

Technical Report 1552 September 1993

Acid Volatile Sulfides and Simultaneously Extracted Copper, Lead, and Zinc In Sediments of Sinclair Inlet, Washington

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Robert K. Johnston

NAVAL COMMAND, CONTROL AND OCEAN SURVEILLANCE CENTER RDT&E DIVISION San Diego, California 92152-5001

K. E. EVANS, CAPT, USN Commanding Officer R. T. SHEARER Executive Director

ADMINISTRATIVE INFORMATION

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Released by	Under authority of
J. Grovhoug, Head	P. F. Seligman, Head
Marine Environmental Quality Branch	Environmental Sciences Division

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EXECUTIVE SUMMARY

INTRODUCTION

As part of an assessment of the availability and mobility of toxic metal contamination in the sediments of Sinclair Inlet, Puget Sound, Washington, acid volatile sulfide (AVS) and simultaneously extracted metal (SEM) concentrations of Cu, Pb, and Zn were measured in conjunction with deployments of a benthic flux chamber. The AVS and SEM measurements were made from sediment cores, collected at each of the benthic flux stations, to assess the geochemical availability and potential toxicity of the metal present. This report describes the procedures and methods used for sampling and analyzing AVS and SEM, presents the results obtained, and evaluates the significance of AVS and SEM concentrations measured in the sediments of Sinclair Inlet.

METHODS

Two sediment cores 10 to 30 cm in depth were collected from each of ten stations located in the Inlet. One core was sectioned into 2-cm intervals to determine the AVS profile. A composite sample was obtained from the other core by extruding and homogenizing the top 10 cm of the core.

The AVS in aliquots of homogenized sediment samples (2 to 10 grams) was released by reacting the sediments with 1 M HC1 and trapping the H₂S volatilized with a sulfide antioxidant buffer (SAOB) solution. The volume of the trap solutions was adjusted to 100 ml of 50% SAOB and 50% and 50% deaireated deionized water (DDIW). The concentrations of S^{-} were determined with a sulfide ion-specific electrode previously calbrated with standards made in 1:1 SAOB:DDIW using a sulfide stock solution that was iodometrically titrated.

The SEM concentrations of Cu, Pb, and Zn were measured in extracts obtained from the 0-10-cm composite and 0-2-cm surface AVS samples. The SEM Cu concentration was also measured in extracts obtained from the AVS core profile samples. The extracts were prepared for metal analysis by decanting 45 or 35 ml of the elutriate from the storage vials into clean, unused centrifuge tubes and centrifuging at 8,000 rpm for 20 minutes. The metals were analyzed by graphite furnace atomic absorption spectrophotometry.

RESULTS

The AVS measured in composite samples ranged from 0.8 to 98.0 μ mol/g. For nine of the ten stations, the composite AVS concentrations were 2.2 to 10.3 times higher than the SEM of Cu, Pb, and Zn, and 1.1 to 6.2 times higher than the divalent metal concentrations (Cd, Cu, Cr, Hg, Ni, Pb, Zn) reported from analysis of bulk sediment grabs collected from the same stations. One station had a composite AVS concentration lower than the composite SEM concentration resulting in a SEM to AVS ratio of 2.5. Seven of the ten stations had high AVS concentrations at the surface (> 20 μ mol/g), which increased at intermediate depths and decreased toward the bottom of the core. Simultaneously extracted Cu profiles were relatively uniform with depth and ranged from 0.1 to 11.2 μ mol/g.

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CONCLUSIONS

The analytical method used to measure AVS in sediments from Sinclair Inlet was able to provide accurate measurements of the reactive sulfide in the sediment cores. The rapid turnaround between sample collection and sample analysis (less than a day for most of the samples) assured a minimum of change in the sediments before analysis.

High concentrations of AVS measured in the sediments for Sinclair Inlet suggest that most of the divalent metals were bound up as nonavailable and nonmobile sulfides.

The core profiles from Sinclair Inlet trace the time course of sulfide geochemistry. At the surface, AVS was relatively high, due to the rapid and sharp decline of oxygen measured in the first few millimeters of sediment, the high amount of organic matter present, and a readily available source of sulphate in the bottom water of the Inlet. The AVS maximum, measured at a 4–6-cm depth in most of the cores, reflects the optimal formation of reactive monosulfides. The decrease in AVS toward the bottom of the core suggests that more sulfide was tied up in less reactive polysulfide and mineral sulfide (pyrite) forms.

There was not a clear relationship between AVS and metal flux measured with the benthic chamber, although a trend of higher flux rates was observed for the metals with lower sulfide solubility (Ni and Zn).

The results reported here are only a snapshot in time, and do not take into consideration seasonal changes of AVS. Not enough information is yet known about the AVS dynamics in sediments. However, these results show that AVS production in the inlet is very high.

High concentrations of sulfide indicate the anoxic nature of the sediments, meaning that most of the metals will reside in a reduced state. As cleanup efforts in the Inlet proceed, it will be important to monitor the sediments. The reduction of pollution stress on the Inlet may increase water quality, attracting benthic organisms which may invade, colonize, and rework the sediments. The sediment reworking will in turn cause the sediments to become more aerobic, destroying AVS and oxidizing and mobilizing metals. The impact to the environment will depend on the rate sediment reworking occurs and on other mediating factors. Future exposure scenarios should be considered as part of any long-term monitoring for Sinclair Inlet.

EX	ECUTIVE SUMMARY iii
IN	TRODUCTION
ME	ETHODS
RE	SULTS
DIS	SCUSSION
CO	NCLUSIONS
RE	FERENCES
AP	PENDIXES:
Α.	Standard Operating Procedure for AVS Determination in Sediments Using Sulfide-Specific Electrode Detection
B.	Raw Data of AVS in Sediment Samples B-1
C.	Raw Data for SEM Concentrations of Cu, Pb, and Zn C-2
D.	Determinations of Organic Matter Content by Loss on Ignition for Sediment Samples From Sinclair Inlet
FIC	GURES
1.	Location of benthic flux stations in Sinclair Inlet, Puget Sound, Washington 1
2.	Apparatus used for measuring acid volatile sulfides
3.	Sulfide electrode calibration curves obtained from July 9 (Julian date 190) to July 19 (Julian date 200), 1991
4.	Concentrations of AVS measured in the top 10 cm of cores collected from the stations in Sinclair Inlet
5.	Vertical profiles of AVS measured in cores collected from Sinclair Inlet
6.	Calibration curves obtained for copper analysis by GFAA
7.	Calibration curves obtained for lead analysis by GFAA
8.	Calibration curves obtained for zinc analysis by GFAA
9.	SEM concentrations of Cu, Pb, and Zn measured in composite (0–10 cm) samples
10.	SEM concentrations of Cu, Pb, and Zn measured in surface (0–2 cm) samples
11.	Depth profiles of SEM copper concentrations
12.	Ratio of SEM (Cu + Pb + Zn) to AVS surface (0–2 cm) and composite (0–10 cm) samples

CONTENTS

5

13.	Ratio of metal to composite (0-10 cm) AVS for SEM and bulk sediment metal concentrations measured during benthic chamber deployments (BFSD Metals, Chadwick et al., 1992) and the site investigation (SI Metals, URS Consultants, Inc., 1992)
14.	Relationship between metal contamination and AVS measured at the stations in Sinclair Inlet
15.	The fraction of organic matter (F_{om}) measured in composite $(0-10 \text{ cm})$ and surface samples $(0-2 \text{ cm})$
16.	Relationship between organic matter and AVS of sediment samples from Sinclair Inlet
17.	Relationship obtained for organic matter and log-transformed AVS concentrations measured in surface (0-2 cm) sediment samples
18.	The relationship between F_{om} measured in surface sediment samples (0-2 cm) and metal flux rates obtained from the benthic flux chamber
19.	Relationship between AVS measured in surface sediment samples (0-2 cm) and metal flux rates obtained from the benthic chamber
20.	The relationship between ratio of SEM to AVS measured in surface sediment samples (0-2 cm) and metal flux rates
21.	The relationship between the ratio of SEM to AVS measured in composite sediment samples (0–10 cm) and metal flux rates
22.	The relationship between the ratio of bulk divalent metal concentrations and AVS measured in composite (0–10 cm) sediment samples and metal flux rates 34
TA	BLES
1.	The concentrations of divalent metals (µmol/g dry weight) measured in sediment grabs collected from the ten stations selected as locations for in-situ flux measurements (Chadwick and Lieberman, 1991) 2
2.	The graphite furnace atomic absorption (GFAA) parameters and furnace programs used for the analysis of copper, lead, and zinc
3.	Quality control criteria used for SEM analysis
4.	Summary of samples collected for AVS and SEM analyses of sediments from Sinclair Inlet
5.	The station location (Sta), core depth of sample (Depth), grams of wet sediment (G_w) , the AVS concentration of the trap (μ M), the dry-to-wet ratio of sediment (R) and the μ mol/g dry weight (μ mol/g) of AVS measured for each sample analyzed from sediment cores collected in Sinclair Inlet
6.	Regression coefficients obtained for sulfide electrode calibration curves measured between July 8 (Julian Date 189) to July 22, 1991 (Julian
	Date 202) 16

7.	Average concentrations of AVS and SEM in composite and surface sediment samples	18
8.	Results of SEM analysis of field blanks (FBs) to determine possible contamination of copper, lead, and zinc	22
9.	Ratios of SEM to bulk metal concentrations (BFSD) metals	!4
10.	Results of regression analysis	0
11.	Summary of benthic flux measurements obtained for Ni, Zn, Cu, and Pb (from Chadwick et al., 1992), and composite (0–10 cm, Comp.) and surface (0–2 cm, Surf.) levels of F_{om} , AVS, bulk metals:AVS (BFSD/AVS), and SEM/AVS (SEM/AVS)	
	(SEM/AVS)	,4
12.	Values of AVS reported from other field studies	8

INTRODUCTION

As part of an assessment of the availability and mobility of toxic metals in the sediments of Sinclair Inlet, Puget sound, Washington, acid volatile sulfides (AVSs) and simultaneously extracted metal (SEM) concentrations of Cu, Pb and Zn were measured in conjunction with deployments of a benthic flux chamber (Chadwick and Lieberman, 1991). The benthic flux chamber was used to measure the in-situ metal flux between the sediments and the water column at ten stations in the Inlet (figure 1). The AVS and SEM measurements were made from sediment cores, collected at each of the benthic flux stations, to assess the geochemical availability and potential toxicity of the metal present. The stations were selected based on previous data (URS Consultants, Inc., 1990, 1992) to encompass the range of metal concentrations in the Inlet (table 1) and provide a representative analysis of the potential for the release of toxic metals from the sediments (Chadwick and Lieberman, 1991). Two replicate stations (a and b) were located in each of five areas of similar metal concentration levels. The areas were determined from the cluster analysis of data from the site inspection study (Chadwick et al., 1992). This report describes the procedures and methods used for sampling and analyzing AVS and SEM, presents the results obtained, and evaluates the significance of AVS and SEM concentrations measured in the sediments of Sinclair Inlet. The results obtained from the flux measurements are reported in Chadwick et al. (1992).



Figure 1. Location of benthic flux stations in Sinclair Inlet, Puget Sound, Washington. Two cores for AVS determination were collected from each station. Table 1. The concentration of metals (μ mol/g dry weight) measured in sediment grabs collected from the ten stations selected as locations for in-situ flux measurements (Chadwick and Lieberman, 1991).

<u>Etatian</u>									
Station	Cd	Cu	Pb	Hg	Ni	Ag	Zn	Sum	Avg
1a	0.01	0.13	0.06	0.00	0.26	0.00	0.55	1.00	
1b	0.00	0.30	0.17	0.00	0.30	0.00	0.87	1.64	1.3
2a	0.01	2.20	0.68	0.00	0.58	0.01	3.49	6.97	
2b	0.03	2.03	0.46	0.00	0.65	0.01	3.57	6.76	6.9
3a	0.03	3.32	0.66	0.01	0.67	0.02	4.09	8.79	
3b	0.04	3.15	1.27	0.00	0.70	0.01	10.59	15.76	12.3
4a	0.02	4.31	0.51	0.01	0.67	0.01	4.56	10.09	
4b	0.04	2.17	0.46	0.00	0.76	0.03	3.07	6.52	8.3
5a	0.03	5.70	1.18	0.01	0.59	0.01	7.48	15.00	
5b	0.05	26.75	2.80	0.01	0.76	0.01	18.77	49.15	32.1

(A) Data complied from measurements made during the Puget Sound Naval Shipyard Site Investigation (SI METALS) conducted fall 1989 (URS Consultants, Inc. 1992).

(B) Concentration of divalent metals measured in bulk sediments sampled during deployments of the benthic flux sampling device (BFSD METALS) July 1992 (Chadwick et al., 1992).

Station]					
Station	Cd	Cu	Pb	Hg	Ni	Zn	Sum	Avg
1a	0.00	0.08	0.05	0.00	0.49	0.40	1.02	
1b	0.00	0.60	0.18	0.00	0.75	1.89	3.43	2.2
2a	0.01	3.02	1.60	0.00	1.38	42.63	48.64	
2b	0.04	2.12	0.28	0.00	0.94	0.66	4.05	26.3
3a	0.01	5.63	1.63	0.01	1.72	5.39	14.39	
3b	0.00	2.00	0.01	0.01	1.69	1.92	5.99	10.2
4a	0.01	3.70	1.35	0.00	1.45	10.19	16.70	
4b	0.01	2.00	0.40	0.00	1.43	1.69	5.54	11.1
5a	0.03	4.04	1.84	0.01	1.12	13.97	21.03	
5b	0.01	4.00.	0.80	0.00	1.58	4.18	10.57	15.8

The importance of AVS in controlling metal toxicity in sediments has been shown in a number of recent studies (Di Toro et al., 1990; Ankley et al., 1991; Carlson et al., 1991; Di Toro et al., 1992). Concentrations of AVS in sediments are operationally defined as the concentration of solid phase sulfide compounds that can be volatilized when treated with cold acid (Di Toro et al., 1990; Boothman and Helmstetter, 1992; Allen, Fu, and Deng 1993). The sulfides obtained from AVS measurements are generally considered to be derived from metal sulfides (MeS) of which iron sulfides (FeS) are the most abundant (Goldhaber and Kaplan, 1974; Cornwell and Morse, 1987). In marine and freshwater sediments, divalent metals form very insoluble compounds when they react with sulfide (Morse et al., 1987). It has been hypothesized that the quantity of AVS represents a "reactive pool" of sulfides (Di Toro et al., 1992) which are able to bind and reduce the availability and toxicity of the metals present (Di Toro et al., 1990).

In sediments, sulfides are produced through the digenesis (breakdown) of organic matter. Under anoxic conditions, sulphate ($SO_4^=$), which is abundant in marine waters (2700 mg/L) (Di Toro et al., 1990), is used as the electron donor to oxidize organic matter:

$$2CH_2O + SO_4^= \rightarrow 2CO_2 + S^= + 2H_2O$$

Since sulfide (S⁼) is very reactive, it forms monosulfides with readily available iron (Fe), resulting in the formation of iron monosulfide solids:

$$Fe^{2+} + S^{=} \leftrightarrows FeS_{solid}$$

The geochemical processes occurring in the sediment will result in the partitioning of sulfides into three fractions: (1) AVS reactive sulfides made up of easily solubilizable iron- and other monosulfides; (2) more resistant and insoluble mineral phase sulfides such as iron pyrite: and (3) organic sulfides which are tightly associated with organic matter (Di Toro et al., 1990). The reactive pool of AVS will disassociate from the solid-phase amorphous iron and manganese sulfides to bind with toxic metals in the sediments, because the toxic metal sulfides are more insoluble (have a lower K_{sp}) than the iron sulfides. The iron sulfide in the sediments is in equilibrium with the dissolved phase; therefore, the presence of toxic metals will cause the iron sulfides to dissolve and the sulfides to bind with metals:

$$Cu^{2+} + FeS_{solid} \rightarrow CuS_{solid} + Fe^{2+}$$

Any metal sulfides formed by divalent metals (NiS, ZnS, CdS, PbS, CuS, HgS, etc.) that have lower metal sulfide solubility than FeS (or MnS) will form sulfides at the expense of iron and manganese sulfide (Di Toro et al., 1990; Cornwell and Morse, 1987; Morse, et al., 1987, Lide, 1990).

The form of the metal within the sediments will determine metal toxicity (Di Toro et al., 1990; Ankley et al., 1991; Carlson et al., 1991; Di Toro et al., 1992) as well as mobility (Goldhaber and Kaplan, 1974; Morse et al., 1987) and therefore risk to the ecosystem. A significant reduction in the toxicity of heavy metals has been observed when AVS concentrations (umol/g dry weight) are equal to or greater than the concentration of toxic metals (Di Toro et al. 1990, Ankley et al., 1991; Carlson et al., 1991). The explanation is that when the ratio of metals to AVS concentrations is less than one, the metals are not biologically available to organisms nor are they mobile, because they are bound up in the reactive pool of AVS (Di Toro et al. 1992). Under conditions where the ratio of toxic metals to AVS ([toxic metals]:[AVS]) is less than one (more AVS than metals), the potential flux of metals from the sediment into the water column

will be reduced or negligible. The actual metal flux will be a function of the form of the metal present, the concentration gradient between the sediment and water column, and the amount of bioturbation occurring in the sediment.

The analysis of AVS requires treating the sediments with a cold acid (HCl) to volatilize the reactive sulfide pool. When the sulfides are volatilized and driven off by the carrier gas (N_2) , any metals that were bound up as metal sulfides are released and remain dissolved in the extract:

 $MeS + 2HCl \rightarrow H_2S + MeCl_{aqueous}$

The concentration of the metals thus extracted is referred to as the simultaneously extracted metal (SEM), which represents the portion of metal activity associated with the reactive pool of AVS (Di Toro et al., 1992). The total extractable metal concentration, usually measured in bulk sediment analysis (see table 1), consists of the AVS-associated metals as well as other, more insoluble, mineral-bound and organic-bound forms of the metals.

METHODS

The AVS concentrations of sediments were determined by using the method described in Appendix A. Ultrapure N₂ gas, passed through an oxygen stripping solution of vanadous chloride, was used in a purge and trap system to evolve H₂S from sediment samples. Aliquots of sediment samples ranging from 2 to 10 grams of wet sediment were extracted by reacting the sediments with 1 M HCl and trapping the H₂S in a series of two traps (Trap1 and Trap2) containing a sulfide antioxidant buffer (SAOB) solution made from 2 M sodium hydroxide, 0.2 M ethylenediaminetetracetec acid (EDTA), and 0.2 M L-asorbic acid (Appendix A). The volume of the trap solutions was adjusted to 100 ml of 50% SAOB and 50% deareated deionized water (DDIW), and the concentration of S⁼ was determined with a sulfide ion electrode previously calibrated with standards made in 1:1 SAOB:DDIW using a sulfide stock solution that was iodometrically titrated. A photograph showing the components of the setup is shown in figure 2.

An inverted graduated cylinder filled with water mounted on a ring stand was used to estimate the N_2 gas flow rate through the system. It was necessary to use the graduated cylinder because flowmeters were not available. The flow rate was measured by timing the displacement of water in the cylinder caused by bubbles exiting from tubing connected to the exit port of the second trap (Trap2). When two reaction vessels were used in parallel (System1 and System2), constant and equivalent flow was maintained by slightly constricting flow through System1 with a ring stand clamp attached to the tubing exiting from Trap2. When the system was configured in this manner, it was possible to maintain a constant flow rate for each analysis. The measured flow rate ranged between 120 and 200 ml/min and was found to be dependent on the ambient air temperature. The second sulfide trap (Trap2) was measured after each analysis to assure that no sulfide was blown past Trap1. No significant concentrations were detected in Trap2 during the analysis of Sinclair Inlet samples.

Sediment cores 10 to 30 cm in depth were collected with a 5-cm-diameter coring d_vice from ten stations in Sinclair Inlet (figure 1). Cores were maintained intact on ice until they were extruded immediately before analysis. Two cores were collected from each station. One core was selected (usually the deepest core) to determine the AVS profile by extruding and homogenizing 2-cm sections sampled from 0-2, 4-6, 8-10, 20-22, and 30-32-cm depths (or until the



(a) Purge and trap system showing two parallel systems consisting of tubing, reaction vessel, and two SAOB traps.



(b) Measurement of SAOB solution with reference and sulfide electrodes.

Figure 2. Apparatus used for measuring acid volatile sulfides.

bottom of the core was reached). The remaining core for each station was used to obtain a composite sample by extruding the 0-10-cm section of the core. Each sample was thoroughly mixed with a rubber spatula and subsampled. About 2 to 6 grams of wet sediment were placed on a preweighed piece of parafilm, weighed on a Sarturius balance, and placed into the reaction vessel by washing with DDIW measured into a syringe. The volume of the reaction vessel was then adjusted to 50 ml with DDIW and placed on the stirring motor and connected to the system.

The dry-to-wet ratio (R) was determined by drying a sample of the sediment at 100° C for approximately 24 hours.

Ultrapure N_2 was purged through the system for a minimum of 10 minutes to remove all free oxygen. After the initial purge, the flow rate and stirring was stopped while 10 ml of 6 M HCl was slowly injected into the septum of the reaction vessel. After it was determined whether any gas was spontaneously formed within the reaction vessel, the gas flow and stirring were turned back on and maintained at a constant rate to trap any sulfide gas evolved by bubbling N_2 through the SAOB traps. After 30 minutes, the gas flow and stirring was stopped and the traps were removed from the system and adjusted to 100 ml volume (1:1 SAOB:DDIW) for measurement with the sulfide electrode. The material (sediment and elutriate) remaining in the reaction vessel was preserved in 80-ml centrifuge tubes for metals analysis to determine the SEM concentration of Cu, Pb and Zn.

The trap solutions were measured with the sulfide electrode for a period of 8–10 minutes. The mV response obtained from the pH meter connected to the sulfide electrodes was recorded at 2-minute intervals until a stable reading was obtained. It was observed that the mV reading would reach a peak and then diminish with time. Therefore, the average mV reading bracketing the peak was used to determine the sulfide concentration. Sulfide concentrations $[S^{=}]$ were determined by converting mV response to concentration by using a regression curve obtained from the electrode response to standards.

Standards, prepared fresh each week from the iodometrically titrated sulfide stock solution, were used to calibrate the sulfide electrode. On each analytical day a new standard curve was generated to calculate the sulfide concentrations for samples analyzed on that day. The concentration of AVS was determined by

AVS
$$= \frac{[S^{=}] \cdot V_{\text{trap}}}{G_{w} \cdot R}$$

where

AVS	=	acid volatile sulfide concentration (umol/g dry weight)
[S=]	=	sulfide concentration determined from mV response
V _{tran}	=	trap volume (0.1 L)
G_{w}	=	grams wet weight of sediment sample
R	=	dry/wet ratio

Blacks and spike recoveries were analyzed in the same manner as the sediment samples. A blank consisting of only DDIW was analyzed to determine if there was any contamination in the system. System recoveries were determined by adding a known spike concentration of the sulfide stock solution, diluted with DDIW, to the reaction vessel before the initial purge. Spikes were added to blanks and to sediments. Recoveries were determined by comparing the $[S^{=}]$ of the blank spike to a bench spike created by adding the same sulfide spike directly to the 1:1 SAOB:DDIW matrix. Recoveries from sediment spike were determined by

$$= \frac{AVS_{spiked} - SAMPLE}{SPIKE}$$

where

Recovery

AVS _{spiked}	-	the concentration of AVS in μ mol/g dry weight determined for the sample spike
		for the sample+spike
SAMPLE	=	the AVS in µmol/g dry weight determined from replicates
		of the spiked sample
SPIKE	=	the concentration of $H_2S(\mu M)$ in the spike added to the sample
		the sample

An aliquot of sediment samples obtained from North Jamestown, Narragansett Bay, RI, was analyzed using this procedure as an intercalibration with a similar AVS procedure used at the EPA Environmental Research Laboratory, Narragansett, RI (ERLN). The values of 5.58 and 5.66 μ mol/g were obtained for the two replicate measures of the aliquot. The concentration of AVS in the North Jamestown sample was independently determined to be approximately 8 μ mol/g (W. Boothman, EPA ERLN).

The SEM concentrations of Cu, Pb, and Zn were measured by graphite furnace atomic absorption (GFAA) spectroscopy in extracts obtained from the 0-10-cm composite and 0-2-cm surface AVS samples. The SEM Cu concentration was also measured in extracts obtained from the AVS core profile samples. The extracts (which consisted of 50 ml DDIW, 10 ml 1 M HCl, and sediment samples) were preserved in tightly sealed and labeled 80-ml centrifuge tubes until SEM analysis was performed in January 1992. The extracts were prepared for metal analysis by decanting 35 or 45 ml of the elutriate from the storage vials in clean, unused centrifuge tubes. The extracts were not filtered; rather the suspended sediment was removed from the elutriate by centrifugation at 8,000 rpm for 20 minutes. Serial dilutions of the extracts were obtained by combining 200 μ l of sample extract with 800 μ l deionized water (DIW) (d_1) and combining 100 μ l of d_1 with 900 μ l of DIW (d_2):

 $d_1 = 200 \ \mu l \ sample \ extract + 800 \ \mu l \ DIW$

 $d_2 = 100 \ \mu l \ d_1 + 900 \ \mu l \ DIW$

The metal analyses were performed with a Perkin Elmer AA 5000 equipped with an HGA 500 furnace and an AS 40 autosampler. A standard uncoated graphite tube was used for the Cu analysis (Lindner and Caso, in preparation) and an platform graphite tube was used for the Pb and Zn analyses (John Andrews, Computer Sciences Corporation, personal communication). The parameters and programs used for the metal analyses are shown in table 2. No matrix modifiers were used.

Table 2. The graphite furnace atomic absorption (GFAA) parameters and furnace programs used for the analysis of copper, lead, and zinc. The AS 40 autosampler was set at a sample volume of 10 μ l for all analyses.

(A) Copper parameters and program (Lindner and Caso, in preparation).

AA 5000 Par	ameters
Sample Time:	10.0 s
Lamp:	Hollow Cathode Lamp, 15-mA, 70-W output
Energy:	68 W
Wavelength	324.8 nm
Slit Length:	0.7 nm

HGA 500 Furnace Program

Step	Description	Temp, °C	Ramp Time, s	Hold Time, s	Gus Flow, ml/min	Read Time, s	Recorder on Time, s	Baseline Set Time, s
1	Dry	110	10	30	100			
2	Char	1150	10	45	100			
3	Cool Down	20	1	20	0			
4	Atomization	2100	0	10	0	0	-3	-11
5	Clean Up	2600	1	3	100		-	

(B) Lead parameters and program (John Andrews, Computer Sciences Corporation, personal communication).

AA 5000 Parameters

Sample Time:	5.0 s
Lamp:	Hollow Cathode Lamp, 10-mA
Energy:	68 W
Wavelength	283.3 nm
Slit Length:	0.7 nm

HGA 500 Furnace Program

Step	Description	Temp, °C	Ramp Time, s	Hold Time, s	Gus Flow, ml/min	Read Time, s	Recorder on Time, s	Baseline Set Time, s
1	Dry	120	10	50	150		*********	
2	Char	850	10	30	150			
3	Cool Down	20	1	15	150			
4	Atomization	1800	0	10	0	0	-3	-11
5	Clean Up	2600	1	5	150			

Table 2. Continued.

AA	5000 Parameter	rs							
Sample Time: Lamp: Energy: Wavelength Slit Length:		5.0 s Hollow Cathode Lamp, 8-mA 60 W 213.9 nm 0.7 nm							
HGA	HGA 500 Furnace Program								
Step	Description	Temp, °C	Ramp Time, s	Hold Time, s	Gus Flow, ml/min	Read Time, s	Recorder on Time, s	Baseline Set Time, s	
1	Dry	120	10	50	150				
2	Char	400	10	30	150				
3	Cool Down	20	1	15	150				
4	Atomization	1000	0	10	30 0	0	-3	-11	
5	Clean Up	2600	1	5	150				

(C) Zinc parameters and program (John Andrews, Computer Sciences Corporation, personal communication).

The sensitivity of the GFAA was determined by using elemental standards made in DIW. A calibration curve was generated with matrix standard additions. Quality assurance QA samples included in each batch consisted of DIW blanks, a sample duplicate, an internal reference sample (e.g., a matrix standard addition whose value was determined from repeated measurements), and a calibration check sample. The QA samples were used to check for laboratory contamination, precision of the method, and instrument drift. In addition, field blanks (FBs) were included as samples in the batch. The FBs were used to control for contamination during the AVS measurement. The control criteria for the QA samples are listed in table 3.

Table 3. Quality contro	l criteria used	for SEM	l analysis.
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Quality Assurance Sample	Criteria
BLANK	No more than 2 times the value of a blank, determined by repeated measurements
DUPLICATE	No more than 10% difference between duplicate samples
INTERNAL REFERENCE	No more than 15% difference of the value of the reference material, determined by repeated measurements
CALIBRATION CHECK	No more than 15% difference of the value of the calibration material, determined by repeated measurements
FIELD BLANK	Used to identify possible contamination problems

The characteristic sensitivity (CS), which relates the instrument response to sensitivity per absorbance unit (ABS), was determined by the following relationship:

Slope = M (ABS • s)/ppb

$$\frac{1}{Slope} = \frac{ppb}{M ABS • s} = \frac{(\mu gl^{-1}) • (10^{-6} \ l\mu l^{-1}) • (SV\mu l) • (1 \ pg \ 10^{6} \mu^{-1})}{M ABS • s} • \frac{(0.0044 \ ABS • s)}{(0.0044 \ ABS • s)}$$

$$= \frac{(SV \ pg) • (0.0044)}{(M) (0.0044 \ ABS • s)}$$

$$= \frac{(SV) • (0.0044)}{M} • \frac{pg}{0.0044 \ ABS • s}$$

where CS, in pg/(0.0044 ABS \bullet s), is defined as

CS = SV(0.0044)/M

and

Slope	= M	×	change in absorbance per unit increase in concentration obtained
			from the regression of DIW standards

SV = sample volume used in the GI	AA	
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0.0044 ABS • s = measurement threshold of the AA 5000

The concentration ($C_{extract}$) of the sample extracts was determined by

Cextract	5	<i>d</i> ₁ /100 μl				
<i>d</i> ₁	z	$(d_2 \bullet 1000 \ \mu)/200 \ \mu)$				
<i>d</i> ₂	z	(CONC • 1000 μl)/100 μl				
CONC	=	(ABS-BLANK)/M _{slope}				
where						
d_1, d_2	=	serial dilutions 1 and 2, respectively				
CONC	=	concentration measured by GFAA				
ABS	=	absorbance units measured on the AA 5000				
M _{slope}	=	slope of the regression curve computed from matrix standard additions				
BLANK	=	value of a blank determined from repeated measurements				
The sediment concentration ($C_{sed} \mu mol/g$) was determined by						

$$C_{\text{sed}} = \frac{(C_{extract} \bullet V_{ext})}{G_* \bullet R \bullet AW}$$

where

 $V_{ext} = total volume of extract$ $= C_{vol} + (W_{sto} - G_w) \cdot (10 \text{ ml}/10.05 \text{ g})$

and		
$C_{\rm vol}$	=	volume of material centrifuged
W _{sto}	=	weight of material remaining in storage vial, determined by tarring the weight of the empty storage vial + label and then measuring the weight of the material (sediment + extract) remaining in the storage vial
G_w	=	amount of sediment extracted (measured weight of sample used to determine AVS)
AW	=	the atomic weight ($\mu g/\mu mol$) of the metal being analyzed
R	=	dry/wet ratio of sediment sample.

The surface (0-2 cm) and composite (0-10) AVS concentrations were compared to the SEM concentrations of Cu, Pb, and Zn measured from the same samples. The composite AVS was also compared to the bulk metal concentrations (BFSD METALs and SI METALs, table 1) to evaluate the availability and potential toxicity from metal contamination of the sediments of Sinclair Inlet. The relationships between metal fluxes and the concentrations of AVS and SEM were evaluated by regression analysis.

The fraction of organic matter in the sediment samples was determined by loss on ignition. Dried sediment samples were homogenized (5 to 12 grams weight) and combusted in a muffle furnace for 6 hours at 450°C. The combusted samples were desiccated for 24 hours and reweighed periodically until a constant weight was obtained. The fraction of organic matter present in the sample was determined as

Fom	=	(Postburn – Dishwt)/(Preburn – Dishwt)
where		
Fom	=	fraction of organic matter
Dishwt	=	weight of dish used for combustion
Preburn	=	weight of sample before ignition
Postburn	=	weight of sample after ignition

The fraction of organic matter was compared to AVS concentrations and metal flux rates to determine the extent of correspondence.

11

RESULTS

The AVS samples were collected and analyzed from each station between July 8 and July 22, 1991 (see Appendix B for raw AVS data), and the SEM analyses were performed January 24-30 1992 (see Appendix C for raw SEM data). A surface composite sample (0-10 cm) and three to five profile samples were analyzed for each station, and a grab sample from station 5a was also analyzed (table 4). Replicates, blanks, and spiked samples were also analyzed periodically throughout the AVS and SEM analyses to provide quality assurance for the procedures. Dry to wet weight ratios showed that the water content of the sediment samples was variable, and it ranged from about 20% (station 1a) to 83% (station 3a) (table 5).

The electrode calibration curve showed very good repeatability throughout the study (figure 3, Appendix A). Titration of the sulfide primary standard stock solution on July 8 (189) and July 16 (197), 1991, yielded concentrations of 4843 and 4853 μ M, respectively. There were only slight variations in the daily calibration curves used to compute sample AVS concentrations (table 6). The AVS data obtained from all the samples and replicates, spiked samples, and blanks are presented in table 5. Blank samples, consisting of only DDIW (sample numbers 1 and 56, Appendix B), showed only trace amounts of AVS. Spiked blank recoveries were all within $\pm 5\%$ of the spiked concentration (table 5). Recoveries for spiked sediment samples were more variable and ranged between a low of 37.9% and a high of 95.2% (table 5). Very good repeatability was obtained from duplicate samples (table 5).

Station	Composite	Profile	Grab
1a	1	3	
1b	1	4	
2a	1	4	
2b	1	5	
3a	1	5	
3b	1	4	
4a	1	4	
4b	1	4	
5a	1	4	1
5b	1	4	
Subtotal	10	41	1
Total Samples	52		

Table 4. Summary of samples collected for AVS and SEM analyses of sediments from Sinclair Inlet.

Table 5. The station location (Sta), core depth of sample (Depth), grams of wet sediment (G_w) , the AVS concentration of the trap (μ M), the dry to wet ratio of sediment (R) and the μ mol/g dry weight (μ mol/g) of AVS measured for each sample analyzed from sediment cores collected in Sinclair Inlet. The average AVS concentrations (Avg) for duplicate samples, AVS concentrations of spikes for blanks and spiked samples (Spike), the percent recovery (RECV), and the calculated AVS concentration (CALC) are also tabulated..

									AVS
		Gw,			AVS	AVS	Spike,		CALC,
Sta	Depth	g	μM	R	µmol/g	Avg, µmol/g	g μM	RECV, %	µmol/g
1a	0–2a	7.18	5.25	0.7123	0.10				
1a	0–2b	7.88	7.39	0.7123	0.13	0.12			
1a	4-6	10.06	34.17	0.7829	0.43				
1a	4-6	10.06	34.17	0.7829	0.43				
1a	8–10	6.02	111.38	0.6102	3.03				
1a	0-10a	13.96	84.40	0.7688	0.79				
1a	0–10b	7.26	54.67	0.7688	0.98	0.88			
1a blk	sys1		495.49				511.00	97.0	
1a blk	sys2		499.32				511.00	97.7	
1 հ	0.2	607	145 41	0 6276	2 24	2.05			
10 15	0-2a	0.02	145.41	0.0370	3.34 2.75	5.05			
10 15	0-20	4.00	03.U3	0.03/0	2.75				
10 15	4-0a 4 65	4.40	138.84	0.7005	4.59	A 66			
10	4-00 8 10a	0.10	202.70	0.7000	4.94	4.00			
10 15	0-10a 8 10b	5.00	209.90	0.5998	11.30	10.00			
10 15	20 22	4.00	240.74	0.3990	0.03	10.09			
10	20-22	5.95	1/0./3	0.7300	5.95 10.46				
10 15 blo	0-10a	4.70	293.43	0.3693	10.40		1726 00	102.2	
10 014	IIKI		1203.00				1230.90	102.5	
2a	02a	6.81	1546.51	0.3209	70.77				
2a*	02b	2.98	1552.48	0.3209	162.35		1190.25	73.6	37.88
2a	4-6a	2.04	1130.69	0.3551	156.09				
2a	46b	2.43	1282.24	0.3551	148.60	152.34			
2a	8-10	1.99	976.78	0.3664	133.96				
2a	18-20	3.73	1993.98	0.3838	139.29				
2a	0–10a	3.48	967.32	0.3779	73.56				
2a	0–10b	2.90	756.97	0.3779	69.07	71.31			
2a* 0-	-10c	0.86	1130.00	0.3779	347.70		1027.16	87.4	31.64
2a bla	nk1		0.47						
2a bla	nk2		1104.72				1139.11	97.0	

* Spiked sediment sample

Table 5. Continued.

		Gua			AVS	AVS Avg.	AVS Spike.		AVS CALC.
Sta	Depth	g	μΜ	R	µmol/g	μmol/g	μM		µmol/g
2b#	0–2a	2.62	155.64	0.2947	20.16				
2b#	0–2b	2.46	699.04	0.2947	96.43				
2ь	0–2c	8.17	1259.02	0.2947	52.30	56.30			
2b#	4-6 a	1.33	385.08	0.3480	83.19				
2b#	46b	1.20	437.64	0.3480	104.79				
2b	4-6c	4.15	838.98	0.3480	58.09	82.02			
2b	810a	5.44	1860.44	0.3786	90.34				
2b	20–22	4.31	2083.08	0.4112	117.54				
2b	0-10a	4.15	335.31	0.3679	21.96				
2ь	0–10b	5.83	477.99	0.3679	22.29	22.13			
2b blk	sys1		467.06				469.47	99.5	
2b blk	tsys2		482.92				469.47	102.9	
3a	0–2a	2.84	383.92	0.1957	69.06				
3a	0–2b	5.96	811.60	0.1957	69.57	69.32			
3a	4-6a	4.01	986.32	0.2336	105.29				
3a*	46b	3.75	1751.89	0.2336	199.99		128.85	73.5%	71.12
3a	8–10a	4.24	596.59	0.2574	54.66				
3a	8–10b	2.16	310.80	0.2574	55.90	55.28			
3a	20-22	4.59	3.10	0.1727	0.39				
3a	30–32	5.25	4.12	0.1727	0.45				
3a	0-9	5.82	67.98	0.3377	3.46				
3a bla	nk1	1	20.92				119.33	101.3%	
3a bla	nk2		1155.66				1183.11	97.7%	
3Ъ	0–2a	2.75	60.97	0.3639	6.09				
3b	0–2b	2.85	65.13	0.3639	6.28	6.19			
3b	4-6a	2.27	141.31	0.5123	12.15				
3b	46b	2.48	164.73	0.5123	12.97	12.56			
3Ъ	8–10	3.81	34.08	0.5225	1.71				
3Ъ	2022	3.94	270.05	0.6860	9.99				
3b	0-10a	2.61	67.06	0.5801	4.43				
3b	0–10b	5.18	125.13	0.5801	4.16	4.30			
3b blk	sys1		396.20				449.80	88.1	
3b blk	sys2		378.79				449.80	84.2	

* Spiked sediment sample # Problem with balance; sample weight may be inaccurate

Table 5. Continued.

Sta	Depth	Gw, g	μΜ	R	AVS µmol/g	AVS Avg, µmol/g	AVS Spike, µM	RECV,	AVS CALC, % μmol/g
4a	0-2a	5.40	533.86	0.2402	41.16				
4a	02b	4.14	369.04	0.2402	37.11	39.13			
4a	4-6 a	3.49	902.35	0.3071	84.19				
4a	46b	4.14	1094.41	0.3071	86.08	85.14			
4a*	46c	2.60	1668.91	0.3071	209.02		1097.03	90.2	71.62
4a	8–10a	5.63	699.44	0.3633	34.20				
4a	2022	6.45	339.00	0.4912	10.70				
4a	0-10	10.90	954.90	0.4349	20.14				
4a bl	ank1		1095.81				1158.14	94.6	
4a bl	ank2		1097.03				1111.43	98.7	
4b	0-2	2.29	125.46	0.2307	23.75				
4b	02b	1.51	80.26	0.2307	23.04	23.39			
4b	46a	1.21	127.98	0.2828	37.40				
4b	46b	2.32	287.36	0.2828	43.80	40.60			
4b	8-10	3.45	329.27	0.3344	28.54				
4b	20-22	3.80	228.63	0.4145	14.52				
4b	0–10a	5.47	693.91	0.3044	41.67				
4b	0–10b	4.94	686.61	0.3044	45.66	43.67			
4b bll	ksys1		449.80				449.80	100.0	
4b bli	ksys2		448.62				449.80	9 9.7	
5a	0-2a	6.97	1698.81	0.3686	66.12				
5a	0–2ь	7.87	1646.83	0.3686	56.77	61.44			
5a	4-6a	5.96	2111.63	0.3899	90.86				
5a*	4-6b	4.47	1992.11	0.3899	114.29		968.00	95.2	58.75
5a	8-10a	6.33	7234.24	0.4072	280.66				
5a	8–10b	8.23	8582.64	0.4072	256.10	268.38			
5a	0–10a	7.27	3023.78	0.4245	97.9 8				
5a bla	nk1		16.96				19.35	87.6	
5a bla	nk2		177.14				181.32		97.7
5a (gr	ab)a	5.84	1395.31	0.3879	61.60				
5a (gr	ab)b	3.30	966.47	0.3879	75.51	68.55			
5a*(g	rab)c	2.09	763.38	0.3879	94.17		242.00	85.8	64.32
5a*(g	rab)d	3.29	966.47	0.3879	75.74		242.00	37.9	56.77
5a blk	sys1		501.21				510.44	98.2	
5a blk	sys2		506.73				510.44	99.3	

* Spiked sediment sample

Table 5. Continued.

Sta	Depth	Gw, g	μΜ	R	AVS µmol/g	AVS Avg, µmol/g	AVS Spike, µM	AVS CALC, RECV, % µmol/g
5b	0–2a	1.15	171.10	0.2821	52.74			
5b	0–2b	2.87	535.70	0.2821	66.17	59.45		
5b	4-6a	2.75	1788.01	0.4032	161.26			
5b	4-6b	2.54	1733.94	0.4032	169.31	165.28		
5b	8-10	2.13	1862.74	0.4455	196.30			
5b	20-22	3.17	1818.93	0.5095	112.62			
5b	0–10a	3.32	857.33	0.3958	65.24			
5b	0-10b	3.64	991.49	0.3958	68.82	67.03		
5b bll	ksys1		367.75				360.29	102.1
5b bll	ksys2		366.80				360.29	101.8

Table 6. Regression coefficients obtained for sulfide electrode calibration curves measured between July 8 (Julian Date 189) to July 22, 1991 (Julian Date 202). New standards were prepared on July 8 and July 17, 1991.

Day	X Coefficient	Constant	r ²	
New Standards Pre	epared			
July 8 (189)	-29.6	-723.1	0.9999	
July 10 (191)	-29.4	-722.8	0.9999	
July 11 (192)	-29.8	-721.0	0.9997	
July 12 (193)	-29.9	-720.5	0.9998	
July 13 (194)	-30.1	-719.0	0.9999	
New Standards Pre	pared			
July 17 (198)	-29.7	-722.7	0.9997	
July 18 (199)	-30.0	-722.0	0.9998	
July 19 (200)	-29.9	-721.8	0.9999	
July 22 (202)	-31.6	-718.7	0.9997	



Figure 3. Sulfide electrode calibration curves obtained from July 9 (Julian date 190) to July 19 (Julian date 200), 1991. New standards were prepared on July 8 and July 16, 1991.

The average AVS concentrations measured in the composite samples ranged from 0.88 (station 1a) to 97.98 μ mol/g (station 5a) (table 7a). The highest composite sulfide measurements were obtained from stations 5a, 2a, and 5b, while the lowest composites measurements were made from samples for stations 1a, 3a, and 3b (figure 4). The highest mean concentrations from the three profile AVS samples, collected from the top 10 cm of the profile core, were obtained from stations 5a, 5b, and 2a, while the lowest were measured for stations 1a. 1b, and 3b. The composite AVS concentrations were generally lower than the mean core profile samples. Station 3a had the largest difference between the mean profile and composite samples (table 5, figure 4). The average surface (0–2 cm) AVS concentrations ranged from 0.12 (station 1a) to 70.77 μ mol/g (station 2a) (table 7b).

The AVS core profiles showed that the highest concentrations (> 160 μ mol/g) were measured for the 8–10-cm section of cores 5a and 5b and the lowest AVS concentrations were measured in cores 1a, 1b, and 3b (figure 5). In general, most cores had high AVS concentrations at the surface (> 20 μ mol/g), increased AVS at intermediate depth (4–6 cm), and decreased AVS toward the bottom of the core.

The calibration curves used to measure SEM concentrations of Cu, Pb, and Zn are shown in figures 6, 7, and 8, respectively. The DIW characteristic sensitivity (CS) for the copper analysis (15.92 pg/0.0044 ABS•s) was very close to the optimal CS (16.00 pg/0.0044 ABS•s) recommended for the instrument (Perkin Elmer AA 5000 operation manual). The sample matrix

slightly reduced the instrument's sensitivity, but a linear calibration curve was obtained for the range of sample concentration analyzed (figure 6). Although there was a much-reduced sensitivity for GFAA analysis of Pb, good linearity was achieved (figure 7). The GFAA analysis of Zn was more problematic. Although the sensitivity was fairly close to the optimal instrument sensitivity, it was very difficult to obtain a linear response for the sample matrix. For this reason, the DIW standard curve, which was linear from 0–300 ppb, was used as the calibration curve for Zn (figure 8). This was done because most of the sample extract concentrations were below 250 ppb (table B-2 (C)). The lack of sensitivity, matrix interference, and possible Zn contamination problems could mean that there is about a 36% error in the Zn concentrations presented here.

The quality assurance criteria were achieved for each oatch for the blank, duplicate, and internal reference samples. Two of the calibration check samples exceeded the QA criteria (copper batches RUN3 and RUN4, Appendix B). An analysis of field blanks (table 8) showed an indication of Zn contamination in samples analyzed for AVS on July 11, 1993 (Julian Date 192). However, the contamination appeared to be isolated and was not present throughout the analysis. Only trace levels of Cu and Pb were detected in the field blanks.

Table 7. Average concentrations of AVS and SEM in composite and surface sediment samples.

	AVS		SEM (µmol/g)				
Station	µmol/g	Cu	Pb	Zn	Sum SEM	SEM/AVS	
1a	0.88	0.13	0.08	0.44	0.65	0.74	
1b	10.46	0.43	0.24	1.48	2.15	0.21	
2a	71.31	2.97	0.90	6.98	10.85	0.15	
2b	22.13	3.02	1.26	5.68	9.96	0.45	
3a	3.46	3.35	0.83	4.47	8.65	2.50	
3b	4.30	1.16	0.29	2.32	3.77	0.88	
4a	20.14	2.00	0.74	2.53	5.27	0.26	
4b	43.67	1.64	0.50	3.13	5.27	0.12	
5a	97.98	2.67	1.36	5.50	9.53	0.10	
5b	67.03	6.99	1.65	8.08	16.72	0.25	

(A) Composite 0–10 cm depth.

Table 7. Continued.

	AVS		SEM (µm	ol/g)		SEM/AVS
Station	µmol/g	Cu	Pb	Zn	Sum SEM	
1a 1b	0.12 3.05	0.10 0.29	0.08 0.40	0.50 1.17	0.68 1.86	5.67 0.61
2a 2b	70.77 56.30	3.10 3.32	0.72 0.64	4.26 3.32	8.08 7.28	0.11 0.13
3a 3b	69.32 6.19	5.66	1.19	7.66	14.51	0.21
4 a	39.13	4.17	1.01	6.23	11.41	0.29
4b	23.39	1.58	0.43	4.26	6.27	0.27
5a 5b	61.44 59.45	2.87 5.83	0.93 1.02	6.07 9.54	9.87 16.39	0.16 0.28

(B) Surface 0-2 cm depth.



Figure 4. Concentrations of AVS measured in the top 10 cm of cores collected from the stations in Sinclair Inle. Data are presented for the 0-10-cm composite samples and the mean of the 0-2, 4-6, and 8-10-cm profile samples.



Figure 5. Vertical profiles of AVS measured in cores collected from Sinclair Inlet.



Figure 6. Calibration curves obtained for copper analysis by GFAA.



Figure 7. Calibration curves obtained for lead analysis by GFAA.



Figure 8. Calibration curves obtained for zinc analysis by GFAA.

Table 8. Results of SEM analysis of field blanks (FBs) to determine possible contamination of copper, lead, and zinc. The sample identification number (ID#), the Julian date and time (Date) the sample was taken, the absorbance reading for the AA 5000 (ABS), and the extract concentration (CONC) in μ g/L are given for each FB sample analyzed.

ID#	Date	ABS	CONC, µg/L	
7	190 2300	0.026	0.43	
26	192 1210	0.002	0.04	
16	192 2000	0.027	0.53	
10	193 1800	0.007	0.06	
42	194 1100	0.003	0.06	
24	198 0015	0.005	0.02	
34	200 1700A	0.003	0.06	
18	200 1700B	0.001	0.02	

(A) Copper FB (ABS 0.004 is indistinguishable from a DIW blank)

Table 8. Continued.

ID#	Date	ABS	CONC, μg/L	
26	192 1210	0.001	0.04	
16	192 2000	0.003	0.13	
10	193 1800	0.001	0.04	
24	198 0015	0.000	0.00	
18	200 1700B	0.000	0.00	

(B) Lead FB (ABS 0.001 is indistinguishable from a DIW blank)

(C) Zinc FB (ABS 0.059 is indistinguishable from a DIW blank)

 ID#	Date	ABS	CONC, μg/L	
26 16	192 1210 192 2000	0.435 0.177	6.83* 2.14*	
10 24 18	193 1800 198 0015 200 1700B	0.085 0.085 0.086	0.46 0.46 0.48	

* Indicates possible contamination

Average SEM concentrations in the composite samples ranged from 0.13 to 6.99 μ mol/g for Cu, 0.08 to 1.65 μ mol/g for Pb, and 0.44 to 8.08 μ mol/g for Zn (table 7). The levels of composite SEM concentrations were in the order of 5b > 2a > 2b > 5a > 3a (figure 9). Only station 3a had higher SEM than AVS (figure 9, table 7). The concentrations of SEM were higher than those of bulk sediment metals at stations 1a, 2b, and 5b for Cu, Pb, and Zn, at stations 4b for Pb and Zn, at station 1b for Pb, and at station 3b for Zn (table 9).

The concentrations of SEM measured in composite (0-10 cm) and surface (0-2 cm) samples were predominantly Zn and Cn (figures 9 and 10). Surface SEM concentrations were higher than composite SEM concentrations at stations 1a, 3a, 4a, 4b, and 5a (table 7). Except for some variations detected at stations 5a and 5b, the SEM Cu concentrations determined for the core profiles showed that copper was uniformly distributed to a depth of about 20 cm (figure 11).

The ratio between SEM concentrations of Cu, Pb, and Zn and the AVS showed that AVS was in excess of metals at all the stations, except for the surface sample at station 1a, and the composite sample from station 3a (figure 12). When the bulk metal concentrations were compared to the average AVS from composite samples, only stations 1a, 3a, and 3b were found to have higher metal than AVS concentrations (figure 13).



Figure 9. SEM concentrations of Cu, Pb, and Zn measured in composite (0-10 cm) samples. Levels of AVS for stations with AVS less than 18 µmol/g are also indicated.

Station	Cu	Pb	Zn	
1a	1.62	1.60	1.10	
1b	0.71	1.33	0.73	
2a	0.98	0.56	0.16	
2b	1.42	4.50	8.61	
3a	0.59	0.51	0.83	
3b	0.58	0.78	1.21	
4 a	0.54	0.55	0.07	
4b	0.82	1.25	1.85	
5a	0.66	0.74	0.39	
5b	1.73	2.06	1.93	

Table 9. Ratios of SEM to bulk metal concentrations (BFSD metals). The ratio of SEM measured in 0–10-cm composite samples to bulk metal concentrations measured at the same stations ([SEM]/[BFSD metal]) are calculated for Cu, Pb, and Zn.



Figure 10. SEM concentrations of Cu, Pb, and Zn measured in surface (0-2 cm) samples. Level of AVS for those stations with AVS less than 18 μ mol/g are indicated.



Figure 11. Depth profiles of SEM copper concentrations.



Figure 12. Ratio of SEM (Cu + Pb + Zn) to AVS for surface (0-2 cm) and composite (0-10 cm) samples.



Figure 13. Ratio of metal to composite (0-10 cm) AVS for SEM and bulk sediment metal concentrations measured during benthic chamber deployments (BFSD Metals, Chadwick et al., 1992) and the site investigation (SI Metals, URS Consultants, Inc., 1992).

The relationship between AVS and metal contamination in the inlet is shown by figure 14. The majority of stations, especially those with metal contamination levels greater than 10 μ mol/g, had AVS concentrations in excess of metal concentrations. Stations 1a, 3a, and 3b had metal concentrations in excess of AVS concentrations. Of these stations, only station 3a had SEM concentrations high enough to place it in the range of increasing toxicity (METAL/AVS) > 1.5. Borderline stations such as 1a and 1b, which were near the range of metals in excess, had relatively low metal concentrations.



Figure 14. Relationship between metal contamination and AVS measured at the stations in Sinclair Inlet. The distribution is shown for the average AVS measured in composites samples versus the SEM and bulk sediment metal concentrations. The ratio [metal]:[AVS] = 1.5 is shown to denote the range of increasing toxicity.

The fraction of organic matter determined for the sediment samples ranged from 0.01 (station 1a) to 0.17 (station 2a) (see Appendix D for determinations of organic matter content). Organic matter levels greater than 10% were measured in surface samples from stations 2a, 2b, 4a, 4b, 5a, and 5b (figure 15). The regression between the fraction of organic matter (F_{om}) and AVS was significant (probability p < 0.05) for all data (table 10, figure 16). Additionally, AVS concentrations were best predicted using the transformed (logarithm base 10) surface F_{om} data (table 10, figure 17).


Figure 15. The fraction of organic matter (F_{om}) measured in composite (0-10 cm) and surface samples (0-2 cm).

The F_{om} , AVS, and SEM/AVS data were not capable of predicting the flux rates obtained from the benthic chamber (table 10B). The flux rates of Ni, Cu, Pb, and Zn seemed to be independent of F_{om} (figure 18), AVS (figure 19), SEM/AVS (figures 20 and 21), and BFSD/AVS (figure 22). The regressions between SEM Cu concentrations and Cu flux rates were significant ($p \le 0.05$) for both the surface and composite data (table 10). A summary of the divalent metal flux rates (Chadwick et al., 1992), organic matter content, AVS concentrations, and ratios of bulk metals and SEM to AVS measured in the Inlet is presented in table 11.

Table 10. Results of regression analysis.

Independent Variable	Dependent Variable	р	Equation	r ²
All Data: F _{om} F _{om}	AVS Log (AVS)	0.0007 <0.0001	$\begin{array}{llllllllllllllllllllllllllllllllllll$	0.23 0.51
Surface (0–2cm): F _{om} F _{om}	AVS Log (AVS)	0.03 0.0004	AVS = $523.4 (F_{om}) - 13.9$ Log (AVS) = $21.2 (F_{om}) - 0.9$	0.53 0.85

(A) Regression between F_{om} and AVS

(B) Regression between $F_{\text{om}},$ AVS, SEM/AVS, $\text{Cu}_{\text{SEM}},$ Zn_{SEM} and flux rates

Independent Variable	Dependent Variable	р	Equation	r ²
Surface (0-2 cm)				
Fom	Sum Flux	0.47		0.07
AVS	Sum Flux	0.20		0.22
SEM/AVS	Sum Flux	0.11		0.35
SEM/AVS	Zn Flux	0.16		0.30
SEM/AVS	Cu Flux	0.34		0.15
Cu _{SEM}	Cu Flux	0.05	Cu Flux = $1.1 (Cu_{SEM}) - 2.2$	0.44
ZnSEM	Zn Flux	0.25		0.16
Composite (0-10 c	cm)			
Fom	Sum Flux	0.24		0.16
AVS	Sum Flux	0.08		0.32
SEM/AVS	Sum Flux	0.97		<.01
BFSD/AVS	Sum Flux	0.99		<.01
Cusem	Cu Flux	0.03	Cu Flux = 1.4 (Cu _{SEM}) - 3.0	0.47
Zn _{SEM}	Zn Flux	0.25		0.16



Figure 16. Relationship between organic matter and AVS of sediment samples from Sinclair Inlet. AVS = $628.7 (F_{om}) - 6.9$, $r^2 = 0.23$.



Figure 17. Relationship obtained for organic matter and log-transformed AVS concentrations measured in surface (0–2 cm) sediment samples. Log (AVS) = 21.2 (F_{om}) – 0.9, r² = 0.85.



Figure 18. The relationship between F_{om} measured in surface sediment samples (0-2 cm) and metal flux rates obtained from the benthic flux chamber.



Figure 19. Relationship between AVS measured in surface sediment samples (0-2 cm) and metal flux rates obtained from the benthic chamber.



Figure 20. The relationship between ratio of SEM to AVS measured in surface sediment samples (0-2 cm) and metal flux rates.



Figure 21. The relationship between the ratio of SEM to AVS measured in composite sediment samples (0-10 cm) and metal flux rates.



Figure 22. The relationship between the ratio of bulk divalent metal concentrations and AVS measured in composite (0-10 cm) sediment samples and metal flux rates.

Table 11. Summary of benthic flux measurements obtained for Ni, Zn, Cu, and Pb (from Chadwick et al., 1992), and composite (0-10 cm, Comp.) and surface (0-2 cm, Surf.) levels of F_{om} , AVS, bulk metals to AVS (BFSD/AVS), and SEM to AVS (SEM/AVS).

	Fl	uxes µr	nol m ⁻²	d-1	F _c)m	AVS (µmol/g)		BFSD/AVS	SEM/AVS	
Station	Ni	Zn	Cu	Pb	Comp.	Surf.	Comp.	Surf.	Comp.	Comp.	Surf.
1a	-0.4	-3.4	-2.8	-0.6	0.01	0.02	0.9	0.1	1.2	0.7	5.7
1b	0.4	11.7	2.2		0.06	0.04	10.5	3.1	0.3	0.2	0.6
2a	0.1	4.7*	-1.8	-0.1	0.09	0.12	71.3	70.8	0.7	0.2	0.1
2b	5.2*	-0.8	-1.9		0.70	0.13	22.1	56.3	0.2	0.5	0.1
3a 3b	-1.3 4.1*	11.3* 2.0	4.7* -6.2	-0.1	0.10 0.04	0.09	3.5 4.3	69.3 6.2	4.2 1.4	2.5 0.9	0.2
4a	-0.5	3.7	1.1	0.0	0.15	0.13	20.1	39.1	0.8	0.3	0.3
4b	5.2*	1.9	-0.6		0.16	0.13	43.7	23.4	0.1	0.1	0.3
5a	9.6*	12.8*	1.4	-0.1	0.11	0.11	98.0	61.4	0.2	0.1	0.1
5b	1.0	11.2*	7.7		0.09	0.11	67.0	59.5	0.2	0.3	0.3
Average for Inlet	2.3	5.5	0.4	-0.2	0.09	0.10	34.1	38.9	0.9	0.6	0.9

* Indicates statistically significant release (Chadwick et al., 1992).

DISCUSSION

The benthic flux stations were selected to provide a representative sample of contamination levels in Sinclair Inlet (Chadwick and Lieberman, 1991). The station groups were selected based on the weight of metal contamination levels ($\mu g/g$). A somewhat different distribution is obtained when the contamination levels are displayed on a molar concentration basis ($\mu mol/g$) (table 1). The molar concentration has more relevance when assessing the importance of AVS interaction to the availability and flux of metals.

The proximity of the stations (figure 1) and the wide variation in contamination levels and AVS concentrations (table 1, figure 4) indicate that the Inlet sediments are highly variable and heterogeneous. Another example of the spatial variability of the Inlet sediments is seen in the dry:wet ratio measured for the sediments (table 6). Differences in water content of the sediment cores are indicative of different depositional environments. The presence of numerous bulkheads, piers, and pilings in and around the shipyard and their interaction with tidal currents and eddy patterns can create vastly different sedimentation regimes at stations only 20 to 30 meters apart (Håkanson, 1992). The sampling regime used to evaluate the AVS distribution (two cores per station, one for a composite sample, one for profile samples) was used to obtain representative measures of AVS relative to the range of contamination levels present. The AVS data provided background information to evaluate the flux measurements.

The quality control data used to evaluate the accuracy and precision of the AVS measurements showed that the sampling method provided acceptable data. The sulfide electrode response was very stable and showed good linearity and repeatability for standard concentrations ranging from 2.9 to 1050 µM S⁼. Very good precision was obtained for duplicate samples which spanned a wide range of AVS concentrations. The accuracy of the sampling method was also tested with spiked samples. Blank spike recoveries were within $\pm 5\%$ of the spiked concentration. However, spiked sediment samples rarely resulted in better than $\pm 10\%$ recovery of the spiked concentration. A problem with spiking sediments is that one is not always sure what the sediment AVS concentration was to which the spike was added. Since there is not much possibility of sulfide contamination during the AVS analysis, the biggest quality assurance concern is to assure that AVS is not lost during the analysis. It is possible that the free sulfide spike (made in DDIW) added to the sediment reacts with sediment monosulfides present, converting a portion of the sulfide into non-AVS forms (Morse et al., 1987; Berner, 1970). Underestimates of AVS due to interferences caused by elemental sulphur, produced by reactions between ferric iron and sulfide, have been reported in the literature (Morse et al., 1987). This interference would not occur in the spiked blanks because no sediment is present. The reaction between the sulfide in the spike and the monosulfides in the sediments would have to occur during the initial purge step (nominally 10 minutes) before the sample is acidified to release the AVS. Similar problems with spiked sediment recoveries have been encountered by other workers (W. Boothman, EPA ERLN, personal communication; Boothman and Helmstetter, 1992).

A wide range of AVS was measured from the different stations. Clearly stations 5a and 5b had the highest AVS (figures 4 and 5) and stations 1a, 1b, and 3a had the lowest. The low AVS at stations 1a and 1b may be explained by the sandy and what appeared to be well-oxygenated sediments sampled at those locations. Stations 1a and 1b were closer to the Sound and subject to more mixing (figure 1). The high AVS measured at the other stations around the shipyard indicates that the anoxic sediments favored the formation of sulfides. Station 3b was an anomaly.

Oxygen microprofiles obtained from station 3b showed that the aerobic sediments extended down to about 0.5 cm, which was the deepest aerobic layer obtained from the four oxygen microprofiles measured (stations 2b, 3b, 4a, and 5b) (Chadwick et al., 1992). Oxygenated sediments at depth could account for the low AVS measured from the cores collected at station 3b.

The differences between the composite AVS and the mean profile AVS (figure 4) may be due to the variability in the sediment samples. Dry to wet ratios (R) obtained for composites were similar to the average R of the profile samples, suggesting that the composite samples were well homogenized. It appears that homogenizing the core has the effect of reducing AVS, because the higher AVS material at the intermediate core depth (4-6 cm) is diluted by the lower AVS material at the surface and deeper in the core (figure 5).

The core profiles, which showed higher AVS at intermediate depths, are indicative of the sulphur chemistry in sediments. The major processes which control the form and abundance of sulphur in anoxic sediments include (1) the reduction of sulfate to sulfide, (2) the reaction of sulfides with iron (and other metals) to form monosulfides, and (3) the reaction between the monosulfides and elemental sulphur to form pyrite (Morse et al., 1987). The rate of these reactions will be controlled by the amount of organic matter and bacteria available to reduce sulphate, the rate of diffusion of sulphate from the overlying water into the sediment, the concentration and activity of metals (mainly iron), and the evolution of elemental sulphur (Morse et al., 1987). The core profiles from Sinclair Inlet trace the time course of the above reactions. At the surface, AVS is relatively high, because of the rapid and sharp decline of oxygen measured in the first few millemeters of sediment, the high amount of organic matter present, and a readily available source of sulphate in the bottom water of the inlet (Chadwick et al., 1992). The high water content of the surface sediment suggests that the diffusion of sulphate into the sediment is not impeded. The AVS maximum, measured at a 4-6-cm depth in most of the cores, reflects the optimal formation of reactive monosulfides. The decrease in AVS toward the bottom of the core suggests that more sulfide is tied up in less reactive polysulfide and mineral sulfide (pyrite) forms (Morse et al., 1987; Cornwell and Morse, 1987). These profiles are comparable to AVS core profiles commonly measured from a variety of marine environments (Di Toro et al., 1990; Boothman and Helmstetter, 1992).

The relationship of toxic metal contamination to AVS was provided by the analysis of SEM concentrations of Cu, Pb, and Zn. These three metals accounted for about 70–90% of the total toxic metals present at stations in the Inlet (table 1). Since the monosulfides formed by CuS, PbS, and ZnS have lower solubility products (K_{sp}) than FeS (Di Toro et al., 1990), they are associated with the reactive pool of AVS. After Fe, Ni and Zn would be metals most likely to be released (Di Toro et al., 1992), because the K_{sp} of these sulfides are higher than those for the other compounds (e.g., the K_{sp} of MnS > FeS > NiS > ZnS > CdS > PbS > CuS > HgS (Di Toro et al., 1990)).

Copper, which was one of the more abundant contaminants, was also measured in the core profile samples to provide information on the relative distribution of the metal in the top 30 cm of the sediments sampled. If Cu can be considered as a tracer of metal contamination (i.e., Cu has the same source and is involved in the same depositional processes as the other metals), then it appears that the metal concentrations are fairly uniform in the top 20 to 30 cm of the sediment. This could be an indication of constant loading over time, or evidence that the sediments have been reworked by bioturbation, prop wash, diffusion, etc. The method used for GFAA Cu analysis provided very good sensitivity for the range of Cu concentrations measured during this study. The method used was developed as part of an extensive program to evaluate the leaching rates of copper from bottom paints and was optimally suited for measuring relatively high Cu concentrations in a seawater-like matrix (Lindner and Caso, in preparation). The sensitivity obtained for the GFAA Pb analysis was much reduced, although the response was linear over the range of samples analyzed. The GFAA analysis of Zn was hampered by contamination problems (table 8) and a nonlinear response. The DIW standard curve, rather than the matrix standard addition curve, was used to calibrate Zn response because the DIW standard curve was linear over a greater concentration range (0 to 250 ppb). It was thought that this would provide a better method of comparing the relative Zn levels observed at the stations. The methodology used requires further development to improve the accuracy of the Pb and Zn analysis.

Since it was not possible to obtain an accurate and comprehensive analysis of all the SEM concentrations, bulk sediment metal concentrations, determined from grab samples collected from the same stations (Chadwick et al., 1992; URS Consultants, Inc., 1990), were used as surrogates for SEM. This provides a much more conservative analysis of metal to AVS ratios because bulk sediment metal analysis uses a strong acid digestion step and therefore releases more tightly bound, and not biologically available, solid phase metals (Di Toro et al., 1990; Ankley et al., 1991; Di Toro et al., 1992). By normalizing the bulk sediment metals with the composite AVS, it is possible to evaluate the potential availability of toxic metal contamination in the Inlet.

The data presented here show that AVSs were very abundant and are available to react and bind with the toxic metals. Recent studies have shown that when the metal:AVS ratio is less than or equal to one, organisms are not sensitive to toxic metal exposure. Di Toro et al., (1990) showed that AVS normalization explained the observed metal toxicity to amphipods (*Ampelisca* sp., and *Rhepoxynius hudsoni*) exposed to sediment contaminated with Cd. The theoretical and technical basis of AVS-mediated toxicity, presented by Di Toro et al. (1990), has been supported by other studies. Ankley et al. (1991) reported that toxicity and bioaccumulation of Ni and Cd to the amphipod *Hyalella azteca* and oligochaete *Lumbriculus variegatus* were greatly reduced when the SEM:AVS ratio was less than one. Similarly, Carlson et al. (1991) showed that AVS reduces the toxicity and bioavailablity of metals in freshwater systems. Further work indicates that the SEM:AVS ratio may be used to predict the acute toxicity of metals in sediments (Di Toro et al. 1992) and can therefore be used to develop sediment quality criteria (Di Toro et al., 1991; Burton, 1992; Power and Chapman, 1992).

The levels of AVS measured in the sediments of Sinclair Inlet are very high and compare to the high end of values reported in the literature from field studies (table 12). The high AVS means that the sediments have a high buffering capacity, enabling toxic metals to be assimilated and detoxified (Di Toro et al., 1992; Burton, 1992). The AVS neutralizes the toxic metal activity in the aqueous and solid phases of the sediment (Di Toro et al., 1990), and therefore can be used determine the absolute metal toxicity to aquatic organisms when AVSs are present in sufficient quantities (e.g., SEM/AVS < 1) (Di Toro et al., 1992). However, AVS has been shown to vary seasonally, with the lowest concentrations measured during winter (Di Toro et al., 1990; Boothman and Helmstetter, 1992). As such, the July 1991 AVS measurements made for Sinclair Inlet should be viewed as probably being higher than the average yearly concentration. If the levels of AVS can be viewed as determining the boundary between low and high metal activity (Di Toro et al., 1992), the high level of AVS (many times higher than metal concentrations)

indicates that during this sampling period a very high metal assimilation capacity existed in the sediments of the Inlet.

There appeared to be no direct correlation between the flux rates measured for Ni, Zn, Cu, and Pb and the AVS measured in composite and surface samples, nor for the bulk sediment metal concentrations normalized by AVS (table 11). Chadwick et al. (1992) reported significant positive correlations between the bulk sediment concentrations of Cu and Zn and flux of Cu and Zn.

Location	Depth (cm)	AVS Range (µmol/g)	Source
Long Island Sound, NY	0-10	8.4 – 17.4	Di Toro et al., 1990
Sapelo Island	0-10	14.6 - 43.2	Di Toro et al., 1990
Lake Mendota, WI	0–10	8.7 – 112.0	Nriagu & Coker, 1976
Freshwater marsh, NY	grab	0.1 - 75.5	Ankley et al., 1992
Narragansett Bay, RI	grab	1.9 - 53.6	Johnston, unpublished
Black Rock Harbor, CT		175.0 ± 41.0	Di Toro et al., 1990
Hudson River, MU		12.6 ± 2.8	Di Toro et al., 1990
Pequaywan lake, MN		42.8 ± 8.0	Carlson et al., 1990
East River, WI		8.8 ± 1.4	Carlson et al., 1990
W. Bearskin Lake, MN		3.6 ± 1.6	Carlson et al., 1990
Lake Ontario, Canada	0-10	27.1	Nriagu, 1968
Pettaquamscutt Cove, RI	0-1	<5.0	Boothman & Helmstetter, 1992
Pettaquamscutt Cove, RI	25	15.0 - 35.0	Boothman & Helmstetter, 1992
Pettaquamscutt Cove, RI	6-15	10.0 - 15.0	Boothman & Helmstetter, 1992

Tab	le	12.	Values of	AVS	reported	from	other	field	studies.
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The AVS data alone do not explain the observed flux rates; however, the AVS analysis does provide some indications of the metal availability in the sediment. The flux chamber measured the in situ interaction between the sediment surface and bottom water. Independent measurements of dissolved oxygen and $S^{=}$ showed steep gradients in the top 0.5 cm of sediment (Chadwick et al., 1992). Therefore, it would be necessary to measure AVS in the top 0 to 0.5 cm, or at most 0 to 1.0 cm, in order to compare AVS more directly to the flux measured by the benthic chamber. In this sense, the AVS measured in the deeper sediment and in composite samples may relate more to the long-term availability and in-place toxicity of the sediments than to the instantaneous metal flux measured at a particular point in time by the benthic chamber. Another factor could be that the flux rates are governed by the redox potential at the sediment interface, which is controlled by the dissolved O₂ concentration (Riley and Chester, 1971), rather than AVS reactions. The sharp dissolved O₂ gradient in the very surface of the sediment suggests that O₂ diffusion rates are high and are probably independent of AVS processes. In addition, the variability factor (discussed above) could result in vastly different sediment conditions (e.g., AVS and metal flux) over relatively short distances at the same sampling sites. The AVS and flux rate measurements do indicate certain trends in the metal availability and toxicity in the sediments. The very low AVS measured at station 1a suggests that other processes are controlling the metal distribution. The excess metal concentrations measured at stations 3a and 3b (figure 14) are consistent with statistically significant release rates at stations 3a (Zn, and Cu) and 3b (Ni) (table 11). Station 4a, which was on the borderline of excess metals, also showed a statistically significant release of Ni (table 11). Stations 5a and 5b clearly had excessive levels of AVS present (table 5), yet significant release rates were measured (table 11). These were the most contaminated stations, and the increased flux signal may be more related to diffusion in pore water and other non-AVS metal species.

A wide variety of factors will influence metal mobility and availability. Chemical reactions, including cation exchange, oxidation reduction, acid-base reactions, and interaction with organic complexing agents; biological transformations mediated by benthic micro- and macroorganisms; and physical advection, diffusion, and convection mechanisms will affect the availability and flux of metals from the sediment (Burgess and Scott, 1992). If the ten Sinclair Inlet stations are taken as replicate measures of sediment conditions in the Inlet, the average release rates and AVS concentrations can be evaluated (table 11). On average, Ni and Zn had the highest flux rates, which is consistent with the predicted release based on sulfide solubility (Di Toro et al., 1990; Di Toro et al., 1992; Chadwick et al., 1992). The average AVSs measured for composite and surface samples are comparable to the high range of AVS reported in the literature (table 12). The conservative estimate of bulk/AVS results in the ratio of 0.9 and SEM/AVS (without the Ni contribution to SEM) results in a ratio of about 0.5 to 0.9. While the conservative estimate indcates that the metal/AVS ratio is nearing the toxic threshold (figure 14), it assumes that all the metals are available. A more reasonable conclusion is that the "real" metal/AVS ratio lies somewhere between the two and that, on average, the AVS pool is sufficient to reduce the mobility and toxicity of the metals present (table 11).

It should also be noted that the results reported here are only a snapshot in time. Other workers have reported large seasonal variations in sediment AVS concentrations (Di Toro et al., 1990; Schubauer-Berigan, 1992; Boothman and Helmstetter, 1992). Not enough information is yet known about the AVS dynamics in sediments. However, these results show that AVS production in the Inlet is very high and that AVSs are abundant in most of the cores analyzed.

Finally, because the high concentrations of sulfides indicate the anoxic nature of the sediments, the metals will most likely reside in a reduced state. As cleanup efforts in the Inlet proceed, the sediments should be monitored. As water quality increases and high loads of organic matter (biological oxygen demand) are reduced, benthic organisms may invade the bottom of the Inlet and begin to colonize and rework the sediments. This will cause the sediments to become aerobic, destroying the AVS and allowing the metals to oxidize and become more mobile. In any case, there will be a large pool of anoxic sediments and AVS in the Inlet for some time. Even as the upper sediments are reworked and oxidized, the large concentration of AVS at depth will act as a sink (or source) for dissolved metals as they diffuse through the sediment, possibly resulting in a reduction of metal availability in the aerobic layer (Di Toro et al., 1992). The impact to the environment will depend on the rate this occurs and on other mediating factors. Future exposure scenarios should be considered as part of any long-term management plan for Sinclair Inlet.

CONCLUSIONS

The analytical method used to measure AVS in sediments from Sinclair Inlet was able to provide accurate measurements of the reactive sulfide in the sediment cores. The rapid turnaround between sample collection and sample analysis (less than a day for most of the samples) assured a minimum of change in the sediments before they were analyzed.

High concentrations of AVS measured in the sediments for Sinclair Inlet suggest that most of the divalent metals are bound up as nonavailable and nonmobile sulfides.

The core profiles from Sinclair Inlet trace the time course of sulfide geochemistry. At the surface, AVS is relatively high, due to the rapid and sharp decline of oxygen measured in the first few millimeters of sediment, the high amount of organic matter present, and a readily available source of sulphate in the bottom water of the Inlet. The AVS maximum, measured at a 4-6-cm depth in most of the cores, reflects the optimal formation of reactive monosulfides. The decrease in AVS toward the bottom of the core suggests that more sulfide is tied up in less reactive polysulfide and mineral sulfide (pyrite) forms.

There was not a clear relationship between AVS and metal flux measured with the benthic chamber, although a trend of higher flux rates was observed for metals with lower sulfide solubility (Ni and Zn).

The results reported here are only a snapshot in time and do not take into consideration seasonal changes of AVS. Not enough information is yet known about the AVS dynamics in sediments. However, these results show that AVS production in the Inlet is very high.

High concentrations of sulfide indicate the anoxic nature of the sediments, meaning that most metals will reside in a reduced state. As cleanup efforts in the Inlet proceed, the sediments should be monitored. As water quality increases and high loads of organic matter (biological oxygen demand) are reduced, benthic organisms may invade the bottom of the Inlet and begin to colonize and rework the sediments. This will cause the sediments to become aerobic, destroying AVS and oxidizing metals, making them more mobile. The impact to the environment will depend on the rate this occurs and on other mediating factors. Future exposure scenarios should be considered as part of any long-term management plan for Sinclair Inlet.

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Appendix A

STANDARD OPERATING PROCEDURE FOR AVS DETERMINATION IN SEDIMENTS USING SULFIDE-SPECIFIC ELECTRODE DETECTION

Warren S. Boothman U.S. Environmental Protection Agency Environmental Research Laboratory Narragansett, R.I.

I. Introduction

Acid volatile sulfides (AVS) are amorphous or moderately crystalline metal monosulfides, primarily FeS. They have been hypothesized as the single most important factor controlling the availability of heavy metals to benthic organisms in anoxic sediments (1). The molar ratio of extractable metal to AVS is postulated as an indicator of metal availability. However, methods used to determine AVS have been quite varied in both reagents and conditions. In order to insure that data used by different investigators to test the AVS hypothesis be comparable, a common protocol for the sampling and analysis of AVS needs to be established. The method described follows the findings of Cornwell and Morse (2). Comparisons of this method with a gravimetric method used at Manhattan College have reportedly given equivalent results (D. Hansen, personal communication).

II. Sample collection and storage

The accurate determination of acid volatile sulfides in sediments places a number of rather stringent requirements on the handling of samples after collection. Sulfide ion is thermodynamically unstable in the presence of dissolved oxygen, and so sediments from anoxic environments must be preserved in such a way as to protect any sulfides present from reaction with air. Storage containers must exclude or minimize air space above the sediments; if possible, purging of container headspace with dry, oxygen-free nitrogen gas would be helpful. Sediments should be kept cold or frozen during storage and transportation. Wet sediments may be stored at 4°C for short time, but anoxic sediments stored at 4°C for 20 days show significant changes in metals' partitioning, suggestive of oxidation of the sediment (3). Freezing of sediments seems to cause the least change in the speciation of metals (and by implication sulfides) in anoxic sediments; comparison of metals' extractability in fresh sediments and sediments stored for 20 days at -30°C showed essentially no significant differences. Drying of sediments, either in air or by freeze-drying, has been shown to reduce the concentration of AVS measured in anoxic sediments and should be avoided. The loss of AVS may be due to oxidation or formation of more crystalline (and non-acid volatile) sulfides

III. Determination of Acid Volatile Sulfides

The classification of sulfides as "acid volatile" is an operational definition, that is, the extent to which mineral sulfide phases are volatilized by the analysis will depend on the analytical conditions employed, e.g. acid concentration, time, etc. For AVS data obtained by different investigators to be comparable, the reaction conditions utilized to volatilize sediment sulfides must be similar; the methods used to isolate and quantify the volatilized sulfides, on the other hand, may vary according to instrumental or laboratory availability. In the method presented here, acid volatile sulfides are determined by reaction of sediment sulfides with 1M HCl to form gaseous H_2S and purging the evolved H₂S with nitrogen. The purged H₂S is then trapped in sulfide anti-oxidant buffer (SAOB) (4), diluted to volume and the S⁼ concentration measured with a sulfide ionspecific electrode (Orion 94-16A). Overall sulfide recovery is determined by analysis of aliquots of a working sulfide standard solution and sediments which have been previously well characterized.

A. Volatilization and trapping of sediment sulfides



Figure 1. Apparatus for AVS determination: $1.N_2$ cylinder; 2.Gas washing bottles: (a) oxygen scrubbing solution, (b) deionized water; 3.Three-way stopcock; 4.Purge flow controller; 5.Reaction flask; 6.Magnetic stirrer; 7.Sulfide traps.

The apparatus used for the volatilization and trapping of acid volatile sulfides in sediments, illustrated in Figure 1, is an adaptation of the system developed by Allen and coworkers at the University of Deleware (5) and DiToro et al.(1). This configuration of the glassware allows the acidification of the sediment while minimizing the entrainment of laboratory air into reaction vessel. It also allows the use of more than one analytical setup at a time while providing purge gas control for each individual analysis.

To prevent oxidation of sulfides due to oxygen in the analytical train, the apparatus should be purged with oxygenfree nitrogen for at least 30 minutes prior to initiating analyses.

- 1. <u>Wet</u> sediment (ca. 10 g) is weighed into a 250-ml standard taper round bottom flask.
- 2. Fifty milliliters (50 ml) of deaerated deionized water (DDIW) is added to cover the sediment, a magnetic stir bar placed in the flask and the flask is placed into the sampling apparatus. Impinger (trap) bottles should be filled with 50 ml of SAOB and 30 ml of DDIW.
- 3. Initiate purge gas flow at 100 ml min⁻¹ to remove any entrained air from the headspace and purge for 10 minutes. Reduce flow to 40 ml min⁻¹.
- 4. Halt purge gas flow and slowly inject 10 ml of 6M HCl (over approximately 15 sec.) through the septum sidearm, resulting in a concentration of 1.0 M HCl (neglecting the water content of the sediment).
- 5. Resume purge gas flow of 40 ml min⁻¹ and stir sediments vigorously. Purge and trap generated H_2S for desired time (usually 30 minutes).
- 6. Stop purge flow, rinse impingers with DDIW into bottles and remove bottles from apparatus.
- B. <u>Measurement of sulfides by ion-specific electrode</u> Note: Sulfide electrode and meter should be calibrated prior to performing sediment analyses using sulfide standards prepared in SAOB diluted 1:1 with DDIW.
 - 1. Pour bottle contents into 100-ml volumetric flask. Rinse bottle with DDIW, adding rinse to the vol. flask. Dilute to volume with DDIW.
 - 2. Pour contents of volumetric flask into 150-ml beaker, add magnetic stirring bar and place on stirrer. Begin stirring with minimum agitation to avoid entrainment of air into solution and minimize oxidation of sample during the measurement.
 - 3. Rinse sulfide and reference electrodes into waste container and blot dry with absorbent tissue. Immerse electrodes in sample solution.
 - 4. Allow electrode response to stabilize (8-10 minutes), then take measurement of sulfide concentration (C_s2-) . Reading may be directly in concentration units, if the meter is in **concentration** mode and a 2-point calibration has been performed, or in millivolts. If the millivolt reading is used, convert millivolts to concentration using the calibration curve obtained from standard solutions.
- C. <u>Calculation of AVS concentration in sediments</u>
 - 1. The sediment dry weight/wet weight ratio (R) must be determined separately. Acid volatile sulfides can be oxidized or altered to non-acid volatile forms during various drying processes.

2. AVS concentration in a sample is calculated using the formula:

AVS(μ mol/g dry sediment) = $\frac{(C_s^{2-}) \times V_{trap}}{g \text{ wet sediment x R}}$ $= \frac{0.1 \times C_s^{2-}}{g \text{ wet sediment x R}}$

when $C_{s2-} = \mu \text{mole liter}^{-1}$ (μM) and $V_{\text{trap}} = 100 \text{ ml}$.

- D. <u>Calibration of sulfide-specific electrode</u>
 - Direct concentration (2-point calibration)

 a. select CONC mode on meter.
 - b. press the CAL button on the meter. The CAL1 light should come on.
 - c. immerse electrodes in first calibration standard as for sample (III.B.3-4) and allow response to stabilize. Adjust the concentration displayed to match the standard concentration, using the t and t keys. Press the ENTER key.
 - d. After a few seconds, the CAL1 light should go off and the CAL2 light should come on. Repeat step c for the second calibration standard. After pressing ENTER, the SAMPLE light should come back on, indicating calibration is complete and providing direct readout of concentration.
 - e. Press **SLOPE** and verify that a value near the theoretical slope (-29.6 mV/decade) is displayed. If the value is far off, repeat the calibration or prepare new standards.
 - 2. multipoint calibration
 - a. select MV mode on meter.
 - b. immerse electrodes in first calibration standard as for sample (III.B.3-4) and allow response to stabilize. Record the electrode response. Repeat for other standards.
 - c. The calibration curve is obtained by linear regression of millivolts against log concentration.
 - 3. Sulfide calibration standards Calibration standards are prepared from the primary sulfide stock solution (IV.D.). The primary stock concentration <u>must</u> standardized by iodometric titration (IV.D.2) before preparing standards.
 - a. Prepare 700 ml of diluent by mixing 350 ml of SAOB (IV.B) with an equal volume of deaerated deionized water (IV.A).
 - b. Prepare a working stock solution (approx. 1500 μ M). Pour 50 ml of the 1:1 SAOB diluent in a 100-ml volumetric flask. Pipette an appropriate volume of the primary stock and an equal volume of SAOB into

the flask. Dilute to volume with the diluent.
c. Prepare calibration standard solutions by dilution of
the working stock solution. An example of the
calculation of standard concentrations is illustrated
below.

Primary Stock Con	ncentra	tion	55.	07	umol/ml
-			17	66	ug/ml
Working stock pro	eparati	on			
Primary stock	aliquo	t 2.7	50 ml		
Working stock	volume	1	100		
Working stock	concen	tration	15	14	uM
-			48	.6	ug/ml
Working Stock	Calibra	tion Sta	indard		
Volume Vo	olume	Concer	ntration		
ml	ml	uM	ug/ml		
0.200	100	3.03	0.097		
0.500	100	7.57	0.243		
2.00	100	30.3	0.97		
5.00	100	75.7	2.43		
20.00	100	303	9.7		
50.00	100	757	24.3		

IV. Preparation of reagents and stock solutions

A. Deaerated Deionized Water (DDIW)

Dissolved oxygen-free water for preparation of standards, reagents, etc. should be prepared daily by bubbling nitrogen gas (N2) vigorously through 2.5 1 of deionized water for a minimum of 1 hour. The nitrogen gas should be stripped of traces of oxygen by passing through a gas washing bottle filled with vanadous chloride solution (IV.C.).

B. Sulfide Anti-oxidant Buffer Reagent (SAOB)

(2M NaOH, 0.2M EDTA, 0.2M ascorbic acid)

- 1. Dissolve 80.00 g NaOH slowly in 700 ml DDIW.
- 2. When cool, add 74.45 g EDTA (disodium form) and stir until dissolved.
- 3. Add 35.23 g ascorbic acid and stir until dissolved.
- 4. Pour solution into a 1.00-1 volumetric flask and dilute to volume with DDIW.

C. Vanadous Chloride Oxygen-stripping Solution

- 1. Weigh 4 g of ammonium metavanadate (NH_4VO_3) in a 100-ml beaker. Add 50 ml concentrated HCl and heat to near boiling. Cool and dilute to 500 ml with deionized water. Pour the solution into a 500-ml gas washing bottle.
- 2. Prepare amalgamated zinc by covering 15 g of zinc metal with deionized water, adding 3 drops (150 μ 1) of concentrated HCL, and adding a small amount of mercury. Stir slightly to mix in the mercury.
- 3. Add the amalgamated zinc to the vanadous chloride solution in the gas washing bottle. The solution should

be green or blue. Bubble nitrogen through the solution until the color becomes purple. When the solution returns to a blue or green color, the oxygen stripping capacity has been exhausted; it may be replenished by the addition of more amalgamated zinc or a slight amount of conc. HCL.

D. Sulfide stock solution

A sulfide stock solution should be prepared and maintained for use in quality assurance and calibration. An aqueous solution of $Na_2S \cdot 9H_2O$ of sufficient concentration may be used as a stock for secondary stocks for spiking sediments, calibrating, etc. The concentration of this stock should be determined before each use by iodometric titration or other standardization techniques.

1. Preparation

- a. Wash crystals of $Na_2S \cdot 9H_2O$ with deionized water and blot dry.
- b. Weigh approximately 12 g of $Na_2S \cdot 9H_2O$ and dissolve in 900 ml DDIW.
- c. Pour into a 1.00-1 volumetric flask and dilute to volume with DDIW.
- 2. Standardization
 - a. Pipette 10.00 ml of standard iodine solution into each of two 125-ml Erlenmeyer flasks.
 - b. Pipette 2.000 ml of sulfide stock into one flask. Pipette 2.000 ml of DDIW as a blank into the other flask.
 - c. Add 5.00 ml of 6M HCl into each flask, swirl slightly, then cover and place in the dark for 5 minutes.
 - d. Titrate each with 0.025N thiosulfate solution, adding soluble starch indicator when the yellow iodine color fades. The end point is reached when the blue color disappears.
 - e. The sulfide concentration may be calculated from:

Sulfide = $\frac{(T_{blank} - T_{sample}) \times N_{S2O3}2-}{V_{aample}} \times \frac{1 \text{ mole } S^{2-}}{2 \text{ equiv } S^{2-}} \times \frac{1000 \ \mu\text{moles}}{1 \text{ mmole}}$

where T = volume of titrant used for the blank and sample (ml)

 $N = \text{concentration of } S_2O_32 - \text{titrant}$ V = volume of sample used (2.00 ml)

- E. Standard Iodine solution (0.025N): Dissolve 20-25 g KI in 100 ml deionized water. Weigh 3.2 g I_2 and dissolve in KI solution. Dilute to 1.00-1 with deionized water. This solution may be standardized against the thiosulfate solution.
- F. Thiosulfate titrant (0.025N) may be purchased commercially or prepared in the laboratory. If prepared in the lab, it should be standardized against potassium dichromate.

- 1. <u>Preparation</u>: Weigh approx. 6.2 g of $Na_2S_2O_3 \cdot 5H_2O$ into 500ml beaker. Add 0.1 g Na_2CO_3 and dissolve in 400 ml DDIW. Pour into 1.00-1 vol. flask and dilute to volume with DDIW.
- 2. <u>Standardization</u>
 - a. Weigh 0.2 g dry K_2CrO_7 into a 500-ml Erlenmeyer flask and dissolve in 50 ml deionized water.
 - b. Dissolve 3 g of KI in 50 ml of deionized water, add 5 ml of 6M HCl, and add to KI solution. Swirl, cover and store in dark for 5 minutes. Add 200 ml deionized water and titrate with the thiosulfate solution, adding starch indicator when the yellow iodine color fades, until the blue color fades to pale green.
- 3. Calculate the thiosulfate concentration as follows:

-	g $K_2 CrO_7$	1 mole K ₂ CrO ₇	6 equiv K_2CrO_7	1000 ml
$N(S_2O_3^{2})$	≈ X	X	· · · · · · · · · · · · · · · · · · ·	<
- • ·	ml $S_2 O_3^{2}$	294.19 g K ₂ CrO ₇	1 mole K ₂ CrO ₇	1 1

G. Boluble starch indicator is prepared by dissolving 1.0 g starch in 100 ml boiling deionized water.

V. Preparation of sulfide electrode and meter

The sulfide and reference electrodes and meter should be used and maintained as per the manufacturer's specifications. The instructions below give the brief description; see the meter or electrode operating manuals for detailed instructions.

- 1. Clean the inner sleeve of the double junction reference electrode with deionized water and fill the inner sleeve with the appropriate filling solution (saturated AgCl).
- 2. Clean and dry the outer sleeve walls with deionized water. Moisten the gasket of the outer sleeve with the outer sleeve filling solution (10% KNO₃), slide the sleeve on over the inner sleeve and screw on the end cap. Fill the outer sleeve with filling solution.
- 3. Inspect the surface of the sulfide electrode and polish if necessary (see electrode operating instructions for details).
- 4. Connect the electrodes to the meter.
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Appendix B

RAW DATA OF AVS IN SEDIMENT SAMPLES

Standard curves were generated daily using the log base 10 (log[S--]) of the standardized SAOB concentrations ([S--]uW), versus the electrode response (mV). Regression were calculated using QUATROPRO spreadsheet software.

July 8, 1991 STANDARD CURVE

	[S]uM	log[S]	۳V	Regression Outpu	t:
	2.900	0.462398	-736.8	Constant	-723,16
	- 7.26	5 0.861236	-748.6	Std Err of Y Est	0 15816
	29.058	3 1.463266	-766.7	R Squared	0.99998
	72.64	5 1.861206	-778.5	No. of Observations	6
	290.580	2.463266	-796.0	Degrees of Freedom	4
	1050.450	3.021375	-812.7	- -	
				X Coefficient(s) -:	29.6386
				Std Err of Coef.	0.0734
(=	-29.6386(10	X) - 723.16	63		

logX = (Y+723.163)/-29.6386

Sample Log: For each entry the sample identifier is given. If the entry is a sediment sample the wet weight (g) of sediment is also given. Results obtained from measurements of the trap solutions are tabulated for sulfide electrode response (mV) and the sulfide concentration in uM ([S--]) calculated using the current day standard curve. If the sample was spiked the spike concentration in uM is given (SpkAdd) and the recovery ratio is determined for the measured versus added concentrations (Neas/Add) and the system spike versus the bench spike (Sys/Ben) of the same spike concentration. The AVS per unit gram wet sediment is also calculated (uM/g).

		a or			Recovery	Recovery
No. 1	Sample ID wet wt (g) SysBlank	Ĩ₩V	[5]	SpkAdd 0	Weas/Add	Sys/Ben
	Trap2	-588.5	0.00			
	Trap1	-545.9	0.00			
2	SvsSpk1		••••	19.37		
	Trap2	-540.0	0.00			
	Trap1	-759.6	16.96		0.88 (0.88
3	BenSpk1	-761.3	19.35	19.37	1.00	
4	BenSpk2	-790.1	181.32	193.70	0.94	
5	SvsSpk2			193.70	••••	
	Trap2	-554.6	0.00			
	Trap1	-789.8	177.14		0.91	0.98
6	sin190-5a4-avs-0-2a	7.0		1723.41	[Total recv. a	after 1hr]
	trap2	-715.2	0.54		••••••	·····,
	trap1-30min	-818.9	1698.81		0.99 [Mea:	s. of 30min
	trap2+15min	-718.5	0.70		reci	overv, after
	trap1+15min	-760.2	17.77		0.01 ano	ther 15min.+
	trap2+15min(again)	-717.0	0.62		ano	ther 15min
	trap1+15min(again)	-747.9	6.83		0.00 of (extraction]
7	sin190-5a4-avs0-2b	7.9				
	trap2	-725.4	1.19			
	trap1	-818.5	1646.83			
	·	-814.1	1170.02			M / _
8	sin190-5a4-avs4-6a	6.0			u 354	4 .30
	trap2	-728.4	1,50			
	trap1	-821.7	2111.63			
0	cin100.524.2vs4.8b	4 5		968 0		
3	+snike	4.J		300.0	40	8 38
	trap2	- 729 . 7	1.66		40	0.00

	trap1.2114	-821.3	2047.02			
	trap1.2122	-820.6	1938.67			
	avg	-821.0	1992.11			
10	sin190-5a4-avs8-10a	6.3				1142.85
	trap2	-735.7	2.65			
	trap1.2213	-838.3	7668.27			
	trap1.2221	-836.8	6824.77			
	avo	-837.6	7234.24			
			0.00			
11	sin190-5a4-avs8-10b	8.2				1042.85
	trap2	-740.2	3.76			
	trap1.2309	-840.4	9027.16			
	trap1.2317	-839.1	8159.99			
	avg	-839.8	8582.63			
12	sin190-5a4-2300svsspk					
	spike			1937.2		
	trap2	-740.2	3.76			
	trap1.0003	-816.0	1356.13			
	trap1.0012	-815.5	1304.46		0.67	
	trap1.jul10.0930	-815.3	1284.35			

July 10, 1020 Standard Curve

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	[S]uM 2.997.2029.0072.6290.51050.4Y = -29.4267(1logX = (Y + 722.8)	log[S] 00 0.462 65 0.861 58 1.463 45 1.861 80 2.463 50 3.021 0gX) - 722.82 21)/-29.4267	W -736.10 -748.30 -766.00 -778.00 -795.10 -811.60	R Constant Std Err of R Squared No. of Obs Degrees of X Coeffici Std Err of	egression Y Est ervations Freedom ent(s) Coef.	Output: -722.82 0.30329 0.99991 6.00 4.00 -29.43 0.14
No.	Sample ID wet wt	g or (g) ∎V	[\$}	SpkAdd	Recovery Neas/Add	Recovery Sys/Ben
13	benchspike(121.1u 1220 1217 1246 avg	M) -784.3 -783.4 -784.1 -783.9	122.81 114.46 120.90 119.33	121.1	0.99	
14	sin190-3a4-avs121 trap2 trap1.1259 trap1.1327 avg	Osyspike -693.5 -784.3 -783.9 -784.1	0.10 122.81 119.02 120.90 120.92	121.1	1.00	1.01
15	sin190-3a4-avs0-2 trap2 trap1.1414 trap1.1424 trap1.1447 avg	2.8 -671.5 -798.7 -799.0 -798.9 -798.9	0.02 378.95 387.95 384.92 383.92 0.00			135.18
16	sin190-3a4-avs0-2 trap2 trap1.1503 trap1.1508 trap1.1514 avg	2b 6.0 -688.2 -808.4 -808.6 -808.3 -808.3	0.07 809.49 822.25 803.18 811.60			136.17
17	sin190-3a4-avs0-6 trap2 trap1.1553 trap1.1557 trap1.1603 trap1.1611 avg	6a 4.0 -690.3 -811.0 -811.2 -810.8 -810.7 -810.9	0.08 992.12 1007.77 976.71 969.10 986.32			245.96
18	Bench spike 1613 1626 avg	-813.0 -812.3 -812.7	1160.19 1098.35 1128.85	1211.0	0.93	
19	sin190-3a4-avs4- trap2 trap1.1707 trap1.1710 trap1.1716 avg	6b(s 3.8 -695.5 -819.1 -818.4 -817.3 -818.3	0.12 1869.93 1770.26 1624.26 1751.89		0.74 0.86	

20	sin190-3a4-avs8-10a trap2 trap1.1738 trap1.1748 avg	4.2 -702.0 -804.7 -804.3 -804.5	0.20 606.00 587.33 596.59			140.71
21	sin190-3a4-avs8-10b trap2 trap1.1828 trap1.1834 trap1.1839 avg	2.2 -700.6 -796.5 -796.1 -795.9 -796.2	0.18 319.02 309.19 304.39 310.80			143.89
22	sin190-3a4-avs20-22a trap2 trap1.1917 trap1.1923 trap1.1928 avg	4.6 -695.2 -737.1 -737.8 -736.9 -737.3	0.12 3.06 3.23 3.01 3.10			0.67
23	sin190-3a4-avs30-32a trap2 trap1.2002 trap1.2007 trap1.2012 avg	5.3 -697.8 -740.9 -741.1 -740.7 -740.9	0.14 4.12 4.18 4.05 4.12			0.78
24	benchspike 2027 2032 2043 avg	-813.3 -813.2 -812.3 -813.3	1187.75 1178.49 1098.35 1183.11	1211.0	0.98	
25	sin190-3a4-avs2000sys trap2 trap1.2049 trap1.2034 trap1.2104 avg avg'	pike -699.8 -812.9 -813.0 -812.0 -812.6 -813.0	0.17 1151.15 1160.19 1072.87 1127.38 1155.66 0.00	1211.0	0.95 0.98	0.93 0.95

JULY 11, 1991 STANDARD CURVE

[S]ulli		log[S]	mV	Regression	Output	:
• •	2.900	Ū.462	-734.10	Constant	•	-720.97
	7.265	0.861	-746.90	Std Err of Y Est		0.53371
	29.058	1.463	-765.10	R Squared		0.99972
	72.645	1,861	-777.00	No. of Observations		6.00
	290.580	2.463	-794.20	Degrees of Freedom		4.00
1	1050.450	3.021	-810.80	-		
				X Coefficient(s)	-29.83	
				Std Err of Coef.	0.25	
Y = -29.8	3312(logX	() - 720.9	71			
logX = (Y +	720.971)	/-29.8312				

No. 26	Sample ID wet wt (g) benchspike0930	g or mV	[S]	SpkAdd 1211.0	Recovery Neas/Add	uM/g or Recovery Sys/Ben
	0932 0934 0958	-812.4 -812.6 -812.1	1161.12 1179.19 1134.55			
	avg	-812.4	1158.14		0.96	
27	sin190-4a1-avs0900sys	pike		1211.0		
	trap2 trap1 1007	-684.9	0.06			
	trap1.1013	-811.7	1100.05			
	avg	-811.7	1095.81		0.95	0.90
28	sin190-4a1-avs0-2a	5.4				98.86
	trap2	-593.2	0.00			
	trap1.1057	-802.5	540.77			
	trap1.1100	-802.3	532.40			
	avg	-802.3	533.86			
29	sin190-4a1-avs0-2b	4.1				89.14
	trap2	-691.8	0.11			
	trap1.1143	-797.6	370.47			
	trap1.1143	-797.5	367.62			
	avg	-797.6	369.04			
30	sin190-4a1-avs4-6a	3.5				258.55
	mtrap2	-694.0	0.12			
	trap1.1227	-809.1	900.03			
	trap1,1230	- 609.2	907.00			
	avg	-809.1	902.35			
31	sin190-4a1-avs4-6b	4.1				264.35
	trap2	-701,4	0.22			-
	trap1.1314	-811.7	1100.05			
	trap1.1317	-811.8	1108.58			
	trap1.1324	-811.4	1074.87			
	avy	-01(.0	1094.41			
32	sin190-4a1-avs4-6cspk	2.6	_	1211.00	uM spike	0.85
	trap2	-703.8	0.27			
	Trap1.1400	-816.9	1643.34			
	trap1.1402	-817.2	1681.84			
	LTap1.1409	-01/.2	1081.84			
	~ * 9	-017,1	1000.30			

33	sin190-4a1-avs8-10a trap2 trap1.1448 trap1.1450 trap1.1455 avg	5.6 -704.7 -805.8 -806.0 -805.7 -805.8	0.28 697.64 708.50 692.28 699.44			124.23
34	sin190-4a1-avs20-22 trap2 trap1.1540 trap1.1551 avg	6.5 -704.7 -796.6 -796.3 -796.5	0.28 342.95 335.10 339.00			52.56
35	sin190-4a2-avs0-10a trap2 trap1.1630 trap1.1632 trap1.1640 avg	10.9 -707.8 -810.0 -810.1 -809.5 -809.9	0.36 964.77 972.25 928.25 954.90			87.61
36	sin190-3a3-avs0-9 trap2 trap1.1729 trap1.1733 trap1.1741 avg	5.8 -707.7 -775.7 -775.9 -775.3 -775.6	0.36 68.33 69.39 66.25 67.98			11.68
37	sin190-5a5-avs0-10a træp2 træp1.1818 træp1.1820 avg	7.3 -720.4 -824.9 -824.7	0.96 3047.21 3000.53 0.00 3023.78			415.93
38	benchspike 1750 1752 1753 avg	-811.8 -812.0 -811.7 -811.8	1108.58 1125.82 1100.05 1111.43	1211.0 uM	spike 0.92	
39	sin190-5af-avs1800sys trap2 trap1.1903	3pike -719.0 -811.4 -811.5 -811.6 -811.3	0.86 1074.87 1083.20 1091.59 1066.61	1211.0 uM	spike	
		-811.5 -812.9	1079.03		U.97	0.89

July 12, 1991	sunny, ho	t and wai	rm -	
[S]uM	log[\$]	WV.	Regression	Output:
2.900	Ŏ.46Ż	-734.00	Constant	-720.54
7.265	0.861	-746.30	Std Err of Y Est	0.34758
29.058	1.463	-764.60	R Squared	0.99988
72.645	1.861	-776.60	No. of Observations	6.00
290.580	2.463	-794.10	Degrees of Freedom	4.00
1050.450	3.021	-810.60	0	
			X Coefficient(s)	-29.90
			Std Err of Coef.	0.16
Y = -29.8998(100)	() - 720.5	39		
log X = (Y + 720, 539)	1-29.8998			

No.	Sample ID wet wt (g)	g or mV	[S]	SpkAdd	Recovery Neas/Add	uM/g or Recovery Sys/Ben
40	benchspike			1210.8 uM	spike	
		-812.9	1227.49		-	
		-813.1	1246.54			
	avg	-813.0	1236.98		1.02	
41	svspike			1210.8		
	1335	-813.4	1275.67			
	1346	-813.3	1265.89			
	1348	-813.2	1256.18			
	avg	-813.3	1265.89		1.02	1.05
42	sin1b193-a2(0-2)a	6.8				21.32
	trap2	-703.3	0.27			
	trap1.1527	-785.0	143.19			
	trap1.1532	-785.4	147.67			
	Trap1.1534	-/85.2	145.41			
	avg	-/60.2	145.41			
43	sin1b193-a2(0-2b)	4.9				17.53
	trap2	-707.3	0.36			
	trap1.1622	-778.4	86.13			
	trap1.1624	-778.2	84.82			
	trap1.1631	-778.1	84.16			
	avg	-778.2	85.03			
44	sin1b193-a2(4-6a)	4.5				30.99
	trap2	-706.0	0.33			
	trap1.1707	-784.6	138.84			
	trap1.1709	-784.8	141.00			
	trap1.1/16	-784.4	136.72			
	avg	-784.0	138.84			
45	einth103.a2(1.6b)	Q 1	0.00			34 00
40	tran2	-705 6	0 32			34.90
	trani 1751	.703.8	281 98			
	tran1 1755	.793 9	284 16			
	trap1,1756	-793.8	281.98			
	avg	-793.8	282.70			
46	sin1b193-a2(8-10a)	3_1				68.15
	trap2	-704.8	0.30			
	trap1.1841	-790.4	217.02			
	trap1.1851	-789.9	208.82			
	trap1.1858	-789.6	204.06			
	avg	-790.0	209.90			

47	sin1b193-a2(8-10b) trap2 trap1.1928 trap1.1933 trap1.1935 avg	4.7 -707.1 -792.0 -792.2 -792.0 -792.1	0.36 245.48 249.29 245.48 246.74		52.95
48	sin1b193-a2(20-22a) trap2 trap1.2012 trap1.2014 trap1.2019 avg	6.0 -706.9 -787.8 -787.8 -787.6 -787.7	0.35 177.64 177.64 174.93 176.73		2 9 .70
49	sin2a193-a2(0-2a) trap2 trap1.2053 trap1.2100 trap1.2109 avg	6.8 -712.6 -816.2 -816.1 -815.4 -815.9	0.54 1582.65 1570.51 1488.09 1546.51		227.09
50	sin2a193-a2(0-2b)spk trap2 trap1.2140 trap1.2148	3.0 -711.0 -816.1 -815.8	1570.51 1534.65	1210.75	0.72 114.67 0.48 sys/ben 0.73
	avg 2253	-816.0 -815.8	1552.48 1534.65		sys/ben 0.72
51	benchspike 2137 2137 2139 avg 2255	-812.6 -812.7 -812.6 -812.6 -812.5	1199.45 1208.73 1199.45 1202.54 1190.25	1210.8	
52	sin2a193-a2(4-6a) trap2 trap1.2231 trap1.2232 trap1.2232 avg	2.0 -712.7 -811.8 -811.9 -811.8 -811.8	0.55 1127.79 1136.51 1127.79 1130.69		554.26
53	sin2a1193-a2(4-6b) trap2 trap1.2321 trap1.2324 trap1.2327 avg	2.4 -722.9 -813.6 -813.5 -813.3 -813.5	1.20 1295.47 1285.54 1265.89 1282.24		527.67
54	sin2a193-a2(8-10a) trap2 trap1.0007 trap1.0011 trap1.0019 avg	2.0 -723.1 -810.2 -810.0 -809.6 -809.9	1.22 997.05 981.81 952.03 976.78		490.84
55	sin2a193-2a-(18-10) trap2	3.7 -722.8 -819.1 -819.3 -819.2 -819.2	1.19 1978.68 2009.40 1993.98 1993.98		534 <i>.</i> 58

B-8

	July 13, 1991 STANDAR cool, cloudy sho [S]uM 2.900 7.265 29.058 72.645 290.580 1050.450	NDS Dwers Log[S] 0.462 0.861 1.463 1.861 2.463 3.021	mV -732.80 -744.80 -763.20 -775.40 -792.90 -809.70	F Constant Std Err of R Squared No. of Obs Degrees of X Coeffici Std Err of	egression Y Est ervations Freedom ent(s)	Output: -719.05 0.27228 0.99993 6.00 4.00 -30.05 0.13
	Y = -30.0491(logX) logX = (Y + 719.053)/	- 719.0 -30.0491	53		0061.	0.10
	JULY 13 1991					
No . 56	Sample ID wet wt (g) sin194-a100 (blank)	g or mV	[S]	SpkAdd	Recovery Neas/Add	um/g or Recovery Sys/Ben
	ci api	-703.2	0.47			
57	benchspike 1147 1148 1149 avg 1240	-810.9 -811.2 -811.1 -811.1 -811.1	1139.11 1165.60 1156.70 1153.75 1139.11	1211.0 uM	spike 0.95	
58	systemspike1100 trap2 trap1.1151 trap1.1153 trap1.1159 avg 1240	1211.0 -707.2 -811.2 -811.1 -810.9 -811.1 -810.5	uM Spike 0.40 1165.60 1156.70 1139.11 1153.75 1104.72		1.00 0.97	0.95
59	sin2a193-a1(0-10)a trap2 trap1.1258 1300 1302 avg	3.5 -709.7 -808.7 -808.9 -808.7 -808.8	0.49 962.39 977.25 962.39 967.32			277.96
60	sin2a193-a1(0-10)b trap2 trap1.1346 1348 1351 avg	2.9 -711.9 -805.5 -805.7 -805.5 -805.6	0.58 753.11 764.74 753.11 756.97			261.02
61	sin2a193-a(0-10)cspk trap2 trap1.1430	0.9 -710.3 -810.5 -810.8	0.51 1104.72 1130.41 0.00	1211.0	0.84 usin	g sys
62	sin1b193-a3(0-10)a trap2 trap1 trap1.1521	4.8 -710.2 -793.3 -793.1 -793.2	0.51 295.70 291.21 293.45 0.00			61.65

63	systemspike1500 trap2=trap1 1604 1606 1610	-810.0 -809.7 -809.5 -809.0 -809.6	1063.20 1039.03 1023.23 984.77 1027.16 0.00	1211.0 uM s	pike
	JULY 16 1991, STANDAR	D CURVE	1919; cl	oudy with clearing	, cool temp
	[S]uM 10 2.900 7.265 29.058 72 645 290.580 1050.450 Y = -29.6386(logX) logX = (Y + 723.163)/	• g[S] 0.462 0.861 1.463 1.861 2.463 3.021 - 723.1 -29.6386	EV -736.80 -748.60 -766.70 -778.50 -796.00 -812.70	Regressio Constant Std Err of Y Est R Squared No. of Observation Degrees of Freedom X Coefficient(s) Std Err of Coef.	n Output: -723.16 0.15816 0.99998 s 6.00 4.00 -29.64 0.07
No. 64	Sample ID wet wt (g) bench spike:1915 1917 1920 1928 avg	g or wV -812.0 -812.3 -812.1 -812.2	[S] 993.90 1017.33 1001.65 1009.46	Recover SpkAdd Meas/Ad 1211.0 uM spike	uM/g or y Recovery d Sys/Ben
65	sys-spike sys1-trap2 sys1-trap1	-468.3 -812.1	0.00 1001.65	1211.0 uM spike	0.99
66	sys2-trap2 sys2-trap1	-712.5 -811.9 -812.0 -812.1	0.44 986.21 993.90 1001.65		
	ary	-01210	aa 3.80	0.98	0.99

JULY 17, 1991: partly cloudy, temp still cool

	[S]uN 7 29 72 291 1052	log[S] .911 0.464 .280 0.862 .118 1.464 .795 1.862 .180 2.464 .616 3.022	EV -735.90 -748.70 -766.60 -778.40 -795.60 -812.40	Regressio Constant Std Err of Y Est R Squared No. of Observation Degrees of Freedom X Coefficient(s) Std Err of Coef.	m Output: -722.73 0.47312 0.99978 s 6.30 4.00 -29.71 0.22
No. 67	Sample ID wet w benchspike 1054	g or t (g) mV -801.4 -801.2	[S] 457.00 449.80	Y = -29.0348(lo logX = (Y + 724.17 Recover SpkAdd Meas/Ad 485.30 uM spike	gX)-724)/-29.034 uM/gor y Recovery d Sys/Ben
	avg 1253	-801.0 -801.2 -801.5	442.75 449.80 460.63	0.93 0.95	
68	sys1spike trao2	-587.0 -801.4 -801.2 -801.0	0.00 457.00 449.80 442.73	485.3 uM spike	
	avg 1253	-801.2 -801.2	449.80 449.80		1.00 0.98
69	sys2spike trap2 trap1	-711.7 -801.2 -801.3 -801.0	0.37 449.80 453.39 442 73	485.3 uM spike	
	avg 1253	-801.2 -800.5	448.62 425.51	0.97 0.92	1.00 0.95
70	sin4b197-a2(0-2) trap2 trap1)a 2.3 -515.8 -785.0 -785.2	0.00 124.47 126.46		54.79 54.79
	avg 1354	-785.1 -785.1	125.46 125.46		
71	sin4b197-a2(0-2) trap2 trap1)b 1.5 -710.9 -779.3 -779.6 -779.5	0.35 79.21 81.11 80.47		53.15 52.87
	avg 1354	-779.5 -779.4	80.26 79.84		
72	sin4b197-a2(4-6) trap2 trap1	a 1.2 -585.1 -785.4 -785.3	0.00 128.48 127.47		105.76 107.03
	avg 1520	-785.4 -785.5	127.98 129.51		

73	sin4b197-a2(4-6)b trap2 trap1 avg	2.3 -710.2 -795.4 -795.7 -795.5	0.33 283.96 290.80 297.36	123.86 123.37
	1520	-795.5	286.22	
74	sin4b197-a2(8-10) trap2 trap1	3.5 -548.2 -797.1 -797.3 -797.4	0.00 324.95 330.14 332.77	95.44
	avg	-797.3	329.27	
75	sin4b197-a2(20-22) trap2 trap1	3.8 -711.5 -792.8 -792.6 -792.6 -792.7	0.37 231.05 227.42 227.42 228.62	60 <i>.</i> 16
76	nindh107 at(0,10)a		220.02	
70	sin4bi97-ai(0-10)a trap2 trap1 avg	5.5 -674.0 -806.6 -806.7 -806.7 -806.7	0.02 690.25 695.75 695.75 693.91	126.86
77	sin4b197-a1(0-10)b trap2 trap1	4.9 -711.4 -806.4 -806.7	0.36 679.39 695.75	138.99
	avg	-806.5	684.80 686.61	
78	sin3b197-a1(0-2)a trap2 trap1 avg	2.8 -594.1 -775.9 -776.1 -776.0 -776.0	0.00 60.49 61.45 60.97 60.97	22.17
79	sin3b197-a1(0-2)b trap2 trap1	2.9 -711.6 -776.7 -777.0 -776.8 -776.8	0.37 64.45 66.00 64.96 65.13	22.85
80	sin3b197-a1(4-6)a trap2 trap1	2.3 -567.7 -786.8 -786.7 -786.3	0.00 143.57 142.44 137.99	62.25 63.75
	avg 2055	-786.6 -786.9	141.31 144.71	
81	sin3b197-a1(4-6)b trap2 trap1 avg	2.5 -710.7 -788.4 -788.6 -788.6 -788.5	0.34 162.99 165.60 165.60 164.73	66.42 65.72
	2055	-788.4	162 99	

B-12

82	sin3b197-a1(8-10) trap2 trap1	3.8 -593.7 -768.5 -768.9 -768.6	0.00 33.63 34.72 33.90			8.95
	avg	-768.7	34.08			
83	sin3b197-a1(20-22) trap2 trap1	3.9 -708.8 -794.8 -795.0 -794.5	0.30 270.77 275.10 264.40			68.54
	avg	-794.8	270.05			
84	sin3b197-a2(0-10)a trap2 trap1 avg 2350	2.6 -632.1 -776.9 -777.4 -777.3 -777.2	0.00 65.48 68.13 67.59 67.05 66.52			25.49
	2000	-111.1	00.52			
85	sin3b197-a2(0-10)b trap2 trap1	5.2 -707.7 -784.9 -785.0	0.27 123.49 124.47 127.47			24.16
	avg 2350	-785.1 -784.3	125.13			
86	sin5b197-a2(0-10)a trap2 trap1	3.3 -680.4 -809.2 -809.3 -809.5	0.00 0.03 848.31 855.06 868.73			258.23
	avg	-809.3	857.33			
87	sin5b197-a2(0-10)b trap2 trap1	3.6 -708.4 -811.3 -811.2 -811.0	0.29 1002.03 994.12 978.47			272.39
	avg	-811.2	991.49			
88	sin198-a(0015)a 0101	-799.5 -799.7 -799.6 -799.6	393.07 399.36 396.20 396.20	485.3	uM spiko	e system1 0.86
80	sin198.2(0015)h			A95 3	uM anti-	
03	0106	-799.0 -799.1 -799.0 -799.0	377.79 380.80 377.79 378.79	400.0	um spik(e systemz
		100.0	010113			0.02

	JULY 18, 1991 [S]uW 2.911 7.280 29.118 72.795 291.180 1052.616	Sunny, ge log[S] 0.464 0.862 1.464 1.862 2.464 3.022	tting war W -735.50 -748.00 -766.00 -778.20 -795.60 -812.50	rmer F Constant Std Err of R Squared No. of Obs Degrees of X Coeffici Std Err of Y = -29 logX = (Y	egression Y Est ervations Freedom ent(s) Coef. 9.9932(log) + 721.951)	Output: -721.95 0.32663 0.99990 6.00 4.00 -29.99 0.15 () - 721 () -29.99
No.	Sample ID wet wt (g)	g or mV	[S]	SpkAdd	Recovery Neas/Add	uM/g or Recovery Sys/Ben
90	Dench Spike	-798.0 -798.9 -799.0 -798.6	343.20 367.75 370.58 360.29	403.0	0.74	
91	sysspike1048-sys1 trap2 trap1	-525.8 -798.8 -798.9 -799.0	0.00 364.93 367.75 370 58	10.0 ml	july 17 sp	bike
	avg	-798.9	367.75		1.02	
92	sysspike1048-sys2 trap2 trap1	-708.8 -798.8 -798.9 -798.9	0.36 364.93 367.75 367.75	10.0 ml	.july 17 s	spike
	avg	-798.9	366.81		1.02	1.00
93	sin5b197-a1(0-2)a trap2 trap1 avg	1.2 -625.3 -788.7 -789.1 -789.0 -788.9	0.00 168.06 173.30 171.98 171.10			148.78
94	sin5b197-a1(0-2)b trap2 trap1 avg	2.9 -708.6 -803.7 -803.9 -803.8 -803.8	0.36 531.60 539.82 535.70 535.70			186.65
95	sin5b197-a1(4-6)a trap2 trap1 avg	2.8 -648.0 -819.4 -819.5 -819.6 -819.5	0.00 1774.34 1788.01 1801.79 1788.01			650.19
96	sin5b197-a1(4-6)b trap2 trap1 avg	2.5 -708.0 -819.1 -819.2 -819.0 -819.1	0.34 1733.94 1747.30 1720.68 1733.94			682.65
97	sin5b197-a1(8-10) trap2 trap1	2.1 -678.1 -819.9 -820.2 -820.0	0.03 1843.77 1886.73 1857.98		874.53	
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	avg	-820.0	1862.74			
98	sin5b197-a1(20-22) trap2 trap1	3.2 -710.0 -819.8 -819.8 -819.6 -819.7	0.40 1825.46 1829.67 1801.79 1818.93		573.80	
99	sin2b199-a2(0-2)a trap2 trap1 avg	2,6 -689.0 -787.6 -787.7 -787.8 -787.7	0.08 154.45 155.64 156.84 155.64		59.41	
100	sin2b199-a2(0-2)b trap2 trap1	2.5 -707.0 -807.2 -807.3 -807.3	0.32 695.47 700.83 700.83		284.16	
	avg	-807.3	699.04			
101	sin2b199-a2(4-6)a trap2 avg	1.3 -799.4 -799.5 -799.6 -799.5 -799.2	0.00 382.14 385.08 388.05 385.08 376.31		289.54	
102	sin2b199-a2(4-6)b trap2 avg	1.2 -708.9 -801.1 -801.2 -801.2 -801.2 -800.4	0.37 435.41 438.77 438.77 437.64 412.63 0.00		364.70	
103	bench spike 2300	-802.5 -802.6 -802.7 -802.8	484.81 488.55 492.32 496 11	485.3 uM spike		
	avg 2200	-802.7	490.43 470.15			

	JULY 19, 1991: clea	r and su	nny, still	. cool		
	[S]uM 1 2.911 7.280 29.118 72.795 291.180 1052.616	og[S] 0.464 0.862 1.464 1.862 2.464 3.022	■V -735.40 G -747.60 S -765.70 R -777.70 N -795.20 D -812.00	R constant itd Err of Squared lo. of Obs legrees of Coeffici	egression Y Est ervations Freedom ent(s)	Output: -721.78 0.23592 0.99995 6.00 4.00
			5	TO ERR OT	COET.	0.11
			1	.ogX = (Y	+ 721.781)	/-29.87
No. 104	Sample ID wet wt (g) bench spike 7/18 210	g or w V - 802.5 - 802.7	[S] 503.19 511.00	SpkAdd 485.3	Recovery Meas/Add	uM/g or Recovery Sys/Ben
105	sysspike0930 trap2 trap1	-659.0 -802.1 -802.3 -802.5	0.01 487.91 495.49 503.19	485.3		
	avg	-802.3	495.49		1	. 02
106	sysspike0930-sys2 trap2 trap1	-708.4 -802.2 -802.5 -802.4	0.36 491.69 503.19 499.32	485.3 uM	spike	
	avg	-802.4	498.04		1.03	1.01
107	sin1a199-a1(0-2)a trap2 trap1 avg	7.2 -679.5 -743.2 -743.4 -743.3	0.04 5.21 5.29 5.25 0.00		0.73	
108	sin1a199-a1(0-2)b trap2 trap1 avg	7.9 -706.7 -747.5 -747.9 -747.8 -747.7	0.31 7.26 7.49 7.43 7.39		0.94	
109	sin2b197-a1(0-10)a trap2 trap1 avg	4.2 -678.0 -797.0 -797.4 -797.3 -797.2	0.03 329.34 339.65 337.04 335.31		80.80	
110	sin2b197-a1(0-10)b trap2 trap1	5.8 -704.7 -801.7 -801.9 -801.9	0.27 473.10 480.45 480.45		81.99	
	avg	-801.8	477.99			

B-16

111	sin1a199-a2(0-10)a trap2 trap1 avg	14.0 -680.0 -779.0 -779.5 -779.5 -779.3	0.04 82.26 85.49 85.49 84.40	6.05	
112	sin1a199-a2(0-10)b trap2 trap1 avg	7.3 -704.7 -773.7 -773.9 -773.5 -773.7	0.00 0.27 54.67 55.52 53.84 54.67	7.53	
113	sin2b199-a2(20-22) trap2 trap1 avg	4.3 -688.4 -821.0 -821.0 -820.8 -820.9	0.08 2093.81 2093.81 2061.79 2083.08	483.31	
114	sin2b199-a2(4-6) trap2 trap1 avg	4.2 -708.3 -809.1 -809.2 -809.1 -809.1	0.35 836.83 843.30 836.83 838.98	202.16	
115	sin1a199-a1(8-10) trap2 trap1 avg	6.0 -691.8 -782.6 -783.0 -783.2 -782.9	0.10 108.56 111.96 113.70 111.38	18.50	
116	sin2b199-a2(8-10) trap2 trap1 avg	5.4 -707.4 -819.4 -819.5 -819.5 -819.5	0.33 1850.91 1865.23 1865.23 1860.44	341.99	
117	sinta199-a1(4-6) trap2 trap1 avg	10.1 -684.6 -767.2 -767.7 -767.9 -767.6	0.06 33.13 34.43 34.97 34.17	3.40	
118	sin2b199-a2(0-2)c trap2 trap1 avg	8.2 -710.8 -814.3 -814.5 -814.4 -814.4	0.43 1249.35 1268.76 1259.01 1259.01	154.10	59.40
119	benchspike1713 1716	-801.5 -801.7 -801.6 -801.6	465.86 473.10 469.47 469.47	483.5 uM spike	
	avg	-801.6	469.47		

120	benchspike2100 7/18		0.00			
	1720	-802.6	507.08			
		-803.0	522.95			
		-803.1	527.00			
	avg	-802.9	518.94			
121	sysspike1730sys1			483.5 uM spike		
	trap2	-570.9	0.00	• · ·		
	trapi	-801.3	458,74			
		-801.7	473.10			
		-801.6	469.47			
	avg	-801.5	467.06	0.97	0.99	
122	sysspike1730sys2			483.5 uM spike		
	trap2	-708.1	0.35			
	trapi	-801.9	480.45			
		-802.2	491.69			
		-801.8	476.76			
	2VQ	-802.0	482.92	1.00	1.03	1.03

Appendix C

RAW DATA OF SEM CONCENTRATIONS OF Cu, Pb, AND Zn

SEM concentrations of Cu, Pb, nad Zn in sediment samples collected from Sinclair Inlet, WA, were analyzed on January 24–30, 1992, at the Ocean Sciences Laboratory of NCCOSC, RDT&E Division, San Diego, CA. Table B-1 provides the raw and calculated data for copper. Table B-2 provides the raw and calculated data for lead and zinc. A Summary of all the SEM data is provided in table B-3. Table C-1. Raw and calculated data for the determination of SEM concentrations of copper measured by GFAA. For each batch the the extract (C_{extract}), the weight of the sample (G_w), the dry to wet weight ratio of the sample (R), the centrifuged volume (C_{vol}), the weight of storage vial and sample (W_{sto}), the calculated volume of the extract (V_{ext}), the mass of the metal in the extract (M_{ext}), and the concentration of the sediment sample (Csed) in µg/g and µmol/g (dry weight) are given. (A) Batch number RUN1 24 JAN 1992. (B) Batch number RUN2 25 JAN 1992. (C) Batch number RUN3 25 JAN 1992. (D) Batch number RUN4 25 JAN 1992. Quality sample sequence number (seq#), the sample identification number (1D#), the station location (Sta), the core depth of sample in cm curve (CONC), the concentration (ppb) of the second dilution (d₂), the concentration of the first dilution (d₁), the concentration of (depth), the absorbance units measured with the GFAA (ABS), the concentration (ppb) determined from the regression standard Assurance (QA) data are presented for each batch.

(A) (OPPER	RUNI 2	4 JAN 1	992	CONC) = (ABS-I	3LANK)/0.	()02542		BLANK	= 0.004					
					CONC	d2	ں م	extract	°,	2	C _{vol}	Weta	Vext	Mavi	Cerd	Cert
*pas	1D#	Sta	depth	ABS	[bpb]	[ddd]	[ddd]	[mdd]	8	d/w	Į	In	Ĩ	рв	pg/g	pumol/g
-	NB	Blank		0.006	2.36											
2	8	Blank		0.004	1.57											
3	-	4a	4-6h	0.220	84.97	849.72	4248.62	4.25	4.14	0.3071	45.0	30.40	71.13	302.20	237.69	3.74
4	2	3a	02h	0.367	142.80	1428.01	7140.05	7.14	5.96	0.1957	45.0	22.63	61.59	439.73	377.01	5.93
v,	e	41	0-2a	0.070	25.96	259.64	1298.19	1.30	6.82	0.6376	45.0	23.45	61.55	79.90	18.37	0.29
9	Ś	\$	0-2c	0.431	167.98	1679.78	8398.90	8.40	8.17	0.2947	45.0	23.77	60.52	508.32	211.12	3.32
7	6	la	0-2a	0.030	10.23	102.28	511.41	0.51	7.18	0.7123	45.0	27.58	65.30	33.39	6.53	0.10
œ	14	Sa	0-2a	0.172	60.09	660.90	6608.97	6.61	6.97	0.3686	45.0	32.19	70.09	463.25	180.31	2.84
6	7	FB190	2300	0.026	8.65	86.55	432.73	0.43								
10	œ	3a	4-6а	0.248	95.99	959.87	4799.37	4.80	4.01	0.233	6 45.0	31.05	16.17	345.10	368.41	5.80
Ξ	6	la	0-10	0.068	25.18	251.77	1258.85	1.26	13.96	0.7688	45.0	41.05	71.96	90.58	8.44	0.13
12	01	FB193	1800	0.007	1.18	11.80	59.01	0.06								
13		2a	0-10	0.172	60.09	660.90	3304.48	3.30	3.48	0.3779	45.0	33.80	75.17	248.40	188.88	2.97
14	DUP12	3a	010	0.276	107.00	1070.02	5350.12	5.35	5.82	0.3377	45.0	38.79	77.81	416.27	211.80	3.33
15	12	За	0-10	0.279	108.18	1081.83	5409.13	5.41	5.82	0.3377	45.0	38.79	77.81	420.86	214.13	3.37
16	13	4a	0-10	0.422	164.44	1644.37	8221.87	8.22	10.90	0.4349	45.0	39.49	73.45	603.88	127.39	2.00
17	14	5a	0-10	0.355	138.08	1380.80	6904.01	6.90	7.27	0.4245	45.0	38.36	75.94	524.26	169.88	2.67
18	15	4a	0-2h	0.201	77.50	774.98	3874.90	3.87	4.14	0.2402	45.0	31.68	72.40	280.55	282.13	4.44
61	4	5a	0-2a	0.341	132.57	1325.73	6628.64	6.63	6.97	0.3686	45.0	32.19	70.09	464.63	180.85	2.85
20	SI	120ref		0.344	133.75											
21	B2	Blank		0.003	1.18											
) VÒ	heck:															
	14 D11P	actual/n	eference	1-161/6	1.5	72/.173)i i	99.4% 1.6%								
	2			· _ (+ 1 / 7	1	- (117. MM	-	11010								

14actual/reference.172/.173DUP:(DUP12/12) - 1(.276/.279) - 1S1 120:actual/reference.344/.332BLANKS:values $\leq 2x(0.004)$

= 103.6% YES

Table C-1. Continued.

umol/g Csed 9.75 1.16 1.44 6.99 1.45 4.42 0.39 0.12 9.43 4.14 0.43 1.84 4.22 6.21 92.40 268.36 280.99 77.39 116.68 91.62 24.95 599.25 263.06 27.11 73.85 444.34 619.73 394.71 C sed µg/g 32.19 2008.23 246.15 58.22 2076.87 76.07 175.45 152.55 583.88 235.08 538.05 111.81 91.67 917.24 Mext µg 56.28 56.54 49.59 70.30 69.56 51.03 56.28 65.59 71.62 59.84 57.55 53.12 69.43 a K V 26.17 29.62 41.56 27.57 27.60 16.17 39.23 23.68 24.97 40.75 29.62 40.23 36.70 39.23 W sto ml 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 35.0 BLANK = 0.0040.5895 0.3958 0.2307 0.2336 0.2336 0.6102 0.7829 0.3633 0.3044 0.3044 0.3899 0.5801 0.4072 0.4072 2 4.76 2.61 4.94 5.47 3.32 1.51 3.75 6.02 10.06 8.23 5.63 5.96 *28.23 یہ ٹی 4.6% CONC = (ABS - BLANK)/0.002542C_{extract} 98.8% 106.0% = 101.5% 18.45 1.14 0.06 1.06 0.02 1.87 0.65 3.34 3.50 1.32 35.68 13.99 19.20 18.00 18.15 3.05 2.87 0.02 0.04 7.75 YES ŧI 11 11 3343.82 1140.83 1868.61 649.10 39.34 59.01 19.67 19.67 3501.18 1317.86 13985.05 19197.48 17997.64 18155.00 062.16 3048.78 2871.75 35680.57 7749.80 18450.04 [ddd] 10326.51 Ð (.174/.182) -11.80 3839.50 7.87 668.76 228.17 [dqq] 212.43 373.72 609.76 574.35 2065.30 3.93 129.82 700.24 263.57 3599.53 3.93 7136.11 3631.00 1549.96 2797.01 3690.01 .171/.173 352/.332 .505/.498 ď CONC [qdd] 57.44 206.53 55.00 279.70 1.18 383.95 59.95 363.10 369.00 0.39 12.98 0.79 66.88 70.02 26.36 22.82 97.09 21.24 0.39 37.37 60.98 713.61 38.47 values ≤ 2x(0.004) (DUP27/27) - 1 actual/reference actual/reference actual/reference 0.919 0.001 0.159 0.529 0.037 0.174 1.818 0.980 0.004 0.352 0.058 0.150 0.005 0.002 0.182 0.062 0.398 0.715 0.505 0.003 0.927 0.942 ABS 0.071 65.70 -0.001 (B) COPPER RUNI 25 JAN 1992 0-10 0.171 0-10 0-2b 8-10 8-10 8-10 8-10 8-10 depth 0-10 0-10 0-10 0015 4-6b 494 8-10 8-10 4-6a 4 18 FB200 1700b 20 3b FB192 1210 21 14 Reference 14 Reference 10 23 56 11 24 FB198 11 24 FB198 13 26 FB197 14 DUP27 3 15 27 16 28 16 28 17 29 17 29 rerun 30(5ul) 5a rerun 30(5ul) 5a 5a Sta 4 33a 33a 4a 4a **180ref** 64 5a rerun 30(5ul)5a **BLANKS**: **B2** Blank Instr. zero S1 120ref **B** Blank S1 120: SI 180: QA Check: DUP: 1D# (avg)*2 1 33 S1 4 5 3 Seque 20 55 <u>6</u> C 4 Ś 5

* Sample number 30 was off scale. At the end of the batch, the sample was rerun using a 5-µl sample volume. The average of the 3 reanalyses was multiplied by 2 to obtain a comparable concentration.

							Ta	ble C-1. (Continue	d.						
(C)	OPPE	R RUN3	25 JAN	1992	CON	C = (ABS	176)/0.0	02542 + 6	9.23682	-	(NO BLA	NK COR	RECTIO	(z		
#bas	1D#	Sta	depth	ABS	CONC [ppb]	d ₂ [ppb]	d, [ppb]	C _{extract} [ppm]	ഄഀഄ	R d/w		W _{sto}	V ml	M ext µg	C _{sed} µg/g	C _{sed} µmol/g
	£	İnstr. Rlank	Zero	0.000	27.6											
- ~1	25	120ref		0.389	153.03											
ŝ	33	2h	8-10	0.213	83.79	837.92	4189.61	4.19	5.44	0.3786	35.0	37.33	66.73	279.58	135.75	2.14
4	34	FB200	1730a	0.003	1.18	11.80	59.01	0.06								
S	35	Sb	4-6a	0.375	147.52	1475.22	7376.08	7.38	5.96	0.3899	35.0	28.82	57.75	425.94	183.29	2.88
Q	36	5h	0-2a	0.103	40.52	405.19	2025.96	2.03	1.15	0.2821	35.0	26.57	60.29	122.15	376.53	5.93
7	37	4b	4-6a	0.038	14.95	149.49	747.44	0.75	1.21	0.2828	35.0	29.47	63.12	47.18	137.87	2.17
œ	38	1b	20-22	0.135	53.11	531.08	2655.39	2.66	5.95	0.7566	35.0	38.39	67.28	178.65	39.68	0.62
6	39	3a	8-10b	0.140	55.07	550.75	2753.74	2.75	2.16	0.2574	35.0	37.99	70.65	194.56	349.93	5.51
01	4	Referen	ce	0.188	73.96	739.58	3697.88	3.70								
-	40	2h	20-22	0.494	194.34	1943.35	9716.76	9.72	4.31	0.4112	35.0	40.79	71.30	692.79	390.91	6.15
12	41	4h	8-10	0.084	33.04	330.45	1652.24	1.65	3.45	0.3344	35.0	32.24	63.65	105.16	91.15	1.43
E	42	FB194	0011	0.003	1.18	11.80	59.01	0.06								
14	43	1	8-10a	0.049	19.28	192.76	963.81	0.96	4.76	0.5895	35.0	40.91	70.97	68.40	24.38	0.38
15	44	3a	02a	0.136	53.50	535.01	2675.06	2.68	2.84	0.1957	35.0	39.24	71.22	190.51	342.78	5.39
16	45	5a	4-6b	0.453	178.21	1782.06	8910.31	8.91	4.47	0.3899	35.0	42.83	73.17	651.96	374.08	5.89
17	46	4b	20-22	0.038	14.95	149.49	747.44	0.75	3.80	0.4145	35.0	28.30	59.38	44.38	28.18	0.44
18 DI	JP47	2a	4-6a	0.124	48.78	487.80	2439.02	2.44	2.04	0.3551	35.0	28.50	61.33	149.58	2()6.49	3.25
16	47	2a	4-6a	0.128	50.35	503.54	2517.70	2.52	2.04	0.3551	35.0	28.50	61.33	154.41	213.15	3.35
20	48	2b	4-60	0.253	99.53	995.28	4976.40	4.98	4.15	0.3480	35.0	30.32	61.04	303.76	210.33	3.31
21	49	2a	0-2a	0.327	128.64	1286.39	6431.94	6.43	6.81	0.3209	35.0	38.93	66.96	430.68	197.08	3.10
22	50	5b	20-22	0.481	189.22	1892.21	9461.05	9.46	3.17	0.5095	35.0	16.29	48.05	454.65	281.50	4.43
23	51	4a	02a	0.267	105.04	1050.35	5251.77	5.25	5.40	0.2402	35.0	31.75	61.22	321.51	247.87	3.90
24	52	Sh	8-10a	0.529	208.10	2081.04	10405.19	10.41	2.13	0.4455	35.0	21.25	54.02	562.14	592.40	9.32
25	53	4a	20-22	0.353	138.87	1388.67	6943.35	6.94	6.45	0.4912	35.0	43.23	71.60	497.12	156.91	2.47
26	SIL	80ref		0.549	215.97											
27	B 3 E	3lank		0.005	1.97											
28	S1 L	80ref		0.562	221.09											
av	g S1 18	0		0.556	218.53											
0 V C	heck:															
	14:		actual/r	eference	-	88/.173	"	108.7%								
	DUP:		(DUP4	7/47) - 1		24/.128) -	"	3.1%								
	SI 120:		actual/r	eference	Ē	89/.332	I	117.2%	* *							
-	SI 180:		actual/r	eference	ŝ	56/.498	11	111.6%								
	BLANK	S:	values <	2×(0.004	<u>-</u>		7	ES								(Contd)
** Ex	ceeded	OA crite	ria.													•

C-4

)(a)	COPPEI	R RUN4 2	I NAL 2	992	CONC	(ABS)	0.002542				(NO BL	ANK. CO	RRECTI	(NO		
#bəs	1D#	Sta	depth	ABS	CONC [ppb]	d, [ppb]	d, b bbb	cextract [ppm]	6° 00	R w/b		W _{sto}	V ext ml	M _{ext} µg	C _{sed} µg/g	C _{sed} µmol/g
0-0	lnstr. B S I	Zero Blank 120ref		0.002 0.006 0.410	2.36 161.29											
÷.	54	Sa	8-10a	1.697	667.58	6675.85	33379.23	33.38	6.33	0.4072	35.0	21.12	49.72	1659.50	643.82	10.13
4	55	За	8-10a	0.333	131.00	1309.99	6549.96	6.55	4.20	0.2574	35.0	43.56	74.16	485.77	449.34	7.07
ŝ	56	3 b	8-10	0.218	85.76	857.59	4287.96	4.29	3.81	0.5225	35.0	31.81	62.86	269.54	135.40	2.13
ę	57	3b	4-6a	0.141	55.47	554.68	2773.41	2.77	2.27	0.5123	35.0	60.61	51.74	143.49	123.38	1.94
7	58	4h	4-6b	0.074	29.11	291.11	1455.55	1.46	2.32	0.2828	35.0	27.70	60.25	87.70	133.67	2.10
œ	59	4b	02a	0.052	20.46	204.56	1022.82	1.02	2.29	0.2307	35.0	23.38	55.99	57.26	108.39	1.71
6	14 Ref	erence		0.198	77.89	778.91	3894.57	3.89								
10	60	2p	0-2a	0.135	53.11	531.08	2655.39	2.66	2.62	0.2947	35.0	29.48	61.73	163.91	212.28	3.34
=	61	2 a	8-10a	0.149	58.62	586.15	2930.76	2.93	5.44	0.3786	35.0	30.87	60.30	176.74	85.81	1.35
12	62	За	20-22	0.244	95.99	959.87	4799.37	4.80	4.59	0.1727	35.0	40.83	71.06	341.04	430.23	6.77
9	63	2a	18-20	0.162	63.73	637.29	3186.47	3.19	3.73	0.3838	35.0	34.71	65.83	209.75	146.52	2.31
14	£	5a	0-2b	0.462	181.75	1817.47	9087.33	9.09	7.87	0.3686	35.0	31.52	58.53	531.90	183.36	2.89
15	65	ЗЬ	0-2a	0.154	60.58	605.82	3029.11	3.03	2.75	0.3639	35.0	28.69	60.81	184.20	184.07	2.90
16	DUP66	5 2h	4-6a	0.081	31.86	318.65	1593.23	1.59	1.20	0.3480	35.0	20.26	53.97	85.98	205.89	3.24
17	6 6	2b	4- 6a	0.082	32.26	322.58	1612.90	1.61	1.20	0.3480	35.0	20.26	53.97	87.04	208.43	3.28
18	16 FE	3192 2000	_	0.027	10.62	106.22	531.08	0.53								
19	67	3a	30-32	0.471	185.29	1852.87	9264.36	9.26	5.25	0.1727	35.0	40.17	69.75	646.15	712.66	11.21
20	68	Зh	20-22	0.126	49.57	495.67	2478.36	2.48	4.59	0.1727	35.0	25.80	56.10	139.05	175.41	2.76
21	SI	180ref		0.574	225.81											
22	B 3	Blank		0.004	1.57	15.74	78.68	0.08								
rerun	54(5ul)5a	8-10a	0.908	357.20	3571.99	17859.95	17.86								
rerun	54(5ul)5a	8-10a	0.840	330.45	3304.48	16522.42	16.52								
rerun	54(5ul)5a	8-10a	0.809	318.25	3182.53	15912.67	15.91								
rerun	54(5ul)5a	8-10a	0.858	337.53	3375.30	16876.48	16.88								
avg	54(5ul))5a	8-10a	0.854	335.86	3358.58	16792.88	16.79 *2	2 6.33	0.4072	35.0	21.12	49.72	1669.76	647.80	10.19
rerun	S1 120)ref		0.411												
rerun	121 120	Jret		17.5.0												
0 VO	Theck:															
I	14:		actual/r	eference	51.	9.173	H	114.5%								
	DUP:		(DUP6	5/66) - 1	30.)	31/.082) -	"	1.2%								
	SI 120		actual/r	eference	4	10/.332	11	123.5%*	*							
	SI 180. Ri Ani	ن لا	actual/r	eference	75. M	14/.498	11 ≥	115.3% FS								
			-				-	3								
 San 	nple num	ther 54 wa	s off scale	. At the e	nd of the b	atch the sat	nple was rei	un usinga 5	i-jul sample	e volume. 1	The averag	e of the 4	reanalyses	was multif	ofied by	

Table C-1. Continued.

2 to obtain a comparable concentration.

** QA criteria exceeded.

5

(Contd)

C-5

sediment sample (Csed) in µg/g and µmol/g (dry weight) are given. (A) Lead batch number RUN1 29 JAN 1992. (B) Lead batch number RUN2 30 JAN 1992. (C) Zinc batch number RUN1 30 JAN 1992. Quality Assurance (QA) data are presented for each batch. extract (Cextract), the weight of the sample (Gw), the dry to wet weight ratio of the sample (R), the centrifuged volume (Cvol), the weight of storage core depth of sample in cm (depth), the absorbance units measured with the GFAA (ABS), the concentration (ppb) determined from the regression standard curve (CONC), the concentration (ppb) of the second dilution (d_2), the concentration of the first dilution (d_1), the concentration of the Absorption (GFAA). For each hatch the sample sequence number (seq#), the sample identification number (ID#), the station location (Sta), the vial and sample (W_{sto}), the calculated volume of the extract (V_{ext}), the mass of the metal in the extract (M_{ext}), and the concentration of the Table C-2. Raw and calculated data for the determination of SEM concentrations of lead and zinc measured by Graphite Furnace Atomic

E	LEAD RI	UNI 29	IUN 92		CONC	C = ABS/0	.001185	C	NO BLA	NK COR	RECTION	.				
					CONC	d,	d, C	extract	°°	R	C _{vol}	W _{sto}	Vext	Mext	C _{sed}	C _{sed}
seq#	#CII	Sta	depth	ABS	[ddd]	[qdd]	[qdd]	[mdd]	8	d/w	lm	Im	m	Bri	µ2/2	µmol/g
_	MB	Blank		0.001	0.84											
7	6	1a	0-10	0.065	54.85	548.52	2742.62	2.74	13.96	0.7123	35.0	41.05	61.96	169.92	17.09	0.08
m	10	FB190	1800	0.001	0.84	8.44	42.19	0.04								
4	11	2a	0-10	0.077	64.98	649.79	3248.95	3.25	3.48	0.3779	45.0	3.80	75.17	244.22	185.71	0.00
ŝ	12	За	0-10	0.103	86.92	869.20	4345.99	4.35	5.82	0.3377	45.0	38.79	77.81	338.14	172.05	0.83
9	13	4a	0-10	0.233	196.62	1966.24	9831.22	9.83	10.90	0.4349	45.0	39.49	73.45	722.08	152.32	0.74
7	14	5a	0-10	0.271	228.69	2286.92	11434.60	11.43	7.27	0.4245	45.0	38.36	75.94	868.29	281.35	1.36
×	15	4a	0-2b	0.080	67.51	675.11	3375.53	3.38	4.14	0.2402	45.0	31.68	72.40	244.40	245.77	1.19
6	ĪW	120ref		0.171	144.30	1443.04	7215.19	7.22								
10	16	FB190	2000	0.003	2.53	25.32	126.58	0.13								
11	17	1b	0-10	0.047	39.66	396.62	1983.12	1.98	4.76	0.5895	35.0	41.56	71.62	142.03	50.61	0.24
12	18	FB200	1700	0.000	0.00	0.00	0.00	0.00								
13	61	5h	01-10	0.149	125.74	1257.38	6286.92	6.29	4.15	0.3679	35.0	32.67	63.38	98.45	260.98	1.26
14	20	3h	0-10	0.036	30.38	303.80	1518.99	1.52	2.61	0.5801	35.0	27.57	59.84	90.89	60.03	0.29
15	21	4h	0-10	0.067	56.54	565.40	2827.00	2.83	4.94	0.3044	35.0	27.60	57.55	162.69	108.19	0.52
16	22	4 1	0-10	0.076	64.14	641.35	3206.75	3.21	5.47	0.3044	35.0	23.68	53.12	170.34	102.30	0.49
17	DUP27	3a	4-6h	0.076	64.14	641.35	3206.75	3.21	3.75	0.2336	35.0	39.23	70.30	225.45	257.36	1.24
<u>e</u> .	27	3 a	4-6b	0.072	60.76	607.59	3037.97	3.04	3.75	0.2336	35.0	39.23	70.30	213.58	243.81	1.18
61	23	5h	010	0.188	158.65	1586.50	7932.49	7.93	3.32	0.3958	35.0	24.97	56.54	448.52	341.33	1.65
20	24	FB198	0015	0.000	0.00	0.00	0.00	0.00								
21	25	4h	0-2h	0.015	12.66	126.58	632.91	0.63	1.51	0.2307	35.0	16.17	49.59	31.38	90.09	0.43
22	26	FB190	1210	0.001	0.84	8.44	42.19	0.04								
23	S	2h	0–2c	0.125	105.49	1054.85	5274.26	5.27	8.17	0.2947	45.0	23.77	60.52	319.2	132.58	0.64
24	9	la	()-2a	0.029	24.47	244.73	1223.63	1.22	7.18	0.7123	45.0	27.58	65.30	19.90	15.62	0.08
53	4	5a	02a	0.173	145.99	1459.92	7299.58	7.30	6.97	0.3686	45.0	32.19	70.09	511.66	199.16	0.96
26	2	3a	0-26	0.100	84.39	843.88	4219.41	4.22	5.96	0.1957	45.0	22.63	61.59	259.86	222.79	1.08
27	09	2h	()-2a	0.037	31.22	312.24	1561.18	1.56	2.62	0.2947	35.0	29.48	61.73	96.37	124.81	0.60
28	B2	Blank		0.000	0.00	0.00	0.00	0.00								
VO	("heck:															
	DUP:		(DUP2	1-(727)-1	0.)	76/.072) -	11	4.6%								
	MI 120:		actual/1	eference		71/.171	n	106.0%								

(Contd)

YES

values ≤ 2x(0.001)

BLANKS:

Table C-2. Continued.

(B) LEAD RUN2 0201 JAN 30 1992. CONC = ABS/0.001185

(NO BLANK CORRECTION)

M _{ext} C _{sed} C _{sed} µg µg/g µmol/g		111.67 25.68 0.40	50.25 270.34		68 60 211 73		100102 130 31			(1'00) 10°C+C		227 31 175 25	216.98 167.28		
V ml		61.55	00.90 71.22	61 22	60.29		11 13	55 00	58.53	cr.oc		61 22	61.22		
W _{sto}		23.45	39.24	31.75	26.57		28 50	23.38	31 57	4		31.75	31.75		
		45.0 35.0	35.0	35.0	35.0		35.0	35.0	35.0			35.0	35.0		
R d/w		0.6376	0.1957	0.2402	0.2821		0.3551	0.2307	0 3686			0.24 12	0.2407		
یہ <mark>ٹ</mark> ی		6.82 6.81	2.84	5.40	1.15		2.04	2.29	7.87			5.40	5.40		
Cextract [ppm]	0.04 7.34	18.1	2.11	3.63	1.14		1.65	0.84	9.32	10.46	0.00	3.71	3.54		101.8%
[bb.]	42.19 7341.77	1814.35	2109.70	3628.69	1139.24		1645.57	843.88	9324.89	10464.14	0.00	3713.08	3544.30		#
d, [ppb]	8.44 1468.35	362.87 978 90	421.94	725.74	227.85		329.11	168.78	R64.98	2092.83	0.00	742.62	708.86		4/.171
CONC	0.84 146.84	36.29 97.89	42.19	72.57	22.78		32.91	16.88	186.50	209.28	0.00	74.26	70.89		.17
ABS	0.001 0.174	0.043	0.050	0.086	0.027	0.51	0.039	0.020	0.22:	0.248	0.000	0.088	0.084		eference
depth		0-2a 0-2a	0-2a	0-7a	0-2a	~	4-fia	()2a	0-2b			()2a	8° -		(lost) actual/r
Sta	Blank 12()ref	1 1 2a	За	4a	5h		2a	4h	5a	18()ref	Blank	4a	4a		<u>.</u>
P	MB	49	4.4	51	36	dup47	47	6	64	Ī	h2	n ^c i	n51	Check	DUP: MI 120
seq	(4	m 4	v,	9	-	œ	6	2	-	12	Ξ	reru	reru	Š	

(Con(d)

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Table

(C) ZINC	RUN 130	361 NV[2	CONC =	(ABS - B	lank)/0.002	274 (Blan	k = 0.05	95)						
seq# 1D#	Sta	depth	ABS	CONC [pph]	d ₂ [pph]	d, [pph]	C _{extract} [ppm]	^{هو} 5	R d/w		w _{sto}	v ml	M _{ext} µg	C sed µg/g	С _{sed} µmol/g
0 B	BLANK		0.075	27.29											
I MB	BLANF	<u> </u>	0.044	16.01											
2 9		0-10	0.312	91.89	918.85	4594.25	4.59	13.96	0.7123	35.0	41.05	61.96	284.64	28.62	0.44
3 10	FB190	18(0)	0.085	9.28	92.79	463.97	0.46								
4 ==	2a	0-10	0.498	159.57	1595.71	7978.53	7.98	3.48	0.3779	45.0	33.80	75.17	599.74	456.04	6.98
<u>s</u> 12	3a	0-10	0.465	147.56	1475.62	7378.09	7.38	5.82	0.3377	45.0	38.79	77.81	574.06	292.08	4.47
6 13	4a	0-10	0.646	213.43	2134.28	10671.40	10.67	10.90	0.4349	45.0	39.49	73.45	783.79	165.34	2.53
7 14	Sa	0-10	0.863	292.39	2923.94	14619.72	14.62	7.27	0.4245	45.0	38.36	75.94	1110.15	359.73	5.50
8 15	4a	02h	0.427	133.73	1337.34	6686.68	6.69	4.14	0.2402	45.0	31.68	72.40	484.14	486.85	7.45
9 16	0618.4	2()()	0.177	42.76	427.58	2137.92	2.14								
10 17	41	01-10	0.268	75.87	758.73	3793.67	3.79	4.76	0.5895	35.0	41.56	71.62	271.69	96.82	1.48
IM II	180ref		0.523	168.67	1686.68	8433.41	8.43								
12 18	FB200	1700	0.086	9.64	96.43	482.17	0.48								
13 21	44	0-10	0.376	115.17	1151.75	5758.73	5.76	4.94	0.3044	35.0	27.60	57.55	331.40	220.38	3.37
14 22	4h	01-0	0.385	118.45	1184.50	5922.49	5.92	5.47	0.3044	35.0	23.68	53.12	314.60	188.94	2.89
15 23	5b	01-0	0.734	245.45	2454.51	12272.56	12.27	3.32	0.3958	35.0	24.97	56.54	693.92	528.07	8.08
16 24	FB198	0015	0.085	9.28	92.79	463.97	0.46								
17 25	44	()-2h	0.174	41.67	416.67	2083.33	2.08	1.51	0.2307	35.0	16.17	49.59	103.31	296.55	4.54
18 26	0618.1	1210	0.435	136.64	1366.45	6832.24	6.8.3								
19 19	2h	0-10	0.551	178.86	1788.57	8942.87	8.94	4.15	0.3679	35.0	32.67	63.38	566.78	371.23	5.68
20 20	46	01-10	0.27	76.60	766.01	3830.06	3.83	2.61	0.5801	35.0	27.57	59.84	229.17	151.36	2.32
21 DUP.	27 3a	4-6h	0.353	106.80	1068.05	5340.25	5.34	3.75	0.2336	35.0	39.23	70.30	375.44	428.58	6.56
22 27	3a	4-6h	0.341	102.44	1024.38	5121.91	5.12	3.75	0.2336	5.0	39.23	70.30	360.09	411.06	6.29
23 5	2h	()-2c	0.534	172.67	1726.71	8633.55	8.63	8.17	0.2947	45.0	23.77	60.52	522.52	217.02	3.32
24 6	la	()-2a	0.201	51.49	514.92	2574.60	2.57	7.18	0.7123	45.0	27.58	65.30	168.12	32.87	0.50
25 4	5a	0Za	0.928	316.05	3160.48	15802.40	15.80	6.97	0.3686	45.0	32.19	70.09	1107.66	431.14	6.59
26 2	За	0-2h	0.436	137.01	1370.09	6850.44	6.85	5.96	0.1957	45.0	22.63	61.59	421.90	361.72	5.53
27 3	41	()- Za	0.356	107.90	1078.97	5394.83	5.30	6.82	0.6376	45.0	23.45	61.55	332 ()4	76.36	1.17
28 49	2a	() Za	0.559	181.77	1817.69	9088.43	6()(6	6.81	0.3209	35.0	38.93	66.96	608.56	278.48	4.26
29 44	За	()-2a	0.334	99,89	16.800	4994.54	4,00	2.84	0.1057	35.0	39.24	71.22	355 71	640.00	9.79
30 51	4a	02a	0.441	1,38.83	1388.28	6941.41	6.94	5.40	6.2402	35.0	31.75	61.22	424.95	327.62	5.01
96 16	ł?	()-2a	0.244	67.14	671.40	3356.99	3.36	1.15	0.2821	35.0	26.57	60.29	202 40	623.91	9.54
32 59	4h	()2a	0.194	48.94	489.45	2447.23	2.45	2.29	0.2307	35.0	23.38	55.99	137.01	259.34	7.97
33 64	5a	02h	1.048	359.72	3597.16	17985.81	17,99	7.87	0.3686	35.0	31.52	58.53	1052.75	362.91	5.55
OA Check															
DUP	_	(DUP)	1 - 12012	ř. T	53/ 341)		3500								
N I W	80.	actual/r	eference	j v	23/ 545	-	96.0%								
BLA	NKS	values	≤ 2x(0.0	(505)		Y	'ES								(Contd)

Station	Depth	SEM Metal	Sed (nom)	Sed	AVS	SEM/AVS
Station	(((((((((((((((((((((((((((((((((((((((Ivicial	(ppm)	(µ1101/g)	(μποι/g)	JLWI/AV 3
1a 1a 1a	0–10 0–10 0–10	Cu Pb Zn	8.44 17.09 28.62	0.13 0.08 0.44	0.79	0.827
1a 1a 1a	0-2 0-2 0-2	Cu Pb Zn	6.53 15.62 32.87	0.10 0.08 0.50	0.10	6.809
1a 1a	46 810	Cu Cu	7.39 24.95	0.12 0.39	0.43 3.03	0.271 0.130
1b 1b 1b	0-10 0-10 0-10	Cu Pb Zn	27.11 50.61 96.82	0.43 0.24 1.48	10.46	0.206
1b 1b 1b	02 02 02	Cu Pb Zn	18.37 25.68 76.36	0.29 0.40 1.17	3.34	0.557
1b 1b	08–10 20–22	Cu Cu	24.38 39.68	0.38 0.62	10.46 3.93	
2a 2a 2a	0-10 0-10 0-10	Cu Pb Zn	188.88 185.71 456.04	2.97 0.90 6.98	73.56	0.147
2a 2a 2a	0-2 0-2 0-2	Cu Pb Zn	197.08 149.97 278.48	3.10 0.72 4.26	70.77	0.114
2a 2a 2a	046 046 046	Cu Cu Pb	206.49 213.15 139.31	3.25 3.35 0.67	156.09	
2a 2a	08–10 18–20	Cu Cu	85.81 146.52	1.35 2.31	133.96	
2b 2b 2b	0-10 0-10 0-10	Cu Pb Zn	191.88 260.98 372.23	3.02 1.26 5.68	21.96	0.453
2b 2b 2b	0-2 0-2 0-2	Cu Pb Zn	211.12 132.58 217.02	3.32 0.64 3.32	52.30	0.139
2b 2b	0-2 0-2	Cu Pb	212.28 124.81	3.34 0.60	20.16	
2b 2b 2b 2b 2b	46 46 810 2022	Cu Cu Cu Cu	208.43 205.89 210.33 135.75 390.91	3.28 3.24 3.31 2.14 6.15	83.19 83.19 82.02 90.34 117.54	

Table C-3. SEM for Cu, Pb, and Zn and AVS concentrations measured in sediment cores collected from Sinclair Inlet. The SEM/AVS value is calculated for composite and surface samples.

. .	Depth	SEM	Sed	Sed	AVS	
Station	(cm)	Metal	(ppm)	(µmol/g)	(µmol/g)	SEM/AVS
3a 3a 3a 3a	0-10m 0-10m 0-10m 0-10m	Cu Cu Pb Zn	214.13 211.80 172.05 292.08	3.37 3.33 0.83 4.47	3.46	2.500
3a 3a 3a	0-2 0-2 0-2	Cu Pb Zn	377.01 222.79 361.72	5.93 1.08 5.53	69.57	0.180
3a 3a 3a	0-2 0-2 0-2	Cu Pb Zn	342.78 270.34 640.00	5.39 1.30 9.79	69.06	0.239
3a	46	Cu	368.41	5.80	105.29	
3a 3a 3a 3a 3a 3a	4-6 4-6 4-6 4-6 4-6 4-6	Cu Pb Pb Zn Zn	268.36 280.99 243.81 257.36 428.58 411.06	4.22 4.42 1.18 1.24 6.56 6.29	71.12	
3a 3a	8–10 8–10	Cu Cu	449.34 349.93	7.07 5.51	54.66 55.90	
3a 3a	20–22 30–32	Cu Cu	430.23 712.66	6.77 11.21	0.39	
3b 3b 3b	010 010 010	Cu Pb Zn	73.85 60.03 151.36	1.16 0.29 2.32	67.06	0.056
3b 3b 3b 3b	0–2 04–6 08–10 20–22	Cu Cu Cu Cu	184.07 123.38 135.40 175.41	2.90 1.94 2.13 2.76	6.09 12.15 1.71 9.99	
4a 4a 4a	0-10 0-10 0-10	Cu Pb Zn	127.39 152.32 165.34	2.00 0.74 2.53	20.14	0.262
4a 4a 4a 4a 4a	0-2 0-2 0-2 0-2 0-2	Cu Pb Pb Pb Zn	247.87 171.27 175.25 167.28 327.62	3.90 0.83 0.85 0.81 5.01	41.16	0.237
4a 4a 4a	0-2 0-2 0-2	Cu Pb Zn	282.13 245.77 486.85	4.44 1.19 7.45	37.11	0.352
4a 4a 4a	46 810 2022	Cu Cu Cu	237.69 263.06 156.91	3.74 4.14 2.47	86.08 34.20 10.70	

Table C-3. SEM for Cu, Pb, and Zn and AVS concentrations measured in sediment cores collected from Sinclair Inlet. The SEM/AVS value is calculated for composite and surface samples. Continued.

	Depth	SEM	Sed	Sed	AVS	
Station	(cm)	Metal	(ppm)	(µmol/g)	(µmol/g)	SEM/AVS
4b 4b 4b	0-10 0-10 0-10	Cu Pb Za	91.62 102.30	1.44 0.49 2.89	41.67	0.116
40 46 46 46	0-10 0-10 0-10 0-10	Zn Cu Pb Zn	108.94 116.68 108.19 220.38	1.84 0.52 3.37	45.66	0.125
4b 4b 4b	0-2 0-2 0-2	Cu Pb Zn	92.40 90.09 296.55	1.45 0.43 4.54	23.04	0.279
4b 4b 4b	0-2 0-2 0-2	Cu Pb Zn	108.39 89.43 259.34	1.71 0.43 3.97	23.75	0.257
4b 4b 4b 4b	46 46 810 2022	Cu Cu Cu Cu	137.87 133.67 91.15 28.18	2.17 2.10 1.43 0.44	37.40 43.80 28.54 14.52	
5a 5a 5a	0-10 0-10 0-10	Cu Pb Zn	169.88 281.35 359.73	2.67 1.36 5.50	97.98	0.097
5a 5a 5a 5a	0-2 0-2 0-2 0-2	Cu Cu Pb Zn	180.31 180.85 199.16 431.14	2.84 2.85 0.96 6.59	66.12	0.157
5a 5a 5a	0-2 0-2 0-2	Cu Pb Zn	183.36 188.15 362.91	2.89 0.91 5.55	56.77	0.165
5a 5a	46 46	Cu Cu	394.71 374.08	6.21 5.89	90.86 58.75	
5a 5a 5a 5a 5a average	8-10 8-10 8-10 8-10 8-10	Cu Cu Cu Cu Cu	306.92 344.48 643.82 323.90 318.69 387.56	4.83 5.42 10.13 5.10 5.02 6.10	280 66	
5a 5a 5a 5a 5a average	8-10 8-10 8-10 8-10 8-10	Cu Cu Cu Cu	599.25 302.27 304.91 322.42 619.73 429.71	9.43 4.76 4.80 5.07 9.75 6.76	268.38	
5b 5b 5b	0-10 0-10 0-10	Cu Pb Zn	444.34 341.33 528.07	6.99 1.65 8.08	65.24	0.256
5b 5b 5b 5b 5b 5b	0-2 0-2 0-2 4-6 8-10 20-22	Cu Pb Zn Cu Cu Cu	376.53 211.73 623.91 183.29 592.40 281.50	5.93 1.02 9.54 2.88 9.32 4.43	52.74 90.86 196.30 112.62	0.313

Table C-3. SEM for Cu, Pb, and Zn and AVS concentrations measured in sediment cores collected from Sinclair Inlet. The SEM/AVS value is calculated for composite and surface samples. Continued.

Appendix D

				Dish	g	g		AVS
Station	Depth	Rep	Dish #	wt	Preburn	Postburn	Fom	µmol/g
1a	0-2	al	34	1.0	11.40	11.20	0.019	0.12
18	0-10	a2	3	1.4	9.61	9.52	0.012	0.88
10	0-10	a2	2	1.4	15.82	15.68	0.010	0.88
10	4-0	a1	/	1.0	9.69	9.57	0.014	0.43
14	0-10	aı	18	1.0	6.88	6.52	0.062	3.03
Ib 1b	0-10	a3	20	1.0	8.91	8.41	0.064	10.46
10	0-2	82	20	1.0	9.57	9.23	0.040	3.05
10	20-22	az -2	30	1.0	11.61	11.28	0.032	3.93
10	4-0 9 10	82	13	1.0	12.20	11.69	0.046	4.66
10	0-10	81	11	1.5	9.82	9.12	0.084	10.09
2a	0-10	al	10	1.0	7.85	7.27	0.085	71.31
28	0-2	82	24	1.0	6.10	5.51	0.116	70.77
28	18-20	82	31	1.0	8.41	7.69	0.097	139.29
28	4-0	az	1	1.4	6.39	5.71	0.138	152.34
28	8-10	82	29	1.0	7.06	6.05	0.167	133.96
2b	0-10	al	6	1.5	5.82	5.52	0.069	22 13
2b	0–2	a2	9	1.0	5.36	4.82	0.125	56.30
26	20-22	a 2	6	1.0	6.59	5.89	0.125	117.54
26	4-6	a 2	13	1.5	8.12	7.32	0.121	82.02
26	8-10	a 2	15	1.2	9.22	8.24	0.123	90.34
3a	0-9	a 3	23	1.0	8.30	7.60	0.096	3.46
3a	20-22	4	28	1.0	5.28	4.97	0.074	0.39
3a	30-32	4	35	1.0	6.61	6.31	0.054	0.45
3a	4-6	a 2	14	0.9	6.60	5.90	0.122	105.29
38	8-10	84	11	1.0	5.50	5.05	0.101	55.28
3b	0-10	a 2	3	1.5	10.44	10.10	0.038	4 4 3
3b	0-10	a 2	2	1.5	10.91	10.59	0.034	4 4 3
36	0-2	al	4	1.5	8.95	8.31	C.J86	6.19
30	20-22	al	33	1.0	12.19	11.71	0.043	9.99
30	4-6	al	22	1.0	9.49	8.91	0.069	12.56
30	8-10	al	4	1.4	9.75	9.21	0.065	1.71
4a	0-10	a	8	1.5	7.19	6.33	0.151	20.14
4a	0-2	al	17	1.0	5.10	4.58	0.128	39.13
4a	20-22	a	25	1.0	6.89	6.39	0.085	10.70
4a	4-6	al	27	1.0	6.20	5.51	0.133	85.14
48	8-10	a	21	1.0	6.98	6.27	0.119	34.20
4b	0-10	al	12	1.5	11.44	9.89	0.156	43.67
45	0-10	al	36	1.0	6.62	5.72	0.160	43.67
40	0-2	a2	1	1.5	4.88	4.45	0.127	23.39
40	20-22	a 2	14	1.0	6.02	5.60	0.084	14.52
4D	4-6	82	5	1.0	5.60	4.88	0.158	40.60
40	8-10	82	5	1.5	6.34	5.75	0.122	28.54
Sa	0-10	a5	32	1.0	7.89	7.11	0.113	97,98
5a	4-6	a	10	1.5	6.32	5.90	0.087	90.86
28	grab	a 3	8	1.0	7.22	6.54	0.110	68.55
56	0-10	a2	19	1.0	6.75	6.26	0.085	67.02
5b	0-2	al	16	1.0	6.30	5.71	0112	50 45
5b	20-22	al	15	1.0	8.31	7.80	0.070	112.62
5b	4-6	al	9	1.5	10.85	10.00	0.091	165.28
50	8-10	al	12	1.0	7.40	6.91	0.077	196.30

DETERMINATIONS OF ORGANIC MATTER CONTENT BY LOSS ON IGNITION FOR SEDIMENT SAMPLES FROM SINCLAIR INLET

D-1

REPO		Form Approved OMB No. 0704-0188				
Public reporting burden for this collection of info manifaring the data needed, and completing and suggestions for reducing this burden, to Wash 22202-4302, and to the Office of Management a	rmation is estimated to average 1 i d reviewing the collection of informa hington Headquarters Services. Di nd Budget, Paperwork Reduction F	hour per response, includi ition. Send comments rega inectorate for information Project (0704-0188), Washi	ng the time for reviews rding this burden estim Operations and Repo ngton, DC 20503.	ng instruction ate or any othe rts, 1215 Jeff	s searching existing data sources, gathering and er aspect of this collection of information including erson Davis Highway, Suite 1204, Arkington, VA	
1. AGENCY USE ONLY (Leave blank) 2. REPORT DATE			3. REPORT TYPE AND DATES COVERED			
September 1993				Final: June 1991–January 1992		
4. TITLE AND SUBTITLE ACID VOLATILE SULFIDES AND SIMULTANEOUSLY EXTRACTED COPPER, LEAD, AND ZINC IN SEDIMENTS OF SINCLAIR INLET, WASHINGTON					5 FUNDING NUMBERS ME8352	
Robert K. Johnston						
7. PERFORMING ORGANIZATION NAME(S) AND						
Naval Command, Control and RDT&E Division San Diego, CA 92152–5001	Ocean Surveillance Ce	enter (NCCOSC)		TR 15	52	
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13. ABSTRACT (Maximum 200 words)		ו••				
Acid volatile sulfides (AV (Zn) were measured in the m flux of metals between the se used to measure metal fluxes sediment concentrations in the Two sediment cores (10 a profile (2-cm increments), an- sured in composite samples r 2.2 to 10.3 times higher than metal concentrations (Cd, Cu One station had a composite Seven of the ten stations had and decreased towards the bo ranged from 0.1 to 11.2 µmol, chamber.	(S) and simultaneously arine sediments of Sind diment and bottom wa with a benthic chambe he Inlet. to 30 cm depth) were co d the other was homog anged from 0.8 to 98.0 the SEM concentration the SEM concentration the AVS concentration high AVS concentration bitom of the core. Simu- /g. There was not a cle	extracted metal (clair Inlet, Puget ter of the Inlet. 7 er. The stations re- collected from each enized to obtain a μ mol/g. For nine has of Cu, Pb, and 2 in) measured in b yer than the comp ons at the surface iltaneously extracted ar relationship be	SEM) concentr Sound, Washir The AVS sampl epresented a rate in station; one concentric a composite AVS of the ten stat Zn, and 1.1 to 6 ulk sediment g osite SEM concentric (> 20 µmol/g), cted Cu profiles etween AVS an	ations of ogton, as es were c inge of pr ore was s S value fo ions comp 2 times rabs colle centratio which in s were rel d metal f	copper (Cu), lead (Pb), and zinc part of a project to evaluate the ollected from the same stations reviously measured heavy metal selected to determine the AVS or the top 10 cm. The AVS mea- posite AVS concentrations were higher than the sum of divalent exted from the same stations. n, resulting in SEM/AVS = 2.5. iccreased at intermediate depths atively uniform with depth and hux measured with the benthic	
14. SUBJECT TERMS					15 NUMBER OF PAGES	
acid volatile sulfide (AVS)leadorganic mattercadmiummanganesesediment corecoppermetal fluxsilver					91	
uivalent metal iron	mercury nickel	simultaneously (zinc	extracted meta	i (SEM)	16 PRICE CODE	
17 SECURITY CLASSIFICATION	18 SECURITY CLASSIFICATION	ſ	19 SECURITY CLASS	FICATION	20 LIMITATION OF ABSTRACT	
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Robert K. Johnston	(619) 553-2773	Code 522