



# The effectiveness of steel foundry by-products in the treatment of stormwater

Jiyang Ren

A thesis submitted in partial fulfilment of the degree of Master of Applied Science

Auckland University of Technology Auckland New Zealand

June 2006

## Acknowledgements

This study is completed with the support of many people, one way or another, over the last two years. I am very grateful to my wife and daughter for their understanding and support, and wish to convey my gratitude to:

- My primary supervisor, Dr. John Robertson (Auckland University of Technology), for his advice on experiment design, assistance in the laboratory and the finalization of the report
- Professor John Buckeridge (Auckland University of Technology), who had been my primary supervisor, for his invaluable support and guidance
- Dr. Kevin Tilbrook (Auckland University of Technology) who had been my secondary supervisor and helped me to design the field experiments
- The SteelServ Ltd. who propose the project and provide aggregate for testing. Special thanks to Mr. Bill Bourke and Mr. Hendrik Wortman for their suggestions and support
- Mr. Christopher Whyburd and Ms. Yan Wang (Auckland University of Technology) for their assistance with the Flame Atomic Absorption Spectrometer and other instruments
- Mr. Michael Lindgreen (Metro Water Limited), Ms. Jane Puddephatt and Ms. Danielle Hancock (Auckland Regional Council) for their suggestions on stormwater sample collection

## Abstract

The capacity and efficiency of melter slag (provided by New Zealand Steel) to remove heavy metals and suspended solids from stormwater samples are studied in this thesis. A series of batch tests were carried out to investigate the adsorption-desorption mechanism of the slag to remove heavy metals (Cd<sup>2+</sup>, Zn<sup>2+</sup> and Cu<sup>2+</sup>) from working solutions. The results showed that all the tested metallic ions could be removed by mixing the melter slag with the working solutions. Adsorption and ion exchange are the dominant mechanisms in this process. The adsorption capacity follows the descending order of Cu<sup>2+</sup> > Zn<sup>2+</sup>  $\approx$  Cd<sup>2+</sup>. Varied binding energy of different metallic ions to the slag resulted in competitive adsorption between ions.

A variety of substances: inorganic salts (KCl, NaCl, KNO<sub>3</sub> and sea water), organic acids (citric and tartaric) and inorganic acids (nitric and carbonic), were tested as desorbing agents to recover the used slag. Citric acid in sea water was found to be the best in terms of desorption efficiency and cost-effectiveness.

The column tests were carried out to simulate the filtration bed in laboratory conditions. Stormwater samples were collected from residential and commercial areas in Auckland. The synthetic stormwater samples were prepared from clay and metallic ion solution. Both samples were used as feeding solutions in the tests. The removal efficiencies of suspended solids and metallic ions  $(Zn^{2+}, Cd^{2+} \text{ and } Cu^{2+})$  vary depending on different operation conditions. They are in inverse proportion to the grain size of the melter slag and the influent flow rate. The higher removal efficiency of  $Zn^{2+}$  than that of suspended solids infers that metallic ions are removed by both precipitation and adsorption.

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## **Statement of Originality**

'I hereby declare that this submission is my own work and that, to the best of my knowledge and belief, it contains no material previously published or written by another person or material which to a substantial extent has been accepted for the qualityfication of any other degree or diploma of a university or other institution of higher learning, except where due acknowledgements is made in the acknowledgments.'

.....Signed

Jiyang Ren

.....Date

## **Chapter 1** Introduction

## 1.1 Overview

### 1.1.1 Stormwater

The engineering design of stormwater management systems, even though relatively sophisticated in some situations (Gribbin, 1997), still has much need of improvement (Ng, 2004).

Both water quantity and quality need to be considered in the design and operation of stormwater systems (Auckland City Council, 2002). Urban stormwater, in terms of its pollution, has often been considered less of a problem than domestic sewage, which has been given considerable attention from collection, treatment and reuse perspectives (Henze *et al.*, 2001). City planning and administrative authorities have worked extensively to build up separate collection systems to isolate the stormwater from sewage which is directed to and treated at wastewater treatment plants (Salvato *et al.*, 2003). Stormwater, on the other hand, is mostly discharged directly to natural water systems (e.g. rivers, lakes, streams and wetlands, groundwater, harbours as well as coastal waters) without any treatment (Mara *et al.*, 2003; Ng, 2004).

A number of local and international studies on water quality have identified stormwater discharges as one of significant sources of water pollution (USPEA, 1993; Auckland Healthcare Services Ltd., 2000). An investigation made in Auckland (ARC Stormwater Liaison Group, 1998; Ng, 2004) identified that the pollutants in stormwater derive from:

- Vehicle uses and emissions;
- Runoff from roads, car parks, paved surfaces and roofs;
- Sedimentation from earthworks during urban development;
- Infiltration of stormwater into sewerage systems and subsequent overflows;

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- Runoff from poorly managed industrial and commercial sites;
- Poor domestic chemical disposal practices; and
- Landfill leachates.

The awareness of pollution from stormwater was initially raised following deposition of sediment in waterways. The elevated suspended solids in runoff after heavy rainfall, either from erosion by stormwater or run-off from paved surfaces, are regarded as one of the major sources for sediments in receiving waters. Blocked, leaking and illegally connected sewage pipes, together with inadequate sewer capacity and pump failures add wastewater contaminants to the mix (ARC, 2004).

Suspended solids increase the turbidity of receiving water bodies, thereby reducing the penetration of light, resulting in decreased activity and growth of photosynthetic organisms. The increased turbidity also detracts from the aesthetics of natural waters. Solids that settle in the receiving water pose long-term threats resulting from their oxygen demand and gradual accumulation of toxic substances (Adams *et al.*, 2000; Ng and Buckeridge, 2000).

The increased application of fertilizers and pesticides on farmland, the expansion of artificial pavement accompanied by urbanization and the change of land use with the industrialization have led to heavy metal accumulation in soils (Ng and Buckeridge, 2002).

Studies in the United States and Canada indicate that heavy metals, including lead, zinc, and copper, were the most prevalent toxic contaminants found in urban runoff. Other toxic pollutants found in stormwater include phthalate esters (plasticizer compounds), phenols and creosols (wood preservatives), pesticides and herbicides, oils and greases, and polycyclic aromatic hydrocarbons (PAHs) (Adams *et al.*, 2000; Ng, 2004).

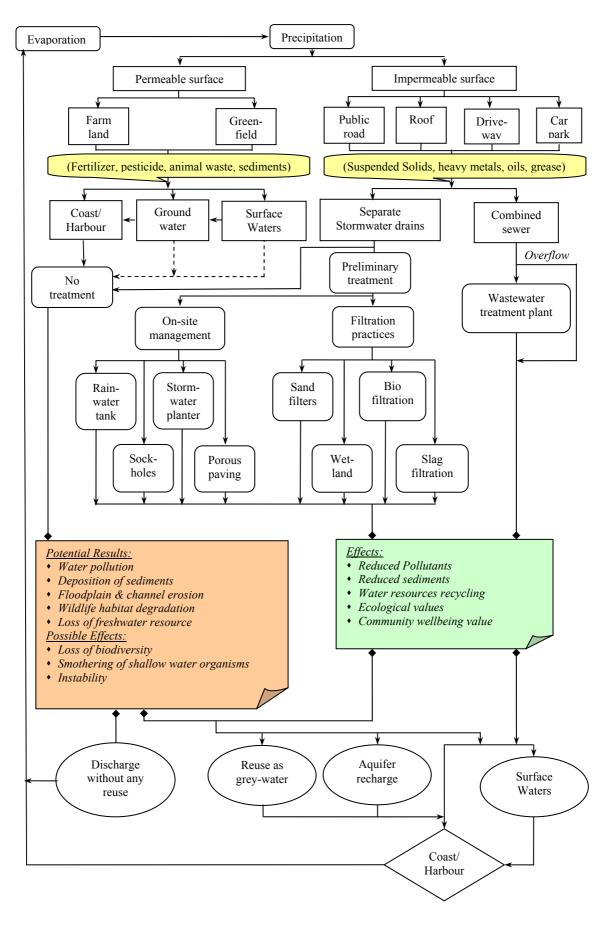


Figure 1-1 Stormwater development, collection and treatment

The primary sources of heavy metals in urban area are traffic-related activities (eg exhaust from motor vehicles) and atmospheric fallout (Sorme and Lagerkvist, 2002) as well as the wear and tear of the road pavement (Ng, 2004).

An investigation in Auckland (Williamson and Kelly, 2003) applied environmental response criteria (ERC) to evaluate the pollution by each element. The red, amber and green ERC represent unhealthy, degraded health and healthy environments respectively. The report showed that zinc exceeded the Red more frequently than lead or copper. However, copper and lead fell within the Amber range more frequently than zinc. Zinc and copper concentrations are generally increasing in estuarine sediments whereas lead concentrations are decreasing. The latter is believed to be the result of following an international trend by introducing lead-free petrol in 1996 (Kingett Mitchell Limited, 2003). PAH levels were Green except at some sites with older catchments such as Meola Stream (sewage overflows) and Motions Creek (overflows from combined drainage system), and sites near historic gas works (Chelsea, Little Shoal Bay).

Even though the implementation of source control is often more cost-effective than traditional end-of-pipe approaches, the residual solids and toxic substances collected in the runoff still need to be removed in order that appropriate (e.g. Local Authority) water quality standards be achieved against the standard or quote to eliminate the potential hazard to the receiving water bodies.

In the United States, most stormwater discharges are considered point sources and have been regulated under the *National Pollutant Discharge Elimination System* (NPDES) since 1994.

A traditional approach where stormwater is collected in underground pipes and disposed of as soon as possible at some convenient downstream location is, given the subsequent flooding and pollution problems is not acceptable in developed countries and in many cases, no longer acceptable in developing countries (Nathanson, 2000).

Potable water shortage is now a global problem due to the increased water demand as well as the deterioration of ecosystems. It was found (Kingsley, 2005) that the flow

into catchments in the Perth area had been decreased by two-thirds in the last 30 years and that Perth, which now survives on fossil water, will run out of its available sustainable natural water resources by 2015 if nothing is done to slow usage rates. In fact, stormwater is increasingly being viewed as a natural resource for use in a beneficial manner, rather than as a waste material to be disposed of quickly. For example, the stormwater in on-site storage or detention basins can percolate to the groundwater and help to recharge an aquifer (Nathanson, 2000).

Environment-friendly management of stormwater has been included in the agenda of local authorities such as Auckland City Council (ACC). It is generally recognized that a high percentage of heavy metals and toxic contaminants have a strong affinity for the suspended sediments present in runoff. This association is fortuitous in terms of control and treatment of runoff since it is relatively straightforward to separate suspended solids and hence the pollutants attached to them (Adams *et al.*, 2000). Ng (2004) studied the particle size distribution in stormwater in Auckland and concluded that particles larger than 100  $\mu$ m make up around 90% of the total road sediment load. The removal of coarse sediments (size over 100  $\mu$ m) would ensure removal of 75% to 90% of all deleterious metal contaminants.

The ACC plans to remove 27% of the suspended sediments in 70% of stormwater catchments by 2020. The detention pond in Waiatarua Basin and mechanical filtration systems at Central Park in Penrose have been built up to facilitate the achievement of the goal (Auckland City Council, 2005).

## 1.1.2 Slags

From the metallurgical point of view, slag is not just an unwanted by-product in most metallurgical processes (Bramming and Wikstram, 2002). It is used in the smelting furnaces to separate the impurities, such as gravel, sand, clay and stones, from the molten iron (Clair, 1964). It plays an important role in all metal smelting processes and the slag composition must ensure satisfactory composition of the metal being produced (Davies and Oelmann, 1985). The functions of a slag in molten metal processing are as follows:

- 1) To protect the melt from contamination from the furnace atmosphere;
- 2) To take up (and remove) unwanted impurities present in the melt;
- To control the supply of refining agents to the melt through additions to the slag; and
- 4) To insulate the melt.

Smelting slags are usually dominated by silicates, but may also contain calcium, magnesium, aluminium, phosphorous and iron etc.

Slag can be classified as basic or acidic according to its nature. A basic slag is one which contains an excess of basic oxide, while an acidic slag contains excess acidic oxide. The characteristics of a slag are to some extent a function of the alkalinity of the slag, which is expressed as the ratio of the mass of basic oxides to that of acidic oxides in the slag. The most important basic oxide is CaO, and the principal acidic oxide is SiO<sub>2</sub>.



Figure 1-2 Mechanical screening of slag at NZ Steel (Auckland)

Considering the energy crisis, global warming and the environmental deterioration, a large effort has been made in modern metallurgical industries to minimize heat loss, slag consumption and evolution of waste slag (Emi and Seetharaman, 2000). Slag volume reduction has always been an important part in the development of the operation of blast furnace. The low slag volume operation in blast furnaces requires a tight control over both raw material quality and process. The improved efficiency of the process could, on the other hand, result in a number of advantages such as lower cost of production, higher productivity and improved hot metal quality (Bramming and Wikstram, 2002).

Blast furnace slag has been put to many commercial uses (Dennis, 1963). According to the manner in which it is cooled from the molten state, three general types may be recognized.

*Air-Cooled slag*: This is the normal product from smelting. The slag is simply poured from the ladle car on to the slag dump. After crushing and screening, it may be used for railway ballast, road-making or as aggregate for cement.

*Granulated slag*: This is prepared either by running the molten slag into water, or the slag is subjected to a high pressure water jet as it falls into the pit. The principal use is for making cement or concrete blocks.

*Foamed or expanded slag*: This is the product obtained by running slag into a controlled amount of water or by blowing steam through the molten slag. The amount of water used is less than that required for granulation. The result is relatively dry cellular product. It is an effective non-conductor of heat and sound and is used for thermal and acoustic insulation.

#### 1.2 Background

It was estimated in 2004 (U.S. Geological Survey, 2005) that the iron and steel slag output was on the order of 220 to 380 million tonnes in the world, based on typical ratios of slag to crude iron and steel output. Almost  $2 \times 10^7$  tonnes of slag were

consumed in the U.S in 2004. The major uses were in construction sector as aggregates for asphaltic paving, fill, and road bases, and as a feed for cement kilns. Some slags were also returned to the furnaces as ferrous and flux feed.

A certain portion of slag is disposed of, either as land fill or stock-piled outdoors, besides those consumed or recycled. Research on the application of iron and steel making slags in wastewater treatment provides another option for using this material. The mechanisms and efficiency of the removal of heavy metals like arsenic (Ahn *et al.*, 2003), lead (Dimitrova and Mehandgiev, 1998; Dimitrova,, 2002) copper (Feng *et al.*, 2004), dye removal/decolorization (Konduru *et al.*, 1997; Gupta *et al.*, 2003) from different sources had been studied.

*SteelServ Ltd.* of Auckland, as part of the *Multiserv* international group of companies, requested the University to examine the use of iron and steel making slags as a media for assisting the removal of suspended solids and heavy metals from stormwater. It is recognized that this is an international problem – not unique to New Zealand – and one that may provide an outlet and constructive end use for surplus steel industry coproducts.

As a by-product from metallurgical industry, most slags contain pollutants to some extent, especially the heavy metals. However, the *SteelServ* melter slag is a by-product of titanomagnetite. Leaching trials have shown that there is no release of heavy metals from this material when it is subjected to prolonged leaching (New Zealand Steel, 1992). Therefore it is likely to be a non-toxic aggregate for the treatment of stormwater.

More recently, the removal of heavy metal contaminants from stormwater has become a strategic imperative for Territorial and Regional councils, through heightened awareness of their impact on the environment. It is contended here that the slag should be regarded as a candidate to assist with these initiatives.

A field test carried out by *SteelServ* in a filtration bed showed that the zinc concentration of the effluent from the industry dropped from 100 mg  $L^{-1}$  to 10 mg  $L^{-1}$  after draining through the two-step beds, i.e. over 90% of zinc can be removed from

the effluent of the factory. However, a batch test carried out recently (Menzlin, 2005) showed that the *SteelServ* melter slag was no better than river sand or silica sand (which were used as controls in the test) to remove Zn (up to 3 ppm) from feeding water at low concentration. The hydroxide precipitation plays a key role in the process.

As constructed wetlands have been widely used to process the sewage and stormwater (Cameron *et al.*, 2003), the massive land coverage makes them unsuitable for use in urban areas. The availability of a cheap, compact, efficient heavy metal adsorbing, and environmentally-safe aggregate is seen as a solution to the problem.

## **1.3 Research aims and objectives**

The overall research aim of the project is:

To develop a better understanding of the nature and ability of slag to remove solids and heavy metals from stormwater without any deleterious side effects on the water quality.

Following objectives are designed to achieve the above aim:

- 1) Build on previous scientific study undertaken in New Zealand and internationally with these materials;
- In conjunction with *SteelServ Ltd.*, determine where and in what role slagbased products may play a part in assisting with improving storm water quality;
- 3) Organise, trials and experiments. Monitor and report on their outcomes;
- 4) Confirm the safety and suitability of these materials for such end uses.

## 1.4 Literature review

The ability of removing heavy metals from contaminated water by metallurgical industry slags has been investigated in a number of studies.

Kang *et al.* (2004) used steel slag as an adsorbent to remove lead from wastewater. The impact of temperature and pH on the kinetic and equilibrium characteristics of Pb adsorption on the slag was also studied.

Dimitrova (2002) carried out a column test which used granular blast furnace slag as an adsorbent for lead removal. It was found that the presence of Na and Ca could impede the uptake of Pb. The fact revealed that ion exchange played a dominant role in the process. Solubilisation and hydrolysis of the calcium silicates and aluminosilicates of the slag is accompanied by the adsorption of metals presented in the solution.

The adsorption of nickel from wastewater by steel converter slag is considered to be enhanced by the presence of magnetite (Ortiz *et al.*, 2000). It works under a wide range of conditions in terms of initial metal concentration, retention time, adsorbent dosage and temperature.

Feng *et al.* (2004) compared the iron and steel making slag in terms of their capacity to remove copper and lead from aqueous solutions. The iron slag presents higher efficiency due to its higher surface area, porosity and ion-exchange ability.

There is little research on desorption of heavy metal from saturated aggregates. EDTA, which is able to chelate metals, was proved to desorb lead rapidly from the adsorbent (Kang *et al.*, 2004).

Desorption of heavy metals from contaminated soils can be achieved by applying organic acids, which function as chelating agents, e.g. citric, malic, acetic, tartrate acids (Gao *et al.*, 2003; Qin *et al.*, 2004), or inorganic salts like sodium nitrate (Arias *et al.*, 2005) and calcium chloride (Qin *et al.*, 2004). In general, the organic acids showed higher ability to desorb metals than inorganic salts. However the disposal of the solution following desorption, and the cost, are key considerations of promoting the application. These aspects were not discussed in the above papers.

The side effect of introducing such aggregates (which are usually regarded as waste) into environment is always a concern. Kozanoglou *et al.* (1997) found that higher

levels of Fe and Cr in biota from the sampling sites situated near the ferro-nickel smelting plant or the dumping site of the metalliferous slag in contrast with the obviously lower levels of the more distant sites. Molluscs showed generally higher concentrations of Fe and Cr in their tissues than plants.

The leaching trials were conducted on melter slag by New Zealand Steel (1992). Water with different pH's for a period of over two years was used to leach the slag continuously. The results showed that there were no major impact on water quality when the test rig feed water was neutral or basic. Iron and manganese were leached out under acid conditions. The investigation on the reaction of slag to a marine environment (New Zealand Steel, 2002) showed the medium extraction concentration of Ca, S, K, B, Mg initially and dropped to below detection levels rapidly. Mo, Cu, Mn, Zn, Pb, Cd and Ni were below detection levels from the beginning. Fe and Al showed low but consistent discharge value even after ten cycles.

It can be concluded from the review that the capacity of heavy metal removal by slag has been verified by many studies. It is currently applied in constructed wetland systems as an adsorption substrate. However, the application of melter slag in stormwater purification is under-researched in terms of heavy metal and suspended solids removal. These, and an evaluation of the recovery of the used aggregates, are the basis of this study.

## **Chapter 2** Materials and Methods

#### 2.1 Selection of metal analyzing instruments

The analysis of the metallic elements in solution was one of the essential tasks in this study. Spectroscopic techniques, which process the samples in a rapid and effective way (Rouessac, 2000), are now widely used in laboratories for analyzing metal elements. Both flame atomic absorption spectrometer (Flame AAS, Perkin-Elmer 3110) and Inductively Coupled Plasma – Atomic Emission Spectrometer (ICP-AES, Varian Liberty AX-Sequential) were available in the AUT chemistry laboratory. The AAS was selected in this study for the following considerations:

- acceptable detection limit;
- simplicity; and
- lower operational cost

Element	Flame AAS	ICP-AES
Ca	1	0.08
Cd	0.5	1
Cu	1	0.9
Zn	0.8	1

Table 2.1 Atomic spectrometric detection limits ( $\mu g L^{-1}$ )

Source: Modern Methods for Trace Element Determination (Vandecasteele and Block, 1993; Oguz, E. 2005)

Varied detection limit can be achieved by using different instrument. They are, nevertheless, in the ppb level.

One of the advantages of ICP-AES was the ability of simultaneous multi-element analysis. However due to the limited budget of this study only three metallic elements were considered. The increased workload through using AAS, which processes the sample at a wavelength that is specific and characteristic of the element under consideration, was acceptable.



Figure 2.1 The Perkin-Elmer 3110 flame atomic absorption spectrometer used for all metal analysis conducted at AUT

### 2.2 Stock standard solutions preparation

Stock standard solutions were used for making the standards to calibrate the measurement of metallic ions by AAS. They were also used to prepare the working solutions for the batch tests and column tests. The nominal concentration of the stock solutions used in the study was  $1000 \text{ mg L}^{-1}$ .

The standard solutions are preferably to be prepared by metals (User Manual of AAS). However considering the absence of the metals in the laboratory, the metallic salts (reagent grade) were used as alternative in this study. The stocks for each element were prepared as follows:

a) Cadmium

Dissolve 2.283 g of reagent grade cadmium chloride into a 1L volumetric flask to volume with 1% hydrochloric acid.

b) Copper

Dissolve 2.953 g of copper nitrate  $Cu(NO_3)_2$  (reagent grade) into a 1L volumetric flask to volume with 1% nitric acid.

c) Zinc

Dissolve 1.000 g of Zinc metal (reagent grade) in 40 mL hydrochloric acid and dilute into a 1L volumetric flask to volume with deionized water.

d) Calcium

Dissolve 2.497 g of dried calcium carbonate  $CaCO_3$  in a volume of 1:4 nitric acid. Dilute into a 1L volumetric flask to volume with deionized water.

The stock solutions were made from nitric or hydrochloric acid and stored in a sealed volumetric flask. All the working solutions and standards hereafter were diluted from these stock solutions to ensure the consistency of the results.

## 2.3 Calibration curve

A set of standards ranging  $0.2 \sim 20 \text{ mg L}^{-1}$  for each element were prepared and measured by AAS. Three standards for each element were selected for making calibration curve. They were determined by the following criteria:

- The reading from the spectrometer should be within  $0.1 \sim 0.8$
- Linear response of the reading to the concentration

## 2.4 Slag aggregate

The original melter slag sample provided by SteelServ had a grain size range of 5 - 10 mm. It was then crushed by Wiley mill and sieved in a sieve stack into the following three grain size categories:

0.60 ~ 1.00 mm (fine) 1.00 ~ 2.00 mm (medium) 2.00 ~ 3.35 mm. (coarse)

The sieved slag was then washed by deionized water to remove the small particles and dried in the oven (Wilton Utility) at 105°C for 24 hours. The samples were stored in large uncovered beakers until required in the experiments.

Component	Weight percentage (%)	Component	Weight percentage (%)
Fe	0~10	MgO	11 ~ 15
CaO	12~20	P <sub>2</sub> O <sub>5</sub>	< 0.005
SiO <sub>2</sub>	9~15	$V_2O_5$	0.1 ~ 0.5
TiO <sub>2</sub>	27 ~ 42	MnO	0.5 ~ 1.7
Al <sub>2</sub> O <sub>3</sub>	15 ~ 21	Cr <sub>2</sub> O <sub>3</sub>	0.2 ~ 0.6

 Table 2.2
 Chemical composition of the slag

Source: Results of Leaching Trials Conducted on Melter Slag, NZ Steel, 1992

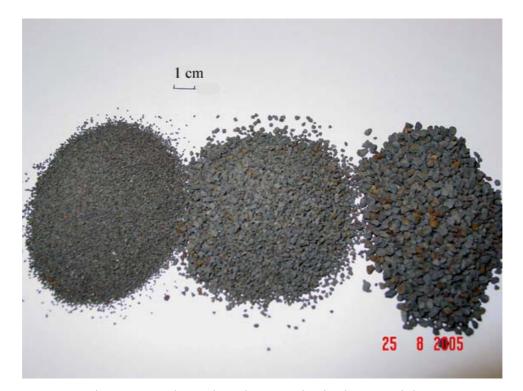


Figure 2.2 The melter slag samples in three particle sizes used as adsorbent in the study

## 2.5 Adsorption kinetics

The first experiment was designed to verify the capacity of the slag to remove heavy metals.

Hypothesis 1That melter Slag has the capacity to remove zinc and<br/>copper ions from wastewater, and that this ability is not<br/>simply a function of the alkalinity of the slag.

A working solution (around 20 mg L<sup>-1</sup>) containing known levels of zinc or copper was prepared by diluting the stock solution with deionized water. Mix 150 mL of this "synthetic wastewater" with slag samples of different grain size in 250 mL conical flask. A magnetic stirrer was used to homogenize the solution. The concentrations of  $Cu^{2+}$  and  $Zn^{2+}$  in the solution were measured by AAS at intervals of 15 minutes. The trial was stopped when equilibrium was reached, i.e. at the point of saturation of the aggregate or the end point of the adsorption. The experiment was conducted at 23°C.



Figure 2.3 Adsorption batch experiments

Since the batch tests on slag were supposed to be conducted under acidic or neutral surroundings, the marble  $(92.4\% \text{ of } CaCO_3)$ , which is considered to be a more

alkaline material, was used in a series of control tests. The control tests were carried out under the same conditions as those on slag in terms of particle size, initial pH and concentration of the working solution and the liquid/solid ratio.

The pH of the solution was measured by a pH meter (MeterLab PHM201) at the time of AAS analysis. The initial and final concentrations of calcium  $(Ca^{2+})$  in the solution were also measured.

## **Background to Experimental Procedure**

The adsorption capacity of the aggregate depends on the temperature (Oguz, 2005), the nature of the adsorbate (such as pH, concentration, and turbidity) (Ricou *et al.*, 1999; Pan *et al.*, 2003; Oguz, 2005) and the nature of adsorbent (such as grain size and specific surface area) (Sansalone, 1999).

## 2.6 Adsorption isotherm

Batch adsorption tests were then carried out to investigate the nature of adsorption.

The equilibrium relationship between the amount of substance adsorbed and that remaining in solution is defined for a given set of conditions by an equation known as an adsorption isotherm. Depending on the theory used to model the adsorption, Langmuir and Freundilich are most commonly used for the application of adsorbent in water treatment (Binnie *et al.*, 2002; Pan *et al.*, 2003).

Hypothesis 2	That the Freundlich theory applies to the adsorption
	isotherm, i.e. that the equation for adsorption is:
	$x/m = Kc_q^{1/n}$
	where:
	x = weight of material adsorbed
	m = weight of adsorbent
	$c_q$ = concentration of material remaining in solution once
	equilibrium has been reached
	K and n are constants.

The adsorption capacity of different aggregates or the same aggregate in different particle sizes can be evaluated by comparing the constants in each isotherm. It will be discussed in Chapter 4.

The analytes include  $Zn^{2+}$ ,  $Cu^{2+}$  and  $Cd^{2+}$ . Samples of about 1.0, 2.0, 3.0, 4.0 and 5.0g of melter slag at different particle sizes were prepared for testing. The "synthetic wastewater" solutions which contain 20 mg L<sup>-1</sup> of one of the metallic ions were adjusted to pH=4 with 10% NaOH or 10% HNO<sub>3</sub>. The slags were mixed with 150 mL of the metal solutions in 250 mL conical flasks at room temperature (23°C) and stirred on magnetic stirrer for 24 hours. The concentrations of metallic ion and calcium were measured by AAS before and after the test. The pH of the solutions was also measured using pH meter. All batch tests were carried out three times.

## 2.7 Desorption

The batch tests were conducted to confirm whether or not the adsorption process is reversible.

**Hypothesis 3** That adsorption is a reversible process and that used slag can be recovered by exposing the slag to inorganic acid salts or organic acids.

The used slag that was saturated by  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  ions in the batch tests was washed with deionized water and dried in oven at 105°C for 3 hours. The slag was then weighed.

There is little study on desorption of heavy metals from slag aggregate. Lowmolecular-weight organic acids had been used for desorbing copper, cadmium and lead from contaminated soils (Qin *et al.*, 2004). Although synthetic chelators such as diethylenetriaminepentaacetic acid (DTPA), ethylenediamine tetraacetic acid (EDTA) and nitrilotriaceticacid (NTA) had the verified capacity to extract heavy metals from soils (Feng *et al.*, 2005; Jeon *et al.*, 2005), they were not selected in this study due to the toxicity of the formulated chelate complex to the receiving water bodies (Knepper, 2003).

Potassium chloride, sodium chloride, sodium nitrate, nitric acid (ion exchange) citric acid, tartaric acid (chelators) and carbonic acid (generated by blowing carbon dioxide into deionized water) and solutions in different concentration were prepared from corresponding salt or acid solid (AR grade). Sea water, which is supposed to be one alternative desorption agent was collected at Herne Bay (Auckland). The supernatant<sup>1</sup> was used for the tests. The inorganic salts were used for comparison since they were main background electrolytes in environment (Qin *et al.*, 2004).

The desorption mechanism of each agent is to be discussed in Chapter 4.

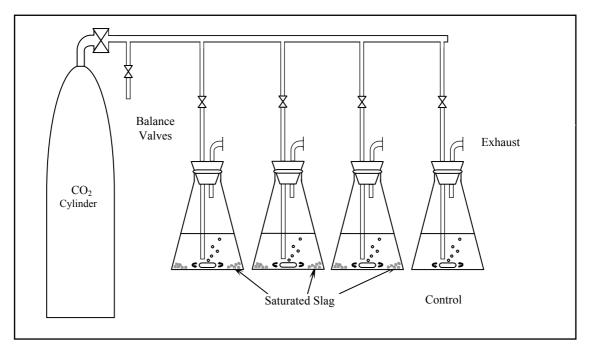


Figure 2.4 Desorption of metallic ions by carbonic acid

The slag was mixed with 150 mL of the test solutions or the combination of these for 24 hours. The concentrations of metallic ions released to the aqueous phase were measured thereafter. As the adsorbed ions in the slag was known, the desorption

<sup>&</sup>lt;sup>1</sup> The seawater sample was kept at 4°C for one night and the fluid above the sediment was siphoned off.

efficiency in the first 24 hours (fist run,  $\eta_1$ ) can be expressed as the percentage of released ions in the adsorbed metals in the slag.

Similar to the above process, the second run was repeated except for extending the retention time to 96 hours. The accumulative efficiency ( $\eta_2$ ) was calculated and the effectiveness of the first run can be given as the percentage of  $\eta_1$  in  $\eta_2$ . The results are given in Section 3.5 (P.39).

#### 2.8 Column experiments

#### 2.8.1 Stormwater sampling

The working solutions used in the batch experiments differed from real stormwater runoff due to the absence of solids. The presence of solids, composed of both organic and inorganic substances, could change the characteristics of the solution from the physical, chemical and biological point of view (Asaf *et al.*, 2004; Zanders, 2005). The stormwater samples collected from different function areas were used in the column experiments. Since the first flush is regarded as containing elevated level of pollutants (Lee *et al.*, 2002; Ng, 2004), collection of such runoff is essential to the analysis in the study.

As the study was carried out during a wet winter (July and August) in Auckland, it was only possible to collect on two "first flush occasions".

Si	ite	Function area	Sampling point	Days since last storm
Oaklay	Site 1	Residential	Outfall of stormwater	
Oakley – Creek	Site 2	Construction site with unpaved surface	drains	2
Hobson S	St. (CBD)	Commercial and high traffic	Gratings before entering the drainage	3

 Table 2.3
 Description of stormwater sampling sites

#### 2.8.2 Sample conservation:

Plastic collection containers, which had been rinsed with deionized water and diluted nitric acid, were used for sampling. Considering the formulation of unsoluable oxides by some metals at high pH, it was recommended (Eaton *et al.*, 1995) that conservation of the samples be carried out at pH 2 (adjustment with nitric acid). However such conservation may mobilize the metallic components that are attached to the organic and/or inorganic particles. On the other hand, more metallic ions in the sample without conservation may be attached to the solids during the storage (Arias *et al.*, 2005). The allocation of heavy metals between solution and suspended solids may therefore be changed from the original status. The determination of the removal efficiency of heavy metals by slag will, as a result, be affected. The problem is minimized by analyzing the sample immediately after the sampling. Also the presence of nutrients and microorganisms in the stormwater sample (Ng, 2004) may change both the physical and chemical properties; the sample was therefore kept at 4°C to reduce the biological reaction by microorganisms. The column test was carried out within 24 hours of sampling to minimize the possible error.

#### 2.8.3 Sample analysis

The method of nitric acid digestion accompanied by sample dry ashing was applied to metallic elements analysis in the study (Hseu, 2004).

- 200 mL of stormwater sample was filtered through pre-weighted filter paper (Whatman, No.42). The filtrate was kept for the analysis of soluble metallic elements.
- The filter paper is then dried at 105±2°C in drying oven for 16 hours. The SS content is determined by comparing the increase of the weight of the filter paper.
- Place the filter paper (ash-less) in the muffle furnace (McGregor NL-200), which is set to 200°C and raise the temperature by 50°C increment until the temperature reaches 350°C and there is little or no smoking. Finally set the

temperature to 450°C and ash about 16 hours.

- Remove the samples from the furnace and allow them to cool. Wet the ash with DI water and add 4 mL of HNO<sub>3</sub> (70%, reagent grade) to cover the ash. Cover with a watch glass and reflux on a hot plate for about 1 hour. Then move the watch glass and reduce the heat to gently evaporate the acid. Return the samples to the furnace and ash at 375°C for 1 hour until a white ash is obtained. This ash represented the suspended solids in the sample.
- Add 5 mL 10% hydrochloric acid and gently boil the solution, cool and make up to 25 mL with deionized water.

A blank was prepared in the same manner.

The  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  concentrations in the digested sample and filtrate were measured by AAS.

The same samples were also measured directly by AAS without any filtration and digestion to show the efficiency of the digestion method.

## 2.8.4 Column tests with stormwater samples

The column experiments were carried out to determine whether other components within the slag had caused adsorption.

Hypothesis 4Removal of heavy metals from stormwater by slag is not<br/>simply through precipitation of cations as hydroxides<br/>and/or subsequent particle settlement

The glass columns (5 mm ID) which were packed with melter slag were used in these experiments. The columns were in U-shaped to ensure that the aggregate was submerged throughout the experiment (Figure 2.5). Three different particle sizes of slag (0.6 - 1.0, 1.0 - 2.0 and 2.0-3.35 mm) were tested. The bulk densities of each

size of slag were 1.67, 1.64 and 1.44 g/cm<sup>3</sup> respectively. The slag bed height for each size of slag was 19.2, 19.6 and 22.2 cm respectively at the weight of 6 g.

The stormwater sample, which was stirred by magnetic stirrer to mix the solution, was stored in a 1L beaker and pumped by peristaltic pump (Alitea) through the column with the linear flow rates of 1.96, 3.92 and 8.92 cm/min.

A blank column was used as control in the experiment.

The effluent samples from each column were collected at intervals of 1 hour for 4 hours and analyzed without previous filtration. The influent was also analyzed.

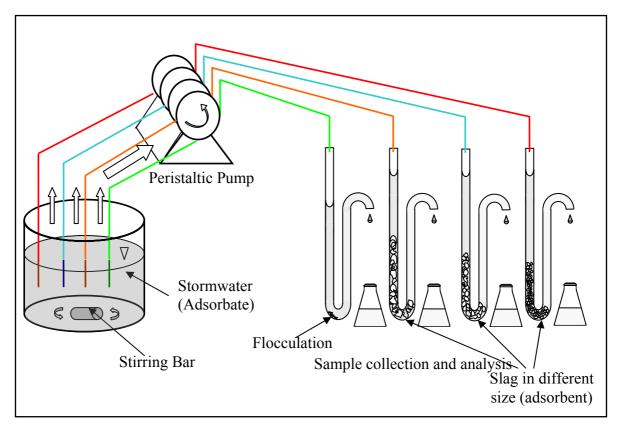


Figure 2.5 Layout of column experiment

The analysis of the sample showed that the concentrations of  $Cu^{2+}$  and  $Cd^{2+}$  are extremely low (Table 3.13). It was hard to determine the removal efficiency of

metallic ions from the stormwater samples. Therefore a set of supplementary column tests which used synthetic stormwater samples as filtrate were carried out.

### 2.8.5 Column tests with synthetic stormwater samples

The synethetic stormwater samples were prepared by mixing clay  $(29\pm2\%)$  of moisture) with deionized water and metallic solutions which were originated from stock standard solutions. The clay is quite easily mixed to give a uniform milky suspension. The concentrations of SS and metallic ions were adjusted as follows:

Sample	SS(mg/kg)	$Cd \ (\mu g \ L^{-1})$	$Zn \ (\mu g \ L^{-1})$	Cu (µg L <sup>-1</sup> )
1	100	100	400	200
2	200	100	600	200
3	400	100	800	200

 Table 2.4
 Concentration of each element in the synthetic stormwater samples

The sample was stabilized at room temperature for 1 hour before it was used in the column test. The variables in the supplementary tests include the particle size of the slag ( $0.6 \sim 1.0$ ,  $1.0 \sim 2.0$  and  $2.0 \sim 3.35$  mm), flow rate of the feeding sample (1.96, 3.92 and 8.92 cm/min) and the concentration of each element in the samples.

The efficiencies of the SS and heavy metal removal were evaluated by comparing the difference between the influent and effluent.

Although turbidity is the typical parameter for evaluating the clarity of the water sample, the scattering of the sample creats an apparent absorbance which can be measured by spectrophotometer (Ultrospec 2100 pro). It was applied in this study as an alternative due to its simplicity.

The wave scanning of the sample is shown in Figure 2.6.

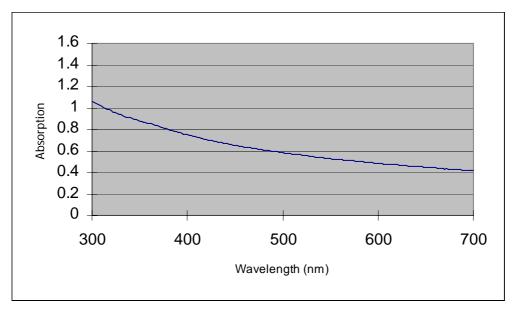


Figure 2.6 Spectrum of stormwater sample

The wavelength was set to be 400 nm to measure the absorption of the samples. The linear regression of the absorbance of the samples to the corresponding SS (mg/kg) showed that they are closely related ( $R^2$ =0.9965) within the tested SS region. Therefore the absorbance was used to represent the SS level in this study.

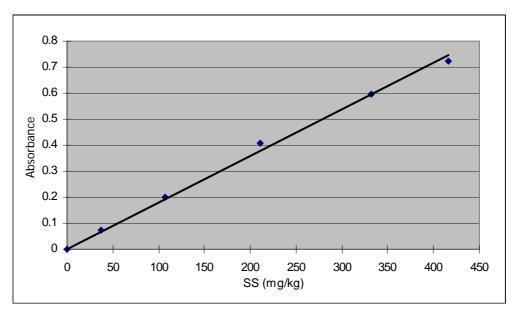


Figure 2.7 Relationship between suspended solids and absorbance of synthetic stormwater sample

The tests followed the same process described in the above section.

## Chapter 3 Results

#### 3.1 Adsorption kinetics

It was observed (Oguz, 2005) that the adsorption capacity of metallic ion on adsorbent was proportional to active sites on its surface. Adsorption kinetics describes the temporal rate of solute adsorption. It is one of the key characteristics in evaluating the efficiency of adsorption.

Figure 3.1 and 3.2 are composite results, showing the removal efficiency of  $Zn^{2+}$  and  $Cu^{2+}$  from working solutions by melter slag at different grain sizes (Fine –  $\Phi$  0.6 ~1.0 mm; Medium –  $\Phi$  1.0 ~2.0 mm and Coarse –  $\Phi$  2.0 ~3.5 mm). The mass of slag used in the tests was 50g.

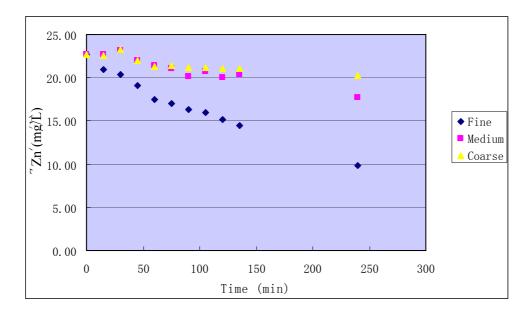


Figure 3.1 Trends of  $Zn^{2+}$  removal over time by melter slag (50g)

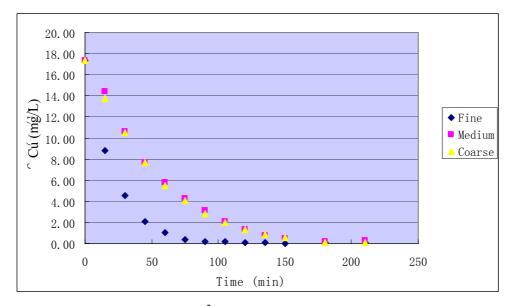


Figure 3.2 Trends of  $Cu^{2+}$  removal over time by melter slag (50g)

Figure 3.3 and 3.4 show the results of control tests which used marble as adsorbent under the same conditions as slag.

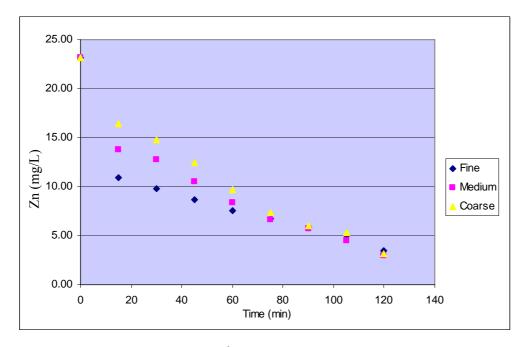


Figure 3.3 Trends of  $Zn^{2+}$  removal over time by marble (50g)

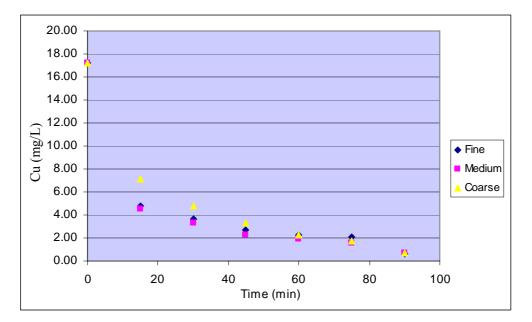


Figure 3.4 Trends of  $Cu^{2+}$  removal over time by marble (50g)

The removal efficiencies for  $Zn^{2+}$  by fine, medium and coarse slag in 4 hours were 56%, 22% and 11% respectively, while over 99% of  $Cu^{2+}$  can be removed by slag at all tested grain sizes. A second experiment which increased the mass of slag to 100g was carried out (Figure 3.5) to see whether more  $Zn^{2+}$  can be adsorbed by elevated amount of slag.

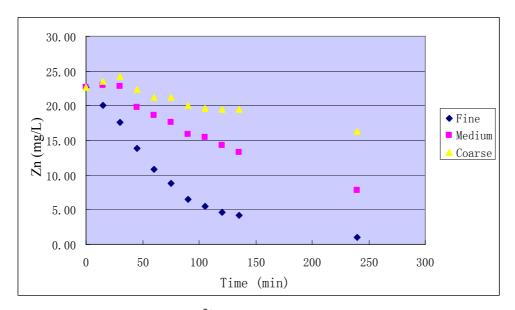


Figure 3.5 Trends of  $Zn^{2+}$  removal over time by melter slag (100g)

The  $Zn^{2+}$  removal efficiency was raised to 95%, 66% and 28% respectively for fine, medium and coarse slag in the test.

A solution consisting around 20 mg  $L^{-1}$  of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  respectively was mixed with 3g of slag (fine size). The removal of each ion from this multi-ion solution over time was shown in the following figure.

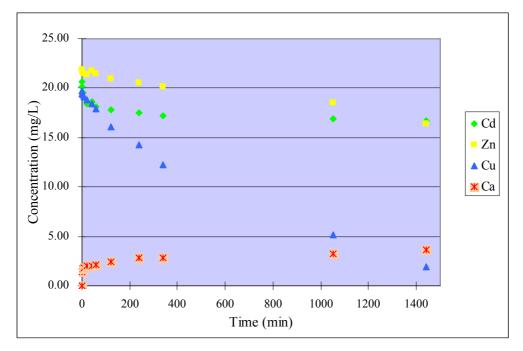


Figure 3.6 Metallic ions removal by slag (fine) from multi-ion solution

## 3.2 Alkalinity effect of the slag

The melter slag tested in the study can be classified as alkaline due to the dominant composition of basic oxide, i.e. CaO (12%) and MgO (13.1%) over that of acidic oxide , i.e.  $SiO_2$  (14%).

It was detected that the pH of the solution had been increasing during the adsorption process, accompanied by an increase of  $Ca^{2+}$  in solution. The concentrations of  $Ca^{2+}$  and pH of the solution before and after the tests were given in Table 3.1.

			e		1		Unit	: mg L <sup>-1</sup>
Analyte	Mass of	Items	Fine		Мес	lium	Coarse	
Slag (g)	Slag (g)	nems	Initial	Final	Initial	Final	Initial	Final
Cu <sup>2+</sup> 50	50	Ca <sup>2+</sup>	0.14	11.13	0.14	9.46	0.14	6.82
Cu	50	pН	3.03	5.87	3.03	5.63	3.03	5.35
	50	Ca <sup>2+</sup>	0.34	25.61	0.34	19.74	0.34	8.95
Zn <sup>2+</sup>	50	pН	3.14	5.98	3.14	4.91	3.14	4.45
Zfl	100	Ca <sup>2+</sup>	0.34	48.74	0.34	39.49	0.34	24.06
	100	pН	3.14	6.52	3.14	6.36	3.14	5.92

Table 3.1 Change of  $Ca^{2+}$  and pH in the solution

The cause of the alkalinity effect by the slag will be further discussed in Section 4.1.2 (Page 56).

## 3.3 Competitive effect of the metallic ions

Since the affinity to adsorbent differs between each ion, it is assumed that the competitive effect between different ions exist during the adsorption process. The adsorption of  $Cu^{2+}$  by the slag that was saturated with  $Zn^{2+}$  is shown in Table 3.2.

The competitive effect of the elements for adsorption was verified by a reverse test which mixed the  $Zn^{2+}$  solution with the slag that was saturated with  $Cu^{2+}$ . The results are shown in Table 3.3.

It was shown that the slag still had the ability to remove  $Zn^{2+}$  even though it had been saturated by  $Cu^{2+}$ . However there was no  $Cu^{2+}$  released from the slag at any instance.

Mass of	Items	Fi	ne	Меа	lium	Сос	arse	
slag (g)	nems	Initial	Final	Initial	Final	Initial	Final	
	pН	3.04	4.72	3.04	4.47	3.04	4.35	
	$Cu^{2+}(mg L^{-1})$	17.32	1.17	17.32	3.39	17.32	3.07	
50	$Zn^{2+}(mg L^{-1})$	0	1.91	0	1.73	0	1.12	
50	$Ca^{2+}(mg L^{-1})$	0.07	1.04	0.07	0.36	0.07	0.27	
	Cu <sup>2+</sup> adsorbed (mg)	2.42		2.	2.09		14	
	Zn <sup>2+</sup> released (%)	8.51		12.87		21.	21.03	
	pН	3.04	6.30	3.04	4.91	3.04	4.66	
	$Cu^{2+}(mg L^{-1})$	17.32	0.29	17.32	1.10	17.32	1.68	
100	$Zn^{2+}(mg L^{-1})$	0	1.21	0	1.99	0	2.08	
100	$Ca^{2+}(mg L^{-1})$	0.07	4.37	0.07	2.98	0.07	1.50	
	Cu <sup>2+</sup> adsorbed (mg)	2.55		2.43		2.35		
	$Zn^{2+}$ released (%)	5.	35	8.81		9.38		

Table 3.2  $Cu^{2+}$  removal by slag which is saturated with  $Zn^{2+}$ 

**Note:** A parallel sample using deionized water as control should have been included in these early tests. However the later desorption tests of the study showed that deionized water had little effect on the release of adsorbed metal ions from slag. Therefore the presence of  $Zn^{2+}$  in the solution can be expressed as the effect of ion-exchange by  $Cu^{2+}$  which has stronger affinity to the slag.

			-				
Mass of	Items	Fi	ne	Mea	lium	Сос	arse
slag (g)	nems	Initial	Final	Initial	Final	Initial	Final
	pН	4.15	5.74	4.15	5.62	4.15	5.20
	$Cu^{2+}(mg L^{-1})$	0	0	0	0	0	0
25	$Zn^{2+}(mg L^{-1})$	18.65	3.50	18.65	6.00	18.65	10.16
23	$Ca^{2+}(mg L^{-1})$	0	5.47	0	5.29	0	2.85
	Zn <sup>2+</sup> adsorbed (mg)	2.27		1.90		1.27	
	Cu <sup>2+</sup> released (%)	0		0		0	
	pН	4.15	6.53	4.15	6.43	4.15	5.99
	$Cu^{2+}(mg L^{-1})$	0	0	0	0	0	0
50	$Zn^{2+}(mg L^{-1})$	18.65	1.42	18.65	1.75	18.65	4.16
30	$Ca^{2+}(mg L^{-1})$	0	13.07	0	12.78	0	7.90
	Zn <sup>2+</sup> adsorbed (mg)	2.59		2.54		2.17	
	Cu <sup>2+</sup> released (%)	(	)	0		0	

Table 3.3  $Zn^{2+}$  removal by slag which is saturated with  $Cu^{2+}$ 

# 3.4 Adsorption isotherm

The results of the adsorption experiments are presented in Appendix A.

# 3.4.1 Adsorption capacity

The capacity of slag aggregate to remove metallic ions from a working solution can be calculated from the reduction of ion concentration in the solution after the adsorption tests. The working solutions containing each metallic ion and the mixture of the three elements in the same concentration were used in the batch tests.

r	Tuble 5.4 Ausorption cupacity of meter stag to cach for									
I.	tems	Metallic ions Adsorption Co	apacity in 24 hours (mg/g)							
11	ems	Single-ion solution	Multi-ion solution							
	Fine	$0.19 \pm 0.02$	$0.07 \pm 0.02$							
$\mathrm{Cd}^{2+}$	Medium	$0.15 \pm 0.02$	$0.07 \pm 0.01$							
	Coarse	$0.09 \pm 0.02$	$0.04 \pm 0.01$							
	Fine	$0.28 \pm 0.03$	$0.22 \pm 0.06$							
Zn <sup>2+</sup>	Medium	$0.14 \pm 0.04$	$0.14 \pm 0.03$							
	Coarse	$0.09 \pm 0.02$	$0.09 \pm 0.04$							
	Fine	$0.66 \pm 0.09$	$1.00 \pm 0.27$							
Cu <sup>2+</sup>	Medium	$0.79 \pm 0.19$	$0.90 \pm 0.17$							
	Coarse	$0.46 \pm 0.08$	$0.64 \pm 0.08$							

 Table 3.4
 Adsorption capacity of melter slag to each ion

Table 3.5 Adsorption efficiency of melter slag to each ion

I	tems	Metallic ions Adsorption e	efficiency in 24 hours (%)		
11	ems	Single-ion solution	Multi-ion solution		
	Fine	$19 \pm 6$	$6 \pm 2$		
$\mathrm{Cd}^{2+}$	Medium	$12 \pm 3$	$6 \pm 1$		
	Coarse	$8\pm 2$	$3 \pm 1$		
	Fine	$24 \pm 6$	$19 \pm 5$		
$Zn^{2+}$	Medium	$12 \pm 3$	$12 \pm 4$		
	Coarse	$9\pm3$	$8 \pm 4$		
	Fine	69 ± 13	$79\pm 8$		
$\mathrm{Cu}^{2^+}$	Medium	$70 \pm 12$	$76 \pm 11$		
	Coarse	$46 \pm 13$	$60 \pm 14$		

### 3.4.2 Adsorption isotherm

The equilibrium relationship between the amount of substance adsorbed and that remaining in solution is defined for a given set of conditions by an equation known as an adsorption isotherm. Depending on the theory used to model the adsorption, Langmuir and Freundilich formulae are most commonly used for the application of adsorbent in water treatment (Binnie *et al.*, 2002).

The Freundilich theory assumes that the equation for adsorption is:

$$x/m = K \times c_q^{1/n} \tag{1}$$

where

x = weight of material adsorbed

m = weight of adsorbent

 $c_q$  = concentration of material remaining in solution once equilibrium has been reached

K and n are constants.

Logarithmic transformation of equation (1) gives following equation.

$$ln(x/m) = lnK + (1/n) \times lnc_q$$
<sup>(2)</sup>

Plotting ln(x/m) against  $lnc_q$  on graph paper will give a straight line with a slope of 1/n and an intercept of lnK where  $lnc_q=1$ .

The results of metallic ion concentrations from three duplicates for each element  $(Zn^{2+}, Cu^{2+} \text{ and } Cd^{2+})$  are averaged respectively and used for plotting the isotherm curves. The linear isotherm derived from the curves is shown in Table 3.6.

	T.		Freundlich isotherm	
	Items		Formula	$R^2$
		Fine	$\ln(x/m) = -0.96 - 0.26 \times \ln Cq$	0.4205
n	Cd	Medium	$\ln(x/m) = -4.39 + 0.82 \times \ln Cq$	0.6062
utio	ntic	Coarse	$\ln(x/m) = -16.97 + 4.90 \times \ln Cq$	0.3614
solı	Single-ion solution	Fine	$\ln(x/m) = -3.61 + 0.83 \times \ln Cq$	0.5660
on		Medium	$\ln(x/m) = -7.82 + 2.01 \times \ln Cq$	0.2577
le-i		Coarse	$\ln(x/m) = -1.36 - 0.35 \times \ln Cq$	0.0127
[gui		Fine	$\ln(x/m) = -0.71 + 0.20 \times \ln Cq$	0.8816
S		Medium	$\ln(x/m) = -0.93 + 0.41 \times \ln Cq$	0.7776
		Coarse	$\ln(x/m) = -1.07 + 0.12 \times \ln Cq$	0.1991
		Fine	$\ln(x/m) = -24.70 + 7.44 \times \ln Cq$	0.3220
с	Cd	Medium	$\ln(x/m) = -49.13 + 15.48 \times \ln Cq$	0.7578
Itio		Coarse	$\ln(x/m) = -31.32 + 9.48 \times \ln Cq$	0.1585
solu		Fine	$\ln(x/m) = -32.49 + 10.50 \times \ln Cq$	0.9608
s uc	Zn	Medium	$\ln(x/m) = -6.59 + 1.54 \times \ln Cq$	0.6308
ti-i(		Coarse	$\ln(x/m) = 14.54 - 5.68 \times \ln Cq$	0.4906
Multi-ion solution		Fine	$\ln(x/m) = -56.64 + 19.17 \times \ln Cq$	0.9168
2	Cu	Medium	$\ln(x/m) = -0.55 + 0.32 \times \ln Cq$	0.6623
		Coarse	$\ln(x/m) = -0.63 + 0.08 \times \ln Cq$	0.0943

Table 3.6 Adsorption isotherm of the slag to each ion

**Note:** *x* -- mass of metal adsorbed by slag (mg)

m -- mass of slag (g)

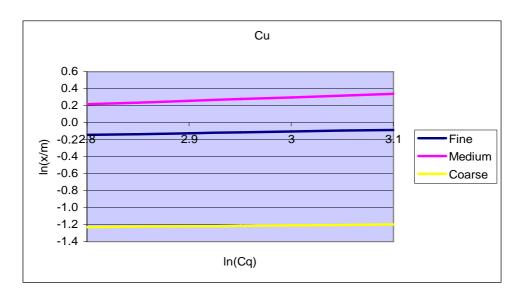
Cq - concentration of metal remaining in solution once equilibrium has been reached

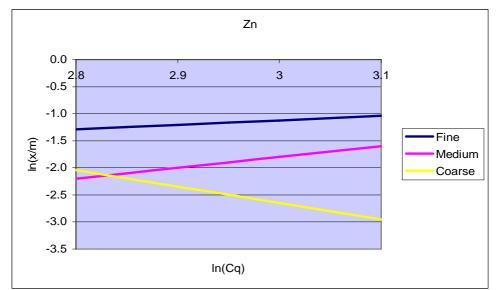
The adsorption isotherm curves under cross conditions are shown in Figure 3.4 to 3.7.

Note: ln(x/m) – Logarithmic transformation of the weight of metallic ion adsorbed by unit mass of slag

ln(Cq) – logarithmic transformation of concentration of metal remaining in solution once equilibrium has been reached

The analysis of the isotherms under different conditions is given in Section 4.1.4 (Page 60).





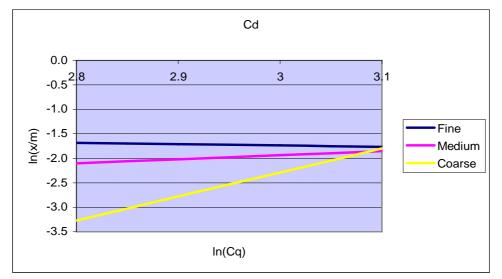
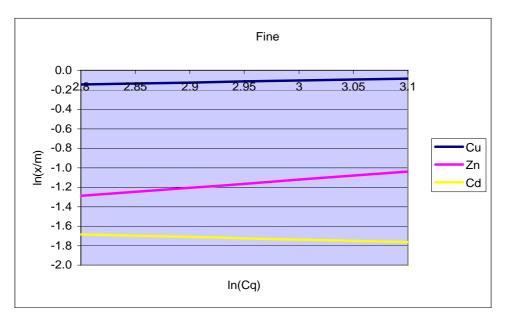
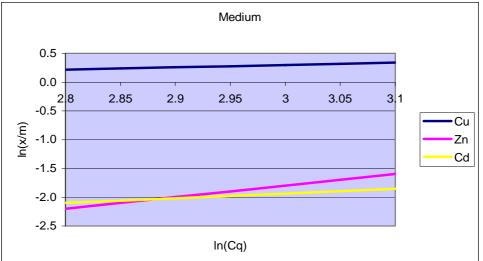


Figure 3.7 Adsorption isotherm of slag to each metallic ion (single-ion solution)





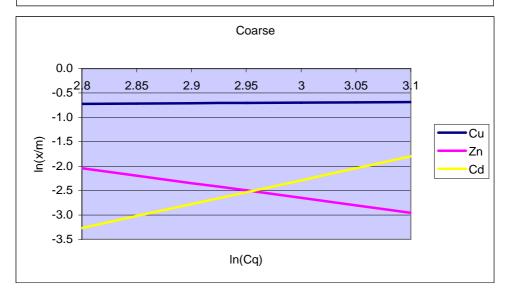


Figure 3.8 Adsorption isotherm of slag in different grain size (single-ion solution)

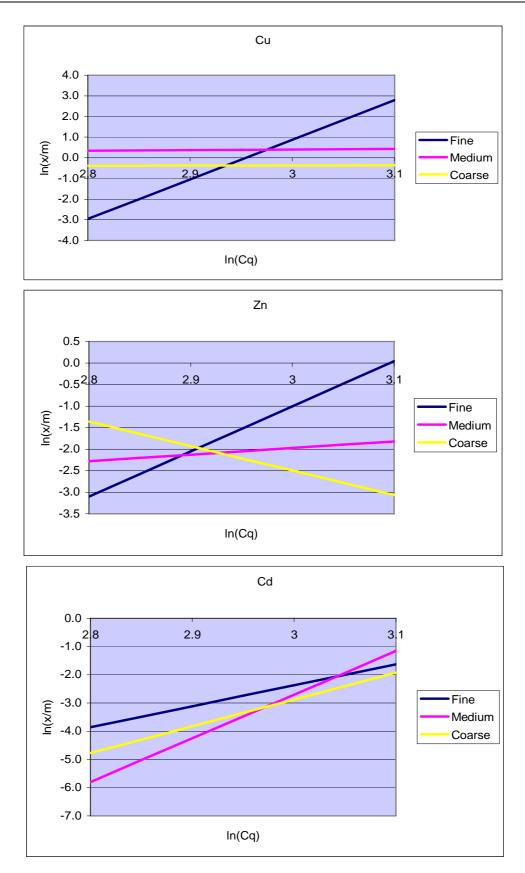
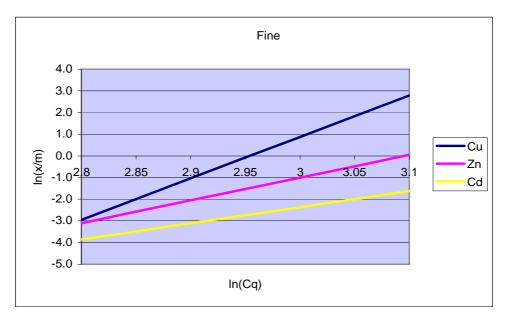
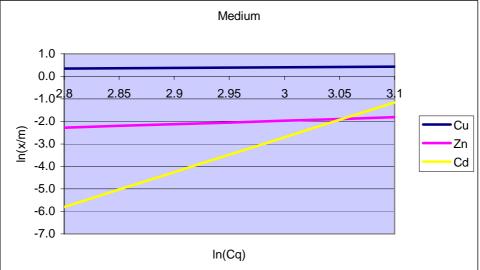
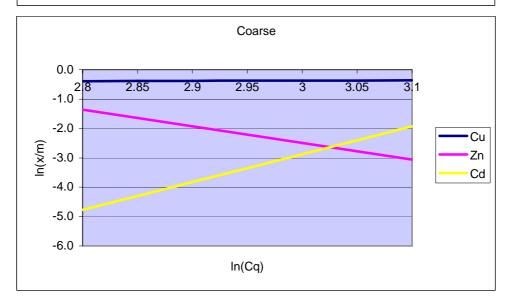
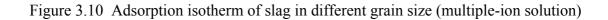


Figure 3.9 Adsorption isotherm of slag to each metallic ion (multiple-ion solution)









It was found that pH and  $Ca^{2+}$  concentrations in the solution at equilibrium are mutually dependent. This follows an exponential distribution under most conditions within the tested pH range.

Adsorbate	Grain size	Formula	$R^2$
	Fine	$b = 5.91a^{0.06}$	0.9829
$Zn^{2+}$	Medium	$b = 5.65a^{0.10}$	0.9749
	Coarse	$b = 5.76a^{0.09}$	0.9661
	Fine	$b = 6.32a^{0.04}$	0.9699
$\mathrm{Cd}^{2+}$	Medium	$b = 5.86a^{0.07}$	0.9839
	Coarse	$b = 5.74a^{0.11}$	0.9720
	Fine	$b = 4.77a^{0.03}$	0.5973
Cu <sup>2+</sup>	Medium	$b = 4.53a^{0.03}$	0.5696
	Coarse	$b = 4.71a^{0.01}$	0.8154
	1 •	, ,· · 1,·	

Table 3.7 Relationship between pH and equilibrium concentration of  $Ca^{2+}$ 

Note: a –calcium concentration in solution

b – pH of the equilibrium solution

The alkaline effect of the aggregate is discussed in Section 4.1.2 (Page 57).

### 3.5 Desorption

Desorption is the reverse process to adsorption. The used slag can be recovered by separating the adsorbed ions from the adsorbent with the assistance of desorption agent. The most commonly applied mechanisms for desorption are:

- Competitive adsorption ٠
- Ion-exchange ٠
- Chelation

A range of inorganic salt, sea water, organic and inorganic acid was selected as desorbent. The efficiencies of each substance or the combination to recover the used slag are shown in the following tables.

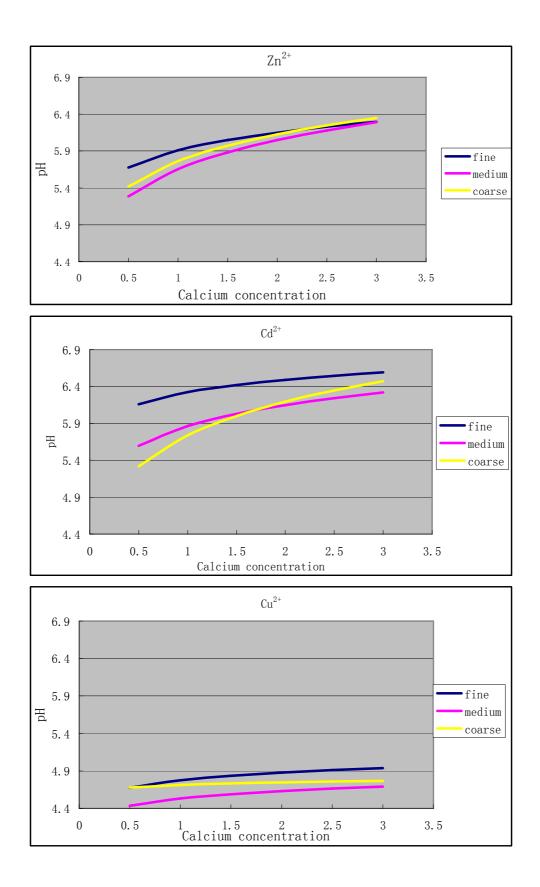


Figure 3.11 Variation of pH in the solutions with the change of  $Ca^{2+}$  concentration

			5	0		<u> </u>		,	
Metallic ion	Cd			Zn			Cu		
Desorption Agent	KCl	NaCl	NaNO <sub>3</sub>	KCl	NaCl	NaNO <sub>3</sub>	KCl	NaCl	NaNO <sub>3</sub>
η <sub>1</sub> (%)	41 ± 6	$40 \pm 9$	$46 \pm 22$	10 ± 5	9 ± 3	$7 \pm 1$	0	0	0
$\eta_2$ (%)	$48\pm8$	47 ± 12	50 ± 23	11 ± 6	$10 \pm 3$	8 ± 1	0	0	0
$\eta_1/\eta_2$ (%)	85 ± 4	84 ± 3	90 ± 2	93 ± 2	88 ± 4	88 ± 4		-	_

Table 3.8 Recovery of used slag with inorganic salts (0.01M)

In which:  $\eta_1 = \text{Recovery in } 24 \text{ h} (1 \text{ st run}, \%)$ 

 $\eta_2$  = Accumulative recovery in 96 h (2nd run, %)

 $\eta_1/\eta_2$  = Efficiency of the 1st run (%)

Table 3.9 Recovery of used slag with inorganic salt and organic acid

Metallic ion	Cd				Zn			Си		
Desorption Agent	KCl	0.02 M Citric Acid	KCl + 0.02M Citric Acid	KCl	0.02 M Citric Acid	KCl + 0.02M Citric Acid	0.02 M Citric Acid	KCl + 0.02M Tartaric Acid	KCl + 0.02M Citric Acid	
η <sub>1</sub> (%)	$19 \pm 4$	46 ± 13	46 ± 5	8 ± 3	65 ± 26	108 ± 12	23 ± 15	27±12	32 ± 10	
$\eta_2$ (%)	$26\pm 8$	52 ± 12	52± 5	$10 \pm 3$	69 ± 29	118 ± 10	35±23	39 ± 19	39 ± 12	
$\eta_1/\eta_2$ (%)	72 ± 5	88 ± 7	88 ± 5	81 ± 4	94 ± 3	92 ± 3	69 ± 18	$70\pm 8$	81 ± 15	

Table 3.10 Recovery of used slag with seawater and organic acid

Metallic ion	Cd			Zn			Си		
Desorption Agent	SW	SW + 0.02M Citric Acid	SW + 0.02M Tartaric Acid	SW	SW + 0.02M Citric Acid	SW + 0.02M Tartaric Acid	SW	SW + 0.02M Citric Acid	SW + 0.02M Tartaric Acid
η <sub>1</sub> (%)	41 ± 16	50 ± 15	49 ± 15	3 ± 1	56 ± 21	97 ± 21	4 ± 1	$50\pm9$	49 ± 13
$\eta_2(\%)$	50 ± 18	55 ± 16	53 ± 17	5 ± 2	61 ± 22	106 ± 23	6 ± 2	54±11	52 ± 13
$\eta_1/\eta_2$ (%)	81±5	92 ± 3	94 ± 2	$56 \pm 9$	91 ± 3	91 ± 2	64 ± 2	94 ± 2	94 ± 2

Note: SW - Seawater

Table 5.11 Recovery of used stag with carbonic acid										
Items	Ca	!	Zn	!	Си					
nems	$DI + CO_2$	$SW + CO_2$	$DI + CO_2$	$SW + CO_2$	$DI + CO_2$	$SW + CO_2$				
Recovery in 1 h (%)	33	39	42	27	11	18				
Recovery in 4 h (%)	44	51	66	42	16	23				
Recovery in 24 h (%)	61	66	124	84	18	25				

Table 3.11 Recovery of used slag with carbonic acid

Note: DI - Deionized water

Li anu a	C	<sup>2</sup> d	Z	In	Си		
Items	5% HNO <sub>3</sub>	10% HNO <sub>3</sub>	5% HNO <sub>3</sub>	10% HNO <sub>3</sub>	5% HNO <sub>3</sub>	10% HNO <sub>3</sub>	
Recovery in 24 h (%)	50 ± 31	65 ± 28	$63 \pm 40$	54 ± 26	85 ± 47	$60 \pm 5$	

Table 3.12 Recovery of used slag with nitric acid

### 3.6 Column test

The column tests were conducted to determine the removal efficiency of suspended solids and metallic ions by slag which is used as filtrate in the glass column. Stormwater samples from residential and commercial areas were collected and tested as analyte in this study. Also the column tests which used synthetic stormwater samples as filtrate were carried out to determine the removal efficiency of SS,  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  which were adjusted to higher level ( $Cd^{2+}$  and  $Zn^{2+}$ ) or similar to actual stormwater sample (SS and  $Cu^{2+}$ ).

# 3.6.1 Stormwater sample analysis

The concentrations of suspended solids and metallic ions in the stormwater samples are as follows:

	aa	$Cd^{2+}$ (µg $L^{-1}$ )		$Zn^{2+}$ (µg $L^{-1}$ )			$Cu^{2+}$ (µg $L^{-1}$ )			
Items	SS (mg/kg)	Dig	gested	Non-	Dige	ested	non-	Dige	ested	Non-
	(	Cs	Cx	digestion	Cs	Cx	digestion	Cs	Cx	digestion
Oakley Creek	Oakley Creek Site 1 4.98	0	1	0	61	99	52	0	4	0
Site 1			1	0	160		53	4		0
Oakley Creek	79.01	0	2	0	142	172	154	0	42	32
Site 2	/9.01	2		0	314		134	4	2	52
Hobson St.	421.01	0	0	0	193	689	200	0	134	(7
	431.81		0	0	882		386	134		67

Table 3.13 Characteristics of the stormwater samples

**Note:** *Cs* = soluble ions in the solution

Cx = extractable ions attached to suspended solids

# 3.6.2 Effect of nitric acid digestion

The suspended solids, either organic or inorganic, in the sample may adsorb part of the metallic ions (McFarlane *et al.*, 2005). For the tested elements ( $Cd^{2+}$ ,  $Cu^{2+}$  and  $Zn^{2+}$ ) there are few significant chemical interference effects for environmental samples in th air-aceteylee flame (Haswell, 1991). The physical characteristics of the samples are different from those of standards in terms of the surface tension and viscosity. The solids exist in the droplets of the sample after nebulization and may cause physical interferences (Lajunen, 1992). Also the elements attached to the solids may not be atomized in the flame of AAS due to the low temperature (2450 K) and short residence time. Therefore it is accurate to digest the sample to obtain the actural concentration of the elements in the samples. The filtrate of the column test was analyzed with and without nitric acid digestion. The comparisons of the results are as follows:

Ita	Items	Stormwater	Contro	ol column	Coars	e slag	Mediur	n slag	Fine	slag
ne	ms	Stormwater	С.	η	С.	η	С.	η	С.	η
pl	H	7.00	7	7.30	7.3	31	7.3	37	7.	46
Cd <sup>2+</sup>	N/D	0.000	0.000	/	0.000	/	0.000	/	0.000	/
Cu	DGT	0.000	0.000	/	0.000	/	0.000	/	0.000	/
Zn <sup>2+</sup>	N/D	0.386	0.346	10.29	0.267	30.88	0.182	52.94	0.125	67.65
ZII	DGT	0.882	0.681	22.74	0.341	61.37	0.346	60.73	0.204	76.82
Cu <sup>2+</sup>	N/D	0.067	0.067	0.00	0.067	0.00	0.067	0.00	0.034	50.00
Cu	DGT	0.134	0.134	0.00	0.067	49.95	0.067	49.95	0.000	100.00

Table 3.14 Efficiency of nitric acid digestion

Note:

C = concentration of the elements (mg L<sup>-1</sup>)

 $\eta$  = removal efficiency (%) N/D = non-digestion DGT = nitric acid digested

3.6.3 Removal efficiency of metallic ions in column test

Although there are differences between the analysis with and without digestion of the filtrate, the analysis of the sample was conducted without digestion for the following considerations:

- There is insufficient space in the muffle furnace in the lab to process that much analyte (16 samples in one batch);
- The removal efficiency obtained from the digestion analysis are in consistent with those without digestion, i.e. the finer the slag, the smaller the difference in removal efficiency.

The removal efficiency of each stormwater sample in the column experiments are presented in the following tables. The interpretation of the results are given in Section 4.3.

Itom	~	Ì	Removal efficiency (%	%)
Item.	S	$Cd^{2+}$	$Zn^{2+}$	$Cu^{2+}$
	1 <sup>st</sup> h	_	77.78	-
Oakley	2 <sup>nd</sup> h	_	77.78	-
Creek Site 1	3 <sup>rd</sup> h	_	77.78	-
(3.9cm/min)	4 <sup>th</sup> h	_	77.78	—
	Average	_	78	-
	1 <sup>st</sup> h	_	84.62	100.00
Oakley	2 <sup>nd</sup> h	_	76.92	100.00
Creek Site 2	3 <sup>rd</sup> h	_	76.92	100.00
(3.9cm/min)	4 <sup>th</sup> h	_	76.92	100.00
	Average	_	$79 \pm 6$	100.00
	1 <sup>st</sup> h	_	65.38	100.00
Oakley	2 <sup>nd</sup> h	_	76.92	100.00
Creek Site 2	3 <sup>rd</sup> h	_	76.92	100.00
(8.9cm/min)	4 <sup>th</sup> h	_	73.08	100.00
	Average	_	$73 \pm 9$	100.00
	1 <sup>st</sup> h	_	76.47	100.00
Habaan St	2 <sup>nd</sup> h	_	66.18	50.00
Hobson St. (2.0cm/min)	3 <sup>rd</sup> h	_	64.71	100.00
(2.0011/11111)	4 <sup>th</sup> h	_	72.06	100.00
	Average	—	$70 \pm 9$	$88 \pm 40$
	1 <sup>st</sup> h		67.65	100.00
Hohson St	2 <sup>nd</sup> h	_	55.88	100.00
Hobson St. (3.9cm/min)	3 <sup>rd</sup> h	_	45.59	0.00
(3.3011/11111)	4 <sup>th</sup> h		45.59	50.00
	Average	_	$54 \pm 17$	/

Table 3.15 Metal ion removal efficiency by slag

(fine size  $0.6 \sim 1.0 \text{ mm}$ )

Note: "-" = amount too small to be determined by this technique

			(	512 <b>C</b> 1.0 2.0 mm)
Item	a	R	emoval efficiency (%	6)
nem.		$Cd^{2+}$	$Zn^{2+}$	$Cu^{2+}$
	1 <sup>st</sup> h	_	77.78	_
Oakley	2 <sup>nd</sup> h	_	100	_
Creek Site 1	3 <sup>rd</sup> h	_	88.89	_
(3.9cm/min)	4 <sup>th</sup> h	_	88.89	_
	Average	_	89 ± 14	-
	1 <sup>st</sup> h	_	76.92	100.00
Oakley	2 <sup>nd</sup> h	_	65.38	100.00
Creek Site 2	3 <sup>rd</sup> h	_	73.08	100.00
(3.9cm/min)	4 <sup>th</sup> h	_	73.08	100.00
	Average	_	$72 \pm 8$	100.00
	1 <sup>st</sup> h	_	50.00	100.00
Oakley	2 <sup>nd</sup> h	—	46.15	100.00
Creek Site 2	3 <sup>rd</sup> h	_	53.85	100.00
(8.9cm/min)	4 <sup>th</sup> h	—	50.00	100.00
	Average	—	$50 \pm 5$	100.00
	1 <sup>st</sup> h	_	64.71	100.00
Hohaon St	2 <sup>nd</sup> h	—	55.88	50.00
Hobson St. (2.0cm/min)	3 <sup>rd</sup> h	—	48.53	50.00
(2.0011/11111)	4 <sup>th</sup> h	—	55.88	100.00
	Average	—	$56 \pm 11$	$75 \pm 46$
	1 <sup>st</sup> h	_	52.94	0.00
Hobson St.	2 <sup>nd</sup> h		48.53	100.00
(3.9  cm/min)	3 <sup>rd</sup> h	_	42.65	0.00
(3.3011/11111)	4 <sup>th</sup> h	_	30.88	100.00
	Average	_	$44 \pm 15$	/

Table 3.16 Metal ion removal efficiency by slag

(medium size  $1.0 \sim 2.0 \text{ mm}$ )

			(000150 51	,
Item	g	R	emoval efficiency (%	6)
nem	3	$Cd^{2+}$	$Zn^{2+}$	$Cu^{2+}$
	1 <sup>st</sup> h	—	33.33	-
Oakley	$2^{nd}h$	_	55.56	_
Creek Site 1	3 <sup>rd</sup> h	_	66.67	_
(3.9cm/min)	4 <sup>th</sup> h	_	66.67	-
	Average	_	$56 \pm 25$	_
	1 <sup>st</sup> h	_	73.08	0.00
Oakley	2 <sup>nd</sup> h	_	61.54	0.00
Creek Site 2	3 <sup>rd</sup> h	_	69.23	0.00
(3.9cm/min)	4 <sup>th</sup> h	_	61.54	0.00
	Average	_	$66 \pm 9$	0.00
	1 <sup>st</sup> h	_	23.08	100.00
Oakley	2 <sup>nd</sup> h	_	38.46	100.00
Creek Site 2	3 <sup>rd</sup> h	_	30.77	100.00
(8.9cm/min)	4 <sup>th</sup> h	_	34.62	100.00
	Average	_	$32 \pm 10$	100.00
	1 <sup>st</sup> h	_	35.29	50.00
Habsen St	2 <sup>nd</sup> h	_	35.29	50.00
Hobson St. (2.0cm/min)	3 <sup>rd</sup> h	_	36.76	50.00
(2.0011/11111)	4 <sup>th</sup> h	_	35.29	50.00
	Average		$36 \pm 1$	50.00
	1 <sup>st</sup> h	_	30.88	0.00
Hohson St	2 <sup>nd</sup> h		35.29	50.00
Hobson St.	3 <sup>rd</sup> h	_	29.41	50.00
(3.9cm/min)	4 <sup>th</sup> h	_	14.71	0.00
	Average		$28 \pm 14$	/

Table 3.17 Metal ion removal efficiency by slag

(coarse size  $2.0 \sim 3.35 \text{ mm}$ )

L		R	emoval efficiency (%	<i>(o)</i>
Item.	8	$Cd^{2+}$	$Zn^{2+}$	$Cu^{2+}$
	1 <sup>st</sup> h	_	33.33	_
Oakley	2 <sup>nd</sup> h	_	0	-
Creek Site 1	3 <sup>rd</sup> h	_	44.44	_
(3.9cm/min)	4 <sup>th</sup> h	—	44.44	—
	Average	—	$31 \pm 33$	—
	1 <sup>st</sup> h	—	30.77	0.00
Oakley	$2^{nd}$ h	—	50.00	0.00
Creek Site 2	3 <sup>rd</sup> h	_	50.00	0.00
(3.9cm/min)	4 <sup>th</sup> h	—	46.15	0.00
	Average	—	$44 \pm 15$	0.00
	1 <sup>st</sup> h	_	19.23	50.00
Oakley	$2^{nd}$ h	—	15.38	100.00
Creek Site 2	3 <sup>rd</sup> h	_	26.92	50.00
(8.9cm/min)	4 <sup>th</sup> h	_	23.08	100.00
	Average	—	$21 \pm 8$	$75 \pm 46$
	1 <sup>st</sup> h	—	10.29	50.00
Hobson St.	2 <sup>nd</sup> h	_	11.76	50.00
(2.0  cm/min)	3 <sup>rd</sup> h	—	8.82	50.00
(2.0011/11111)	4 <sup>th</sup> h	_	20.59	50.00
	Average	_	$13 \pm 8$	50.00
	1 <sup>st</sup> h	_	10.29	0.00
Hobson St.	$2^{nd}$ h	_	14.71	50.00
(3.9  cm/min)	3 <sup>rd</sup> h	_	17.65	0.00
(3.3011/11111)	4 <sup>th</sup> h	_	10.29	50.00
	Average	_	$13 \pm 6$	/

 Table 3.18
 Results of the control lane in the column test

### 3.6.4 Column test results with synthetic stormwater samples

The synthetic stormwater sample was prepared from clay and metal solutions. The negatively charged surface structure of natural clay is usually compensated for by exchangeable cations, such as Na<sup>+</sup> and Ca<sup>2+</sup> (Pushpaletha *et al.*, 2005). The ion exchange may occur once the clay is exposed to the solution containing metallic ions, such as Cu<sup>2+</sup>, Cd<sup>2+</sup> and Zn<sup>2+</sup>. The chemical analysis of the samples is shown in Table 3.19.

	Sample	A	В	С
	$\mathrm{Cd}^{^{2+}}$	0.095	0.079	0.024
1	Zn <sup>2+</sup>	0.322	0.290	0.076
	$Cu^{2+}$	0.164	0.143	0.041
	$\mathrm{Cd}^{^{2+}}$	0.076	0.061	0.015
2	Zn <sup>2+</sup>	0.623	0.508	0.142
	$Cu^{2+}$	0.122	0.082	0.020
	$\mathrm{Cd}^{2^+}$	0.122	0.081	0.041
3	Zn <sup>2+</sup>	0.859	0.690	0.252
	Cu <sup>2+</sup>	0.141	0.070	0.018

 Table 3.19
 Chemical analysis of synthetic stormwater samples

**Note:** A = Metallic ions concentration of the original solution which is used to prepare the samples (mg  $L^{-1}$ );

B = Metallic ions concentration of the prepared samples tested directly by AA (mg  $L^{-1}$ );

C = Soluable metallic ions concentration in the samples filtrated by filter paper (Whatman, No. 42) (mg L<sup>-1</sup>)

The nominal concentrations of all the elements in synthetic samples are lower than 1 mg  $L^{-1}$ . The limited ability of the AAS to measure the sample with low concentration results in a significant error. A set of samples with different known level of metallic ions and 400 mg/kg SS were prepared. The concentrations of each element were measured directly by AAS. The results are shown in Table 3.20.

	Nominal		Sample withou	nt clay	Sample with clay			
Element	concentration	Measured Conc. (mg $L^{-1}$ )		Standard deviation	Measured (	Conc. (mg $L^{-1}$ )	Standard deviation	
	$(mg L^{-1})$	Average	Range	$(mg L^{-1})$	Average	Range	$(mg L^{-1})$	
	0.1	0.09	0.07~0.11	002	0.05	0.04~0.08	0.02	
	0.2	0.19	0.17~0.21	0.02	0.12	0.11~0.14	0.02	
Cd	0.5	0.54	0.52~0.57	0.02	0.42	0.40~0.45	0.03	
	1.0	1.00	0.97~1.03	0.03	0.81	0.80~0.82	0.01	
	2.0	2.00	1.96~2.03	0.04	1.72	1.69~1.75	0.03	
	5.0	4.78	4.73~4.82	0.05	4.34	4.29~4.39	0.05	
	0.1	0.09	0.07~0.11	0.02	0.08	0.07~0.10	0.01	
	0.2	0.19	0.17~0.21	0.02	0.16	0.15~0.18	0.01	
Zn	0.5	0.53	0.51~0.55	0.02	0.46	0.45~0.48	0.01	
	1.0	1.02	1.01~1.04	0.01	0.87	0.86~0.89	0.01	
	2.0	2.01	2.00~2.03	0.01	1.90	1.87~1.92	0.03	
	5.0	4.87	4.81~4.93	0.06	4.70	4.64~4.79	0.08	
	0.1	0.08	0.02~0.14	0.06	0.06	0.00~0.12	0.06	
	0.2	0.16	0.10~0.22	0.06	0.14	0.08~0.20	0.06	
Cu	0.5	0.47	0.41~0.53	0.06	0.51	0.45~0.57	0.06	
	1.0	1.00	0.93~1.06	0.06	0.93	0.87~1.00	0.06	
	2.0	2.01	1.95~2.07	0.06	1.97	1.91~2.03	0.06	
	5.0	5.00	4.92~5.08	0.08	4.90	4.82~4.98	0.08	

 Table 3.20
 Measurement errors of the AAS

It is shown in the above table that the lower the concentrion, the larger the error of the measurements. It is valid to the samples with or without clay. The impact of such error on the conclusion will be further discussed in Chapter 4.

The removal efficiencies of SS and metallic ions from synthetic stormwater samples in the column experiments are presented in the following tables.

	(0	6-1.0 mm) from synthetic stormwater samples					
Iten	иc	Removal efficiency (%)					
nen		SS (%)	Cd (%)	Zn (%)	Cu (%)		
	1 <sup>st</sup> h	28.49	80.00	90.41	85.71		
Samula 1	$2^{nd}h$	21.39	60.00	87.67	85.71		
Sample 1 $(44 \text{ m}/\text{h})$	3 <sup>rd</sup> h	21.05	70.00	87.67	100.00		
(44 ml/h)	4 <sup>th</sup> h	11.98	70.00	80.82	85.71		
	Average	$21 \pm 11$	$70 \pm 13$	$87 \pm 7$	$89 \pm 11$		
	1 <sup>st</sup> h	37.39	50.00	61.29	75.00		
Sample 2	$2^{nd}h$	17.17	50.00	70.97	75.00		
Sample 2 $(44 \text{ m}/\text{h})$	3 <sup>rd</sup> h	14.97	37.50	59.14	75.00		
(44 ml/h)	4 <sup>th</sup> h	10.95	50.00	58.07	75.00		
	Average	$20 \pm 19$	$47 \pm 10$	$62 \pm 9$	75		
	1 <sup>st</sup> h	68.40	90.91	86.99	90.00		
Sample 2	$2^{nd}h$	45.28	90.91	85.37	80.00		
Sample 3 (22 ml/h)	3 <sup>rd</sup> h	42.10	90.91	81.30	60.00		
(22 111/11)	4 <sup>th</sup> h	36.82	81.82	82.11	80.00		
	Average	$48 \pm 22$	$89 \pm 7$	$84 \pm 4$	$78 \pm 20$		
	1 <sup>st</sup> h	36.85	90.00	73.81	100.00		
Sample 2	2 <sup>nd</sup> h	35.02	90.00	73.02	75.00		
Sample 3 (44 ml/h)	3 <sup>rd</sup> h	19.39	90.00	58.73	75.00		
(44 1111/11)	4 <sup>th</sup> h	11.29	70.00	46.83	75.00		
	Average	$26 \pm 19$	$85 \pm 16$	$63 \pm 20$	$81 \pm 20$		
	1 <sup>st</sup> h	20.30	54.55	40.17	90.00		
Sampla 2	$2^{nd}h$	10.63	54.55	40.17	90.00		
Sample 3 (100 ml/h)	3 <sup>rd</sup> h	7.90	45.45	29.91	70.00		
	4 <sup>th</sup> h	8.65	36.36	28.21	70.00		
	Average	$12 \pm 9$	$48 \pm 12$	$35 \pm 10$	$80 \pm 18$		

Table 3.21Suspended solids and metal ions removal efficiency by fine slag<br/>(0.6-1.0 mm) from synthetic stormwater samples

	siag	(1.0-2.0 mm) from synthetic stormwater samples					
Iten	MG	Removal efficiency (%)					
nen		SS (%)	Cd (%)	Zn (%)	Cu (%)		
	1 <sup>st</sup> h	21.51	60.00	65.75	71.73		
Samula 1	2 <sup>nd</sup> h	14.97	50.00	60.27	71.73		
Sample 1 (44 ml/h)	3 <sup>rd</sup> h	23.68	50.00	69.86	85.71		
(44 111/11)	4 <sup>th</sup> h	23.96	50.00	58.90	85.71		
	Average	$21 \pm 7$	$53 \pm 8$	$64 \pm 8$	$79 \pm 13$		
	1 <sup>st</sup> h	23.44	50.00	38.71	75.00		
Sample 2	2 <sup>nd</sup> h	17.47	12.50	43.01	75.00		
Sample 2 (44 ml/h)	3 <sup>rd</sup> h	16.17	12.50	36.56	75.00		
(44 111/11)	4 <sup>th</sup> h	15.68	12.50	35.48	75.00		
	Average	$18 \pm 6$	$22 \pm 29$	$38 \pm 5$	75		
	1 <sup>st</sup> h	46.45	54.55	63.41	60.00		
Sample 3	$2^{nd}$ h	38.27	81.82	63.41	70.00		
(22  ml/h)	3 <sup>rd</sup> h	35.15	72.73	59.35	60.00		
(22 111/11)	4 <sup>th</sup> h	32.16	45.45	55.28	60.00		
	Average	$38 \pm 9$	$64 \pm 26$	$60 \pm 6$	$63 \pm 8$		
	1 <sup>st</sup> h	31.51	90.00	46.83	75.00		
Sample 2	2 <sup>nd</sup> h	23.70	80.00	39.68	75.00		
Sample 3 (44 ml/h)	3 <sup>rd</sup> h	13.01	50.00	31.75	75.00		
(44 111/11)	4 <sup>th</sup> h	10.09	60.00	26.98	75.00		
	Average	$20 \pm 15$	$70 \pm 29$	$36 \pm 14$	75		
	1 <sup>st</sup> h	12.62	27.27	23.08	70.00		
Sample 3	2 <sup>nd</sup> h	7.22	27.27	23.08	70.00		
(100  ml/h)	$3^{rd}h$	5.72	27.27	11.11	60.00		
(100 mi/n)	4 <sup>th</sup> h	5.00	9.09	14.53	50.00		
	Average	8 ± 5	$23 \pm 14$	$18 \pm 9$	$63 \pm 15$		

Table 3.22Suspended solids and metal ions removal efficiency by mediumslag (1 0-2 0 mm) from synthetic stormwater samples

(2.0-3.35 mm) from synthetic stormwater samples						
Iten	MG		Removal eff	ficiency (%)		
nen		SS (%)	Cd (%)	Zn (%)	Cu (%)	
	1 <sup>st</sup> h	12.90	30.00	38.36	71.43	
Samela 1	2 <sup>nd</sup> h	24.06	20.00	30.14	85.71	
Sample 1 (44 ml/h)	3 <sup>rd</sup> h	8.42	20.00	50.68	85.71	
(44 111/11)	4 <sup>th</sup> h	15.63	20.00	36.99	85.71	
	Average	$15 \pm 10$	$23 \pm 8$	$39 \pm 13$	$82 \pm 11$	
	1 <sup>st</sup> h	15.73	12.50	22.58	75.00	
Samula 2	2 <sup>nd</sup> h	17.47	12.50	21.51	75.00	
Sample 2 $(44 \text{ m}^{1/h})$	3 <sup>rd</sup> h	12.28	0.00	18.28	50.00	
(44 ml/h)	4 <sup>th</sup> h	10.06	0.00	20.43	50.00	
	Average	$14 \pm 5$	$6 \pm 11$	$21 \pm 3$	$63 \pm 23$	
	1 <sup>st</sup> h	34.52	45.45	28.46	50.00	
Sample 3	$2^{nd}$ h	32.78	54.55	35.77	60.00	
(22  ml/h)	3 <sup>rd</sup> h	33.25	54.55	37.40	50.00	
(22 III/II)	4 <sup>th</sup> h	30.26	45.45	30.08	50.00	
	Average	$33 \pm 3$	$50\pm 8$	$33 \pm 7$	$53 \pm 8$	
	1 <sup>st</sup> h	22.96	60.00	38.10	75.00	
Samula 2	2 <sup>nd</sup> h	24.50	70.00	30.95	75.00	
Sample 3 (44 ml/h)	3 <sup>rd</sup> h	19.52	50.00	24.60	75.00	
(44 111/11)	4 <sup>th</sup> h	11.29	50.00	21.43	50.00	
	Average	$20 \pm 9$	$58 \pm 15$	$29 \pm 12$	$69 \pm 20$	
	1 <sup>st</sup> h	11.66	27.27	7.69	80.00	
Sampla 2	2 <sup>nd</sup> h	11.17	18.18	11.97	80.00	
Sample 3 (100 ml/h)	3 <sup>rd</sup> h	8.45	9.09	9.40	60.00	
	4 <sup>th</sup> h	1.22	9.09	7.69	40.00	
	Average	$8\pm7$	$16 \pm 13$	9 ± 3	$65 \pm 30$	

Table 3.23Suspended solids and metal ions removal efficiency by coarse slag(2.0-3.35 mm) from synthetic stormwater samples

		synthetic stormwater samples					
Iten		Control					
Iten		SS (%)	Cd (%)	Zn (%)	Cu (%)		
	1 <sup>st</sup> h	27.42	20.00	19.18	0.00		
Samula 1	2 <sup>nd</sup> h	21.93	0.00	15.07	14.29		
Sample 1 (44 ml/h)	3 <sup>rd</sup> h	15.26	0.00	20.55	14.29		
(44 111/11)	$4^{th}h$	7.29	0.00	17.81	42.86		
	Average	$18 \pm 13$	/	$18 \pm 4$	/		
	1 <sup>st</sup> h	16.91	0.00	8.60	50.00		
Samula 2	2 <sup>nd</sup> h	16.27	0.00	7.53	25.00		
Sample 2 (44 ml/h)	3 <sup>rd</sup> h	8.98	0.00	9.68	25.00		
(44 111/11)	4 <sup>th</sup> h	12.13	0.00	6.45	0.00		
	Average	$14 \pm 6$	0.00	$8\pm 2$	/		
	1 <sup>st</sup> h	30.46	18.18	13.82	0.00		
Sample 3	2 <sup>nd</sup> h	29.72	27.27	15.45	10.00		
(22  ml/h)	3 <sup>rd</sup> h	31.73	27.27	13.01	0.00		
(22 111/11)	$4^{th}h$	29.26	0.00	8.13	10.00		
	Average	$30 \pm 2$	/	$13 \pm 5$	/		
	1 <sup>st</sup> h	14.42	10.00	38.10	75.00		
Sample 2	2 <sup>nd</sup> h	22.90	10.00	15.87	75.00		
Sample 3 (44 ml/h)	3 <sup>rd</sup> h	11.16	0.00	5.56	0.00		
(44 111/11)	$4^{th}h$	12.08	0.00	3.17	50.00		
	Average	$15 \pm 8$	/	/	/		
	1 <sup>st</sup> h	6.45	9.09	4.27	40.00		
Sample 3	$2^{nd}h$	7.63	9.09	7.69	30.00		
(100  ml/h)	$3^{rd}h$	5.86	0.00	3.42	20.00		
	4 <sup>th</sup> h	6.62	9.09	3.42	20.00		
	Average	$7 \pm 1$	/	$5\pm3$	/		

 Table 3.24
 Results of the control lane in the column test with synthetic stormwater samples

# **Chapter 4 Discussion**

The results derived from the experiments are discussed in this section in light of existing knowledge and understanding of the issue. The heavy metals removal mechanisms and efficiency by the tested slag are the major concerns of the author as they are the basis for further application of the aggregate in stormwater/wastewater purification practice.

### 4.1 Metallic ion removal efficiency

### 4.1.1 Adsorption kinetics

Adsorption is usually composed of two distinct kinetic phases, i.e. a rapid and reversible initial stage followed by a much slower, non-reversible stage (Essington, 2004). The rapid phase usually includes the retention of compounds by easily accessible sites on macro-particles, and on the edges of slag. The slower reaction phase that follows the initial phase generally involves the formation of inner-sphere surface complexes with bonds that have a covalent character.

The batch tests showed that the temporal variations of  $Zn^{2+}$  and  $Cu^{2+}$  concentration in the working solutions after mixing with the slag were similar, i.e. an initial drop sharply followed by a gentle decreasing slope until becoming stable (Figure 3.1 and 3.2).  $Cu^{2+}$  reached equilibrium faster than  $Zn^{2+}$ . Also the  $Cu^{2+}$  concentration in the solution that reached equilibrium is lower than  $Zn^{2+}$  at the same solid/liquid ratio.

The tests on  $Zn^{2+}$  adsorption showed that by increasing the mass of slag from 50g/150mL to 100g/150mL (adsorbent/adsorbate ratio), it is possible to shorten the

The adsorption capacity of adsorbate is proportional to active sites on the surface of the aggregate (Oguz, 2004). Throughout these experiments, the removal capacity of slag is in proportion to its grain size. The fine slag had the greatest capacity to remove both  $Zn^{2+}$  and  $Cu^{2+}$  at any adsorbent/adsorbate ratio. A surface analysis showed that slag had a porous structure (Figure 4.1). Although the surface area of the slag in different sizes was not measured<sup>2</sup>, the finer size of the slag undoubtedly presented a larger surface area and consequently provided more active sites for reaction.

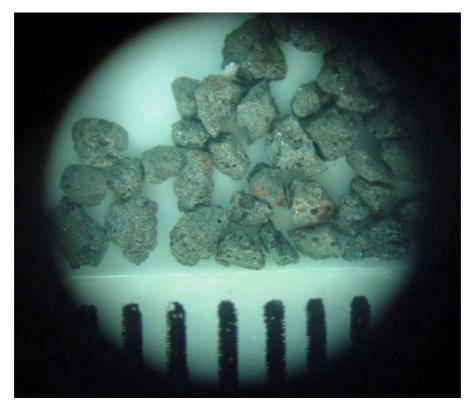


Figure 4.1 Porous structure of the slag (fine) viewed from microscope (Scale with 1 mm intervals)

<sup>&</sup>lt;sup>2</sup> Surface area is determined by gas absorption method. Such facility is not available in AUT chemistry lab

### 4.1.2 Heavy Metal ion removal mechanism

The leaching tests (New Zealand Steel, 1992) indicated that calcium and silicon were continuously released in water, with a consequent increase of pH of the effluent. The alkaline effect of the slag is a result of the release of hydroxyl ions by components such as gehlenite and  $2CaO \cdot SiO_2$  (Dimitrova *et al.*, 2000).

$$Ca_2Al_2SiO_7 + 5H_2O \Leftrightarrow SiO_2 + 2Al(OH)_3 + 2Ca^{2+} + 4OH^-$$
(4.1)

$$2\text{CaO} \cdot \text{SiO}_2 + 2\text{H}_2\text{O} \Leftrightarrow 2\text{Ca}^{2+} + \text{H}_4\text{SiO}_4 + 4\text{OH}^-$$
(4.2)

The above analysis is supported by the observed parallel increase of pH and  $Ca^{2+}$  in the solution (Table 3.7 and Figure 3.11). The extent of the reaction is dependent upon the grain size of the slag and the adsorbent/adsorbate ratio. Thus more OH<sup>-</sup> and Ca<sup>2+</sup> are released when grain size is in the coarse silt/sand range.

The exchange and displacement of the calcium ions in the slag by  $Cu^{2+}$ ,  $Zn^{2+}$  and  $Cd^{2+}$  resulted in an increase of  $Ca^{2+}$  concentration in the solution and a shift of the above equilibrium to the right due to the binding of OH<sup>-</sup> and heavy metal ions to form the complexes (Kang *et al.*, 2004).

One of the concerns of the metallic ions removal by slag is that whether the adsorption effect is simply the reaction of the ions with anions, e.g. to form insoluble hydroxide.

Every dissolved metal ion has a distinct pH at which the optimum hydroxide precipitation will occur (Greenwood and Earnshaw, 1997). The optimum pH for copper and zinc is 8.1 and 10.1 respectively. However the tests on the removal of metallic ions by marble (Table A.13 and 14) showed that:

- The concentration of the tested ions (Zn<sup>2+</sup> and Cu<sup>2+</sup>) in working solutions decreased more significantly by mixing with marble than slag in the first 15 minutes.
- The working solutions were shifted from acidic (pH  $\approx$  3) to neutral or alkaline (pH = 6.95 ~ 7.11) instantly by mixing with marble.
- Different grain sizes of marble presented similar removal efficiency (Figure 3.3 and 3.4).
- A definite decrease in metallic ion concentration after filtering the working solution was observed.

From the above observations, it is deduced that chemical reactions, i.e. dissolving  $CaCO_3$  in solution and followed by formulating ionic species (e.g.  $MCO_3^0$ ,  $MHCO_3^+$ ,  $M(OH)_n$ , etc.) with metallic ions, are the major mechanisms for the removal of heavy metals by marble. Hydroxide precipitation may occur under alkaline or even neutral conditions.

However the tests on slag were carried out in acidic solutions (range:  $3.1 \sim 5.9$ ), although the pH of the solutions had been increasing throughout the process (Table A.15 and 16). From this it is further deduced that the removal of heavy metals by slag in this instance is not due to metal hydroxide precipitation. This contravenes the conclusions reached by Menzlin (See Page 9).

### 4.1.3 Competitive adsorption

Adsorption from solution always involves competition between the components of a

liquid mixture for the adsorption sites (Shaw, 1980).

According to the batch tests (Table 3.4) which mixed the slag with each element for adsorption, the adsorption capacity, expressed as the amount of metallic ion removed from the solution by unit mass of slag, differs for each element. It followed the descending order of  $Cu^{2+} > Zn^{2+} \approx Cd^{2+}$ .

When the three tested elements ( $Cu^{2+}$ ,  $Zn^{2+}$  and  $Cd^{2+}$ ) were mixed in the solution as ions, the  $Cu^{2+}$  removal capacity was raised by 51%, 14% and 39% for fine, medium and coarse slag respectively. However both  $Zn^{2+}$  and  $Cd^{2+}$  adsorption capacities were reduced by 4% ~ 22% and 51% ~ 65% respectively.

The above observations reflect the competitive effect among the elements in the adsorption system. The binding capacity varies among the adsorption sites on the surface of the slag. Competitive effects can be significant even if a small fraction of the surface is occupied, since the sites that are preferentially occupied by one competing adsorbate might be the ones that bind to the other adsorbate most strongly. That is, if there are only a small number of strong binding sites on a solid, a relatively small concentration of strongly competing adsorbates might leave all these sites occupied. In that case, even though there are plenty of sites still available on the surface, the less competitive adsorbate would have to bind to relatively weak sites (Benjamin, 2002).

As shown in section 3.3,  $Cu^{2+}$  was a more competitive adsorbate than  $Zn^{2+}$ . The stronger binding energy of  $Cu^{2+}$  displaced  $Zn^{2+}$  and  $Cd^{2+}$  through ion-exchange. The higher adsorption capacity of  $Zn^{2+}$  as opposed to  $Cd^{2+}$  shows that the affinity of the three elements to the slag follows the descending order of  $Cu^{2+} > Zn^{2+} > Cd^{2+}$ . This assumption accords with the observed elevation of  $Cu^{2+}$  adsorption capacity from the multi-ion solutions and the more significant decrease of  $Cd^{2+}$  adsorption capacity than  $Zn^{2+}$ .

### 4.1.4 Adsorption isotherm

It has been demonstrated that pH, initial metallic ion concentration, temperature and solid/liquid ratio may have an impact on adsorption efficiency (Ortiz *et al.*, 2001; Ahn *et al.*, 2003; Kang *et al.*, 2004; Dimitrova *et al.*, 1998). In this study the pH, temperature and initial metallic ion concentration were set at 4, 23°C, 20 mg  $L^{-1}$  respectively. The variables included the grain size of the slag and solid/liquid ratio.

The adsorption isotherm describes the equilibrium status of a system. The Freundlich isotherm is summarized under different conditions in Table 3.6. In most instances, the amount of metallic ions removed from the solution by slag at equilibrium increased with the increase of solid/liquid ratio. However, the slopes of the curves (1/n) are small, and occasionally even negative. The horizontal lines mean that there is little or no functional dependence between Cq and x/m in these cases. The adsorption capacity, which was measured by metallic ion removal by unit mass of aggregate, decreased as the ratio rose.

Figures 3.5 and 3.6 present the isotherms of each element adsorbed by different size of slag. Usually the larger A and smaller 1/n in the Freundilich isotherm refers to a higher adsorption capacity of an adsorbent (Kang *et al.*, 2004). It is hard to draw any definite conclution, according to the limited data obtained in the experiments, about the removal capacity of the slag under different conditions.

The correlation coefficient  $(R^2)$  of the adsorption isotherms in Table 3.6 showed that the Cu adsorption data fitted the Freundlich model better than Zn and Cd. Also the model fitted the adsorption data for fine slag better than those for coarse slag. The stronger binding energy of the element and the greater adsorption capacity of the aggregate dominated the process and masked the other effects such as pH, temperature and competitive elements etc. The data were derived from batch tests. However the industrial application will be carried out in a continuous-flow system. That would react differently – initially all the substance will be removed, and then the concentration in the treated water will increase steadily with time until there is no removal. Nevertheless, it does permit the comparison of different isotherms.

The distribution of adsorbate between the solution and the adsorbent particles is determined by a number of factors, such as the competition for surface sites among adsorbates, competition for adsorbate between the surface and dissolved ligands, and electrostatic interactions at the surface. The isotherm of the ions adsorbed from multiion solution (Table 3.6) ignored the competitive effect among the ions and was excessively simplified. The multi-site Langmuir model, which assumes that the surface has groups of sites with different affinities for an adsorbate, can be applied to in this case (Benjamin, 2002). However it is out of the study objective and no further discussion will be made hereafter on this issue.

#### 4.2 Desorption of metallic ions from slag

#### 4.2.1 Ion exchange

Potassium chloride, sodium chloride and sodium nitrate were used to test desorption of heavy metals from the slag. The experimental results (Table 3.8) showed that about 40% of  $Cd^{2+}$  was released by all the salts in 24 hours. Less than 10% of  $Zn^{2+}$  was retrieved into aqueous phase from the slag under the same conditions. The inorganic salt had absolutely no effect on the recovery of  $Cu^{2+}$  from contaminated slag.

Over 80% of the recoveries happened in the first 24 hours (out of the 96 hours of experimentation).

Ion exchange is regarded as the key mechanism for inorganic salts to recover the used slag (Qin *et al.*, 2004). As the ion exchange abilities of  $K^+$  and  $Na^+$  are similar, there

is no significant difference of recovery efficiency between the salts. The difference in releasing  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  from the slag, from 40% to zero, could be attributed to the different binding energy of the complexes of each element to the aggregate (Arias *et al.*, 2005).

The recovery experiments using nitric acid as a desorption agent (Table 3.11) showed that 50%, 63% and 85% of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  were mobilized from the slag respectively by 5% HNO<sub>3</sub>. H<sub>3</sub>O<sup>+</sup> presented in the solution worked as competitive ion exchanger to metallic ions on the slag. The increased concentration of HNO<sub>3</sub> from 5% to 10% did not result in an elevation of recovery efficiency. It means that the H<sub>3</sub>O<sup>+</sup> ionic strength in 5% nitric acid is high enough for reaching the equilibrium of the ion exchange reaction.

The variance of the results, indicated by the wide confidence level (95%), made it difficult to determine the order of the recovery efficiency among the three elements. Such variance comes from the non-homogeneous physical structure and chemical composition of the slag and the error of the analysis technology.

# 4.2.2 Complexes with ligands

Metallic ions can combine with numerous other species to form complexes, either dissolved or solid, in aqueous systems (Benjamin, 2002).

#### (1) Chloride

The presence of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  with  $Cl^-$  in solution formed the soluble metal chloride complexes (Pitcher *et al.*, 2004). As these tests were carried out in an acidic environment, the OH<sup>-</sup> complexes of metallic ions can be ignored. The stabilities of each complex at equilibrium are as follows:

$$Cd^{2+} + Cl^{-} \Leftrightarrow CdCl^{+} \qquad \log K_1 = 1.98 \qquad (4.3)$$

$CdCl^{+} + Cl^{-} \Leftrightarrow CdCl_{2}^{0}$	$\log K_2 = 0.62$	(4.4)
$CdCl_2^0 + Cl^- \Leftrightarrow CdCl_3^-$	$\log K_3 = -0.20$	(4.5)
$Zn^{2+} + Cl^{-} \Leftrightarrow ZnCl^{+}$	$\log K_1 = 0.43$	(4.6)
$\operatorname{ZnCl}^+ + \operatorname{Cl}^- \Leftrightarrow \operatorname{ZnCl}_2^0$	$\log K_2 = 0.02$	(4.7)
$ZnCl_{2}^{0} + Cl^{-} \Leftrightarrow ZnCl_{3}^{-}$	$\log K_3 = 0.05$	(4.8)
$ZnCl_3^- + Cl^- \Leftrightarrow ZnCl_4^{2-}$	$\log K_4 = -0.30$	(4.9)
$Cu^{2+} + Cl^{-} \Leftrightarrow CuCl^{+}$	$\log K_1 = 0.43$	(4.10)
$\operatorname{CuCl}^+ + \operatorname{Cl}^- \Leftrightarrow \operatorname{CuCl}_2^0$	$\log K_2 = -0.27$	(4.11)

In which  $K_i$  – equilibrium constant for a reaction between two dissolved species

The higher the equilibrium constant, the more stable the complex. According to the comparison of the equilibrium constant of the complexes formed by each ion, the stability of the complexes can be ranked as  $Cd^{2+} > Zn^{2+} > Cu^{2+}$ . The formulation of more stable complex might reduce the ionic strength of the solution and, as a result, facilitate the release of more metallic ions from solid surface into liquid phase.

#### (2) Carbonic acid

The carbonic acid in solution was produced by blowing carbon dioxide (Air Liquide, food grade) into desorption agent, either deionized water or sea water.

$$\operatorname{CO}_2 + \operatorname{H}_2 \operatorname{O} \Leftrightarrow \operatorname{H}^+ + \operatorname{HCO}_3^- \Leftrightarrow 2\operatorname{H}^+ + \operatorname{CO}_3^{2-}$$
(4.12)

The results of the desorption tests (Table 3.10) indicated that 66%, 84% and 25% of the bound  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  were released by carbonic acid in sea water respectively. As the recovery efficiencies of sea water to each element was 41%, 3%

and 4% respectively, the increased portion of the effect was believed to be due to the carbonic acid.

A number of metallic complexes for each element could be produced in the solution as follows:

$Cd^{2+} + CO_3^{2-} \Leftrightarrow CdCO_3^0$	$\log \beta_1 = 5.4$	(4.13)
$\mathrm{Cd}^{2+} + 3\mathrm{CO}_3^{2-} \Leftrightarrow \mathrm{Cd}(\mathrm{CO}_3)_3^{4-}$	$\log \beta_3 = 6.22$	(4.14)
$Cd^{2+} + HCO_3^{-} \Leftrightarrow CdHCO_3^{+}$	$\log \beta_1 = 12.4$	(4.15)
$Zn^{2+} + CO_3^{2-} \Leftrightarrow ZnCO_3^0$	$\log \beta_1 = 5.3$	(4.16)
$\operatorname{Zn}^{2^+} + 2\operatorname{CO}_3^{2^-} \Leftrightarrow \operatorname{Zn}(\operatorname{CO}_3)_2^{2^-}$	$\log \beta_2 = 9.6$	(4.17)
$Zn^{2+} + HCO_3^- \Leftrightarrow ZnHCO_3^+$	$\log \beta_1 = 12.4$	(4.18)
$Cu^{2+} + CO_3^{2-} \Leftrightarrow CuCO_3^0$	$\log \beta_1 = 6.73$	(4.19)
$\mathrm{Cu}^{2+} + 2\mathrm{CO}_3^{2-} \Leftrightarrow \mathrm{Cu}(\mathrm{CO}_3)_2^{2-}$	$\log \beta_2 = 9.83$	(4.20)
$Cu^{2+} + HCO_3^{-} \Leftrightarrow CuHCO_3^{+}$	$\log \beta_1 = 13.6$	(4.21)

In which:  $\beta_i$  – Equilibrium constant for a reaction between an uncomplexed ion and i ligands to form a complex of the type MeL<sub>i</sub>.

The higher the equilibrium constant, the more stable the complex. Both competitive adsorption of Na<sup>+</sup>, K<sup>+</sup>, Ca<sup>2+</sup> and H<sup>+</sup> etc. to displace metallic ions previously adsorbed by slag and the formation of complexes by the mobilized metallic ions with  $CO_3^{2-}$  and  $HCO_3^{-}$  contributed to the release of metal ions to aqueous phase.

#### 4.2.3 Chelating

Complexes that contain more than one type of ligand are called mixed-ligand complexes. The anions of low-molecular-weight organic acids have been shown to be

effective in desorbing heavy metals from contaminated soils (Qin *et al.*, 2004; Gao *et al.*, 2003). Citric and tartaric acids, which were presented as background substances in the natural environment, were used here as chelating agents (polydentate ions that form strongly bound ring complexs).

The batch tests (Table 3.8) showed that the desorption efficiencies of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  by citric acid were 46%, 65% and 23% respectively. The presence of K<sup>+</sup> in the citric acid could increase the efficiency to a certain extent. The desorption of  $Zn^{2+}$  was even raised to 108%<sup>3</sup>.

The effects of citric and tartaric acid on the  $Cd^{2+}$  and  $Cu^{2+}$  desorption were similar, i.e. around 50%, when using sea water as solution. Again the release of  $Zn^{2+}$  from slag by tartaric acid reached high levels – in this case, 97% in 24 hours.

Citric acid has strong chelating ability with  $Cu^{2+}$ ,  $Cd^{2+}$ , and  $Zn^{2+}$  (Qin *et al.*, 2004; Ottosen *et al.*, 2005). The difference of metallic desorption tendency may be caused by the differences between the complexing ability of citric acid.

It can be concluded that desorption capacity of organic acids can be reinforced with the presence of an inorganic salt. The desorption behaviour of metallic ions was influenced by electrolytes and organic ligands in desorption solution. Desorption of metals can be dominated by the electrolytes  $K^+$ ,  $Na^+$  and  $Ca^{2+}$ , etc, in sea water, which can release metallic ions from used slag by ionic exchange, at no or low levels of citric or tartaric acids in the solution.

<sup>&</sup>lt;sup>3</sup> Desorption efficiency should be no more than 100% since there is no zinc contained in the original slag sample. Such result could be caused by the error of the measuring method and the low concentration of the analyte.

The increase of organic acids (citric or tartaric) raises the competitive ability of organic ligands in the solution for binding  $Cd^{2+}$ ,  $Cu^{2+}$  and  $Zn^{2+}$ , and as a result leads to enhanced desorption capacity of the agent. (Gao *et al.*, 2003).

The assessment of each desorbing agent can also be carried out from the cost perspective<sup>4</sup>.

#### 4.3 Column tests with stormwater sample

Stormwater samples are different from working solutions used in the batch tests in terms of the presence of suspended solids. The surfaces of suspended solids can sometimes acquire an electrical charge, much as dissolved species do. This surface charge can enhance or impede sorption of metallic ions from solution (Benjamin, 2002).

The study on road sediment (Ng, 2004) indicated that the removal of particles over 100  $\mu$ m would ensure average removal of 75% to 90% of heavy metals and phosphorus in stormwater. However the porous slag, which had been shown to have adsorption capacity to remove metallic ions from solution in the previous experiments, was expected to be used in this study more than as a filter to physically separate solids from solution. The adsorption of soluble metallic ions from the feeding stormwater sample was assumed to be occurred during the process.

#### 4.3.1 Chemical analysis of the stormwater samples

Chemical analysis of the actural stormwater sample was carried out with and without digestion. Dry ashing and nitric acid digestion procedures were applied to evaluate recovery efficiency due to its utility (Hseu, 2004).

<sup>&</sup>lt;sup>4</sup> The cost analysis of the desorption tests is presented in Appendix D.

The analysis of the samples (Table 3.12) indicated that the concentrations of each parameter (SS,  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$ ) varied significantly among the sampling sites. Although the Sites 1 and 2 of Oakley Creek were about 300 meters apart, the characteristics of the catchments (well paved residential zone of Site 1 versus mostly unpaved construction field of Site 2) resulted in the elevated levels of all the parameters on Site 2. Hobson Street stands as one of the main accesses to *State Highway 1* from *Auckland* CBD. The further increase of the pollutant concentrations (except for  $Cd^{2+}$  which was not detected) in the Hobson St. sample was believed to be contributed by heavy traffic, roof runoff and ambient dust precipitations of the area.

The distribution of metallic ions between the aqueous and solid phase in the samples differs according to the different elements.  $Cu^{2+}$  and  $Cd^{2+}$  were not detected in the filtrate, i.e. they were mostly attached to suspended solids.  $Zn^{2+}$  existed as both soluble and extractable in the samples. The extractable content of  $Zn^{2+}$  made up 55~78% of the whole lots in the samples. The wide variance of the mobility was due to the affinity of the element to different targets as well as the nature of samples in terms of the turbidity, carbonates, metallic oxides, organic and inorganic residues (Stead-Dexter *et al.*, 2004).

The comparison of the results with and without digestion of the samples showed that the measurement of the sample directly by AAS without digestion was between the concentration of soluble ions and the total (the sum of soluble and extractable ions). Such differences indicated that most of the mobile ions and a part of the element attached to solid in the samples could be atomized and detected by AAS. The ratios of non-digested concentration over digested (assumed to be the "real" concentration of the sample) of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  were 0%, 33~49% and 0~76% respectively.

The chemical analysis of the synthetic stormwater samples (Table 3.19) indicated that a large proportion of metallic ions (66%~80% of  $Cd^{2+}$ , 71%~77% of  $Zn^{2+}$  and 75%

~87% of  $Cu^{2+}$ ) were immobilized by clay. This is the result of the ion exchange as explained in Section 3.6.4.

The measurements of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  in the synthetic stormwater samples by AAS without digestion are 66%~83%, 80%~90% and 50%~87% of the actural concentrations respectively.

Even though the method of direct measuring without digestion could underestimate the metallic ion concentrations, it was applied, as explained in Section 3.6.3, in the column experiments to measure the metallic ions level.

#### 4.3.2 Pollutants removal efficiency in columns

The results of the column experiments were shown in Table  $3.14 \sim 3.22$ .

#### (1) Cadmium

The amounts of  $Cd^{2+}$  in all the samples and the SS and  $Cu^{2+}$  in Oakley Creek Site 1 were so low that they could not be determined by the methods in this study.

The concentration of  $Cd^{2+}$  in the synthetic stormwater sample was elevated to around 100 µg L<sup>-1</sup>.

Different removal efficiencies had been recorded under different conditions. It reached 89% by fine slag at low flow rate. On the contrary, only 16% was removed by coarse slag at high flow rate.

The above analysis could be doubtful considering the large relative error of measuring  $Cd^{2+}$  at low concentration (see Table 3.20). The measured concentration of  $Cd^{2+}$  is about 50% lower than the actural value when it is as low as 100 µg L<sup>-1</sup>. Therefore it is

still hard to draw any definite conclusion on the removal efficiency of  $Cd^{2+}$  in the column test with these results.

(2) Copper

The  $Cu^{2+}$  removal efficiency by fine and medium slag from Oakley Creek Site 2 sample reached 100% during the 4-hour-cycle at both flow rates (3.9 and 8.9 cm/min). No conclusion could be drawn from other experiments. Erratic removal efficiencies of  $Cu^{2+}$  in solution were observed. The very low levels of  $Cu^{2+}$  versus the sensitivity of the analysis methods resulted in this uncertainty.

The readings from AAS to measure the Cu<sup>2+</sup> concentration in Oakley Creek Site 2 and Hobson St. samples were 0.01 and 0.02 absorption unit respectively. These readings convert to concentration of 32 and 67  $\mu$ g L<sup>-1</sup> by using the calibration curve. However the fluctuation of the reading was  $\pm 0.03$  which means that the actual Cu<sup>2+</sup> concentrations could be within 0 ~ 160  $\mu$ g L<sup>-1</sup> range. The margin of error thus makes drawing any definitive conclusion unwise.

Similarly to the case of  $Cd^{2+}$ , the concentration of  $Cu^{2+}$  in the synthetic stormwater samples were raised to around 200 µg L<sup>-1</sup>. Although it is safe to say that a portion of  $Cu^{2+}$  in the influent could be removed by the slag when passing through the column and the finer aggregate presents likely the better performance than the coarse one, the unambiguous  $Cu^{2+}$  removal efficiency by the slag can not be given because of the significant error of the method.

(3) Zinc

 $Zn^{2+}$  removal efficiency varied with the size of slag and flow rates of the feeding solution.

The fine slag performed with the highest efficiency to remove zinc at all occasions, followed in sequence by medium and coarse sizes. This is consistent with the results from batch tests.

The tests on Oakley Creek Site 2 and Hobson St. samples were carried out at two flow rates. The slag removed  $Zn^{2+}$  more efficiently at the lower rate.

There was no decline of removal efficiency in the 4-hour-cycle except for the test with Hobson St. sample at high flow rate (3.9 cm/min). The high SS content in that sample resulted in the rapid accumulation of residues in the column and blocked the flow path. Pressure was introduced to the column to keep a stable flow rate over the last 2 hours. The effluent became cloudy due to the presumed channelling effect and greater fluid velocity and contained more SS and  $Zn^{2+}$  which was attached to the solids.

Similar conclusion can be drawn from the tests with synthetic stormwater sample. The removal efficiency of  $Zn^{2+}$  is in reverse proportion to the grain size of the aggregate and the flow rate of the feeding solution, regardless of the initial ionic concentration of the sample.

 $Zn^{2+}$  concentrations in the samples are higher (400 ~ 800 µg L<sup>-1</sup>) and AAS is more sensitive to this element, the relative error of measuring Zn is within 10%. Therefore the above conclusions are solid.

(4) Suspended solids

The suspended solids in the sample formed as floc through precipitation or aggregation once the sample was kept stable for a while. The sample was stirred during the test to be homogeneous.

The slag exhibited a similar tendency to remove SS as  $Zn^{2+}$ , i.e. the removal efficiency of SS follows inverse proportion to the size of slag as well as the flow rate of the feeding solution. However the removal efficiency of SS is much lower than that of metallic ions. It is assumed that the very small particle size of the suspended solids in the samples results in the low removal efficiency. The removal effect of the aggregate on the actural stormwater runoff, which contains more large particles, would be elevated.

#### (5) Removal mechanism

A certain portion of metallic ions were associated with the suspended solids. Therefore the removal of suspended solids, as a result of precipitation, led to the removal of metallic ions attached to them.

The amount of  $Zn^{2+}$  removed by SS precipitation was significantly lower than the total removal of each element. Thus it is concluded that the remainder of the ions removal was a result of adhesion to the slag.

(6) Control column

A control column, which was blank and used as a quality control measure, was set for each test. The U-shape design of the column and slow flow rate resulted in precipitation of residues at the bottom of the column (as shown in Figure 2.5). The SS removal efficiencies in the control column were around 30%. Unlike the columns filled with slag, the removal efficiency of  $Cd^{2+}$ ,  $Zn^{2+}$  and  $Cu^{2+}$  is usually lower than or similar to that of SS in the control column. This reinforces the above proposal that the removal mechanisms for slag were both precipitation and adsorption (Pitcher, 2004).

#### 4.3.3 Perspective of the application

In New Zealand there is no standard for heavy metals in stormwater runoff. The following standards, either for drinking water quality or receiving water bodies, are quoted here as reference.

Standards	Cadmium	Copper	Zinc
Drinking Water Standard (NZ) MAVs	0.003	2.0	N/A
Drinking Water Standard (US) (MCLs)	0.005	1.3	N/A
National Secondary Drinking Water Regulations (US) (MCLs)	N/A	1.0	5.0
Universal Treatment Standard (US)	0.69	N/A	2.61
Surface Water Standard (CN)	0.005	1.0	2.0

Table 4.1 Water Quality Standards (unit:  $mg L^{-1}$ )

In which: MAVs - Maximum Acceptable Values

MCLs - Maximum Contaminant Levels. The highest level of a contaminant that is allowed in drinking water. MCLs are enforceable standards

The concentration of  $Zn^{2+}$  in the Hobson St. sample can be reduced from 0.4 mg L<sup>-1</sup> to 0.1 ~ 0.2 mg L<sup>-1</sup> after passing through the slag column. It is hard to decide the efficiency of cadmium and copper removal from stormwater by the slag due to the very low level of these elements in the samples. However the tests using synthetic stormwater samples and the results of the batch tests would likely support the deduction that the slag will have similar (Cd<sup>2+</sup>) or even better (Cu<sup>2+</sup>) effect on removing these elements.

The concentration of tested heavy metals in the effluent is lower than the above standards. It makes the melter slag an attractive material to process stormwater runoff. However attention needs to be paid to the disposal of the used slag. The metallic ions adsorbed by the aggregate can be mobilized by a variety of agents. Therefore the inappropriate disposal may cause a serious problem because of the elevated level of contaminants attached to the used slag.

### **Chapter 5** Conclusions

1. The melter slag tested in this study is a rather porous material. Previous leaching trials (New Zealand Steel, 1992) indicated that the heavy metals were bound in the aggregate and the mobilization of these elements into aqueous phase can be ignored, especially when it is mixed with neutral or basic solution.

2. Copper, zinc and cadmium were selected in this study for testing. The batch tests showed that the melter slag had the ability to remove these elements from working solutions. The removal efficiency of each ion by the slag is evaluated by two indices, i.e. the decreasing rate of the ion concentration in solution over time and the amount that it can be removed by unit mass of slag. It follows the descending order of  $Cu^{2+} > Zn^{2+} \approx Cd^{2+}$ .

3. Ion exchange is the dominant mechanism when metallic ions are removed by the slag. The decrease of metallic ions was accompanied by the increase of  $Ca^{2+}$  concentration and pH in the working solutions. The finest grain size slag, which had the largest specific surface area, was the most effective in removing heavy metals.

4. Different elements have varied binding energy to the aggregate. Competitive adsorption was observed when the slag was mixed with solutions containing several kinds of metallic ions. The affinity of the three tested elements to the slag follows the descending order of  $Cu^{2+} > Zn^{2+} > Cd^{2+}$ .

5. Author intended to establish an isotherm model to evaluate the equilibrium status of the treatment system and to compare the capacity of the slag in different sizes to remove different elements from solutions. However only  $Cu^{2+}$  adsorption by

the melter slag in fine size obeys the Freundlich isotherm model. It is hard to draw any definite conclusion from the data in other experiments. Further studies are needed on this issue in the future.

6. Used slag which is saturated by heavy metals can be recovered and treated for reuse. Desorption of ions from the slag may be achieved by applying inorganic salts (ion exchange), inorganic acids (metallic complexes) and organic acids (chelating). A mixture of inorganic salt and organic acid present the best solution for desorption of heavy metals from slag, e.g. citric acid in sea water can mobilize about 50% of each metallic ion in 24 hours. However tartaric acid has a different effect on each ion. The slag contaminated by  $Zn^{2+}$  could almost be completely recovered by tartaric acid in sea water or KCl solution. Tartaric acid and citric acid have similar effect on the desorption of  $Cd^{2+}$  and  $Cu^{2+}$  (around 50% in 24 hours).

7. The slag can be reused after saturation, given that it is recoverable by a range of desorbing agents. A mixture of citric acid and sea water is recommended considering the best desorption efficiency and the lowest costs.

8. Relatively high levels of  $Zn^{2+}$  were found in all the stormwater samples. The SS and  $Cu^{2+}$  varied significantly among different samples.  $Cd^{2+}$  was present only at trace levels, i.e. 1 or 2 µg L<sup>-1</sup>. A large portion of  $Zn^{2+}$  (55 ~ 78%) was attached to suspended solids. All the  $Cu^{2+}$  present was extracted in the samples, i.e.  $Cu^{2+}$  is firmly bound with organic and/or inorganic suspended solids. The affinity of metallic ions to suspended solids ensures that the removal of solids from stormwater runoff can result in a reduction of metallic ions in the effluent.

9. The melter slag effectively removes pollutants, i.e. suspended solids and heavy metals, from stormwater samples based on the mechanisms of physical filtration and chemical adsorption.

10. The pollutant removal efficiency of the slag column is in inverse proportion to the grain size of the melter slag and the influent flow rate.

The melter slag has the potential to act as an adsorbent to process stormwater runoff. Its capacity for removing suspended solids and heavy metals significantly assist meeting targets for water purification standards.

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# Abbreviations

AAS	Atomic Absorption Spectrometer
ACC	Auckland City Council
ARC	Auckland Regional Council
$C_q$	Concentration of material remaining in solution once equilibrium has been reached
ERC	Environmental Response Criteria
ICP-AES	Inductively Coupled Plasma – Atomic Emission Spectrometer
т	Weight of adsorbent
$R^2$	Correlation coefficient
SS	Suspended Solid
x	Weight of metallic ion adsorbed by aggregate
USEPA	United States Environmental Protection Agency

## Glossary

*Adsorption*: Accumulation of a substance at or near an interface relative to its concentration in the bulk solution.

*Apparent adsorption isotherm*: The amount of solute adsorbed at a given temperature per unit mass of adsorbent is plotted against the equilibrium concentration.

*Complex:* An ion formed by an ion combining with an atom or molecule.

*Control experiment*: An experiment that isolates the effect of one variable on a system by holding constant all variables but the one under observation.

*Desorption*: The reverse process of adsorption, i.e., it is the release of an adsorbed substance to the bulk solution.

*Heavy metal:* Metallic elements with high atomic weights, e.g., mercury, chromium, cadmium, arsenic, and lead. They can damage living things at low concentration and tend to accumulate in the food chain.

*First flush:* The initial period of stormwater runoff during which the concentration of pollutants is substantially higher then during later periods.

*Slag:* A by-product of smelting process. Slag is a glassy, sand-like material containing small amounts of lead, zinc, copper and other materials.

*Stock standard solution*: A solution of known concentration for use in volumetric analysis.

*Working solution*: A synthetic solution prepared with certain amount of metallic ion or mixture of different ions. It is used as adsorbate in the batch tests of this study.

# Appendix A

# Adsorption Batch Tests Results

Time				A			В			С	
(h)	Dup.	Items	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.07	19.16	0.15	4.07	19.16	0.15	4.07	19.16	0.15
		C (mg L <sup>-1</sup> )	6.19	17.78	0.59	6.28	16.32	1.25	6.45	15.64	1.60
	1	m (g)		1.0026			2.0047			3.0023	
		x/m (mg/g)		0.22			0.22			0.18	
		$C (mg L^{-1})$	6.25	17.67	0.63	6.33	17.22	1.11	6.5	15.87	1.53
	2	m (g)		1.0033			2.0035			3.0029	
24		x/m (mg/g)		0.24			0.15			0.17	
		$C (mg L^{-1})$	6.28	18.46	0.73	6.36	17.11	1.22	6.44	14.86	2.08
	3	m (g)		1.0056			2.0034			3.0088	
		x/m (mg/g)		0.12			0.16			0.22	
	Ave.	$C (mg L^{-1})$	6.24	17.97	0.65	6.3233	16.88	1.19	6.46333	15.46	1.74
	Ave.	x/m (mg/g)		0.19			0.18			0.19	
		(8-8)		0.17			0.10			0.17	
Time	Dup			0.19 D			0.10 E			Control	
Time (h)	Dup.	Items	pН		Ca <sup>2+</sup>	рН		Ca <sup>2+</sup>	рН		<i>Ca</i> <sup>2+</sup>
	Dup.		<i>pH</i> 4.07	D	<i>Ca</i> <sup>2+</sup> 0.15	<u>рН</u> 4.07	E	<i>Ca</i> <sup>2+</sup> 0.15	<u>рН</u> 4.07	Control	<i>Ca</i> <sup>2+</sup> 0.15
(h)	Dup.	Items	-	D Cd <sup>2+</sup>		-	E Cd <sup>2+</sup>		-	Control Cd <sup>2+</sup>	
(h)	Dup.	<i>Items</i> C (mg L <sup>-1</sup> )	4.07	D Cd <sup>2+</sup> 19.16	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h)		Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)	4.07	D Cd <sup>2+</sup> 19.16 14.63	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15	0.15	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h)		Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)	4.07	D           Cd <sup>2+</sup> 19.16           14.63           4.0066	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070	0.15	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h)		Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)	4.07	<i>D</i> <i>Cd</i> <sup>2+</sup> 19.16 14.63 4.0066 0.17	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21	0.15 2.81	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h)	1	Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )	4.07	D     Cd2+     19.16     14.63     4.0066     0.17     14.29	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21 12.04	0.15 2.81	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h) 0	1	Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)	4.07	D       Cd <sup>2+</sup> 19.16       14.63       4.0066       0.17       14.29       4.007	0.15	4.07	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21 12.04 5.0029	0.15 2.81	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h) 0	1	Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)	4.07 6.58 6.55	D       Cd <sup>2+</sup> 19.16       14.63       4.0066       0.17       14.29       4.007       0.19	0.15 2.26 2.43	4.07 6.59 6.52	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21 12.04 5.0029 0.22	0.15 2.81 2.78	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h) 0	1	Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)	4.07 6.58 6.55	$\begin{array}{c} D \\ \hline Cd^{2+} \\ 19.16 \\ 14.63 \\ 4.0066 \\ 0.17 \\ 14.29 \\ 4.007 \\ 0.19 \\ 14.18 \\ \end{array}$	0.15 2.26 2.43	4.07 6.59 6.52	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21 12.04 5.0029 0.22 12.83	0.15 2.81 2.78	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15
(h) 0	1	Items           C (mg L <sup>-1</sup> )           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)           x/m (mg/g)           C (mg L <sup>-1</sup> )           m (g)	4.07 6.58 6.55	D         Cd <sup>2+</sup> 19.16         14.63         4.0066         0.17         14.29         4.007         0.19         14.18         4.0085	0.15 2.26 2.43	4.07 6.59 6.52	<i>E</i> <i>Cd</i> <sup>2+</sup> 19.16 12.15 5.0070 0.21 12.04 5.0029 0.22 12.83 5.0059	0.15 2.81 2.78	4.07	<i>Control</i> <i>Cd</i> <sup>2+</sup> 19.16	0.15

Table A.1 Cadmium adsorption by slag (fine size)

Note: m – mass of slag (adsorbent)

x/m -- mass of metallic ion (adsorbate) taken up by unit slag

Time	D	L		A			В			С	
<u>(h)</u>	Dup.	Items	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.07	23.07	0.00	4.07	23.07	0.00	4.07	23.07	0.00
		C (mg L <sup>-1</sup> )	5.53	22.51	0.67	5.77	20.71	1.03	6.11	19.32	1.81
	1	m (g)		1.0015			2.0023			3.0033	
		x/m (mg/g)		0.08			0.18			0.19	
		$C (mg L^{-1})$	5.74	21.82	0.59	5.96	21.40	0.99	6.09	20.01	1.62
	2	m (g)		1.0033			2.0019			3.0026	
24		x/m (mg/g)		0.19			0.12			0.15	
		$C (mg L^{-1})$	5.81	21.82	0.71	5.86	20.84	1.10	6.11	20.98	1.70
	3	m (g)		1.0062			2.0036			3.0054	
		x/m (mg/g)		0.19			0.17			0.10	
	Ave.	$C (mg L^{-1})$	5.69	22.05	0.66	5.86	20.98	1.04	6.10	20.10	1.71
	Ave.	x/m (mg/g)		0.15			0.16			0.15	
Time	Dup.	Element		D			Ε			Control	
(h)	Dup.	Liemeni	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.07	23.07	0.00	4.07	23.07	0.00	4.07	23.07	0.00
		$C (mg L^{-1})$	6.07	18.62	1.85	6.19	17.51	2.60	4.07	23.07	0.04
	1	m (g)		4.0006			5.0041				
		x/m (mg/g)		0.17			0.17				
		$C (mg L^{-1})$	6.15	20.57	1.85	6.19	19.04	2.37			
	2	m (g)		4.0012			5.0061				
24		x/m (mg/g)		0.09			0.12				
		$C (mg L^{-1})$	6.29	19.45	2.29	6.18	18.90	2.37			
	3	m (g)		4.0035			5.0044				
	3	m (g) x/m (mg/g)		4.0035 0.14			5.0044 0.13				
	3 Ave.		6.17		2.00	6.19		2.45			

Table A.2 Cadmium adsorption by slag (medium size)

Note: m - mass of slag (adsorbent)

x/m -- mass of metal (adsorbate) taken up by slag

Time	D	T		A			В			С	
<u>(h)</u>	Dup.	Items	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$
0		C (mg L <sup>-1</sup> )	4.03	20.15	0.00	4.03	20.15	0.00	4.03	20.15	0.00
		C (mg L <sup>-1</sup> )	5.30	20.15	0.43	5.44	20.40	0.77	5.79	19.00	1.24
	1	m (g)		1.0011			2.0022			3.0065	
		x/m (mg/g)		0.15			0.06			0.11	
		$C (mg L^{-1})$	5.22	20.40	0.43	5.70	19.76	0.67	5.88	19.00	1.05
	2	m (g)		1.0082			2.0018			3.0092	
24		x/m (mg/g)		0.11			0.11			0.11	
		C (mg L <sup>-1</sup> )	5.09	20.40	0.48	5.70	19.76	0.72	5.78	19.51	1.05
	3	m (g)		1.0017			2.0080			3.0017	
		x/m (mg/g)		0.11			0.10			0.08	
	Ave.	$C (mg L^{-1})$	5.20	20.32	0.45	5.61	19.98	0.72	5.82	19.17	1.12
	Ave.	x/m (mg/g)		0.13			0.09			0.10	
Time	Dup.	Element		D			Ε			Control	
(h)	Dup.	Liemeni	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	$Ca^{2+}$	рН	$Cd^{2+}$	<i>Ca</i> <sup>2+</sup>
0		$C (mg L^{-1})$	4.03	20.15	0.00	4.03	20.15	0.00	4.03	20.15	0.00
		C (mg L <sup>-1</sup> )	5.89	19.64	1.29	6.05	18.62	1.68	4.06	21.17	0.05
	1	m (g)		4.0018			5.0064				
		x/m (mg/g)		0.06			0.08				
		C (mg L <sup>-1</sup> )	5.98	19.76	1.29	5.97	18.62	1.68			
	2	m (g)		4.0073			5.0029				
24		x/m (mg/g)		0.05			0.08				
		C (mg L <sup>-1</sup> )	5.89	19.00	1.29	6.03	18.87	1.63			
	3		5.89		1.29	6.03	18.87 5.0006	1.63			
	3	C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	5.89	19.00	1.29	6.03		1.63			
	3 Ave.	C (mg L <sup>-1</sup> ) m (g)	5.89	19.00 4.0012	1.29	6.03 6.02	5.0006	1.63			

Table A.3 Cadmium adsorption by slag (coarse size)

Note: m - mass of slag (adsorbent)

x/m -- mass of metal (adsorbate) taken up by slag

Time				A			В			С	
(h)	Dup.	Items	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$
0		C (mg L <sup>-1</sup> )	4.13	21.68	0.00	4.13	21.68	0.00	4.13	21.68	0.00
		C (mg L <sup>-1</sup> )	5.84	19.86	0.91	5.92	18.44	1.37	6.17	16.85	1.78
	1	m (g)		1.0029			2.0027			3.0005	
		x/m (mg/g)		0.37			0.29			0.27	
		C (mg L <sup>-1</sup> )	5.81	20.39	0.76	5.97	18.97	1.23	6.14	17.55	1.99
	2	m (g)		1.0039			2.0042			3.0009	
24		x/m (mg/g)		0.29			0.25			0.24	
		$C (mg L^{-1})$	5.82	19.33	0.61	5.98	18.44	1.14	6.12	16.49	1.72
	3	m (g)		1.0026			2.0034			3.0018	
		x/m (mg/g)		0.45			0.29			0.29	
	Ave.	$C (mg L^{-1})$	5.82	19.86	0.76	5.96	18.62	1.25	6.14	16.96	1.83
	Ave.	x/m (mg/g)		0.37			0.28			0.27	
Time	Dup.	Element		D			Ε		_	Control	
(h)	Dup.		рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.13	21.68	0.00	4.13	21.68	0.00	4.13	21.68	0.00
		$C (mg L^{-1})$	6.24	15.25	2.34	6.27	12.77	3.22	4.26	22.34	0.06
	1	m (g)		4.0008			5.0053				
		x/m (mg/g)		0.27			0.29				
		$C (mg L^{-1})$	6.23	15.96	2.28	6.27	13.30	2.63			
	2	m (g)		4.0017			5.0044				
24		x/m (mg/g)		0.21			0.25				
		$C (mg L^{-1})$	6.20	15.78	2.10	6.29		14.54			
	3	m (g)		4.0025			5.0025				
		x/m (mg/g)		0.25			0.23				
	Ave.	$C (mg L^{-1})$	6.22	15.66	2.24	6.28	13.54	3.07			
	Ave.	x/m (mg/g)		0.24			0.26				

Table A.4 Zinc adsorption by slag (fine size)

Note: m – mass of slag (adsorbent)

x/m -- mass of metal (adsorbate) taken up by slag

Time	D	L		A			В			С	
(h)	Dup.	Items	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.06	20.57	0.09	4.06	20.57	0.09	4.06	20.57	0.09
		C (mg L <sup>-1</sup> )	5.53	19.17	0.67	5.83	19.36	1.13	5.92	18.81	1.53
	1	m (g)		1.0073			2.0024			3.0029	
		x/m (mg/g)		0.30			0.14			0.12	
		$C (mg L^{-1})$	5.39	19.36	0.70	5.88	18.44	1.28	6.05		18.08
	2	m (g)		1.0029			2.006			3.0047	
24		x/m (mg/g)		0.27			0.20			0.16	
		$C (mg L^{-1})$	5.47	20.63	0.89	5.83	20.45	1.38	5.94		18.99
	3	m (g)		1.0049			2.0034			3.0051	
		x/m (mg/g)		0.08			0.05			0.11	
	Ave.	$C (mg L^{-1})$	5.46	19.72	0.75	5.85	19.42		5.97	18.63	1.70
	Ave.	x/m (mg/g)		0.22			0.13			0.13	
Time	Dup.	Element		D			E			Control	
(h)	Dup.	Liemeni	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.06	20.57	0.09	4.06	20.57	0.09	4.06	20.57	0.09
		• (8 )	<b>ч</b> .00	20.37	0.09	ч.00	20.07	0.07		20.57	
		C (mg L <sup>-1</sup> )	6.09	18.81	2.05	6.20	16.43	2.66	3.98	21.18	0.09
	1										
	1	$C (mg L^{-1})$		18.81			16.43				
	1	C (mg L <sup>-1</sup> ) m (g)		18.81 4.0078			16.43 5.0060				
	1	C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	6.09	18.81 4.0078 0.09	2.05	6.20	16.43 5.0060 0.14	2.66			
24		C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> )	6.09	18.81 4.0078 0.09 18.08	2.05	6.20	16.43 5.0060 0.14 16.62	2.66			
24		C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	6.09	18.81         4.0078         0.09         18.08         4.0027	2.05	6.20	16.435.00600.1416.625.0072	2.66			
24		C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	6.09	18.81 4.0078 0.09 18.08 4.0027 0.12	2.05	6.20	16.43         5.0060         0.14         16.62         5.0072         0.14	2.66			
24	2	C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> )	6.09	18.81         4.0078         0.09         18.08         4.0027         0.12         17.71	2.05	6.20	16.43         5.0060         0.14         16.62         5.0072         0.14         17.16	2.66			
24	2	C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	6.09	18.81         4.0078         0.09         18.08         4.0027         0.12         17.71         4.005	2.05	6.20	16.43         5.0060         0.14         16.62         5.0072         0.14         17.16         5.0056	2.66			

Table A.5 Zinc adsorption by slag (medium size)

Time				A			В			С	
(h)	Dup.	Items	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$	рН	$Zn^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.11	21.36	0.09	4.11	21.36	0.09	4.11	21.36	0.09
		C (mg L <sup>-1</sup> )	5.47	20.11	0.52	5.8	20.29	0.89	5.92	19.05	1.23
	1	m (g)		1.0020			2.0056			3.0398	
		x/m (mg/g)		0.13			0.05			0.10	
		$C (mg L^{-1})$	5.38	20.11	0.54	5.82	19.05	0.95	5.92	18.52	1.17
	2	m (g)		1.0178			2.0142			3.0179	
24		x/m (mg/g)		0.13			0.14			0.12	
		$C (mg L^{-1})$	5.32	20.64	0.49	5.72	20.46	0.97	5.9	19.23	1.26
	3	m (g)		1.0112			2.0051			3.0021	
		x/m (mg/g)		0.05			0.04	-		0.09	
	Ave.	$C (mg L^{-1})$	5.39	20.29	0.52	5.78	19.93	0.94	5.91	18.93	1.22
	Ave.	x/m (mg/g)		0.10			0.08			0.10	
Time	Dun	Itams		D			E			Control	
Time (h)	Dup.	Items	pН	D $Zn^{2+}$	Ca <sup>2+</sup>	pН	E $Zn^{2+}$	<i>Ca</i> <sup>2+</sup>	pН	Control Zn <sup>2+</sup>	Ca <sup>2+</sup>
	Dup.	<i>Items</i> C (mg L <sup>-1</sup> )	<i>рН</i> 4.11		<i>Ca</i> <sup>2+</sup> 0.09	<i>рН</i> 4.11		<i>Ca</i> <sup>2+</sup> 0.09	<i>рН</i> 4.11		
(h)	Dup.		-	$Zn^{2+}$		-	$Zn^{2+}$		-	$Zn^{2+}$	<i>Ca</i> <sup>2+</sup>
(h)	<i>Dup.</i> 1	C (mg L <sup>-1</sup> )	4.11	<i>Zn<sup>2+</sup></i> 21.36	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.11	Zn <sup>2+</sup> 21.36 19.41	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36 17.99	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g)	4.11	Zn <sup>2+</sup> 21.36 19.41 4.0182	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36 17.99 5.0083	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.11 5.95	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06	0.09	4.11 6.04	<i>Zn</i> <sup>2+</sup> 21.36 17.99 5.0083 0.09	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h)	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.11 5.95	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06 19.23	0.09	4.11 6.04		0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.11 5.95	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06 19.23 4.0188	0.09	4.11 6.04	Zn <sup>2+</sup> 21.36           17.99           5.0083           0.09           16.23           5.0135	0.09	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.11 5.95 6.06	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06 19.23 4.0188 0.07 18.52 4.0025	0.09 1.58 1.49	4.11 6.04 6.04		0.09 1.92 2.03	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.11 5.95 6.06	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06 19.23 4.0188 0.07 18.52	0.09 1.58 1.49	4.11 6.04 6.04	Zn <sup>2+</sup> 21.36           17.99           5.0083           0.09           16.23           5.0135           0.14           18.88	0.09 1.92 2.03	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.11 5.95 6.06	Zn <sup>2+</sup> 21.36 19.41 4.0182 0.06 19.23 4.0188 0.07 18.52 4.0025	0.09 1.58 1.49	4.11 6.04 6.04	Zn <sup>2+</sup> 21.36           17.99           5.0083           0.09           16.23           5.0135           0.14           18.88           5.0071	0.09 1.92 2.03	4.11	<i>Zn</i> <sup>2+</sup> 21.36	<i>Ca<sup>2+</sup></i> 0.09

Table A.6 Zinc adsorption by slag (coarse size)

Time				A			В			С	
(h)	Dup.	Items	рН	$Cu^{2+}$	$Ca^{2+}$	рН	$Cu^{2+}$	$Ca^{2+}$	рН	<i>Cu</i> <sup>2+</sup>	$Ca^{2+}$
0		$C (mg L^{-1})$	4.07	17.78	0.14	4.07	17.78	0.14	4.07	17.78	0.14
		$C (mg L^{-1})$	4.75	13.64	0.81	4.72	4.21	1.75	4.79	4.75	2.53
	1	m (g)		1.0048			2.0074	-		3.0075	
		x/m (mg/g)		0.62			1.01			0.65	
		$C (mg L^{-1})$	4.92	12.80	1.16	4.77	8.49	1.43	4.90	6.45	2.50
	2	m (g)		1.0055			2.0010			3.0060	
24		x/m (mg/g)		0.74			0.70			0.57	
		$C (mg L^{-1})$	4.80	11.43	0.84	4.82	7.92	1.57	4.84	2.81	2.21
	3	m (g)		1.0031			2.0029			3.0039	
		x/m (mg/g)		0.95			0.74			0.75	
	Ave.	C (mg L <sup>-1</sup> )	4.82	12.63	0.94	4.77	6.88	1.58	4.84	4.67	2.42
	Ave.	x/m (mg/g)		0.77			0.82			0.65	
Time											
Time	Dun	Itoms		D			Ε			Control	
(h)	Dup.	Items	рН	$\frac{D}{Cu^{2+}}$	<i>Ca</i> <sup>2+</sup>	pН	<i>E</i> <i>Cu</i> <sup>2+</sup>	<i>Ca</i> <sup>2+</sup>	pН	Control Cu <sup>2+</sup>	Ca <sup>2+</sup>
	Dup.	Items C (mg L <sup>-1</sup> )	<u>рН</u> 4.07		<i>Ca</i> <sup>2+</sup> 0.14	<u>рН</u> 4.07		<i>Ca</i> <sup>2+</sup> 0.14	<i>рН</i> 4.07	L	
(h)	Dup.		-	<i>Cu</i> <sup>2+</sup>		-	<i>Cu</i> <sup>2+</sup>		1	<i>Cu</i> <sup>2+</sup>	<i>Ca</i> <sup>2+</sup>
(h)	<i>Dup.</i> 1	C (mg L <sup>-1</sup> )	4.07	<i>Cu<sup>2+</sup></i> 17.78	0.14	4.07	<i>Cu</i> <sup>2+</sup> 17.78	0.14	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> )	4.07	<i>Cu</i> <sup>2+</sup> 17.78 3.38	0.14	4.07	Cu <sup>2+</sup> 17.78           1.47	0.14	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g)	4.07	Cu <sup>2+</sup> 17.78           3.38           4.0041	0.14	4.07	Cu <sup>2+</sup> 17.78           1.47           5.0029	0.14	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h)		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.07 4.97	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54	0.14 2.82	4.07 5.14	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49	0.14 3.93	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h)	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> )	4.07 4.97	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54           2.01	0.14 2.82	4.07 5.14	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49           0.80	0.14 3.93	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.07 4.97	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54           2.01           4.0018	0.14 2.82	4.07 5.14	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49           0.80           5.0037	0.14 3.93	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.07 4.97 4.99	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54           2.01           4.0018           0.59	0.14 2.82 3.58	4.07 5.14 4.95	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49           0.80           5.0037           0.51	0.14 3.93 3.26	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.07 4.97 4.99	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54           2.01           4.0018           0.59           2.07	0.14 2.82 3.58	4.07 5.14 4.95	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49           0.80           5.0037           0.51           1.00	0.14 3.93 3.26	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14
(h) 0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.07 4.97 4.99	Cu <sup>2+</sup> 17.78           3.38           4.0041           0.54           2.01           4.0018           0.59           2.07           4.0072	0.14 2.82 3.58	4.07 5.14 4.95	Cu <sup>2+</sup> 17.78           1.47           5.0029           0.49           0.80           5.0037           0.51           1.00           5.0016	0.14 3.93 3.26	4.07	<i>Cu</i> <sup>2+</sup> 17.78	<i>Ca<sup>2+</sup></i> 0.14

Table A.7 Copper adsorption by slag (fine size)

Time		-		A			В			С	
(h)	Dup.	Items	рН	$Cu^{2+}$	$Ca^{2+}$	рН	$Cu^{2+}$	$Ca^{2+}$	рН	<i>Cu</i> <sup>2+</sup>	$Ca^{2+}$
0		C (mg L <sup>-1</sup> )	4.04	19.37	0.10	4.04	19.37	0.10	4.04	19.37	0.10
		C (mg L <sup>-1</sup> )	4.43	8.80	1.45	4.73	3.46	2.34	4.70	3.04	3.01
	1	m (g)		1.0049			2.0087			3.0032	
		x/m (mg/g)		1.58			1.19			0.82	
		$C (mg L^{-1})$	4.61	14.95	1.20	4.70	11.22	2.03	4.61	5.15	3.01
	2	m (g)		1.0058			2.0092			3.0030	
24		x/m (mg/g)		0.66			0.61	-		0.71	
		$C (mg L^{-1})$	4.57	9.38	1.30	4.76	8.03	1.82	4.57	5.96	2.91
	3	m (g)		1.0077			2.0069			3.0029	
		x/m (mg/g)		1.49			0.85	-		0.67	
	Ave.	$C (mg L^{-1})$	4.54	11.04	1.32	4.73	7.57	2.06	4.63	4.71	2.98
	Ave.	x/m (mg/g)		1.24			0.88			0.73	
Time	Dup.			D			Е			Control	
(1)											
(h)	2 up.	Items	pН	$Cu^{2+}$	$Ca^{2+}$	pН	<i>Cu</i> <sup>2+</sup>	<i>Ca</i> <sup>2+</sup>	pН	$Cu^{2+}$	<i>Ca</i> <sup>2+</sup>
( <i>n</i> ) 0	Dup.	C (mg L <sup>-1</sup> )	<i>рН</i> 4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10	<i>рН</i> 4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10	<i>pH</i> 4.04		
			1			-				<i>Cu</i> <sup>2+</sup>	<i>Ca</i> <sup>2+</sup>
	1	C (mg L <sup>-1</sup> )	4.04	19.37	0.10	4.04	19.37	0.10	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> )	4.04	19.37 3.88	0.10	4.04	19.37 2.04	0.10	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g)	4.04	19.37 3.88 4.0054	0.10	4.04	19.37 2.04 5.0061	0.10	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.04 4.61	19.37         3.88         4.0054         0.58	0.10	4.04	19.37         2.04         5.0061         0.52	0.10 4.78	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> )	4.04 4.61	19.37         3.88         4.0054         0.58         1.96	0.10	4.04	19.37         2.04         5.0061         0.52         1.11	0.10 4.78	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.04 4.61	19.37         3.88         4.0054         0.58         1.96         4.0062	0.10	4.04	19.37         2.04         5.0061         0.52         1.11         5.0072	0.10 4.78	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.04 4.61 4.73	19.37         3.88         4.0054         0.58         1.96         4.0062         0.65	0.10 4.00 5.14	4.04 4.87 4.77	19.37         2.04         5.0061         0.52         1.11         5.0072         0.55	0.10 4.78 4.62	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.04 4.61 4.73	19.37         3.88         4.0054         0.58         1.96         4.0062         0.65         7.22	0.10 4.00 5.14	4.04 4.87 4.77	19.37         2.04         5.0061         0.52         1.11         5.0072         0.55         1.42	0.10 4.78 4.62	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.04 4.61 4.73	19.37         3.88         4.0054         0.58         1.96         4.0062         0.65         7.22         4.0077	0.10 4.00 5.14	4.04 4.87 4.77	19.37         2.04         5.0061         0.52         1.11         5.0072         0.55         1.42         5.009	0.10 4.78 4.62	4.04	<i>Cu</i> <sup>2+</sup> 19.37	<i>Ca</i> <sup>2+</sup> 0.10

Table A.8 Copper adsorption by slag (medium size)

Time	Dup.	T	A			В			С		
(h)		Items	рН	$Cu^{2+}$	$Ca^{2+}$	рН	$Cu^{2+}$	$Ca^{2+}$	рН	$Cu^{2+}$	$Ca^{2+}$
0		$C (mg L^{-1})$	4.05	19.31	0.00	4.05	19.31	0.00	4.05	19.31	0.00
		C (mg L <sup>-1</sup> )	4.65	16.20	1.25	4.60	13.42	1.54	4.71	12.65	2.44
	1	m (g)	1.0064			2.0074			3.0054		
		x/m (mg/g)	0.46				0.44		0.33		
		$C (mg L^{-1})$	4.67	14.25	0.85	4.91	16.38	1.42	4.79	12.02	2.10
	2	m (g)	1.0037				2.0024			3.0117	
24		x/m (mg/g)	0.76			0.22			0.36		
		$C (mg L^{-1})$	4.77	16.00	0.63	4.70	9.93	1.59	4.82	9.03	1.99
	3	m (g)		1.0072			2.0051		3.0011		
		x/m (mg/g)	0.49		0.70			0.51			
	Ave.	$C (mg L^{-1})$	4.70	15.48	0.91	4.74	13.24	1.52	4.77	11.23	2.18
		x/m (mg/g)	0.57			0.45			0.40		
Time	Dum		D			E			Control		
	Dun	Itoms		D			E			Control	
(h)	Dup.	Items	рН	$Cu^{2+}$	<i>Ca</i> <sup>2+</sup>	рН	$E$ $Cu^{2+}$	<i>Ca</i> <sup>2+</sup>	pН	Control $Cu^{2+}$	<i>Ca</i> <sup>2+</sup>
(h) 0	Dup.	Items C (mg L <sup>-1</sup> )	<i>рН</i> 4.05		<i>Ca</i> <sup>2+</sup> 0.00	<i>рН</i> 4.05	-	<i>Ca</i> <sup>2+</sup> 0.00	<i>pH</i> 4.05		
	Dup.		1	Cu <sup>2+</sup>		-	<i>Cu</i> <sup>2+</sup>		-	<i>Cu</i> <sup>2+</sup>	<i>Ca</i> <sup>2+</sup>
	<i>Dup.</i> 1	C (mg L <sup>-1</sup> )	4.05	<i>Cu</i> <sup>2+</sup> 19.31	0.00	4.05	<i>Cu</i> <sup>2+</sup> 19.31	0.00	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> )	4.05	<i>Cu</i> <sup>2+</sup> 19.31 8.22	0.00	4.05	Cu <sup>2+</sup> 19.31           4.67	0.00	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g)	4.05	Cu <sup>2+</sup> 19.31 8.22 4.0158	0.00	4.05	Cu <sup>2+</sup> 19.31           4.67           5.0177	0.00	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
		C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.05	Cu <sup>2+</sup> 19.31 8.22 4.0158 0.41	0.00 2.96	4.05	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44	0.00 4.66	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> )	4.05	Cu <sup>2+</sup> 19.31           8.22           4.0158           0.41           11.08	0.00 2.96	4.05	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44           1.81	0.00 4.66	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.05	Cu <sup>2+</sup> 19.31           8.22           4.0158           0.41           11.08           4.0102	0.00 2.96	4.05	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44           1.81           5.0129	0.00 4.66	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.05 4.75 4.86	Cu <sup>2+</sup> 19.31           8.22           4.0158           0.41           11.08           4.0102           0.31	0.00 2.96 2.79	4.05 4.82 4.78	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44           1.81           5.0129           0.52	0.00 4.66 3.53	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g)	4.05 4.75 4.86	Cu <sup>2+</sup> 19.31           8.22           4.0158           0.41           11.08           4.0102           0.31           5.30	0.00 2.96 2.79	4.05 4.82 4.78	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44           1.81           5.0129           0.52           6.87	0.00 4.66 3.53	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00
0	1	C (mg L <sup>-1</sup> ) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g) x/m (mg/g) C (mg L <sup>-1</sup> ) m (g)	4.05 4.75 4.86	Cu <sup>2+</sup> 19.31           8.22           4.0158           0.41           11.08           4.0102           0.31           5.30           4.0103	0.00 2.96 2.79	4.05 4.82 4.78	Cu <sup>2+</sup> 19.31           4.67           5.0177           0.44           1.81           5.0129           0.52           6.87           5.0094	0.00 4.66 3.53	4.05	<i>Cu</i> <sup>2+</sup> 19.31	<i>Ca</i> <sup>2+</sup> 0.00

Table A.9 Copper adsorption by slag (coarse size)

Time			A			В	С		
(h)	Dup.	Element	рН	$C (mg L^{-l})$	pН	$C (mg L^{-l})$	pН	$C (mg L^{-1})$	
		$\mathrm{Cd}^{2+}$	4.04	20.54	4.04	20.54		20.54	
0		Zn <sup>2+</sup>		22.48		22.48		22.48	
0		Cu <sup>2+</sup>		20.35		20.35	4.04	20.35	
		Ca <sup>2+</sup>		0.30		0.30		0.30	
		$\mathrm{Cd}^{2+}$	4.79	20.02	4.73	19.48	5.00	19.75	
		Zn <sup>2+</sup>		21.45		16.30		19.62	
		Cu <sup>2+</sup>		11.92		4.06	5.00	6.97	
	1	Ca <sup>2+</sup>		1.71		2.60		3.55	
	1	m (g)		1.0040		2.0026		3.0022	
		x/m (mg/g) Cd		0.04		0.06		0.03	
		x/m (mg/g) Zn		0.12		0.45		0.13	
		x/m (mg/g) Cu		1.26		1.22		0.67	
		$\mathrm{Cd}^{2^+}$	4.81	18.94		19.48		18.67	
	2	$Zn^{2+}$		20.12	4.82	18.13	1 95	19.46	
		Cu <sup>2+</sup>		5.80	4.82	2.63	4.85	4.95	
		Ca <sup>2+</sup>		1.54		2.96		3.78	
		m (g)	1.0061		2.0055		3.0050		
		x/m (mg/g) Cd	0.20		0.06		0.08		
		x/m (mg/g) Zn		0.32	0.31		0.14		
24		x/m (mg/g) Cu	2.17		1.33		0.77		
	3	$\mathrm{Cd}^{2+}$	4.73	19.75		19.75	4.88	18.40	
		Zn <sup>2+</sup>		19.96	4.80	20.45		15.96	
		Cu <sup>2+</sup>		7.79		5.73		1.53	
		Ca <sup>2+</sup>		1.48		2.96		4.61	
		m (g)	1.0069		2.0058		3.0029		
		x/m (mg/g) Cd	0.08		0.04		0.09		
		x/m (mg/g) Zn		0.35	0.14		0.32		
		x/m (mg/g) Cu		1.87		1.09		0.94	
	Ave.	$\mathrm{Cd}^{2^+}$		19.57		19.57	4.91	18.94	
		Zn <sup>2+</sup>	4.78	20.51	4.78	18.29		18.35	
		Cu <sup>2+</sup>	4.70	8.50	4.70	4.14		4.48	
		Ca <sup>2+</sup>		1.58		2.84		3.98	
		x/m (mg/g) Cd		0.11	0.05		0.07		
		x/m (mg/g) Zn	0.26			0.30	0.20		
		x/m (mg/g) Cu		1.77		1.21		0.79	

Table A.10 Cd-Zn-Cu adsorption by slag from multi-ion solution (fine size)

								(Continued)		
Time	Dup.	Element	D		E		Control			
(h)			рН	$C (mg L^{-1})$	рН	$C(mg L^{-1})$	рН	$C (mg L^{-1})$		
0		$Cd^{2+}$		20.54	_	20.54		20.54		
		Zn <sup>2+</sup>	4.04	22.48	4.04	22.48	4.04	22.48		
		Cu <sup>2+</sup>		20.35		20.35	1.01	20.35		
		Ca <sup>2+</sup>		0.30		0.30		0.30		
		$\mathrm{Cd}^{2+}$	4.87	18.94	5.02	18.13	-	20.29		
		Zn <sup>2+</sup>		16.63		15.30	3.99	22.28		
		Cu <sup>2+</sup>	ч.07	1.99		2.06	5.77	20.35		
	1	Ca <sup>2+</sup>		5.32		6.15		0.06		
	1	M (g)	4.0062			5.0016				
		x/m (mg/g) Cd	0.05			0.06				
		x/m (mg/g) Zn	0.21		0.21					
		x/m (mg/g) Cu		0.69	0.55					
		$\mathrm{Cd}^{2+}$		18.94		18.67				
		$Zn^{2+}$	4.88	16.13	5.02	15.63				
	2	Cu <sup>2+</sup>	4.00	1.07	5.02	1.57				
		Ca <sup>2+</sup>		4.91		6.21				
		m (g)	4.0010			5.0051				
		x/m (mg/g) Cd	0.05			0.05				
		x/m (mg/g) Zn	0.16			0.14				
24		x/m (mg/g) Cu	0.72		0.56					
	3	$\mathrm{Cd}^{2+}$		18.67	5.14	18.94				
		Zn <sup>2+</sup>	4.89	17.63		17.29				
		Cu <sup>2+</sup>	4.09	3.49	5.14	2.46				
		Ca <sup>2+</sup>		5.02		5.73				
		m (g)	4.0054			5.0007				
		x/m (mg/g) Cd	0.06			0.04				
		x/m (mg/g) Zn	0.17		0.15					
		x/m (mg/g) Cu	0.63		0.54					
	Ave.	$\mathrm{Cd}^{2+}$		18.85		18.58				
		Zn <sup>2+</sup>	4.88	16.80	5.06	16.08				
		Cu <sup>2+</sup>	7.00	2.18	5.00	2.03				
		Ca <sup>2+</sup>		5.08		6.03				
		x/m (mg/g) Cd	0.05		0.05					
		x/m (mg/g) Zn	0.18			0.17				
		x/m (mg/g) Cu	0.68		0.55					

Table A.10 Cd-Zn-Cu adsorption by slag from multi-ion solution (fine size)

Time	D	Flowent		A		В		С	
(h)	Dup.	Element	рН	$C (mg L^{-l})$	pН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$	
0		$\mathrm{Cd}^{2^+}$		21.34		21.34		21.34	
		$Zn^{2+}$	2.06	22.79	2.06	22.79	2.06	22.79	
		Cu <sup>2+</sup>	- 3.96	20.22	3.96	20.22	3.96	20.22	
		Ca <sup>2+</sup>		0.21		0.21		0.21	
		$Cd^{2+}$	4.80	20.50	4.02	20.38		19.77	
		$Zn^{2+}$		22.02		22.10	4.96	21.01	
		Cu <sup>2+</sup>		8.56	4.92	11.05	4.90	1.85	
	1	Ca <sup>2+</sup>		1.72		2.07		3.24	
	1	m (g)	1.0007			2.0045		3.0002	
		x/m (mg/g) Cd	0.13			0.07		0.08	
		x/m (mg/g) Zn		0.12		0.05		0.09	
		x/m (mg/g) Cu		1.75		0.69		0.92	
	2	$\mathrm{Cd}^{2^+}$	4.90	20.62		20.13	4.98	20.25	
		$Zn^{2+}$		21.63	4.87	20.47		21.48	
		Cu <sup>2+</sup>		11.75	4.07	8.13		2.12	
		Ca <sup>2+</sup>		1.38		2.00		2.89	
		m (g)		1.0055	2.0012		3.0045		
		x/m (mg/g) Cd	0.11		0.09		0.05		
		x/m (mg/g) Zn	0.17		0.17			0.07	
24		x/m (mg/g) Cu		1.26		0.91		0.90	
	3	Cd <sup>2+</sup>		20.62		20.50	4.94	20.50	
		Zn <sup>2+</sup>	4.88	21.32	4.89	19.77		18.45	
		Cu <sup>2+</sup>		11.18		5.64		2.35	
		Ca <sup>2+</sup>		1.24		2.34		3.31	
		m (g)	1.0050		2.0031		3.0017		
		x/m (mg/g) Cd	0.11		0.06		0.04		
		x/m (mg/g) Zn	0.22		0.23		0.22		
		x/m (mg/g) Cu		1.35		1.09		0.89	
	Ave.	Cd <sup>2+</sup>	_	20.58	-	20.34	4.96	20.17	
		Zn <sup>2+</sup>	4.86	21.66	4.89	20.78		20.31	
		Cu <sup>2+</sup>		10.50		8.27		2.10	
		Ca <sup>2+</sup>		1.45		2.14		3.15	
		x/m (mg/g) Cd		0.11		0.08	0.06		
		x/m (mg/g) Zn		0.17		0.15		0.12	
		x/m (mg/g) Cu		1.45		0.89		0.90	

Table A.11 Cd-Zn-Cu adsorption by slag from multi-ion solution (medium size)

	1	.11 Cu-Zii-Cu a				1		`	(Continued)
Time	Dup.	Element		D			E		Control
(h)	Dup.		рН	С	$(mg L^{-1})$	pН	$C(mg L^{-1})$	рН	$C (mg L^{-l})$
		$\mathrm{Cd}^{2+}$			21.34		21.34		21.34
0		$Zn^{2+}$	3.96		22.79	3.96	22.79	3.96	22.79
0		$Cu^{2+}$	5.90		20.22	5.90	20.22	5.90	20.22
		Ca <sup>2+</sup>			0.21		0.21		0.21
		$\mathrm{Cd}^{2+}$		19.17			19.29		21.34
		$Zn^{2+}$	5.00		17.99	5.02	18.06	4.06	22.79
		Cu <sup>2+</sup>	5.00		1.54	5.02	2.42	4.00	20.22
	1	Ca <sup>2+</sup>			3.86		4.75		0.14
	1	m (g)		4.00	52		5.0012		
		x/m (mg/g) Cd		0.0	8		0.06		
		x/m (mg/g) Zn		0.1	8		0.14		
		x/m (mg/g) Cu		0.7	0		0.53		
		$\mathrm{Cd}^{2+}$			19.77		19.77		
		$Zn^{2+}$	5.0	סר	20.00	5.12	15.89		
	Cu <sup>2+</sup>	Cu <sup>2+</sup>	5.0	J9	1.68	5.12	0.77		
	2	Ca <sup>2+</sup>			3.86		7.09		
	2	m (g)		4.00	56		5.0034		
		x/m (mg/g) Cd		0.0	6		0.05		
		x/m (mg/g) Zn		0.1	0		0.13		
24		x/m (mg/g) Cu		0.6	9	0.58			
		$\mathrm{Cd}^{2+}$			20.25		20.13		
		$Zn^{2+}$	4.9	20	20.24	5.06	19.54		
		Cu <sup>2+</sup>	т.,	//	1.58	5.00	1.85		
	3	Ca <sup>2+</sup>			4.48		5.72		
	5	m (g)		4.00	25		5.0037		
		x/m (mg/g) Cd		0.0	4		0.04		
		x/m (mg/g) Zn		0.1	0		0.10		
		x/m (mg/g) Cu		0.7	0		0.55		
		Cd <sup>2+</sup>			19.73		19.73		
		Zn <sup>2+</sup>	51	13	19.41	5.07	17.83		
		Cu <sup>2+</sup>	5.0	5.03 1.60		5.07	1.68		
	Ave.	Ca <sup>2+</sup>			4.06		5.85		
		x/m (mg/g) Cd		0.0	6		0.05		
		x/m (mg/g) Zn		0.1	3		0.12		
		x/m (mg/g) Cu		0.7	0		0.56		

 Table A.11
 Cd-Zn-Cu adsorption by slag from multi-ion solution (medium size)

Time	Dum	Elamont		A		В		С
(h)	Dup.	Element	рН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$
		$\mathrm{Cd}^{2^+}$		20.23		20.23		20.23
0		Zn <sup>2+</sup>	4.00	21.68	4.00	21.68	4.00	21.68
0		Cu <sup>2+</sup>	4.00	20.13	4.00	20.13	4.00	20.13
		Ca <sup>2+</sup>		0.00		0.00		0.00
		$\mathrm{Cd}^{2^+}$		19.44		19.56		19.33
		$Zn^{2+}$	4.99	21.17	5.02	21.17	5.20	20.74
		Cu <sup>2+</sup>	4.99	15.78		11.98	5.20	8.08
	1	Ca <sup>2+</sup>		0.98		1.55		5.69
	1	m (g)		1.0076		2.0026		3.0095
		x/m (mg/g) Cd		0.05		0.02		0.02
		x/m (mg/g) Zn		0.11		0.05		0.06
		x/m (mg/g) Cu		0.65		0.61		0.60
		$\mathrm{Cd}^{2^+}$		19.22		19.22		18.66
		Zn <sup>2+</sup>	4.18	21.17		20.88	4.87	14.77
		$Cu^{2+}$	4.10	16.39	5.10	12.40	4.07	3.31
	2	Ca <sup>2+</sup>		2.53		1.49		1.90
	2 m (g)			1.0055		2.0062		3.0094
		x/m (mg/g) Cd		0.08	0.04			0.06
		x/m (mg/g) Zn		0.11		0.08		0.36
24		x/m (mg/g) Cu		0.56		0.58		0.84
		$Cd^{2+}$		19.44	_	19.44		19.11
		Zn <sup>2+</sup>	5.00	21.75	5.01	21.10	5.11	20.52
		Cu <sup>2+</sup>	5.00	14.15	5.01	9.61	5.11	5.13
	3	Ca <sup>2+</sup>		0.80		2.01		2.70
	5	m (g)		1.0092		2.0075		3.0026
		x/m (mg/g) Cd		0.05		0.03		0.03
		x/m (mg/g) Zn		0.02		0.06		0.07
		x/m (mg/g) Cu		0.89		0.79		0.75
		Cd <sup>2+</sup>		19.37		19.41		19.03
		Zn <sup>2+</sup>	4.72	21.37	5.04	21.05	5.06	18.67
		Cu <sup>2+</sup>		15.44	2.01	11.33	2.00	5.51
	Ave.	Ca <sup>2+</sup>		1.44		1.69		3.43
		x/m (mg/g) Cd		0.06		0.03		0.04
		x/m (mg/g) Zn		0.08		0.06		0.16
		x/m (mg/g) Cu		0.70		0.66		0.73

Table A.12 Cd-Zn-Cu adsorption by slag from multi-ion solution (coarse size)

		.12 Cu-211-Cu ac	1					Continued)
Time	Dup.	Element		D		E		Control
(h)	2 up		pН	$C(mg L^{-1})$	рН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$
		Cd <sup>2+</sup>		20.23		20.23		20.23
0		Zn <sup>2+</sup>	4.00	21.68	4.00	21.68	4.00	21.68
0		Cu <sup>2+</sup>	4.00	20.13	4.00	20.13	4.00	20.13
		Ca <sup>2+</sup>		0.00		0.00		0.00
		$\mathrm{Cd}^{2+}$		19.11		19.11		19.78
		$Zn^{2+}$	5.09	20.74	5.15	20.01	4.12	21.90
		Cu <sup>2+</sup>	5.09	2.79	5.15	6.17	4.12	20.13
	1	Ca <sup>2+</sup>		3.16		2.93		0.00
	1	m (g)		4.0045		5.0002		
		x/m (mg/g) Cd		0.03		0.02		
		x/m (mg/g) Zn		0.04		0.06		
		x/m (mg/g) Cu		0.65		0.42		
		$\mathrm{Cd}^{2^+}$		19.11		18.99		
		$Zn^{2+}$	5.09	19.64	5.27	20.95		
	2	Cu <sup>2+</sup>	5.09	0.52		5.75		
		Ca <sup>2+</sup>		2.76		5.06		
	2	m (g)		4.0030		5.0039		
		x/m (mg/g) Cd	0.03			0.02		
		x/m (mg/g) Zn		0.08		0.03		
24		x/m (mg/g) Cu		0.73	0.43			
		$\mathrm{Cd}^{2+}$		18.88		18.21		
		$Zn^{2+}$	5.05	18.84	5.15	19.72		
		$Cu^{2+}$	5.05	5.26	5.15	2.92		
	3	$Ca^{2+}$		3.05		3.45		
	5	m (g)		4.0094		5.0071		
		x/m (mg/g) Cd		0.03		0.05		
		x/m (mg/g) Zn		0.11		0.07		
		x/m (mg/g) Cu		0.56		0.52		
		$\mathrm{Cd}^{2^+}$		19.03		18.77		
		Zn <sup>2+</sup>	5 00	19.74	5 10	20.23	]	
		Cu <sup>2+</sup>	5.08	2.86	5.19	4.95	]	
	Ave.	Ca <sup>2+</sup>		2.99		3.81		
		x/m (mg/g) Cd		0.03		0.03		
		x/m (mg/g) Zn		0.08		0.05		
		x/m (mg/g) Cu		0.65		0.46	1	

Table A.12Cd-Zn-Cu adsorption by slag from multi-ion solution (coarse size)

		ine in		lium		arse
Time (min)	рН	$C (mg L^{-l})$	рН	$C (mg L^{-l})$	рН	$C (mg L^{-l})$
0	2.88	23.12	2.88	23.12	2.88	23.12
Ca <sup>2+</sup>	2.00	0.04	2.00	0.04	2.00	0.04
15	6.95	10.91	7.00	13.75	7.00	16.48
30	7.15	9.84	7.19	12.80	7.20	14.82
45	7.36	8.65	7.39	10.55	7.38	12.45
60	7.30	7.59	7.35	8.42	7.37	9.72
75	7.38	6.76	7.43	6.64	7.45	7.35
90	7.34	5.81	7.41	5.69	7.44	6.05
105	7.47	4.98	7.51	4.50	7.53	5.33
120		3.47		2.97		3.12
120	7.49	1.58*	7.56	1.58*	7.60	2.26*
Ca <sup>2+</sup>	7.49	148.45	7.30	144.29	7.60	139.67
Ca		8.62*		8.38*		8.40*

Table A.13 Change of  $Zn^{2+}$  in working solution mixed with marble

mass of marble – 50 g Note:

C – concentration of the elements in the solution \* -- concentration of the elements in solutions filtered by filter paper (Whatman, No.42)

Table A.14 Change of Cu in working solution mixed with matole (ditto)								
Time (min)	Fi	ne	Med	lium	Coarse			
	рН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$	рН	$C (mg L^{-l})$		
0	3.19	17.18	3.19	17.18	3.19	17.18		
Ca <sup>2+</sup>	5.19	1.02	5.19	1.02	5.17	1.02		
15	7.11	4.75	7.08	4.54	6.97	7.11		
30	7.29	3.69	7.28	3.26	7.20	4.82		
45	7.44	2.72	7.45	2.29	7.43	3.31		
60	7.51	2.25	7.54	1.89	7.56	2.29		
75	7.56	2.06	7.61	1.54	7.63	1.70		
90		1.63		1.39		1.44		
90	7.58	0.61*	7.63	0.69*	7.66	0.71*		
Ca <sup>2+</sup>		81.40*		75.85*		72.61*		

Table A 14 Change of  $Cu^{2+}$  in working solution mixed with marble (ditto)

1 40		U	in working solution mixed with stag					
Time (min)	Fi	ne	Med	lium	Сос	arse		
1 ime (min)	рН	$C (mg L^{-1})$	pН	$C (mg L^{-1})$	pН	$C (mg L^{-1})$		
0	3.14	22.63	3.14	22.63	3.14	22.63		
Ca <sup>2+</sup>	5.14	0.34	5.14	0.34	5.14	0.34		
15		20.91		22.63		22.51		
30	4.18	20.33	3.86	23.20	3.15	23.32		
45	4.42	19.07	4.2	21.94	3.57	21.94		
60	4.64	17.46	4.34	21.37	3.78	21.25		
75	4.77	17.00	4.45	21.02	4.03	21.37		
90	4.91	16.31	4.48	20.10	4.15	21.14		
105	5.08	15.97	4.55	20.68	4.26	21.14		
120	5.38	15.16	4.68	19.99	4.38	21.02		
135	5.8	14.47	4.73	20.33	4.44	21.02		
Ca <sup>2+</sup>	5.0	25.48	4.75	17.64	4.44	8.17		
240	5.98	9.89	4.91	17.74	4.45	20.24		
Ca <sup>2+</sup>	3.90	25.61	4.91	19.74	4.43	8.95		
24 hours	5.55	0.19	5.35	9.18	5.17	17.31		
Ca <sup>2+</sup>	5.55	43.20	5.55	36.06	3.17	24.15		

Table A.15 Change of  $Zn^{2+}$  in working solution mixed with slag

Note: mass of slag -50 g

C – concentration of the elements in the solution

Table A.16 Change of Cu <sup>-</sup> in working solution mixed with slag									
Time (min)	Fi	ne	Mea	lium	Co	arse			
1 tme (mth)	рН	$C(mg L^{-1})$	рН	$C (mg L^{-1})$	рН	$C (mg L^{-1})$			
0	3.03	17.35	3.03	17.35	3.03	17.35			
15	4.22	8.82	3.55	14.45	3.26	13.74			
30	4.53	4.56	4.07	10.62	3.82	10.55			
45	4.95	2.10	4.4	7.65	4.2	7.70			
60	5.26	1.02	4.57	5.82	4.35	5.51			
75	5.51	0.41	4.73	4.31	4.46	4.12			
90	5.56	0.17	4.9	3.09	4.56	2.87			
105	5.68	0.15	5.1	2.10	4.66	2.02			
120	5.7	0.10	5.26	1.34	4.75	1.32			
135	5.72	0.07	5.44	0.78	4.85	0.85			
150	5.76	0.02	5.49	0.44	4.95	0.54			
180	5.76	0.07	5.52	0.22	5.22	0.07			
210	5.87	0.07	5.63	0.29	5.35	0.07			
Ca <sup>2+</sup>	5.87	11.13	5.05	9.46	5.55	6.82			

Table A.16 Change of  $Cu^{2+}$  in working solution mixed with slag

Note: mass of slag -50 g

C – concentration of the elements in the solution

Table A.17 Change of		Ũ		Į		2
Time (min)		Fine	Medium			oarse
	рН	$C (mg L^{-l})$	рН	$C (mg L^{-1})$	рН	$C(mg L^{-1})$
0		17.32		17.32		17.32
Ca <sup>2+</sup>	3.04	0.07	3.04	0.07	3.04	0.07
Zn <sup>2+</sup>		0.00		0.00		0.00
15	3.43	14.83	3.16	16.25	3.11	16.29
30	3.97	11.79	3.50	14.54	3.25	14.78
45	4.20	9.33	3.80	12.96	3.42	12.84
60	4.27	7.60	3.99	11.64	3.55	11.55
75	4.37	5.24	4.13	9.72	3.75	9.67
90	4.43	3.97	4.20	8.40	3.90	8.43
105	4.50	3.31	4.28	7.40	4.02	7.19
120	4.55	2.48	4.34	6.26	4.13	6.02
135	4.63	1.80	4.40	5.09	4.22	4.92
150	4.66	1.36	4.44	4.12	4.30	3.92
165		1.17		3.39		3.07
Ca <sup>2+</sup>	4.72	1.04	4.47	0.36	4.35	0.27
$Zn^{2+}$		1.91		1.73		1.12
Initial Zn content in slag (mg)	3.37			2.02		0.8
adsorbed Cu <sup>2+</sup> (mg)		2.42	2.09		2.14	
released Zn <sup>2+</sup> (mg)	(	0.29	0.26		0.17	
released Zn (mg) Note: mass of slag – 50 g		0.29		0.20	L L	9.1/

Table A.17 Change of  $Cu^{2+}$  in working solution mixed with slag saturated by  $Zn^{2+}$ 

Note: mass of slag -50 g

C – concentration of the elements in the solution

TT 1 1 A 10	$c_{1}$ $c_{2}$	1. 1	• 1 • 1 1	$(11 - 2^+)$
Table A IX	(hange of Zn <sup>-</sup>	in working solution	mixed with slag s	aturated by Cu <sup>2</sup>
1 4010 1 1.10	Change of En	in working solution		alalated by Ca

	Time (hour)		Fine		ledium	Coarse	
	Time (nour)	рН	$C (mg L^{-1})$	pН	$C(mg L^{-1})$	рН	$C(mgL^{-1})$
	0		18.65		18.65		18.65
	Cu <sup>2+</sup>	4.15	0.00	4.15	0.00	4.15	0.00
	Ca <sup>2+</sup>	0.00			0.00		0.00
	24		1.41		1.75		4.16
	Cu <sup>2+</sup>	6.53	0.00	6.43	0.00	5.99	0.00
	Ca <sup>2+</sup>		13.07		12.78		7.90
Initial (	Initial Cu content in slag (mg)		3.06		3.06	3.05	
ac	adsorbed $Zn^{2+}$ (mg)		2.59 2.54		2.54	2.17	
re	eleased Cu <sup>2+</sup> (mg)	0		0		0	

Note: mass of slag -50 g

C – concentration of the elements in the solution

# Appendix B

# **Desorption Batch Tests Results**

	140	IC D.1 DC	orpeion .	or cuur	mann og	0.0110111	01		
		1			2			3	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)
Feeding solution	5.71	0.00	0.00	5.71	0.00	0.00	5.71	0.00	0.00
m (g)		0.9993			2.0024			2.9969	
adsorbed Cd (mg)		0.21			0.43			0.53	
			С	vernight	;				
24 h	5.95	0.60	0.10	6.01	1.16	0.43	6.08	1.64	0.43
Recovery (%)		43.86			41.00			46.81	
96 h	5.85	0.07	0.09	6.06	0.09	0.20	6.18	0.10	0.44
Recovery (%)		5.04			3.25			2.96	
	4				5			Blank	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)
Feeding solution	5.71	0.00	0.00	5.71	0.00	0.00	5.71	0.00	0.00
m (g)		3.9966			5.0008				
adsorbed Cd (mg)		0.68			1.05				
			С	vernight					
24 h	6.17	1.92	0.62	6.20	2.18	0.96	5.73	0.02	0.19
Recovery (%)		42.47			31.21				
96 h	6.21	0.12	0.43	6.34	0.12	0.72	5.43	0.00	0.24
recovery (%)		2.71			1.75				

Table B 1	Desorption	of Cadmium	by 0.01M KCl
Tuble D.1	Description	or Caumum	

 Table B.2
 Desorption of Cadmium by 0.01M NaCl

		1			2		3		
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)
Feeding solution	5.48	0.00	0.00	5.48	0.00	0.00	5.48	0.000	0.00
m (g)		0.9983			1.9990			3.0004	
adsorbed Cd (mg)		0.22			0.29			0.49	
			С	vernight	t				
24 h	5.75	0.57	0.24	5.86	1.03	0.38	5.98	1.46	0.43
Recovery (%)		38.57			53.25			44.38	
96 h	5.89	0.05	0.23	6.06	0.09	0.20	6.05	0.08	0.17
Recovery (%)		3.11			4.77			2.46	
		4			5			Blank	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)
Feeding solution	5.48	0.00	0.00	5.48	0.00	0.00	5.48	0.00	0.00
m (g)		3.9976			4.9913				
adsorbed Cd (mg)		0.73			1.06				
			С	vernight	ţ				
24 h	6.13	1.67	0.77	6.10	2.02	0.67	5.50	0.02	0.10
Recovery (%)		34.46			28.47				
96 h	6.26	0.11	0.57	6.24	0.14	0.53	5.43	0.00	0.00
recovery (%)		2.31			2.01				

	14010	<u>D.3</u> Dese	ip non o		2			3	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)
Feeding solution	5.63	0.00	0.00	5.63	0.00	0.00	5.63	0.00	0.00
m (g)		0.9996			2.0002			3.0046	
adsorbed Cd (mg)		0.22			0.29			0.49	
		Overnight							
24 h	5.84	0.57	0.24	5.91	1.06	0.29	6.03	1.51	0.57
Recovery (%)		81.94			51.80			35.14	
96 h	5.86	0.01	0.15	6.01	0.06	0.12	6.19	0.09	0.23
Recovery (%)		3.11 4.77					2.46		
		4			5			Blank	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Са (ррт)
Feeding solution	5.63	0.00	0.00	5.63	0.00	0.00	5.63	0.00	0.00
m (g)		4.0032			4.999				
adsorbed Cd (mg)		0.75			0.95				
			С	vernight	ţ				
24 h	6.05	1.50	0.38	6.05	1.90	0.62	5.53	0.02	0.05
Recovery (%)		30.16			30.03	•			
96 h	6.14	0.09	0.15	6.30	0.12	0.23	5.37	0.00	0.00
recovery (%)		1.86			1.82				

Table B.3 Desorption of Cadmium by 0.01M NaNO<sub>3</sub>

#### Table B.4 Desorption of Zinc by 0.01M KCl

		1			2			3	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.81	0.00	0.00	5.81	0.00	0.00	5.81	0.00	0.00
m (g)		1.0014			1.9988			2.9954	
adsorbed Zn (mg)		0.27			0.49			0.24	
			С	vernight	t				
24 h	6.15	0.14	0.52	6.04	0.27	0.49	6.10	0.30	0.58
Recovery (%)		7.93			8.47			18.82	
96 h	6.69	0.01	0.76	6.73	0.02	0.73	6.69	0.02	0.87
Recovery (%)		0.79			0.67			1.34	
		4			5			Blank	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.81	0.00	0.00	5.81	0.00	0.00	5.81	0.00	0.00
m (g)		3.9937			4.9956				
Adsorbed Zn (mg)		0.96			1.34				
			С	vernight	ţ				
24 h	6.20	0.47	0.87	6.22	0.58	1.13	5.54	0.01	0.23
Recovery (%)		7.29			6.48				
96 h	6.86	0.03	1.08	6.91	0.03	1.57	5.57	0.00	0.12
Recovery (%)		0.45			0.32				

	17	ible B.5 L	Jesoiptio		100000000	INI NaC	1		
		1			2			3	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.80	0.00	0.00	5.80	0.00	0.00	5.80	0.00	0.00
m (g)		1.0020			1.9998			2.9943	
adsorbed Zn (mg)	0.19				0.41			0.62	
			С	vernight	;				
24 h	6.09	0.16	0.41	6.11	0.25	0.70	6.18	0.34	0.70
Recovery (%)		12.34			9.33			8.22	
96 h	6.56	0.04	0.52	6.71	0.04	0.58	6.89	0.04	0.93
Recovery (%)	2.80			1.33				1.05	
		4			5			Blank	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.80	0.00	0.00	5.80	0.00	0.00	5.80	0.00	0.00
m (g)		3.9960			4.9969				
Adsorbed Zn (mg)		0.86			1.26				
			С	vernight	;				
24 h	6.16	0.40	0.81	6.11	0.47	0.79	5.56	0.00	0.12
Recovery (%)		7.06			5.60				
96 h	6.96	0.04	0.96	6.98	0.06	0.93	5.52	0.00	0.09
Recovery (%)		0.63			0.69				

Table B 5	Desorption	of Zinc by	0.01M NaCl
Tuble D.5	Description	or Zine U	0.0110110001

#### Table B.6 Desorption of Zinc by 0.01M NaNO<sub>3</sub>

		1			2			3	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.66	0.00	0.00	5.66	0.00	0.00	5.66	0.00	0.00
m (g)		1.0010			2.0006			2.9983	
adsorbed Zn (mg)		0.35			0.49			0.78	
			С	vernight	;				
24 h	5.97	0.14	0.35	5.96	0.25	0.49	6.78	0.38	0.61
Recovery (%)		5.83			7.58			7.23	
96 h	6.55	0.02	0.38	6.64	0.05	0.41	6.78	0.04	0.79
Recovery (%)		0.92			1.56			0.83	
		4			5			Blank	
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)
Feeding solution	5.66	0.00	0.00	5.66	0.00	0.00	5.66	0.00	0.00
m (g)		3.9948			4.9948				
Adsorbed Zn (mg)		0.88			1.07				
			С	vernight					
24 h	6.05	0.43	0.64	6.22	0.58	1.16	5.43	0.00	0.17
Recovery (%)		7.22			8.09				
96 h	6.81	0.06	0.73	6.97	0.04	1.63	5.57	0.00	0.12
Recovery (%)		0.98			0.61				

	1 a	ole B./ D	esorption	10100	pper by b	.01101 KC			
		1			2			3	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)
Feeding solution	5.29	0.00	0.00	5.29	0.00	0.00	5.29	0.00	0.00
m (g)		1.0018			2.0004			2.9997	
adsorbed Cu (mg)		0.62			2.03			1.95	
			С	vernight	;				
24 h	5.71	0.00	0.18	5.78	0.00	0.31	5.86	0.00	0.60
Recovery (%)		0.00	0 0.00				0.00		
96 h	5.80	0.00	0.21	6.01	0.00	0.32	6.18	0.00	0.74
Recovery (%)		0.00			0.00			0.00	
		4		5				Blank	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)
Feeding solution	5.29	0.00	0.00	5.29	0.00	0.00	5.29	0.00	0.00
m (g)		3.9940			4.9962				
Adsorbed Cu (mg)		2.16			2.45				
			С	vernight	;				
24 h	5.76	0.00	0.57	5.91	0.00	0.83	5.54	0.00	0.00
Recovery (%)		0.00			0.00				
96 h	5.99	0.00	0.62	6.26	0.00	1.06	5.63	0.00	0.00
Recovery (%)		0.00			0.00				

Table B.7 D	Desorption of Copper	bv 0.01M KCl

#### Table B.8Desorption of Copper by 0.01M NaCl

		1			2			3	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)
Feeding solution	5.51	0.00	0.00	5.51	0.00	0.00	5.51	0.00	0.00
m (g)		1.0025			1.9981			2.9982	
adsorbed Cu (mg)		0.75			1.39			1.70	
			С	vernight	t				
24 h	5.71	0.00	0.29	5.71	0.00	0.26	5.60	0.00	0.52
Recovery (%)		0.00			0.00			0.00	
96 h	5.90	0.00	0.29	5.89	0.00	0.21	5.86	0.00	0.65
Recovery (%)		0.00			0.00			0.00	
		4			5			Blank	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	pН	Cu (ppm)	Ca (ppm)
Feeding solution	5.51	0.00	0.00	5.51	0.00	0.00	5.51	0.00	0.00
m (g)		3.9965			4.9994				
Adsorbed Cu (mg)		2.37			2.55				
			С	vernight	t				
24 h	5.93	0.00	0.86	5.67	0.00	0.65	5.39	0.00	0.00
Recovery (%)		0.00			0.00				
96 h	6.20	0.00	0.79	5.91	0.00	0.71	5.18	0.00	0.00
Recovery (%)		0.00			0.00				

	1 au	e B.9 Des	sorption	л сөр	$\frac{1}{100}$	I IVI INAIN	03		
		1			2			3	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Са (ррт)
Feeding solution	6.01	0.00	0.00	6.01	0.00	0.00	6.01	0.00	0.00
m (g)		1.0016			1.9989			3.0006	
adsorbed Cu (mg)		0.95 1.48					2.25		
			С	Overnight					
24 h	5.70	0.00	0.18	5.80	0.00	0.39	5.94	0.00	0.42
Recovery (%)		0.00			0.00			0.00	
96 h	5.75	0.00	0.12	6.04	0.00	0.41	6.10	0.00	0.44
Recovery (%)		0.00 0.00					0.00		
		4		5				Blank	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ppm)	Са (ррт)
Feeding solution	6.01	0.00	0.00	6.01	0.00	0.00	6.01	0.00	0.00
m (g)		4.0024			5.0016				
Adsorbed Cu (mg)		2.36			2.52				
			С	vernight	t				
24 h	5.74	0.00	0.65	5.96	0.00	0.62	5.52	0.00	0.00
Recovery (%)		0.00			0.00				
96 h	6.03	0.00	0.74	6.18	0.00	0.53	5.28	0.00	0.00
Recovery (%)		0.00	-		0.00	-			

Table B.9	Desorption of Copper	by 0.01M NaNO <sub>3</sub>

#### Table B.10 Desorption of Cadmium by 0.02M Citric Acid

		1			2			3	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Са (ррт)
Feeding solution	2.48	0.00	0.00	2.48	0.00	0.00	2.48	0.00	0.00
m (g)		0.9999			1.9990			2.9987	
adsorbed Cd (mg)		0.19 0.25					0.46		
		Overnight							
24 h	2.58	0.41	0.11	2.61	0.73	0.14	2.67	1.25	0.23
Recovery (%)		32.40			43.88			40.87	
96 h	2.49	0.10	0.25	2.50	0.16	0.33	2.56	0.12	1.10
Recovery (%)		8.15			9.51			4.07	
		4			5			Blank	
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	pН	Cd (ppm)	Са (ррт)
Feeding solution	2.48	0.00	0.00	2.48	0.00	0.00	2.48	0.00	0.00
m (g)		4.0032			4.999				
adsorbed Cd (mg)		0.75			0.95				
			С	vernight	ţ				
24 h	2.70	1.68	0.17	2.80	1.86	0.29	2.53	0.00	0.00
Recovery (%)		67.05			46.09				
96 h	2.65	0.09	0.38	2.72	0.12	1.12	2.46	0.00	0.00
recovery (%)		3.62			3.09				

		<i>1</i>			2			3		
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.40	0.00	0.00	2.40	0.00	0.00	2.40	0.00	0.00	
m (g)		1.0050			2.0019			3.0023		
adsorbed Cd (mg)		0.19			0.33			0.31		
			С	vernight	;					
24 h	2.54	0.57	0.40	2.63	0.95	0.52	2.68	1.14	0.77	
Recovery (%)		45.90			42.53			54.53		
96 h	2.45	0.11	1.15	2.58	0.22	1.83	2.60	0.09	3.50	
Recovery (%)		9.05			9.68			4.35		
		4		5				Blank		
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	pН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.40	0.00	0.00	2.40	0.00	0.00	2.40	0.00	0.00	
m (g)		3.9976			4.9983					
adsorbed Cd (mg)		0.54			0.63					
			С	vernight						
24 h	2.73	1.61	0.66	2.78	1.72	0.40	2.47	0.00	0.00	
Recovery (%)		44.55			41.31			/		
0(1	2.68	0.12	3.75	2.73	0.18	1.23	2.41	0.00	0.00	
96 h	2.00	0.12	5.15	2.75	0.10					

Table B.11 Desorption of Cadmium by 0.02M Citric Acid + 0.01M KCl

Table B.12 Desorption of Zinc by 0.02M Citric Acid

		1	sorption		2			3		
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	
Feeding solution	2.48	0.00	0.00	2.48	0.00	0.00	2.48	0.00	0.00	
m (g)		1.0014			2.0028			2.9988		
adsorbed Zn (mg)		0.27			0.41			0.47		
			С	vernight	t					
24 h	2.57	0.54	0.06	2.62	1.27	0.29	2.69	2.53	0.23	
Recovery (%)		29.58		46.19			81.52			
96 h	2.46	0.02	0.19	2.52	0.05	1.92	2.60	0.16	0.88	
Recovery (%)		1.22			1.90			5.03		
		4		5				Blank		
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	
Feeding solution	2.48	0.00	0.00	2.48	0.00	0.00	2.48	0.00	0.00	
m (g)		3.9958			4.9976					
Adsorbed Zn (mg)		0.47			0.68					
			С	vernight	t					
24 h	2.73	2.87	0.17	2.75	3.34	0.14	2.53	0.00	0.00	
Recovery (%)		92.51			73.17					
96 h	2.68	0.27	0.55	2.72	0.33	0.36	2.46	0.00	0.00	
Recovery (%)		8.63	-		7.33	-				

		<i>l</i>			2			3		
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	
Feeding solution	2.40	0.00	0.00	2.40	0.00	0.00	2.40	0.00	0.00	
m (g)		1.0038			2.0008			3.0011		
adsorbed Zn (mg)		0.08			0.11			0.33		
			С	vernight						
24 h	2.52	0.69	0.47	2.59	0.81	1.31	2.65	2.34	0.52	
Recovery (%)		126.40			110.60			106.68		
96 h	2.45	0.03	1.18	2.48	0.09	6.65	2.58	0.20	1.89	
Recovery (%)		5.43			12.22		9.17			
		4		5				Blank		
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	
Feeding solution	2.40	0.00	0.00	2.40	0.00	0.00	2.40	0.00	0.00	
m (g)		3.9958			4.9976					
Adsorbed Zn (mg)		0.47			0.69					
			С	vernight						
24 h	2.72	3.61	0.35	2.78	3.78	0.41	2.47	0.00	0.00	
Recovery (%)		104.20			94.12					
96 h	2.72	0.40	0.79	2.76	0.39	0.88	2.41	0.00	0.00	
		11.58			9.63					

Table B.13 Desorption of Zinc by 0.02M Citric Acid + 0.01M KCl

 Table B.14
 Desorption of Copper by 0.02M Citric Acid

		1	÷		2			3	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)
Feeding solution	2.50	0.00	0.00	2.50	0.00	0.00	2.50	0.00	0.00
m (g)		1.0018			2.0004			2.9997	
adsorbed Cu (mg)		0.62			2.03			1.95	
			С	vernight	;				
24 h	2.61	1.83	0.24	2.65	2.47	0.21	2.70	2.71	0.37
Recovery (%)		44.28			18.18			20.82	
96 h	2.45	0.79	0.57	2.48	3.02	0.84	2.58	0.96	2.80
Recovery (%)	19.06			22.24			7.37		
		4		5				Blank	
Items	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)
Feeding solution	2.50	0.00	0.00	2.50	0.00	0.00	2.50	0.00	0.00
m (g)		3.9940			4.9962				
Adsorbed Cu (mg)		2.16			2.45				
			С	vernight					
24 h	2.72	2.82	0.37	2.76	2.01	0.26	2.54	0.00	0.00
Recovery (%)		19.57			12.31				
96 h	2.61	0.86	0.92	2.71	0.45	1.28		0.00	0.00
Recovery (%)		5.95			2.73				

		1	or coppe		2			3		
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	
Feeding solution	2.41	0.00	0.00	2.41	0.00	0.00	2.41	0.00	0.00	
m (g)		1.0025			1.9981			2.9982		
adsorbed Cu (mg)		0.75			1.39			1.70		
			С	vernight	ţ					
24 h	2.54	2.01	1.52	2.62	2.32	0.39	2.67	3.10	0.58	
Recovery (%)		40.38			25.04			27.38		
96 h	2.41	0.62	3.99	2.48	1.34	2.91	2.56	0.72	2.74	
Recovery (%)		12.41			14.40			6.36		
		4			5			Blank		
Items	pН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	
Feeding solution	2.41	0.00	0.00	2.41	0.00	0.00	2.41	0.00	0.00	
m (g)		3.9965			4.9994					
Adsorbed Cu (mg)		2.37			2.55					
			С	vernight	ţ					
24 h	2.75	3.10	0.53	2.76	1.51	0.45	2.45	0.00	0.00	
Recovery (%)		40.38			25.04					
96 h	2.64	0.69	3.10	2.74	0.27	1.30	2.35	0.00	0.00	
Recovery (%)		4.35			1.62					

 Table B.15
 Desorption of Copper by 0.01M KCl + 0.02M Citric Acid

#### Table B.16 Desorption of Copper by 0.01M KCl + 0.02M Tartaric Acid

		1			2			3		
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	
Feeding solution	2.37	0.00	0.00	2.37	0.00	0.00	2.37	0.00	0.00	
m (g)		1.0016			1.9989			3.0006		
adsorbed Cu (mg)		0.95			1.48			2.25		
			C	vernight	ŧ					
24 h	2.52	2.68	0.13	2.59	2.78	0.11	2.65	3.73	0.18	
Recovery (%)		42.21			28.24			24.95		
96 h	2.36	1.27	0.68	2.49	1.37	0.57	2.56	2.06	0.84	
Recovery (%)		19.99			13.91			13.74		
		4		5				Blank		
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	
Feeding solution	2.37	0.00	0.00	2.37	0.00	0.00	2.37	0.00	0.00	
m (g)		3.9965			4.9994					
Adsorbed Cu (mg)		2.37			2.55					
			C	vernight	ţ					
24 h	2.70	3.84	0.21	2.74	2.47	0.21	2.41	0.00	0.13	
Recovery (%)		24.35			14.53					
96 h	2.69	0.86	1.09	2.74	1.03	0.68	2.30	0.00	0.00	
Recovery (%)		5.43			6.06					

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		1			2			3		
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	
Feeding solution	7.97	0.00	597.59	7.97	0.00	597.59	7.97	0.00	597.59	
m (g)		1.0049			2.0071			3.0031		
adsorbed Cu (mg)		0.49			0.94			1.09		
			С	vernight	t					
24 h	8.02	0.19	192.27	8.00	0.23	724.90	7.99	0.23	688.52	
Recovery (%)		5.89		3.71			3.21			
96 h	8.03	0.18		8.05	0.22		8.05 0.22			
Recovery (%)		3.31			2.31		2.00			
		4		5				Blank		
Items	pН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	
Feeding solution	7.97	0.00	597.59	7.97	0.00	597.59	7.97	0.00	597.59	
m (g)		3.9981			5.0008					
Adsorbed Cu (mg)		1.57			1.74					
			С	vernight	t					
24 h	7.97	0.43	795.05	7.98	0.39	698.92	7.96	0.00	828.83	
Recovery (%)		4.08			3.36					
96 h	8.06	0.29		8.06	0.29		7.98	0.07		
Recovery (%)		2.08	-		1.88			-	-	

Table B.17	Desorption	of Copper	by Sea	Water
14010 0.17	Description	or copper	0 9 0 0 0 0	

#### Table B.18 Desorption of Copper by 0.01M Citric Acid in Sea Water

		1			2		3			
Items	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Си (ppm)	Ca (ppm)	
Feeding solution	2.49	0.14	722.10	2.49	0.14	722.10	2.49	0.14	722.10	
m (g)		1.0036			2.0058			2.9963		
adsorbed Cu (mg)		1.58			2.38			2.44		
			С	vernight	t					
24 h	2.55	3.21		2.58	2.20		2.6	3.31		
Recovery (%)		29.06			12.94			19.46		
96 h	2.52	1.56	827.90	2.65	1.13	755.04	2.69	0.33	748.42	
Recovery (%)		13.78			6.41			1.34		
		4			5			Blank		
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	рН	Си (ppm)	Ca (ppm)	
Feeding solution	2.49	0.14	722.10	2.49	0.14	722.10	2.49	0.14	722.10	
m (g)		3.9995			4.9927					
Adsorbed Cu (mg)		2.32			2.59					
			С	vernight	ţ					
24 h	2.63	3.90		2.66	3.52		2.43	0.14		
Recovery (%)		24.34			19.56					
96 h	2.72	0.29	682.19	2.80	0.33	708.68	2.38	0.11	798.10	
Recovery (%)		1.18			1.26					

		1		2	2		3		
Items	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Си (ppm)	Ca (ppm)
Feeding solution	2.13	0.00	722.30	2.13	0.00	722.30	2.13	0.00	722.30
m (g)		1.0055			2.0037			3.002	
adsorbed Cu (mg)		0.33			0.83			1.19	
			С	vernight	;				
24 h	2.23	1.44	761.27	2.29	2.95	769.07	2.33	3.85	753.48
Recovery (%)		64.64			53.28			48.65	
96 h	2.09	0.18		2.16	0.25		2.19	0.22	
Recovery (%)		4.90			4.59	.59 2.76			
		4			5			Blank	
Items	рН	Си (ррт)	Са (ррт)	рН	Си (ррт)	Са (ррт)	рН	Си (ррт)	Ca (ppm)
Feeding solution	2.13	0.00	722.30	2.13	0.00	722.30	2.13	0.00	722.30
m (g)		4.0052			4.9958				
Adsorbed Cu (mg)		1.31			1.50				
			С	vernight	;				
24 h	2.35	3.96	792.45	2.37	3.96	594.99	2.16	0.00	937.95
Recovery (%)		45.27			39.73				
96 h	2.23	0.18		2.28	0.22		2.02	0.07	
Recovery (%)		2.08			2.19				

 Table B.19
 Desorption of Copper by 0.02M Citric Acid in Sea Water

#### Table B.20 Desorption of Copper by 0.03M Citric Acid in Sea Water

		1			2			3	
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ppm)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)
Feeding solution	2.14	0.17	739.15	2.14	0.17	739.15	2.14	0.17	739.15
m (g)		1.0062			2.0018			2.9988	
adsorbed Cu (mg)		1.50			1.70			2.01	
			С	vernight	;				
24 h	2.19	3.76		2.22	3.31		2.26	3.38	
Recovery (%)		37.73			29.28			25.24	
96 h	2.10	0.44	798.10	2.14	0.33	771.60	2.19	0.47	758.36
Recovery (%)		2.92			1.61		2.44		
		4		5				Blank	
Items	рН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	pН	Cu (ppm)	Ca (ppm)
Feeding solution	2.14	0.17	739.15	2.14	0.17	739.15	2.14	0.17	739.15
m (g)		4.0022			5.0020				
Adsorbed Cu (mg)		1.82			2.69				
			С	vernight					
24 h	2.30	3.62		2.33	3.73		2.08	0.00	
Recovery (%)		29.89			20.81				
96 h	2.23	0.25	725.24	2.30	0.36	708.68	2.03	0.15	771.60
Recovery (%)		0.90			1.22				

		<i>1</i>			2		3			
Items	pН	Си (ррт)	Ca (ppm)	рН	Си (ррт)	Ca (ppm)	pН	Си (ррт)	Ca (ppm)	
Feeding solution	2.07	0.00	815.83	2.07	0.00	815.83	2.13	0.00	815.83	
m (g)		1.0076			2.0051			3.0016		
adsorbed Cu (mg)		0.42			0.96			1.23		
			С	vernight	;					
24 h	2.08	1.90	670.34	2.14	3.46	506.65	2.22	3.93	659.94	
Recovery (%)		67.40			54.11			47.71		
96 h	1.82	0.11		1.88	0.22		1.92	0.33		
Recovery (%)		2.57			3.41			3.98		
		4		5				Blank		
Items	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	
Feeding solution	2.07	0.00	815.83	2.07	0.00	815.83	2.13	0.00	815.83	
m (g)		3.9978			5.0010					
Adsorbed Cu (mg)		1.35			1.64					
			С	vernight	;					
24 h	2.27	3.77	527.43	2.26	3.50	678.13	1.99	0.00	639.16	
Recovery (%)		41.78			31.97					
96 h	1.96	0.29		2.00	0.29		1.81	0.04		
Recovery (%)		3.22			2.66					

Table B.21 Desorption of Copper by 0.02M Tartaric Acid in Sea Water

#### Table B.22 Desorption of Copper by 0.02M Tartaric Acid

		1			2		3				
Items	рН	Си (ррт)	Ca (ppm)	рН	Cu (ppm)	Ca (ppm)	pН	Си (ррт)	Ca (ppm)		
Feeding solution	2.36	0.00	0.17	2.36	0.00	0.17	2.36	0.00	0.17		
m (g)		1.005			2.006			3.0016			
adsorbed Cu (mg)		0.40			0.68		1.17				
			C	vernight	ŧ						
24 h	2.57	1.55	0.08	2.62	2.06	0.11	2.65	3.07	0.14		
Recovery (%)		58.71			45.57			39.32			
96 h	2.42	0.11		2.47	0.18		2.50	0.22			
Recovery (%)		4.12			4.02			2.79	2.79		
		4			5			Blank			
Items	рН	Cu	Ca	рН	Cu	Ca	pН	Cu (ppm)	Ca (ppm)		
		(ppm)	(ppm)		(ppm)	(ppm)		(ppm)	(PPIII)		
Feeding solution	2.36	0.00	0.17	2.36	( <i>ppm)</i> 0.00	0.17	2.36	0.00	0.17		
m (g)	2.36			2.36			2.36				
	2.36	0.00		2.36	0.00		2.36				
m (g)	2.36	0.00 4.0025	0.17	2.36 Overnight	0.00 5.004 1.73		2.36				
m (g)	2.36	0.00 4.0025	0.17		0.00 5.004 1.73		2.36 2.50				
m (g) Adsorbed Cu (mg)		0.00 4.0025 1.30	0.17 C	overnight	0.00 5.004 1.73	0.17		0.00	0.17		
m (g) Adsorbed Cu (mg) 24 h		0.00 4.0025 1.30 2.45	0.17 C	overnight	0.00 5.004 1.73 t 2.60	0.17		0.00	0.17		

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		1			2		3				
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)		
Feeding solution	7.93	0.00	688.98	7.93	0.00	688.98	7.93	0.00	688.98		
m (g)	1.0002				1.9985			3.0024			
adsorbed Cd (mg)		0.15			0.11			0.32			
			С	vernight	t						
24 h	7.93	0.18		7.99	0.35		7.97	1.00			
Recovery (%)		16.04			42.76			45.25			
96 h	8.06	0.07		8.05	0.08		8.06	0.20			
Recovery (%)		6.43		10.01				9.08			
	4			5				Blank			
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)		
Feeding solution	7.93	0.00	688.98	7.93	0.00	688.98	7.93	0.00	688.98		
m (g)		3.9972			4.9998						
adsorbed Cd (mg)		0.24			0.45						
			C	vernight	t						
24 h	8.00	0.99		7.99	1.25		7.99	0.02			
Recovery (%)		61.07			41.06						
96 h	8.09	0.17		8.06	0.25		8.02	0.00			
recovery (%)		11.02			8.38						

Table B 23	Desorption	of Cadmium	by Sea Water
1 aoit D.25	Description	or cualifiant	oy bea mater

#### Table B.24 Desorption of Cadmium by 0.02M Citric Acid in Sea Water

		1			2		3			
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.13	0.00	770.35	2.13	0.00	770.35	2.13	0.00	770.35	
m (g)		1.0067			1.9998		3.0058			
adsorbed Cd (mg)		0.11			0.11			0.32		
			С	vernight	t					
24 h	2.29	0.25		2.32	0.48		2.36	0.83		
Recovery (%)		32.75			68.40			38.19		
96 h	2.33	0.07		2.34	0.08		2.36	0.10		
Recovery (%)		4.57			6.55			3.13		
	4				5			Blank		
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	pН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.13	0.00	770.35	2.13	0.00	770.35	2.13	0.00	770.35	
m (g)										
m (g)		4.0019			4.9969					
adsorbed Cd (mg)		4.0019			4.9969 0.38					
			C	vernight	0.38					
	2.40		С	overnight 2.43	0.38		2.24	0.00		
adsorbed Cd (mg)	2.40	0.21	C	-	0.38		2.24	0.00		
adsorbed Cd (mg) 24 h	2.40	0.21	C	-	0.38 1.32		2.24	0.00		

		<i>1</i>			2		3			
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Са (ррт)	рН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.09	0.00	656.43	2.09	0.00	656.43	2.09	0.00	656.43	
m (g)		1.0002			2.0052			2.9963		
adsorbed Cd (mg)		0.11			0.21			0.25		
			С	vernight	;					
24 h	2.12	0.18		2.17	0.75		2.22	0.78		
Recovery (%)		24.21			53.61			47.36		
96 h	2.06	0.03		2.08	0.06		2.11	0.06		
Recovery (%)		1.67			2.56			2.16		
	4			5				Blank		
Items	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	рН	Cd (ppm)	Ca (ppm)	
Feeding solution	2.09	0.00	656.43	2.09	0.00	656.43	2.09	0.00	656.43	
m (g)		3.9975			4.9955					
adsorbed Cd (mg)		0.32			0.34					
			С	vernight	;					
24 h	2.24	1.40		2.29	1.31		2.08	0.00		
Recovery (%)		60.59			51.70					
96 h	2.14	0.13		2.16	0.12		2.05	0.02		
recovery (%)		5.04			4.29					

Table B.25 Desorption of Cadmium by 0.02M Tartaric Acid in Sea Water

#### Table B.26 Desorption of Zinc by Sea Water

		1			2		3				
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)		
Feeding solution	7.93	0.08	688.98	7.93	0.08	688.98	7.93	0.08	688.98		
m (g)		1.0016			2.0041			3.0374			
adsorbed Zn (mg)		0.13			0.11			0.29			
			C	vernight	t						
24 h	7.96	0.04		7.97	0.04		7.96	0.07			
Recovery (%)		2.44			3.06			2.59			
96 h	8.00	0.03		8.05	0.04		8.04	0.04			
Recovery (%)		2.41			4.02			1.83			
	4			5				Blank			
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	pН	Zn (ppm)	Ca (ppm)		
Feeding solution	7.93	0.08	688.98	7.93	0.08	688.98	7.93	0.08	688.98		
m (g)		4.0146			4.9517						
Adsorbed Zn (mg)		0.24			0.44						
			C	vernight	ţ						
24 h	7.95	0.09		7.93	0.09		7.99	0.02			
Recovery (%)		4.07			2.42						
96 h	8.03	0.05		8.04	0.04		8.02	0.01			
Recovery (%)		2.68			1.20						

Tuer		Description		09 0.0			Sea II				
		1			2			3			
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)		
Feeding solution	2.13	0.02	770.35	2.13	0.02	770.35	2.13	0.02	770.35		
m (g)	0.9911				1.947			3.0151			
adsorbed Zn (mg)		0.13			0.28			0.37			
			С	vernight	t						
24 h	2.32	0.24		2.37	0.89		2.44	1.66			
Recovery (%)		25.91			46.30			66.32			
96 h	2.37	0.07		2.41	0.10		2.47	0.20			
Recovery (%)		4.13			3.40			6.61			
	4			5				Blank			
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)		
Feeding solution	2.13	0.02	770.35	2.13	0.02	770.35	2.13	0.02	770.35		
m (g)		4.0156			5.0093						
Adsorbed Zn (mg)		0.26			0.71						
			С	vernight	t						
24 h	2.43	1.46		2.49	2.86		7.99	0.02			
Recovery (%)		81.86	-		59.73	•					
96 h	2.46	0.15		2.57	0.34		2.19	0.04			
Recovery (%)		6.44			6.41						

 Table B.27 Desorption of Zinc by 0.02M Citric Acid in Sea Water

#### Table B.28 Desorption of Zinc by 0.02M Tartaric Acid in Sea Water

		1			2		3				
Items	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)	рН	Zn (ppm)	Ca (ppm)		
Feeding solution	2.09	0.03	656.43	2.09	0.03	656.43	2.09	0.03	656.43		
m (g)	1.0109				2.0029			2.9998			
adsorbed Zn (mg)		0.05			0.08			0.26			
			С	vernight	ţ						
24 h	2.14	0.32		2.19	0.70		2.3	1.50			
Recovery (%)		83.48			126.25			83.48			
96 h	2.09	0.05		2.11	0.09		2.14	0.18			
Recovery (%)		6.03			10.73		8.45				
	4				5			Blank			
Items	pН	Zn (ppm)	Са (ррт)	рН	Zn (ppm)	Са (ррт)	рН	Zn (ppm)	Ca (ppm)		
Feeding solution	2.09	0.03	656.43	2.09	0.03	656.43	2.09	0.03	656.43		
m (g)		3.9982			5.0013						
Adsorbed Zn (mg)		0.37			0.32						
			С	vernight	ţ						
24 h	2.30	1.97		2.34	2.37		2.08	0.03			
Recovery (%)		79.71			112.04						
96 h	2.15	0.25		2.18	0.30		2.05	0.03			
<i>)</i> 0 II	2.15	0.25		2.10	0.50						

		Table	B.29	Desorptio	on of n	netallic io	ons by	carbonic a	acia in	DI water		5
	Ite	M C		1* Conc.		Z Conc.		3 Conc.		4 Conc.		S Conc.
	110		рН	(ppm)	рН	(ppm)	pН	(ppm)	рН	(ppm)	рН	(ppm)
	-i -	Cd		0.00		0.00		0.00		0.00		0.00
Initial	Concen. (mg L <sup>-1</sup> )	Zn	6.18	0.00	6.18	0.00	6.18	0.00	6.18	0.00	6.18	0.00
II	Concen. (mg L <sup>-1</sup> )	Cu	0.10	0.00	0.10	0.00	0.10	0.00	0.10	0.00	0.10	0.00
	m		1	.0040	1	.9995	3	6.0009	Δ	.0023	5	.0000
ads		Cd (mg)		0.11		0.18		0.16		0.24		0.24
		Zn (mg)		0.17		0.35		0.20		0.42		0.65
ads	sorbed	Cu (mg)		1.27		1.81		2.72		2.78		2.92
		~ .				1 hou	r				-	
en.	$(mg L^{-1})$	Cd		0.01	-	0.05		0.10	•	0.23	_	0.35
onc	ng l	Zn	4.65	0.01	3.61	0.34	3.62	0.37	3.66	0.59	3.77	1.05
C	(I)	Cu		0.07		0.92		0.96		1.37		1.44
		Cd (mg)		0.00		0.01		0.02		0.03		0.05
		Zn (mg) Cu (mg)		0.00		0.05 0.14		0.06 0.14		0.09 0.21		0.16
ues	sorbed					4.27						
ery	, –	Cd		1.42				9.48		4.44		22.32
recovery	(%)	Zn		1.05	1	4.48	ź	28.34	2	21.28	2	24.37
re		Cu		0.81		7.64		5.30		7.39		7.39
						4 hou	rs				_	
ü.	.)	Cd		0.00		0.01		0.04		0.05		0.13
once	$(mg L^{-1})$	Zn	4.86	0.01	3.83	0.42	3.86	0.26	3.86	0.25	4.03	0.77
ŭ	(m	Cu		0.03		1.10		0.82		0.96		0.89
des	sorbed	Cd (mg)		0.00		0.00		0.01		0.01		0.02
		Zn (mg)		0.00		0.06		0.04		0.04		0.11
des	sorbed	Cu (mg)		0.01		0.16		0.12		0.14		0.13
ery	,	Cd		0.00		0.85		3.79		3.28		8.53
recovery	(%)	Zn		1.05	1	8.16		19.98		8.99	1	7.71
re		Cu		0.40		9.06		4.54		5.17		4.58
						24 hou	ırs	1				
зп.	()	Cd		0.00		0.03	_	0.08		0.14		0.20
once	$(mg L^{-1})$	Zn	4.90	0.01	4.19	0.54	4.20	0.31	4.3	0.39	4.53	0.99
		Cu		0.21		0.48		0.27		0.41		0.55
		Cd (mg)		0.00		0.00		0.01		0.02		0.03
		Zn (mg) Cu (mg)		0.00		0.08 0.07		0.05		0.06 0.06		0.15
		Cu (ing) Cd		0.00		2.56	1	7.58		9.19		.2.47
very	(%) (%)	Zn		0.53		2.30		23.69		3.82		2.81
recovery	( <sup>0</sup>	Cu		2.43		3.96		1.51		2.22		2.82
		very (%) -		1.42		7.68		20.85		26.91		3.32
Accumulated	reco	Cd very (%) - Zn		2.63		5.81		72.01		14.09		64.88
Acci	reco	very (%) - Cu		3.64	2	20.67		11.35	1	4.77	1	4.79

Table B.29 Desorption of metallic ions by carbonic acid in DI water

Note: \* -- control sample mixed with DI water without CO<sub>2</sub> inflow.

	Table	B.30	Desorptio	on of n	netallic io	ns by		acid in	sea water	r			
Iten	ns		1*		2		3**		4		5		
110		рН	Conc.	рН	Conc.	рН	Conc.)	рН	Conc.	рН	Conc.		
Initial Concen. (mg L <sup>-1</sup> )	Cd	-	0.01		0.01	-	0.01	+	0.01	-	0.01		
Initial Concen mg L <sup>-1</sup>	Zn	7.94	0.01	7.94	0.01	7.94	0.01	7.94	0.01	7.94	0.01		
	Cu		0.00		0.00		0.00		0.00		0.00		
m(g			.0033		.0000		/		.9976		.9983		
adsorbed adsorbed			0.11 0.22		0.13 0.45		/		0.16 0.38		0.18 0.49		
adsorbed			1.36		2.19		/		2.80		2.76		
	× 0/	1			1 hou	r							
n (	Cd		0.06		0.20		0.01		0.27		0.34		
Concen. (mg L <sup>-1</sup> )	Zn	7.85	0.02	4.64	0.51	4.61	0.11	4.63	0.50	4.63	0.70		
U C	Cu		0.14		2.09		0.00		2.26	-	2.70		
desorbed			0.01		0.03		0.00		0.04		0.05		
desorbed			0.00		0.08		0.01		0.07		0.10		
desorbed			0.02		0.31		0.00		0.34		0.41		
recover y (%)	Cd		7.11		21.94		/		23.70		27.31		
y ( <sup>5</sup>	Zn		0.42		6.60	/				18.90			21.06
	Cu		1.52	1	4.33		/		2.12		4.72		
	01		0.00		4 hou	rs	0.01		0.12		0.17		
L-1).	Cd		0.06		0.11		0.01	•	0.13		0.17		
Concen. (mg L <sup>-1</sup> )	Zn	7.88	0.03	4.72	0.26	4.69	0.05	4.71	0.29	4.72	0.42		
Ú C	Cu		0.14		0.38		0.00		0.48		0.68		
desorbed			0.01		0.02		0.00		0.02		0.02		
desorbed			0.00 0.02		0.04		0.01 0.00		0.04 0.07		0.06 0.10		
desorbed	Cu (mg) Cd		7.11		0.06 2.19		/		1.37		3.66		
ery							/						
recovery (%)	Zn		1.25		8.30		/	1	0.77	]	2.41		
re	Cu		1.52		2.58		/		2.57		3.73		
	1				24 hou	irs							
ij.	Cd		0.04		0.09		0.03		0.07		0.09		
Concen. (mg L <sup>-1</sup> )	Zn	7.16	0.04	4.80	0.23	4.73	0.22	4.80	0.42	4.82	0.95		
U C	Cu		0.41		0.65		0.00	Ì	0.10		0.89		
desorbed	Cd (mg)		0.00		0.01		0.00		0.01		0.01		
desorbed			0.00		0.03		0.03		0.06		0.14		
desorbed			0.06		0.10		0.00		0.02		0.13		
ery	Cd		4.27		9.75		/		5.69		6.83		
recovery (%)	Zn		1.66		7.29		/		6.03		28.77		
re	Cu		4.55		4.46		/		0.55		4.84		
ated (%)	Cd		18.49	4	3.88		/	۷	40.76	2	17.79		
Accumulated recovery (%)	Zn		3.32	3	32.19		/	2	45.71	6	52.24		
Acc recc	Cu		7.58	2	21.37		/	1	5.24	2	23.29		

\* -- Control sample mixed with DI water without CO<sub>2</sub> inflow. \*\* -- Blank test, no sample was placed in the flask. Note:

Table B.31 Desorption of metallic ions by 5% nitric acid										
Iter	ms	1	2	3	4	5				
uo	Cd	0.00	0.00	0.00	0.00	0.00				
ial trati L <sup>-1</sup> )	Zn	0.00	0.00	0.00	0.00	0.00				
Initial Concentration $(mg L^{-1})$	Cu	0.00	0.00	0.00	0.00	0.00				
C	Са	0.00	0.00	0.00	0.00	0.00				
m (	(g)	1.0044	2.0036	3.0055	4.0015	5.0016				
adsorbed		0.08	0.08	0.17	0.10	0.12				
adsorbed		0.11	0.15	1.07	0.34	0.14				
adsorbed	Cu (mg)	0.56	0.58	0.84	2.94	2.16				
			24 hours							
ц	Cd	0.13	0.16	0.51	0.54	0.56				
nal htratic L <sup>-1</sup> )	Zn	0.23	0.74	1.98	1.83	0.95				
Final Concentration (mg L <sup>-1</sup> )	Cu	1.81	4.16	7.84	13.03	8.87				
C	Ca	20.89	1.70	1.82	3.08	16.67				
desorbed	Cd (mg)	0.02	0.02	0.08	0.08	0.08				
desorbed	Zn (mg)	0.03	0.11	0.30	0.27	0.14				
desorbed	Cu (mg)	0.27	0.62	1.18	1.95	1.33				
	Cd	22.77	29.27	45.53	79.92	70.87				
recovery (%)	Zn	31.56	72.34	27.70	81.15	100.88				
re	Cu	48.37	108.00	140.22	66.44	61.66				

Table B.31 Desorption of metallic ions by 5% nitric acid

Table B.32 Desorption of metallic ions by 10% nitric acid

Ite		1 J	2	3	4	5
		-				
ion	Cd	0.00	0.00	0.00	0.00	0.00
tial htrati L <sup>-1</sup> )	Zn	0.00	0.00	0.00	0.00	0.00
Initial Concentration $(mg L^{-1})$	Cu	0.00	0.00	0.00	0.00	0.00
Co	Ca	0.00	0.00	0.00	0.00	0.00
m		1.0064	2.0020	3.0076	4.0036	4.9977
adsorbed	Cd (mg)	0.0506	0.0337	0.0674	0.1012	0.1012
adsorbed	Zn (mg)	0.1091	0.1091	0.1746	0.1746	0.2837
adsorbed	Cu (mg)	0.6525	1.2222	1.8066	2.6003	2.0939
			24 hours			
uc	Cd	0.09	0.17	0.36	0.43	0.54
nal ıtratio L <sup>-1</sup> )	Zn	0.16	0.34	0.66	0.86	1.34
Final Concentration $(mg L^{-1})$	Cu	2.58	4.42	7.84	10.38	8.47
CC	Ca	1.38	2.45	2.89	4.28	3.71
desorbed	Cd (mg)	0.01	0.03	0.05	0.06	.08
	Zn (mg)	0.02	0.05	0.10	0.13	0.20
desorbed	Cu (mg)	0.39	0.66	1.18	1.56	1.27
ry	Cd	27.09	77.29	79.32	63.67	79.96
recovery (%)	Zn	22.42	46.50	56.56	73.69	70.78
re	Cu	59.30	54.22	65.06	59.89	60.70

Appendix C

Column Tests Results

Items	Oakley Creek (Site 1)	Oakley Creek (Site 2)	Hobson St.
pH	7.14	6.89	7.00
Absorbance	0.011	1.021	5.177
$SS (mg kg^{-1})$	4.98	79.01	431.81
$Cd^{2+} (mg L^{-1})$	0.000	0.000	0.000
$Zn^{2+}$ (mg L <sup>-1</sup> )	0.053	0.154	0.386
$Cu^{2+} (mg L^{-1})$	0.000	0.032	0.067
$Cd^{2+} (mg L^{-1}) *$	0.001	0.002	0.000
$Zn^{2+} (mg L^{-1}) *$	0.161	0.314	0.882
$Cu^{2+} (mg L^{-1}) *$	0.004	0.042	0.134

Table C.1	Properties of the stormwater samples
-----------	--------------------------------------

Note: \* - Measurement of the ion concentrations by nitric acid digestion

Items	Coarse	Medium	Fine
Mass of slag (g)	5.9711	5.9958	5.9952
Height of column(cm)	22.2	19.6	19.2
Volume of column (mL)	4.15	3.66	3.59
Density of slag (g/mL)	1.44	1.64	1.67

Table C.2 Specifications of the columns

	-	Control	column	Coars	e slag	Mediu	m slag	Fine	slag
	Items	С	η	С	η	С	η	С	η
Mas	s of slag (g)	/		6.0	057	6.0	030	6.0064	
Ret	ention time	/	/	7'2	27"	10'	21"	10'	46"
	pН	7.96	7.86	7.87	8.03	0.147	/	0.070	/
1 <sup>st</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
1 11	Zn <sup>2+</sup>	0.036	33.33	0.036	33.33	0.012	77.78	0.012	77.78
	Cu <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
	pН	7.98	7.93	8.03	8.00	0.003	/	0.048	/
2 <sup>nd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
2 11	Zn <sup>2+</sup>	0.053	0.00	0.024	55.56	0.000	100.00	0.012	77.78
	Cu <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
	pН	8.11	8.09	8.09	8.13	0.155	/	0.128	/
3 <sup>rd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
3 n	Zn <sup>2+</sup>	0.030	44.44	0.018	66.67	0.006	88.89	0.012	77.78
	Cu <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
	pH	8.23	8.21	8.19	8.23	0.029	/	0.083	/
4 <sup>th</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
4 n	Zn <sup>2+</sup>	0.030	44.44	0.018	66.67	0.006	88.89	0.012	77.78
	Cu <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/

 Table C.3
 Removal of heavy metals from Oakley Creek (Site 1) stormwater sample

Note: C – Concentration of the ions (mg L<sup>-1</sup>)

	L	Control	column	Coars	e slag	Mediu	m slag	Fine	slag
	Items		η	С	η	С	η	С	η
Mas	s of slag (g)	,	/		050	6.0	073	5.9997	
Ret	ention time	/	/	11'	40"	11'	25"	11	08"
	pН	7.09	7.21	7.26	7.29				
1 <sup>st</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
1 n	Zn <sup>2+</sup>	0.107	30.77	0.042	73.08	0.036	76.92	0.024	84.62
	Cu <sup>2+</sup>	0.032	0.00	0.032	0.00	0.000	100.00	0.000	100.00
	pН	7.11	7.17	7.34	7.33	0.696	36.50	0.471	57.03
2 <sup>nd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
2 11	Zn <sup>2+</sup>	0.077	50.00	0.059	61.54	0.053	65.38	0.036	76.92
	Cu <sup>2+</sup>	0.032	79.21	0.032	0.00	0.000	100.00	0.000	100.00
	pН	7.26	7.31	7.42	7.38	0.482	54.70	0.459	56.86
3 <sup>rd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
5 11	Zn <sup>2+</sup>	0.077	50.00	0.047	69.23	0.042	73.08	0.036	76.92
	Cu <sup>2+</sup>	0.032	0.00	0.032	0.00	0.000	100.00	0.000	100.00
	pН	7.25	7.25	7.36	7.39				
4 <sup>th</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
4 11	Zn <sup>2+</sup>	0.083	46.15	0.059	61.54	0.042	73.08	0.036	76.92
	Cu <sup>2+</sup>	0.032	0.00	0.032	0.00	0.000	100.00	0.000	100.00

 Table C.4
 Removal of heavy metals from Oakley Creek (Site 2) stormwater sample

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	τ.	Control	column	Coars	se slag	Mediu	m slag	Fine	slag
Items		С	η	С	η	С	η	С	η
Mas	s of slag (g)	,	/		982	5.9	987	5.9272	
Ret	ention time	,	/	5'3	37"	3'2	27"	6'3	30"
	pН	7.09	7.18	7.18	7.15	0.666	38.22	0.512	52.50
1 <sup>st</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
In	Zn <sup>2+</sup>	0.125	19.23	0.119	23.08	0.077	50.00	0.053	65.38
	Cu <sup>2+</sup>	0.032	50.00	0.000	100.00	0.000	100.00	0.000	100.00
	pН	7.32	7.29	7.33	7.33	0.703	34.79	0.448	58.44
2 <sup>nd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
2 11	Zn <sup>2+</sup>	0.131	15.38	0.095	38.46	0.083	46.15	0.036	76.92
	Cu <sup>2+</sup>	0.000	100.00	0.000	100.00	0.000	100.00	0.000	100.00
	pН	7.38	7.39	7.43	7.41	0.566	47.50	0.498	53.80
3 <sup>rd</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
3 N	Zn <sup>2+</sup>	0.113	26.92	0.107	30.77	0.071	53.85	0.036	76.92
	Cu <sup>2+</sup>	0.032	50.00	0.000	100.00	0.000	100.00	0.000	100.00
	pН	7.42	7.38	7.44	7.37	0.518	51.95	0.553	48.70
4 <sup>th</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
4 n	Zn <sup>2+</sup>	0.119	23.08	0.101	34.62	0.077	50.00	0.042	73.08
	Cu <sup>2+</sup>	0.000	100.00	0.000	100.00	0.000	100.00	0.000	100.00

 Table C.5
 Removal of heavy metals from Oakley Creek (Site 2) stormwater sample

Note: C – Concentration of the ions (mg L<sup>-1</sup>)

	L	Control	column	Coars	e slag	Mediu	m slag	Fine	slag
	Items		η	С	η	С	η	С	η
Mas	s of slag (g)	/		6.0	033	6.0	065	6.0024	
Ret	ention time	21'	30"	20'	55"	20	'0"	20'	23"
	pН	7.11	7.18	7.27	7.41	2.208	47.95	1.671	60.61
1 <sup>st</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
1 n	Zn <sup>2+</sup>	0.346	10.29	0.250	35.29	0.136	64.71	0.091	76.47
	Cu <sup>2+</sup>	0.034	50.00	0.034	50.00	0.000	100.00	0.000	100.00
	pН	7.18	7.27	7.30	7.43	2.237	46.86	1.980	52.97
2 <sup>nd</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
2 11	Zn <sup>2+</sup>	0.341	10.29	0.250	35.29	0.170	55.88	0.131	66.18
	Cu <sup>2+</sup>	0.034	50.00	0.034	50.00	0.034	50.00	0.034	50.00
	pН	7.22	7.26	7.28	7.43	2.483	44.18	2.345	47.28
3 <sup>rd</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
3 N	Zn <sup>2+</sup>	0.352	10.29	0.244	36.76	0.199	48.53	0.136	64.71
	Cu <sup>2+</sup>	0.034	50.00	0.034	50.00	0.034	50.00	0.000	100.00
	pН	7.20	7.24	7.31	7.43	2.295	50.53	1.896	59.13
4 <sup>th</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
4 N	Zn <sup>2+</sup>	0.307	10.29	0.250	35.29	0.170	55.88	0.108	72.06
	Cu <sup>2+</sup>	0.034	50.00	0.034	50.00	0.000	100.00	0.000	100.00

Table C.6 Removal of heavy metals from Hobson St. stormwater sample

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	L	Control	column	Coars	se slag	Mediu	m slag	Fine	slag
	Items		η	С	η	С	η	С	η
Mas	s of slag (g)	/		6.0	031	6.0	044	6.0054	
Ret	ention time	12'	55"	7'4	43"	11'	12"	10'	48"
	pН	7.30	7.31	7.37	7.46	3.179	38.59	2.465	52.39
1 <sup>st</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
1 n	Zn <sup>2+</sup>	0.346	10.29	0.267	30.88	0.182	52.94	0.125	67.65
	Cu <sup>2+</sup>	0.067	0.00	0.067	0.00	0.067	0.00	0.034	50.00
	pН	7.24	7.28	7.38	7.41	2.899	43.25	3.614	29.25
2 <sup>nd</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
2 11	Zn <sup>2+</sup>	0.329	14.71	0.250	35.29	0.199	48.53	0.170	55.88
	Cu <sup>2+</sup>	0.034	50.00	0.034	50.00	0.000	100.00	0.000	100.00
	pН	7.27	7.29	7.34	7.39	3.228	32.30	3.868	18.88
3 <sup>rd</sup> h	Cd <sup>2+</sup>	0.000	/	0.000	/	0.000	/	0.000	/
3 N	Zn <sup>2+</sup>	0.318	17.65	0.273	29.41	0.221	42.65	0.210	45.59
	Cu <sup>2+</sup>	0.067	0.00	0.034	50.00	0.067	0.00	0.067	0.00
	pН	7.31	7.32	7.34	7.36	3.460	29.76	3.788	23.10
4 <sup>th</sup> h	$Cd^{2+}$	0.000	/	0.000	/	0.000	/	0.000	/
4 n	Zn <sup>2+</sup>	0.346	10.29	0.329	14.71	0.267	30.88	0.210	45.59
	Cu <sup>2+</sup>	0.034	50.00	0.067	0.00	0.000	100.00	0.034	50.00

Table C.7 Removal of heavy metals from Hobson St. stormwater sample

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	I.t	Control	column	Coars	se slag	Mediu	m slag	Fine	slag
	Items	С	η	С	η	С	η	С	η
Mas	s of slag (g)	,	/	6.0	6.0033		6.0037		007
Ret	ention time	/		11'	27"	11'	33"	11'	08"
	pH	6.	33	5.	91	5.	92	5.	99
	Absorbance	0.135	27.42	0.162	12.90	0.146	21.51	0.133	28.49
1 <sup>st</sup> h	Cd <sup>2+</sup>	0.063	20.00	0.056	30.00	0.032	60.00	0.016	80.00
	Zn <sup>2+</sup>	0.235	19.18	0.179	38.36	0.099	65.75	0.028	90.41
	Cu <sup>2+</sup>	0.143	0.00	0.041	71.43	0.041	71.43	0.020	85.71
	pН	6.	50	6.	31	6.	38	6.	52
	Absorbance	0.146	21.93	0.142	24.06	0.159	14.97	0.147	21.39
$2^{nd} h$	Cd <sup>2+</sup>	0.079	0.00	0.063	20.00	0.040	50.00	0.032	60.00
	Zn <sup>2+</sup>	0.247	15.07	0.203	30.14	0.115	60.27	0.036	87.67
	Cu <sup>2+</sup>	0.123	14.29	0.020	85.71	0.041	71.43	0.020	85.71
	pН	6.	37	6.42		6.	56	6.57	
	Absorbance	0.161	15.26	0.174	8.42	0.145	23.68	0.150	21.05
$3^{rd}h$	$Cd^{2+}$	0.079	0.00	0.063	20.00	0.040	50.00	0.024	70.00
	Zn <sup>2+</sup>	0.231	20.55	0.143	50.68	0.088	69.86	0.036	87.67
	Cu <sup>2+</sup>	0.123	14.29	0.020	85.71	0.020	85.71	0.000	100.00
	pН	6.4	44	6.	37	6.	48	6.	62
	Absorbance	0.178	7.29	0.162	15.63	0.146	23.96	0.169	11.98
$4^{\text{th}}h$	Cd <sup>2+</sup>	0.079	0.00	0.063	20.00	0.040	50.00	0.024	70.00
	Zn <sup>2+</sup>	0.239	17.81	0.183	36.99	0.119	58.90	0.056	80.82
	Cu <sup>2+</sup>	0.082	42.86	0.020	85.71	0.020	85.71	0.020	85.71

Table C.8 Removal of SS and heavy metals from synthetic stormwater sample 1

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	14	Control	column	Coars	se slag	Mediu	m slag	Fine	slag
	Items	С	η	С	η	С	η	С	η
Mas	s of slag (g)		/	6.0	6.0017		5.9994		031
Ret	ention time	/		10'	57"	11	20"	10'	49"
	pН	6.	31	5.	95	5.	84	5.	91
	Absorbance	0.280	16.91	0.284	15.73	0.258	23.44	0.211	37.39
1 <sup>st</sup> h	$Cd^{2+}$	0.061	0.000	0.053	12.500	0.030	50.000	0.030	50.000
	Zn <sup>2+</sup>	0.465	8.602	0.394	22.581	0.312	38.710	0.197	61.290
	Cu <sup>2+</sup>	0.041	50.000	0.020	75.000	0.020	75.000	0.020	75.000
	pН	6.	40	6.	13	6.	18	6.	32
	Absorbance	0.278	16.27	0.274	17.47	0.274	17.47	0.275	17.17
2 <sup>nd</sup> h	$Cd^{2+}$	0.061	0.000	0.053	12.500	0.053	12.500	0.030	50.000
	Zn <sup>2+</sup>	0.470	7.527	0.399	21.505	0.290	43.011	0.148	70.968
	Cu <sup>2+</sup>	0.061	25.000	0.020	75.000	0.020	75.000	0.020	75.000
	pН	6.	39	6.	22	6.	31	6.	43
	Absorbance	0.304	8.98	0.293	12.28	0.280	16.17	0.284	14.97
$3^{rd}h$	$Cd^{2+}$	0.061	0.000	0.061	0.000	0.053	12.500	0.038	37.500
	Zn <sup>2+</sup>	0.459	9.677	0.416	18.280	0.323	36.559	0.208	59.140
	Cu <sup>2+</sup>	0.061	25.000	0.041	50.000	0.020	75.000	0.020	75.000
	pН	6.	35	6.	23	6.	33	6.	45
	Absorbance	0.297	12.13	0.304	10.06	0.285	15.68	0.301	10.95
$4^{th} h$	Cd <sup>2+</sup>	0.061	0.000	0.061	0.000	0.053	12.500	0.030	50.000
	Zn <sup>2+</sup>	0.476	6.452	0.405	20.430	0.328	35.484	0.213	58.065
	Cu <sup>2+</sup>	0.082	0.000	0.041	50.000	0.020	75.000	0.020	75.000

Table C.9 Removal of SS and heavy metals from synthetic stormwater sample 2

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	Itana	Control	column	Coars	se slag	Mediu	m slag	Fine	slag
	Items	С	η	С	η	С	η	С	η
Mas	s of slag (g)		/	6.0	6.0002		5.9992		996
Ret	ention time	/		23'	27"	23'	53"	27'	33"
	pН	6.	26	5.	66	5.	97	6.	01
	Absorbance	0.548	30.46	0.516	34.52	0.422	46.45	0.249	68.40
1 <sup>st</sup> h	$Cd^{2+}$	0.068	18.18	0.046	45.45	0.038	54.55	0.008	90.91
	Zn <sup>2+</sup>	0.585	13.82	0.485	28.46	0.248	63.41	0.088	86.99
	Cu <sup>2+</sup>	0.188	0.00	0.094	50.00	0.075	60.00	0.019	90.00
	pН	6.	35	6.	05	6.	11	6.	31
	Absorbance	0.551	29.72	0.527	32.78	0.484	38.27	0.429	45.28
$2^{nd} h$	$Cd^{2+}$	0.061	27.27	0.038	54.55	0.015	81.82	0.008	90.91
	Zn <sup>2+</sup>	0.574	15.45	0.436	35.77	0.248	63.41	0.099	85.37
	Cu <sup>2+</sup>	0.169	10.00	0.075	60.00	0.056	70.00	0.038	80.00
	pН	6.	31	6.13		6.	24	6.	45
	Absorbance	0.540	31.73	0.528	33.25	0.513	35.15	0.458	42.10
$3^{rd}h$	Cd <sup>2+</sup>	0.061	27.27	0.038	54.55	0.023	72.73	0.008	90.91
	Zn <sup>2+</sup>	0.590	13.01	0.425	37.40	0.276	59.35	0.127	81.30
	Cu <sup>2+</sup>	0.188	0.00	0.094	50.00	0.075	60.00	0.075	60.00
	pН	6.	24	6.	14	6.27		6.50	
	Absorbance	0.561	29.26	0.553	30.26	0.538	32.16	0.501	36.82
$4^{th} h$	Cd <sup>2+</sup>	0.084	0.00	0.046	45.45	0.046	45.45	0.015	81.82
	Zn <sup>2+</sup>	0.623	8.13	0.474	30.08	0.303	55.28	0.121	82.11
	Cu <sup>2+</sup>	0.169	10.00	0.094	50.00	0.075	60.00	0.038	80.00

Table C.10 Removal of SS and heavy metals from synthetic stormwater sample 3

Note: C – Concentration of the ions (mg  $L^{-1}$ )

	Items	Control	column	Coars	e slag	Mediu	m slag	Fine slag	
	nems	С	η	С	η	С	η	С	η
Mas	s of slag (g)		/	6.0033		6.0065		6.0024	
Ret	ention time	/		11	'0"	10'	25"	9'3	35"
	pН	6.	34	5.	63	5.	64	5.	53
	Absorbance	0.641	14.42	0.577	22.96	0.513	31.51	0.473	36.85
$1^{st} h$	$Cd^{2+}$	0.073	10.00	0.033	60.00	0.008	90.00	0.008	90.00
	Zn <sup>2+</sup>	0.427	38.10	0.427	38.10	0.367	46.83	0.181	73.81
	Cu <sup>2+</sup>	0.018	75.00	0.018	75.00	0.018	75.00	0.000	100.00
	pН	6.	12	5.	86	6.	01	6.	09
	Absorbance	0.579	22.90	0.567	24.50	0.573	23.70	0.488	35.02
$2^{nd} h$	$Cd^{2+}$	0.073	10.00	0.024	70.00	0.016	80.00	0.008	90.00
	Zn <sup>2+</sup>	0.580	15.87	0.476	30.95	0.416	39.68	0.186	73.02
	Cu <sup>2+</sup>	0.018	75.00	0.018	75.00	0.018	75.00	0.018	75.00
	pН	6.	00	5.	99	6.	12	6.	23
	Absorbance	0.669	11.16	0.606	19.52	0.655	13.01	0.607	19.39
3 <sup>rd</sup> h	$Cd^{2+}$	0.081	0.00	0.041	50.00	0.041	50.00	0.008	90.00
	Zn <sup>2+</sup>	0.651	5.56	0.520	24.60	0.471	31.75	0.285	58.73
	Cu <sup>2+</sup>	0.070	0.00	0.018	75.00	0.018	75.00	0.018	75.00
	pН	6.	01	5.	94	6.	12	6.	26
	Absorbance	0.662	12.08	0.668	11.29	0.677	10.09	0.668	11.29
$4^{th} h$	$Cd^{2+}$	0.081	0.00	0.041	50.00	0.033	60.00	0.024	70.00
	Zn <sup>2+</sup>	0.668	3.17	0.542	21.43	0.504	26.98	0.367	46.83
	Cu <sup>2+</sup>	0.035	50.00	0.035	50.00	0.018	75.00	0.018	75.00

 Table C.11
 Removal of SS and heavy metals from synthetic stormwater sample 3

Note: C – Concentration of the ions (mg  $L^{-1}$ )

Items		Control column		Coarse slag		Medium slag		Fine slag	
		С	η	С	η	С	η	С	η
Mass of slag (g)		/		6.0059		5.9993		6.0022	
Retention time		/		4' 35″		4' 30″		4' 40″	
1 <sup>st</sup> h	pН	6.76		6.16		5.95		5.99	
	Absorbance	0.682	6.45	0.644	11.66	0.637	12.62	0.581	20.30
	$\mathrm{Cd}^{2+}$	0.078	9.09	0.063	27.27	0.063	27.27	0.039	54.55
	$Zn^{2+}$	0.609	4.27	0.587	7.69	0.489	23.08	0.380	40.17
	Cu <sup>2+</sup>	0.124	40.00	0.041	80.00	0.062	70.00	0.021	90.00
2 <sup>nd</sup> h	pН	6.72		6.43		6.48		6.49	
	Absorbance	0.678	7.63	0.652	11.17	0.681	7.22	0.656	10.63
	$\mathrm{Cd}^{2+}$	0.078	9.09	0.070	18.18	0.063	27.27	0.039	54.55
	Zn <sup>2+</sup>	0.587	7.69	0.560	11.97	0.489	23.08	0.380	40.17
	Cu <sup>2+</sup>	0.145	30.00	0.041	80.00	0.062	70.00	0.021	90.00
3 <sup>rd</sup> h	pН	6.59		6.45		6.53		6.57	
	Absorbance	0.691	5.86	0.672	8.45	0.692	5.72	0.676	7.90
	$\mathrm{Cd}^{2+}$	0.086	0.00	0.078	9.09	0.063	27.27	0.047	45.45
	$Zn^{2+}$	0.614	3.42	0.576	9.40	0.565	11.11	0.446	29.91
	Cu <sup>2+</sup>	0.165	20.00	0.083	60.00	0.083	60.00	0.062	70.00
4 <sup>th</sup> h	pН	6.49		6.37		6.47		6.56	
	Absorbance	0.691	6.62	0.731	1.22	0.703	5.00	0.676	8.65
	$\mathrm{Cd}^{2+}$	0.078	9.09	0.078	9.09	0.078	9.09	0.055	36.36
	$Zn^{2+}$	0.614	3.42	0.587	7.69	0.544	14.53	0.457	28.21
	Cu <sup>2+</sup>	0.165	20.00	0.124	40.00	0.103	50.00	0.062	70.00

 Table C.12
 Removal of SS and heavy metals from synthetic stormwater sample 3

C-11

Flow rate: 8.92 cm/min

Note: C – Concentration of the ions (mg  $L^{-1}$ )

## Appendix D

## Cost analysis of desorption tests

The price of the chemicals used as desorbing agent in the tests is shown in the following table.

Chemical	<i>Citric acid*</i>	Tartaric acid*	KCl*	<i>CO</i> <sub>2</sub> **	Sea Water
Price (\$/kg)	51.00	145.60	40.85	40.00	/

Table D.1 Price of the chemicals used in the desorption test

Note: \* -- Reagent grade

\*\* -- 30 kg CO<sub>2</sub> cylinder (Food Grade)

The costs for preparing 1 L of 0.02M citric and tartaric acid solution are \$0.21 and \$0.44 respectively. Considering the similar effect of citric and tartaric acids on the desorption of heavy metals from slag, citric acid with sea water is recommended because it is more cost-effective.