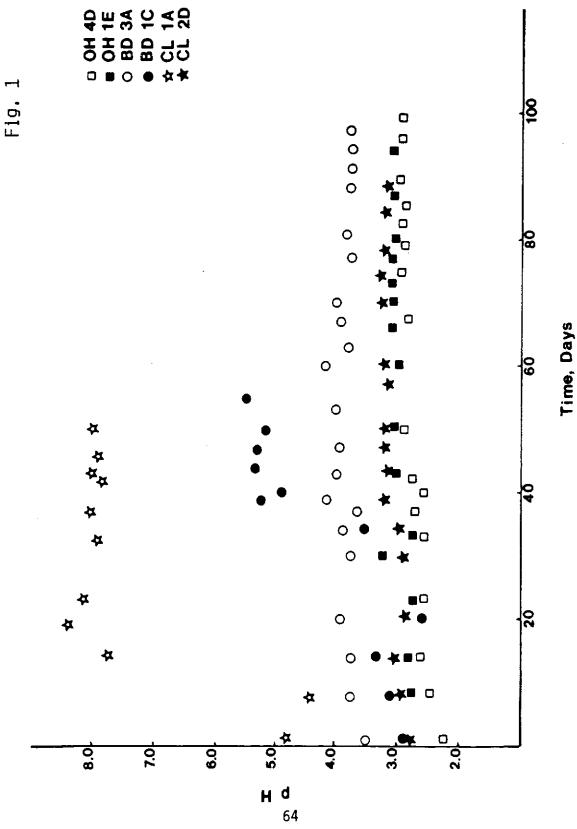
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Ion activities (negative lo	nsn <u>2</u> -
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Table 6a: time.	Nata B

pFe <sup>3+</sup>	6.58	6.19	7.63	14.03	14.30	15.64	15.46	15.89	16.51	15.45	16.75	15.50
pMn <sup>2+</sup>	3.81	3.94	4.12	4.29	4.36	4.53	4.41	4.35	4.48	5.03	4.29	5.95
pAl <sup>3+</sup>	6.13	5.88	6.43	13.23	13.00	13.93	13.63	14.18	14.50	14.67	15.55	13.55
pNa+	3.64	4.06	4.16	4.14	4.21	4.26	4.10	4.12	4.13	3.77	3.85	3.89
pK <sup>+</sup>	3.16	3.45	3.49	3.45	3.49	3.48	3.47	3.47	3.48	3.29	3.40	3.37
pMg <sup>2+</sup>	2.97	3.36	3.49	3.61	3.79	3.89	3.72	3.76	3.86	3.69	3.87	3.79
pCa <sup>2+</sup>	2.12	2.56	2.70	2.90	3.07	3.26	3.05	3.10	3.20	3.30	3.38	3.40
ps0 <mark>4</mark> -	2.10	2.49	2.61	3.37	3.56	3.99	3.84	3.98	4.17	5.03	5.34	4.84
Н		4.36	4.98		7.64							
Rate(ml)	m	m	ო	ო	ო	ო	ო	ო	ო	ო	ო	ო
Date	6-25	6-26	7-2	7-8	7-10	7-13	7-15	7-17	7-27	8-12	8-14	9-14

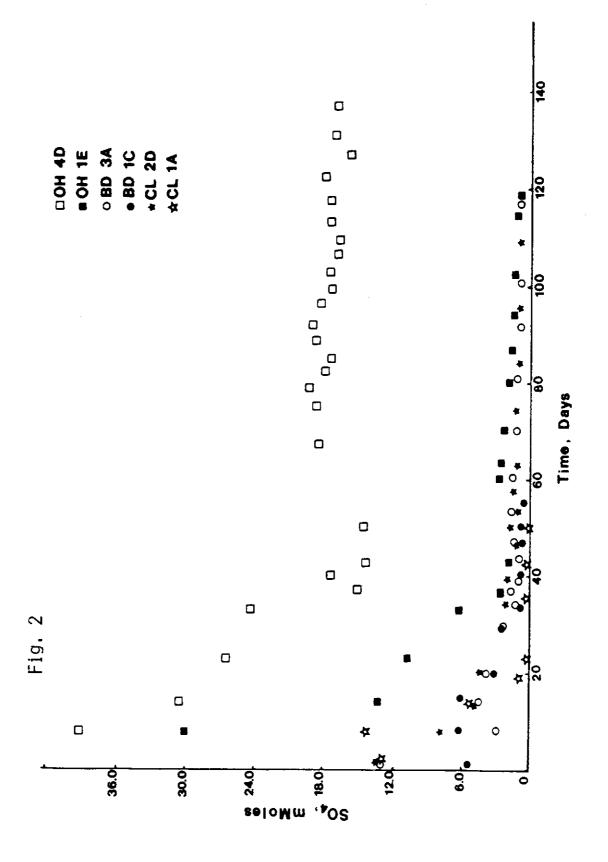
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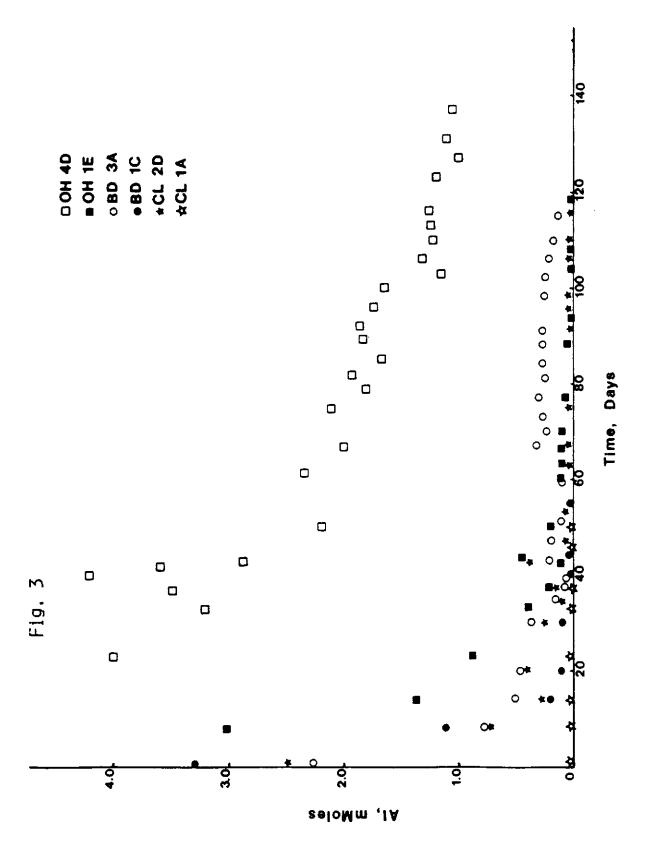
## LIST OF FIGURES

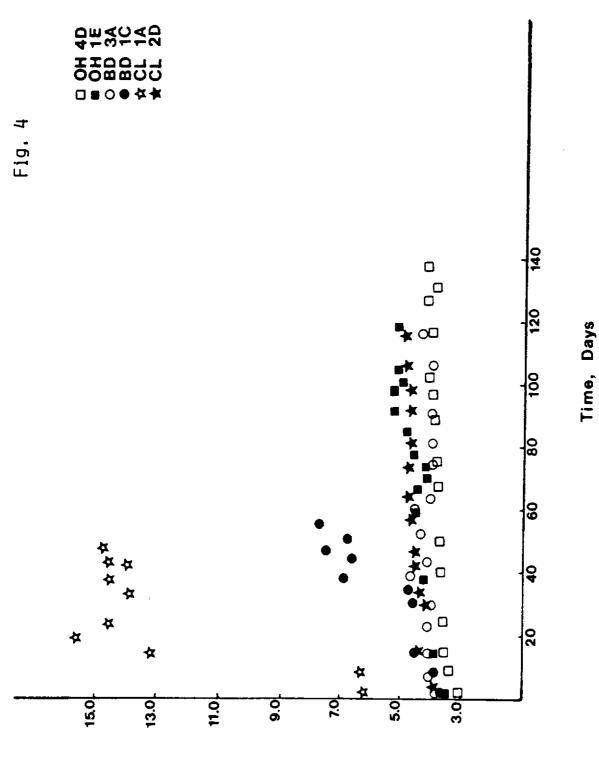
- Fig. 1. Changes in effluent solution pH as a function of time.
- Fig. 2. Sulfate levels released in effluent solutions as a function of time.
- Fig. 3. Aluminum levels released in effluent solutions as a function of time.
- Fig. 4. Activities of  $Al^{3+}$  in effluent solutions as a function of time.
- Fig. 5. Calcium levels released in effluent solutions as a function of time.
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- Fig. 12. Effluent composition points of the CL-2D column in reference to the solubility of common aluminum sulfate and aluminosilicate minerals.
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- Fig. 14. Iron levels released in effluent solutions as a function of time.
- Fig. 15. Activities of  $Fe^{3+}$  in effluent solutions in reference to the solubility of jarosite, goethite, and amorphous  $Fe(OH)_3$ .



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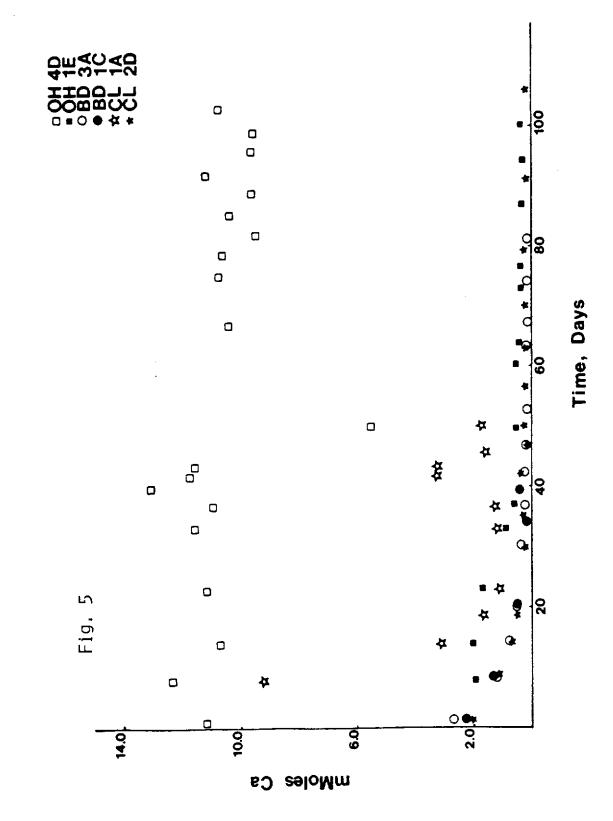


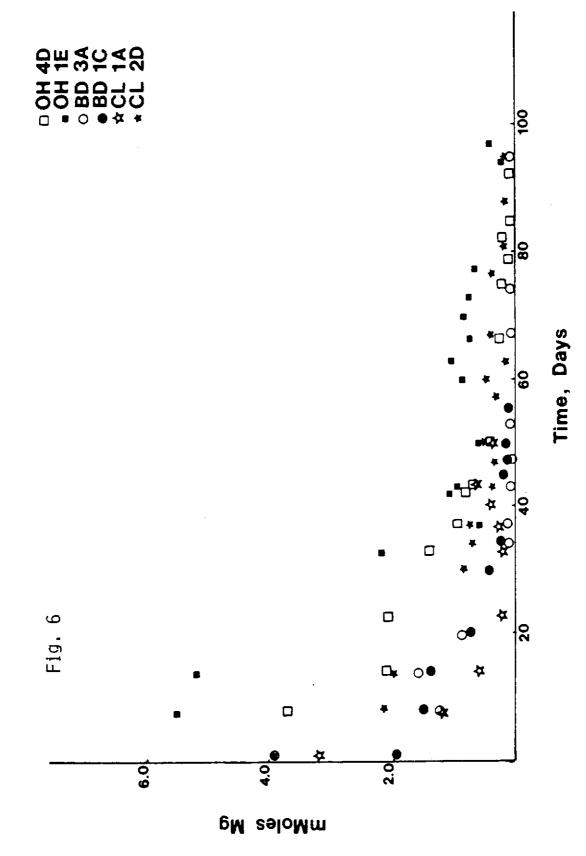


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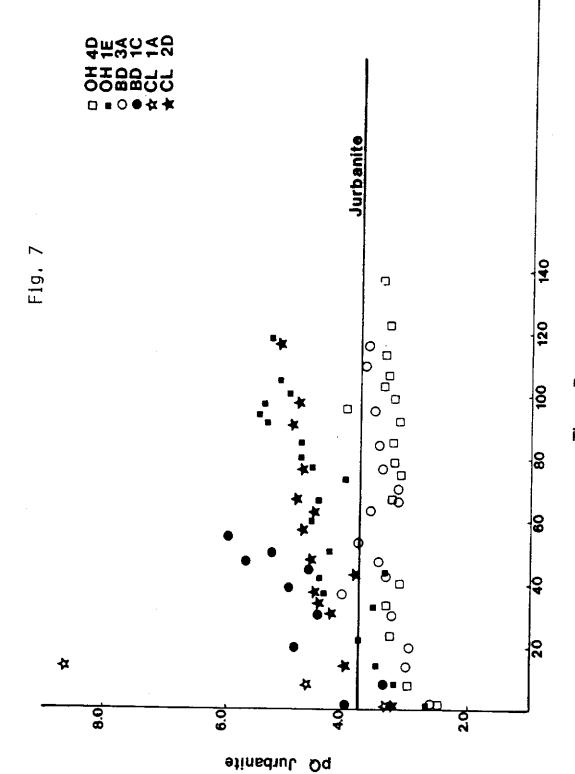
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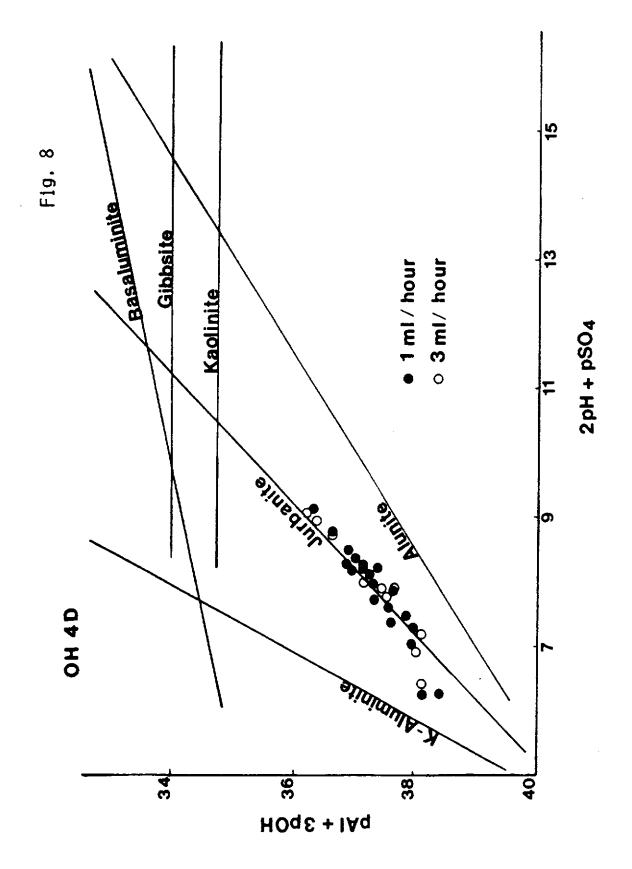


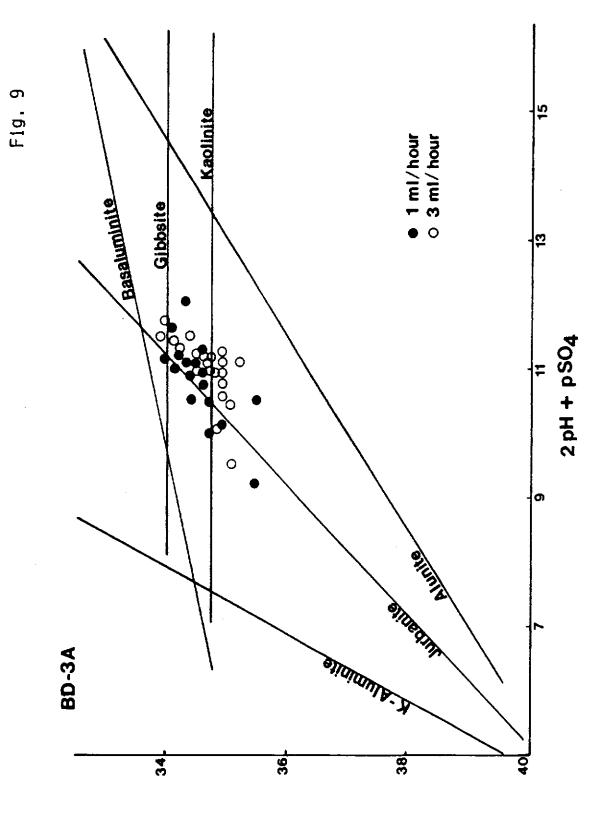


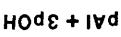


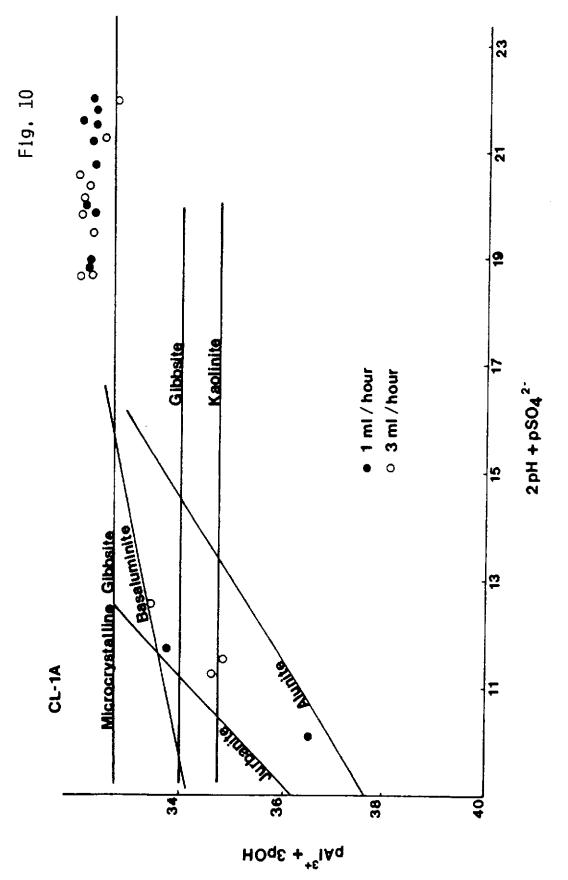


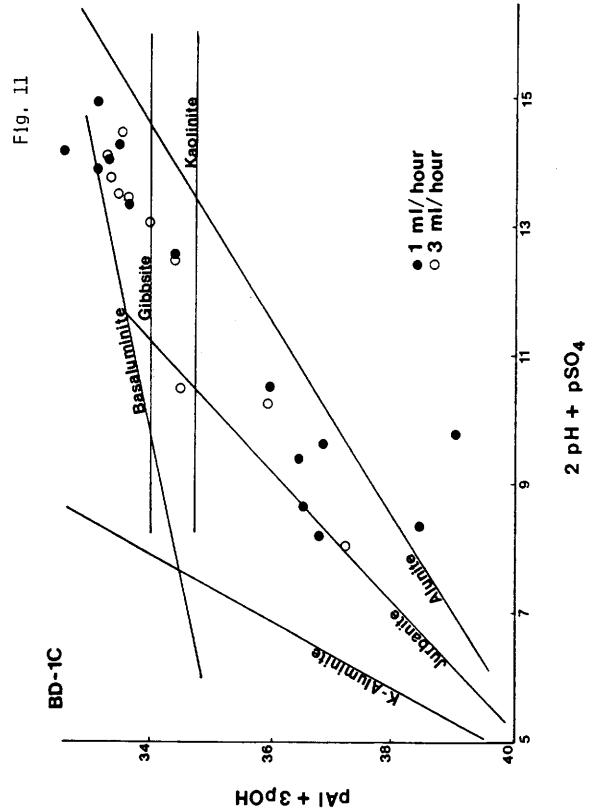
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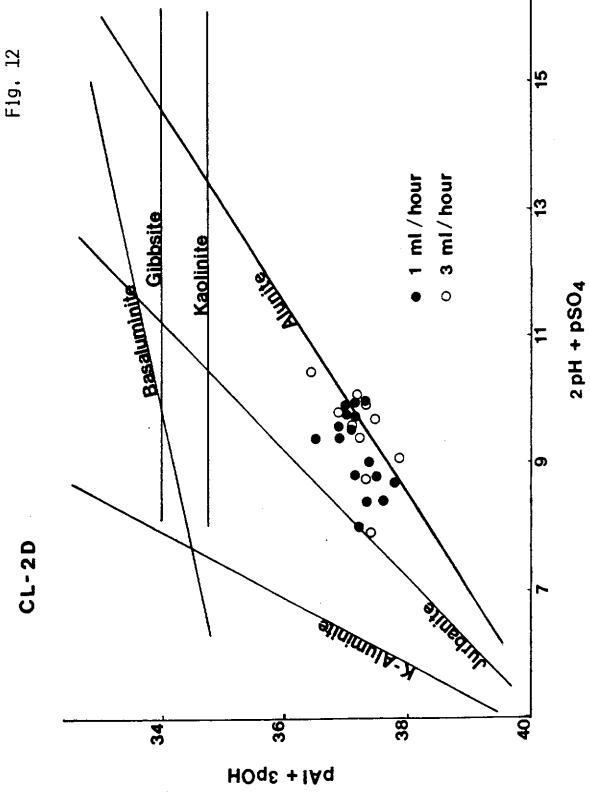


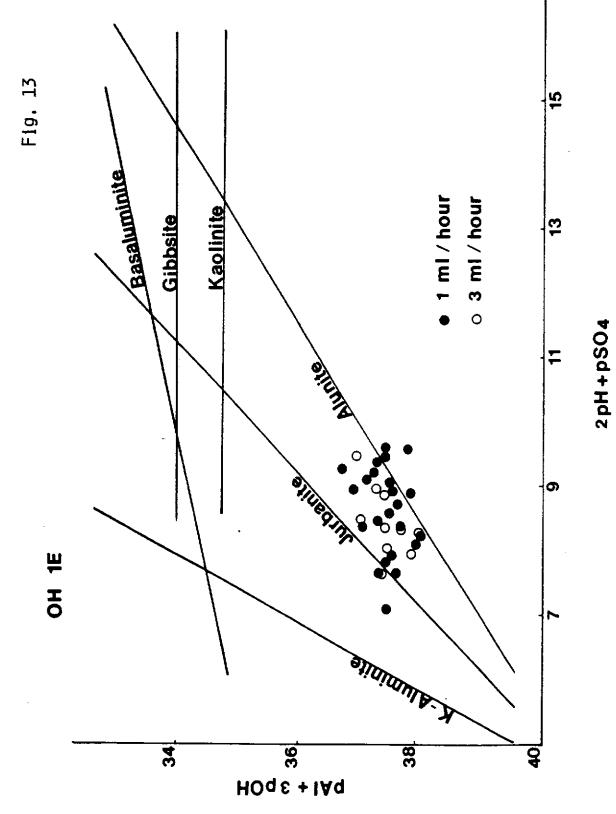


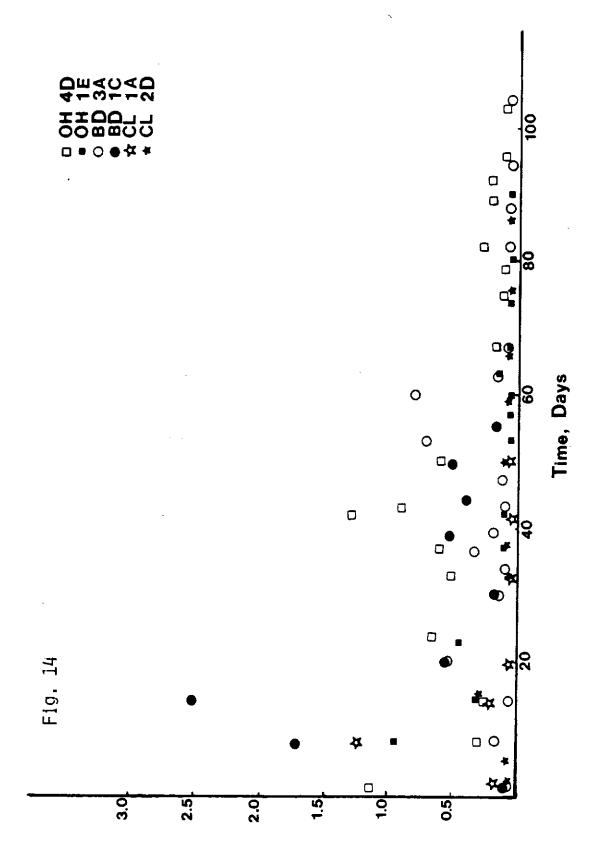












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