

Residues from waste incineration

Final report PSO-5784

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Publication date: 2009

Document Version Publisher's PDF, also known as Version of record

Link back to DTU Orbit

Citation (APA): Astrup, T., Pedersen, A. J., Hyks, J., & Frandsen, F. J. (2009). *Residues from waste incineration: Final report PSO-5784.* Department of Environmental Engineering, Technical University of Denmark (DTU).

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Residues from waste incineration

Final report PSO-5784 Revised version

April 2010

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i

Preface

This project was carried out by DTU Environment and DTU Chemical Engineering in close collaboration with the industrial partners: Amagerforbrænding, Vestforbrænding, AffaldVarme Århus, DONG Energy, FASAN, Babcock & Wilcox Vølund, ALSTOM, and AFATEK. The project was financed partly by the research grant PSO-5784 and partly by the involved industrial partners. The project period was 2005-2009.

The primary objective of the project was to improve the understanding of ash formation (fly ashes and bottom ashes) in waste incinerators and in this context to evaluate the importance of the waste input and the mode of operation.

The collaboration between the involved partners has been very fruitful. The joined forces of DTU Environment and DTU Chemical Engineering in cooperation with the numerous industrial partners has ensured a strong interdisciplinary project, covering all aspects of waste incineration, from the characteristics of the incoming waste, through the combustion characteristics and the ash formation processes in the furnace/boiler system, deposit formation and corrosion aspects, to the final emissions and the quality of the produced combustion residues. It is our conviction that this collaboration has been very profitable for the progresses of the project, and, in particular, for the robustness of our conclusions.

This Final Report is based on scientific contributions by: Thomas Astrup, Jiri Hyks and Christian Riber from DTU Environment; Anne Juul Pedersen, Flemming J. Frandsen, Jacob H. Zeuthen and Jørn Hansen from DTU Chemical Engineering; and Søren Nymann Thomsen from Babcock & Wilcox Vølund. The contributions are greatly acknowledged.

Based on the project results the following scientific journal papers have been prepared:

- (I) Astrup, T., Riber, C. & Pedersen, A.J. (2009): MSW Incinerator Emissions and Residues: Influence of Waste Input and Operational Conditions. *Manuscript*.
- (II) Pedersen, A.J., Frandsen, F.J., Riber, C., Astrup, T., Thomsen, N.S., Lundtorp, K. & Mortensen, L.F. (2009): A Full-Scale Study on the Partitioning of Trace Elements in Municipal Solid Waste Incineration Effects of Firing Different Waste Types. *Energy Fuels*, 2009, 23 (7), 3475–3489.
- (III) Zeuthen, J.H., Pedersen, A.J., Hansen, J., Frandsen, F., Livbjerg, H., Riber, C. & Astrup, T. (2007): Combustion aerosols from municipal waste incineration effect of fuel feedstock and plant operation. *Combustion Science and Technology*. **179**, 2171-2198.
- (IV) Hyks, J. & Astrup, T. (2009): Influence of operational conditions, waste input and ageing on leaching of contaminants from waste incineration bottom ashes. *Chemosphere*, **76**, 1178-1184.
- (V) Frandsen, F. J.; Pedersen A. J.; Hansen, J.; Madsen, O. H.; Lundtorp, K.; Mortensen, L. (2008): Deposit Formation in the FASAN WtE Boiler as a Function of Feedstock Composition and Boiler Operation. *Energy Fuels*, 2009, 23 (7), 3490–3496.

In the text these papers are referred to by their roman numerals. The papers are not included in this public report; however, they are available from the Technical University of Denmark, see note on page vii.

The authors wish to express our sincere gratitude towards all partners in the project as well as others involved for the interest in the project and the willingness to contribute with suggestions and comments. A special thank is forwarded to participants in the two project workshops for their detailed interest in our work, and to the people of FASAN for participating in the measuring campaign.

Last but not least, the authors wish to acknowledge the other members of the project group for their always committed involvement in the project: Uffe Juul Andersen (Amagerforbrænding), Hanne Rasmussen (AffaldVarme Aarhus), Kim Crillesen (Vestforbrænding), Kirsten Reinholdt Bojsen (Vestforbrænding), Kasper Lundtorp (Babcock & Wilcox Vølund), Bo Sander (DONG Energy), Jesper Staal (DONG Energy), Niels Ole Knudsen (Vattenfall), Jens Strandgaard (FASAN), Leif F. Mortensen (FASAN), Lars-Erik Johansson (ALSTOM), Stefan Ahman (ALSTOM), Dan H. Pedersen (ALSTOM), Jens Kallesøe (AFATEK), Ann Birgitte Sørensen (AFATEK). A special thanks to our former DTU colleague Christian Riber (now Rambøll) for his significant contributions to the project.

Summary

The primary objective of the project was to improve the understanding of the formation and characteristics of residues from waste incineration. The project had specific focus on the importance of the waste input and the mode of operation.

The main project work was carried out by DTU Environment and DTU Chemical Engineering in close collaboration with a number of industrial partners. The aim was to cover all aspects of waste incineration: from the characteristics of the incoming waste, through the combustion processes and ash formation in the furnace/boiler system, deposit formation and corrosion aspects, to the final air emissions and the leaching properties of the generated residues.

The project was focused around a full-scale measurement campaign at the FASAN incinerator in Næstved. Samples of bottom ashes and fly ash particles were collected and measurements of gases, aerosols and deposit build-up were carried out. During the campaign, a series of experiments were completed with i) changes in the waste input composition, and ii) changes in furnace operational parameters. Characterization of ash samples were subsequently carried out in the lab. Additional ash samples were collected from a range of Danish waste incinerators during the course of a year to establish a comparable set of ash data for Danish conditions. These samples were characterized with respect to solid composition and leaching properties as well.

In total 12 experiments were carried out at the FASAN incinerator: six with changes in waste input and six with changes in operational conditions, including one reference experiment. In experiments with changes in the waste input six waste material fractions were added one-by-one to a base-load waste defined as the normal waste received at the incinerator, excluding certain industrial waste types to minimize the possibilities for unnecessary variations in the input waste composition. The following waste materials were used (fraction per wet weight in parentheses):

- Road De-icing Salt (0.5 %)
- Shoes (1.6 %)
- Automobile Shredder Waste (14 %)
- Batteries (0.5 %)
- PVC (5.5 %)
- Chromate-Cupper-Arsenate Impregnated Wood (11 %)

During experiments involving changes waste input composition the incinerator was operated as close as possible to standard conditions. During experiments with changes in operational conditions, the incinerator was fed with the base-load waste only. Five experiments were carried out with operational conditions outside normal operation. The parameters changed were:

- Two experiments with changes in oxygen level: Minimum and Maximum O₂
- Two experiments with changes in air supply distribution: Decreased and Increased secondary air
- One reference experiment with standard furnace conditions

• One experiment in which the furnace showed unstable/undefined conditions and incomplete combustion on the grate as an example of a worst case scenario

Overall, it was found that the waste input composition significantly affected the characteristics of the generated residues. A similar correlation between operational conditions and residue characteristics could not be observed. Consequently, the project recommended that optimization of residue quality should focus on controlling the waste input composition. The project results showed that including specific waste materials (and thereby also excluding the same materials) may have significant effects on the residue composition, residue leaching, aerosol and deposit formation, as well as stack emissions. It is specifically recommended to minimize Cl in the input waste. Clean gas concentrations of As, Cd, Cr, Hg, and Sb appeared to be significantly affected by the input waste composition: Impregnated Wood (As, Cr), Batteries (Cd, Hg), and PVC (Sb). Based on the project results, it was found that a significant potential for optimization of emissions from waste incineration exists. Focus is recommended to be on the waste input composition.

Based on mass balance calculations, the following elements appeared enriched in the waste used in the individual experiments with changes in waste composition:

- Road De-icing Salt: Na, Cl
- Shoes: Cd, Cl
- Automobile Shredder Waste: Ba, Cu, Fe, Mn, Mo, Ni, Pb, Sb, Sn, V, Zn
- Batteries: Cd, Co, Hg, S, Sr
- PVC: Cd, Cl
- Chromate-Cupper-Arsenate Impregnated Wood: As, K, Pb, S, Si

In some cases, these enrichments were associated with higher concentrations in the solid residues (e.g. As, Cu, and Mo) while in other cases this was not reflected in the residues (e.g Cr, Ni, and Pb) indicating that other mechanisms were important also. Cl in the waste appeared to have a special role as this element affected several aspects:

- Organic bound Cl appeared released preferably to the gas phase while inorganic bound Cl was found primarily in the bottom ashes
- Cl increased formation of volatile (and more corrosive) Pb species while S had the opposite effect of forming more stable (and less corrosive) Pb species
- Cl appeared to increase the mass-load of aerosols if present as alkali-chloride
- Cl appeared to increase deposition fluxes

In general, no clear correlation between waste composition/operational conditions and the leaching properties could be observed. Only in the case of Cl leaching a significant relation to the solid content of the bottom ashes was found (only in experiments with changes in the waste input). Overall, less than 20 % of the solid Cl in the bottom ash samples appeared leachable. The observed variations in leaching data generally appeared within the ranges observed for typical bottom ashes collected from a range of different Danish plants (generally about one order of magnitude between high and low values for leachate concentrations).

The data provided by the project can form a solid basis for carrying out lifecycle assessments of waste incineration and provide input to authorities for evaluating the consequences of incineration of specific waste materials.

Contents

1.	Introduction Background Objectives Project outline The FASAN experiments Reference residue data Project dissemination Collaboration with other projects	1 2 2 4 6 7 8
2.	Mass flows Mass and energy balance calculations Waste composition Residue composition and element distribution	9 9 9 .11
3.	Flue gas/particle partitioning Overview Sampling positions and methodology Dust and aerosol loads Characterization of fly ash, aerosols and deposits Partitioning of volatile elements Overall link: Aerosols – fly ash – deposits Modeling of heterogeneous condensation	.17 .17 .17 .18 .19 .21 .23 .24
4.	Residue leaching Overview of leaching experiments Cl leaching Evolution of pH during ageing Leaching of metals Comparison with other leaching data Column leaching data	.27 .27 .28 .29 .30 .32
5.	Project conclusions Mass flows and element distribution Flue gas/particle partitioning and corrosion/deposition aspects Residue leaching	.35 .35 .36 .36
6.	Suggestions for further research	.39

Appendices

- (I) Astrup, T., Riber, C. & Pedersen, A.J. (2009): MSW Incinerator Emissions and Residues: Influence of Waste Input and Operational Conditions
- (II) Pedersen, A.J., Frandsen, F.J., Riber, C., Astrup, T., Thomsen, N.S., Lundtorp, K. & Mortensen, L.F. (2009): A Full-Scale Study on the Partitioning of Trace Elements in Municipal Solid Waste Incineration – Effects of Firing Different Waste Types
- (III) Zeuthen, J.H., Pedersen, A.J., Hansen, J., Frandsen, F., Livbjerg, H., Riber, C. & Astrup, T. (2007): Combustion aerosols from municipal waste incineration - effect of fuel feedstock and plant operation
- (IV) Hyks, J. & Astrup, T. (2009): Influence of operational conditions, waste input and ageing on leaching of contaminants from waste incineration bottom ashes
- (V) Frandsen, F.J.; Pedersen A.J.; Hansen, J.; Madsen, O.H.; Lundtorp, K.; Mortensen, L. (2008): Deposit Formation in the FASAN WtE Boiler as a Function of Feedstock Composition and Boiler Operation
- (VI) Operational conditions and flue gas data for the FASAN experiments
- (VII) Reside characteristics: solid composition and leaching
- (VIII) Modeling of heterogeneous condensation

Appendices (I) through (V) are not included in this public version of the report but printed versions of the papers may be freely obtained from the Technical University of Denmark: appendices (I) and (IV) from the library at Department of Environmental Engineering (library@env.dtu.dk), appendices (II), (III), and (V) from Department of Chemical Engineering (chec@kt.dtu.dk).

1. Introduction

Background

Municipal solid waste is receiving increased attention as a fuel for supplementing existing energy production in many countries and waste incineration is in some countries an important treatment option for the solid waste generated by households and industry. Modern waste incinerators offer a proven way to route waste away from landfills and at the same time facilitate recovery of the energy contained in the waste. The most important environmental concerns related to municipal solid waste incinerators are typically associated with emission of specific components to air (e.g. metals, acids, chlorides, and organic compounds), however a major part of the pollution potential is left in the residues, and contaminants released from these solid outputs after final disposal can be significant in a longer time perspective (Astrup et al. I). Today, operation of modern waste incinerators involves complying with strict regulatory limit values for air emissions while at the same time optimizing energy recovery and generating bottom ashes with characteristics useful for utilization as secondary construction materials. Optimal performance of an incinerator within this framework requires an in-depth understanding of the influences from the waste input as well as the furnace operation and combustion conditions.

Stable furnace operation greatly relies on good mixing of the incoming waste in the bunker and continuous feeding of the waste hopper, however waste composition may change over time and individual furnace operators may take different approaches to mixing and feeding. While waste received from households may vary relatively little over the week, waste received from commercial and industrial sectors may vary significantly during the course of a week: a specific plastic processing industry may for example deliver a truckload of waste plastic every Friday thereby significantly affecting the composition of the waste input during weekend. Individual operators may respond differently to such challenges applying different mixing of the waste in the bunker and consequently induce varying furnace conditions. These time-wise variations may then be addressed differently by furnace operators potentially leading to variations in air emissions and residue quality. In order to further optimize furnace operation and to increase robustness of furnace designs, an improved understanding of the responses to variations in waste input composition as well as furnace operational parameters are important.

Existing investigations of municipal solid waste incinerators have so far focused primarily on individual aspects of incinerator performance, e.g. residue characteristics, air emissions, or mass flows in the incinerator, but limited focus has been placed on the potential combined effects caused by variations in waste input or the significance of changes in operational parameters. Within recent years, significant efforts has been carried out in Denmark focusing on characterizing leaching from residues from waste incineration; however the focus has by far in most cases been placed on the residues themselves without linking back to the incinerator.

In order to fully assess environmental consequences from waste incineration (air emissions and residues), the influence from such variations should be quantified. If not we may fail to realize whether emission reductions are within reach or whether choices in waste sorting and management may have negative effects when treating waste thermally. The overall aim of this project is to improve the understanding of the link between the residues and the conditions within the incinerator.

Objectives

The primary objective of the project was to improve the understanding of ash formation (fly ashes and bottom ashes) in waste incinerators and in this context to evaluate the importance of the waste input and the mode of operation. Specifically this involved to:

- Investigate the importance of specific waste fractions for generation of solid residues
- Investigate the importance of operation conditions for generation of solid residues
- Improve the understanding of gas-particle partitioning in waste incinerators
- Evaluate the potential for optimization of plant operation with respect to improvement of residue quality
- Evaluate the consequences of sorting out or including specific waste fractions in the waste input with respect to solid residue generation
- Evaluate the potential for selective/alternative capture of fly ashes
- Evaluate the overall potential for optimization without affecting negatively the overall plant performance
- Establish a coherent and systematic dataset describing waste composition and residue characteristics from a range of Danish waste incinerators as well as variations in these data between plants and over time for a selected plant

Project outline

Overview

The project was focused around a full-scale measurement campaign at the FASAN incinerator in Næstved. Samples of bottom ashes and fly ash particles were collected and measurements of gases and aerosols were carried out. During the campaign, a series of experiments were completed with i) changes in the waste input composition, and ii) changes in furnace operational parameters. Characterization of ash samples were subsequently carried out in the lab. Additional ash samples were collected from a range of waste incinerators during the course of a year. These were characterized in the lab as well.

Experimental approach

A "base-load waste" was defined per weight as approximately 80 % municipal solid waste from households and approximately 20 % "combustible waste" from recycling stations. This waste was identical to the waste normally received at the incinerator but without the industrial waste fractions which could potentially cause unnecessary fluctuations in the input composition during the experiments. The base-load waste had an average lower heating value (LHV) around 9 GJ/ton.

During experiments involving changes in operational conditions, the incinerator was fed with base-load waste only. During experiments involving changes in waste input composition, known amounts of specific material fractions were added to the base-load waste while ensuring that the LHV of the combined waste was within the incinerators design range.

The FASAN incinerator

The experiments were carried out at the FASAN incinerator in Næstved. The incinerator furnace (waste capacity of 8 ton/hour at a LHV of 12 GJ/tonne, corresponding to a thermal boiler capacity of about 96 GJ/hour) was designed by Babcock & Wilcox Vølund and equipped with an air-cooled grate system. The furnace was commissioned in 2005. Flue gas cleaning was carried out by an ALSTOM NID (Novel Integrated Desulphurization) system with injection of hydrated lime and a subsequent bag filter system.

Measurements and sampling

Data and parameters collected during the measurement campaign at FASAN are outlined in Table 1. Gaseous emissions: CO_2 , CO, O_2 , NO_x , Cl, SO_2 , metals, dust and aerosols. Solid residues: major and minor inorganic elements and inorganic/organic carbon. Furnace operation: mass loaded to the incinerator, produced energy, generation of solid outputs. Log data from the furnace operational system were collected as 10 minutes average values (see Astrup et al. I for details).

Stack gases after flue gas cleaning (see Figure 1) were sampled using standard techniques for determination of stationary source emissions (gas extraction, filtering and wash bottles for metal capture). Major gas components (O_2 , CO_2 , CO, NO and SO_2) were continuously measured at the top of the furnace at the entrance to the superheater section (temperatures around 490°C-590°C) using gas extraction probes and standard gas analyzers. Fly ash particles were sampled isokinetic using a steel probe placed in three positions across the main flue gas duct after the boiler section (see Figure 1), please refer to Pedersen et al. (**II**) for details. Aerosols were measured by Low Pressure Cascade Impactor (LPI) and Scanning Mobility Particle Sizer (SMPS) to allow for characterization of aerosol mass, particle size distribution, and chemical composition. Please refer to Zeuthen et al. (**III**) for details on aerosols.

Furnace	State	Flow rate input, grate velocity, air distribution, temperatures				
Gas	State Major components Minor components	Temperature, pressure, flow rate CO ₂ , H ₂ O, O ₂ , SO ₂ , NO _x , HCl Dioxin, metals, salts, CO, HF, dust, aerosols				
Residues	State Major components Minor components	Temperature, flow rate, pH, conductivity H ₂ O, Cl, Fe, Al, Cu, TOC Other metals, salts, SO ₄				
Auxiliary materials	State Major components	Energy content, flow rates H ₂ O, Metals, salts				
Energy	State	Production rate				
Ambient	State	Temperature, humidity				

Table 1. Parameters monitored during the full scale experiments at FASAN.



Figure 1. Overview of the FASAN incinerator as well as gas and residue sampling points.

Special attention was addressed to sampling of bottom ashes as these materials are highly heterogeneous. In the project an automated sampler was constructed and used to frequently take small samples from a falling stream of bottom ashes between two conveyer belts. During the course of one experiment (about 6-8 hours) typically about 40-60 small samples were acquired and combined into a single, primary sample of about 200 kg. This primary sample was sieved to remove large metal and inert objects above 20 mm (corresponding to about 10-15 % by weight), and reduced in size by a riffle sample splitter in order to obtain a lab sample of around 20 kg.

Prior to sampling and measurements mapping of system residence times (e.g. from input of the waste to output of the bottom ashes) were carried out by introducing batches of iron pipes in the waste hopper. All sampling and measurements were subsequently planned to accommodate various residence times of fly ashes and bottom ashes, and address the same specific waste input in the individual experiments, see Astrup et al. (I) for details.

The FASAN experiments

In total 12 experiments were carried out: six with changes in waste input and six with changes in operational conditions, including one reference experiment.

Changes in waste input

In experiments with changes in the waste input (experiments W1-W6) six waste material fractions were added one-by-one to the base-load waste (fraction per wet weight in parentheses):

- Road De-icing Salt (0.5 %)
- Shoes (1.6 %)
- Automobile Shredder Waste (14 %)
- Batteries (0.5 %)
- PVC (5.5 %)
- Chromate-Cupper-Arsenate Impregnated Wood (11 %)

Table 2. Operational data collected during experiments (see Astrup et al. I and Pedersen et al. II for details). O_2 levels are average values downstream the boiler (% vol., wet). Temperatures are given at the bottom of the furnace (T1), top of the furnace (T2) and downstream the boiler (T3). The air flows are given on a normalized basis (to the average of all experiments)

0	Operational Conditions (OC) Experiments					Waste Input (WI) Experiments						
	Ref	Undef.	Min.	Max.	Dec.	Inc.	Salt	Shoes	Shredder	Batteries	PVC	Wood
-		cond.	02	02	Sec.	Sec.			waste			
O ₂ level (vol.%, wet)	7.3	8.3	7.2	9.1	7.9	7.5	7.6	8.0	7.5	7.9	7.9	7.8
T1 (° C)	1037	985	1018	991	1010	1011	1014	1011	1028	1017	1010	1018
T2 (° C)	865	928	912	890	912	1179	920	1124	919	963	1112	1024
T3 (° C)	175	173	176	184	175	175	175	177	175	176	175	176
Primary Air	0.90	1.08	1.00	1.15	1.14	0.78	1.03	0.92	1.01	1.14	0.91	0.93
Secondary Air	1.09	0.79	0.61	1.44	0.73	1.22	1.02	0.99	1.05	0.93	1.12	1.00
Total Air Flow	0.96	1.00	0.88	1.24	1.01	0.91	1.03	0.94	1.02	1.07	0.98	0.95
Air distribution	65/35	76/24	79/21	65/35	78/22	59/41	70/30	68/32	69/31	74/26	65/35	68/32

During these experiments, plant operational parameters were maintained as close as possible to standard values for the incinerator, and the LHV of the mixed waste was kept around 8-10 GJ/ton.

Prior to each experiment, a section of the waste bunker was prepared with alternating layers of base-load waste and the individual fraction used in the particular experiment, thereby allowing the crane grab to vertically cut through several layers at a time. This was done to ensure best possible mixing of the waste fractions with the base-load waste. Feeding was done at typical rates for the furnace (around 8-12 ton/hour). Between the experiments with changes in waste input, the furnace was fed with base-load waste only. The waste composition in these experiments is discussed in Chapter 2. Mass flows.

Changes in operational conditions

Six experiments were carried out with operational conditions outside normal operation (experiments OC1-OC6). The parameters changed were:

- Two experiments with changes in oxygen level: Minimum and Maximum O₂
- Two experiments with changes in air supply distribution: Decreased and Increased secondary air
- One reference experiment with standard furnace conditions ("Reference")
- One experiment in which the furnace showed unstable and incomplete combustion on the grate as an example of a worst case scenario ("Undefined conditions")

More reference experiments were planned but could not be completed within the timeframe of the experiments due to operational instabilities. During the OC experiments only base-load waste was fed to the incinerator. Table 2 outlines the experiments and experimental conditions.

Changes in operational conditions were done by changing appropriate "furnace set-points", i.e. sets of operational parameters that the control system and the operators seek to maintain. As it is not possible to completely fix furnace parameters, this approach allowed the operators to continuously make small adjustments to the furnace exactly as during normal operation. Other changes than oxygen level and air distribution were originally planned but it was difficult during the experiments to maintain steady-state conditions at these alternative set-points. Consequently, the project focused on these two parameters. In all cases, changes to the operational conditions were attempted "as large as possible" while still being possible from a practical point of view.

Solid samples collected

During all 12 experiments at FASAN the following solid residue samples were collected:

- Fly ash particles sampled from the flue gas duct
- Bottom ashes sampled from the ash conveyer: "fresh"
- Bottom ashes same as above but carbonated in the lab: "lab-scale aged"
- Bottom ashes aged for 3 months at AFATEK's facilities in Næstved: "naturally aged"

It was not possible to representatively sample fly ashes after flue gas cleaning at FASAN as large quantities of fly ashes were kept in the bag filter for recycling purposes in the ALSTOM NID system. Due to an estimated residence time in the filter of 1-2 days, it was not possible to relate samples collected here to specific waste on the grate. Consequently, it was decided to focus on samples of fly ash particles taken from the flue gas duct. As these samples were only about 5-20 g per experiment, only a limited number of analyses could be carried out on fly ashes (and leaching experiments could not be done).

Reference residue data

In addition to the residue samples collected during the FASAN experiments, a range of bottom ash and fly ash samples were collected from several Danish waste incinerators. Further, bottom ash data from regulatory compliance testing were supplied by a range of incinerators. The following samples and data were included in the project as a supplement to the FASAN experiments:

- One-year average fly ash and APC residue samples from: Amagerforbrænding, Nordforbrænding, KARA, Odense, Vestforbrænding, Århus (solid composition from Amagerforbrænding, leaching data for all)
- One-year average bottom ash samples from: Amagerforbrænding, Nordforbrænding, KARA, Vestforbrænding, Århus (solid composition, leaching data)
- Bottom ash samples over time (2006-2008): Amagerforbrænding and Århus (solid composition)
- Bottom ash leaching data over time (2006-2008) from regulatory compliance leaching tests: Amagerforbrænding, Nordforbrænding, KARA, Vestforbrænding, Århus, FASAN

Fly ash/APC residue and bottom ash samples were collected during the course of approximately one year for each incinerator within the period: 2006-2008. All samples consisted of a range of subsamples collected as grab samples from each

individual ash container or ash silo, e.g. collected during emptying of the silo or before transport of the container. All subsamples were then collected and mixed thoroughly. The mixed sample covering the entire year was finally transported to the lab and represented an "averaged one-year" sample for each incinerator. Due to limited possibilities for obtaining representative samples at some of the plants, not all of the planned ash samples could be obtained within the scope of the project.

The reference samples were characterized with respect to leaching in batch and column experiments and used for comparison with data from the FASAN experiments. The bottom ash leaching data covering a period of about two years as supplied by the incinerators were also used for comparison (see Hyks and Astrup **IV** for details). Additional ash samples collected over time for two Danish plants were characterized with respect to total content. These data as well as the leaching data were included in Appendix VII.

Project dissemination

Two project workshops were carried out within the project period: one workshop before the FASAN experiments (June 2006) and one workshop towards the end of the project (April 2009). At both workshops, external stakeholders and interested parties from Danish companies, organizations, and public entities were invited.

At the first workshop, plans for the full-scale experiments and subsequent lab activities were presented, and approaches within the project discussed. At the second workshop, all major results were presented and discussed with the invitees. Suggestions for potential follow up activities were also discussed at the second workshop (see Chapter 6. Suggestions for further research).

In addition to the more detailed discussions at the workshops, project results and experiences have been presented at a range of Danish meetings (e.g. DAKOFA, DAFONET) as well as in relation with other research projects. Project results have further been presented at a number of international conferences:

- Astrup, T. (2008): Full scale test incineration of special waste fractions: residue quality and environmental aspects. Proceedings Venice 2008, Second International Symposium on Energy from Biomass and Waste Venice, Italy; 17-20 November 2008
- Riber, C.; Astrup, T.; Christensen, T.H. (2007): Second generation methodology for characterization of solid waste fractions. Sardinia 2007, Eleventh International Waste Management and Landfill Symposium, 1-5 October, S. Margherita di Pula (Cagliari), Italy.
- Pedersen, A.J.; Frandsen, F.; Riber, C.; Astrup, T.; Thomsen, S.N.; Lundtorp, K.; Mortensen, L.F.: A Full-Scale Study on the Partitioning of Trace Elements in Municipal Solid Waste Incineration – Effects of Firing Different Waste Types. 'IMPACTS OF FUEL QUALITY ON POWER PRODUCTION & ENVIRONMENT', September 29th – October 3rd, 2008, Banff, Canada.
- Frandsen, F.J.; Pedersen A.J.; Hansen, J.; Madsen, O.H.; Lundtorp, K.; Mortensen, L. (2008): Deposit Formation in the FASAN WtE Boiler as a Function of Feedstock Composition and Boiler Operation. 'IMPACTS OF FUEL QUALITY ON POWER PRODUCTION & ENVIRONMENT', September 29th – October 3rd, 2008, Banff, Canada.

- Pedersen, A.J.; Frandsen, F.J.; Astrup, T.; Riber, C.; van Lith, S.; Steinsen, S.D.; Holgersen, L.B. (2008): Partitioning of Trace Elements in a WtE boiler; The Influence of Different Waste Types. In: Proceedings from Twenty-Seventh Annual International Conference on Thermal Treatment Technologies, May 12-16, 2008, Montreal Quebec, Canada, Paper #35
- Pedersen, A.J.; Frandsen, F.J (2008): Partitioning of trace elements in a WtE boiler and the effects of different waste types. In: Proceedings from 235th American Chemical Society National Meeting and Exposition, April 6-10, 2008, New Orleans, Louisiana, USA.
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Collaboration with other projects

During the project period, the project partners have utilized the experiences gained in this project in relation to other Danish projects. In a number of cases, the project has collaborated with other projects on specific aspects related to the performed experiments. Examples of such collaboration are given below:

- PSO-7170: fly ash samples from the FASAN experiments were made available for this project for determination of dioxin levels. Further, log data from the FASAN experiments were supplied in relation to a dioxin formation model developed in this project. At the time of writing no results were available from these activities.
- CHEC consortia (ITOB CHEC Industrial Consortium on Thermal Conversion of Fuels): this project installed deposit probes in the boiler section during the FASAN experiments (Reported in Frandsen et al. V). The CHEC consortia also conducted laboratory scale release experiments on waste fractions from FASAN.
- Several projects with Danish waste incinerators on residue quality and leaching: experiences gained in relation to the FASAN experiments have been used for designing and evaluating experiments in other projects, and results from PSO-5784 have been used as a reference for comparison in several other projects.

2. Mass flows

Mass and energy balance calculations

The composition of the waste input cannot be determined directly in a simple manner (e.g. digestion of solid samples) due to the heterogeneity of mixed waste; however, the composition may be determined indirectly based on mass and water balance calculations combining the outputs from the incinerator to obtain the input (the waste) assuming mass conservation in the system. This approach has been discussed by several authors and has been applied in several cases on waste incinerators (Astrup et al. **I**).

In this project the system boundaries for the mass balances included the waste input (waste hopper) to the sampling of fly ash particles and bottom ashes (see Figure 1): outputs were fly ashes (including aerosols), flue gas, and bottom ashes, while the input was the waste. In the following overall substance balance (BA: bottom ash, FA: fly ash), M denotes masses while C denotes element concentrations:

$$M_{waste} \times C_{waste} = M_{BA} \times C_{BA} + M_{FA} \times C_{FA} + M_{flue gas} \times C_{flue gas}$$
(1)

Masses were recalculated into dry basis to account for variations in water content of the input waste. This was done based on a water balance for the furnace system using data for moisture in the gas phase and water flows in the plant; only the water content in the waste input was estimated. This was done by assuming all H in the waste was oxidized to H_2O in the furnace (Astrup et al. I).

Waste composition

The composition of the base-load waste was determined using equation (1) as an average of the waste compositions during experiments with changed operational conditions (OC1-OC6), see Table 3. As the mixing and preparation of the base-load waste was kept as stable as possible during experiments, the variations observed in waste composition during the operational experiments may be regarded as "typical" variations for this incinerator during normal operation. Worth noticing is the observed standard deviation of the composition data: for many elements this is one order of magnitude below the average value. This is rather accurate considering the relatively large heterogeneity of solid waste.

To assess whether the addition of special waste fractions was reflected by the overall waste compositions in experiments with changed waste input (W1-W6), we used the criteria that calculated element concentrations in the individual waste fraction should be higher than the mean value plus two times the standard deviation of the base-load waste (OC experiments). This approach allowed us to identify those elements which were most likely present in high concentrations in the waste fractions compared with the base-load waste. Table 3 shows the affected elements for each experiment, including the calculated element concentrations.

g/i	tonne of waste.	•								
Base-load waste			Waste fractions							
	Average	Standard Dev.	Salt	Shoes	Shredder	Batteries	PVC	Wood		
As	11	2.0						400		
Ba	480	320			6,200					
Cd	6.3	2.2		250		480	120			
Cl	15000	1400	1,100,000	230,000			410,000			
Co	13	8.1				2,200				
Cu	2600	1900			26,000					
Fe	22000	11000			170,000					
Hg	0.33	0.14				240				
ĸ	4300	510						9,800		
Mn	330	100			1,600					
Mo	5.1	3.1			48					
Na	11000	1600	230,000							
Ni	110	76			1,100					
Pb	570	200			2,900			6,900		
S	3500	760				120,000		34,000		
Sb	45	23			260					
Si	62000	7200						140,000		
Sn	48	33			640					
Sr	130	33				5,200				
V	14	4.4			60					
Zn	1600	450			19,000					

Table 3. Chemical composition of the base-load waste and the individual waste fractions (see Astrup et al. I for details). Data for the individual fractions are per weight of the fractions. All data are in g/tonne of waste.

Salt was as expected contributing with considerable quantities of Na and Cl. The calculated Cl content appeared to be slightly above 1 kg/kg clearly indicating that these data should be used with caution, especially in the cases Salt and Batteries. These two waste fractions were added in particular low concentrations (0.5 % w/w) relative to the base-load waste thereby indicating that a higher uncertainty would be associated with the calculated waste composition (i.e., the lower input level of the specific waste fraction, the lower the accuracy of the indirectly calculated chemical composition). Similar considerations should be addressed elements present only in small concentrations likely associated with specific waste materials: the most important example is Hg in batteries which is likely also affected by the relatively low input level (0.5 %). In conclusion, although the waste compositions listed in Table 3 are considered state-of-the-art the data should not be used un-critically.

Our primary focus was to identify which elements appeared to be enriched in the individual waste fractions. By evaluating differences in calculated element concentrations between the individual experiments, it can be seen that the waste fractions likely contributed with specific elements. Shoes were adding Cd and Cl to the waste mix. Shredder Waste contributed with a range of metals likely originating from small pieces of metals in the waste fraction. Batteries appeared to contribute with high amounts of Cd, Co, Hg, S, and Sr relative to the base-load waste. PVC Waste contributed with Cd and Cl, while Impregnated Wood added As, K, Pb, S, and Si. In conclusion, it appeared that the experiments with different waste fractions significantly altered the waste input composition of both heavy metals, organic and inorganic Cl.

Residue composition and element distribution

Evaluating the consequences for the solid residues, it is necessary to account for the concentration levels in the actual ashes as well as the amount of an element routed to the individual ashes. In Figure 2 and 3, element concentrations in bottom ashes and fly ashes are shown as a function of the fraction of the input found in either of the ashes. Relating these data to the amount of an element fed in with the waste can provide information about to which extent the solid residues were affected by the changes induced in the experiments. In the figures, error bars indicate two times the standard deviation around the mean of the OC experiments. Based on this, it was assessed whether the variations were likely caused by fluctuations in operational conditions rather than changes in waste input (Astrup et al. I).

As was the case for the chemical compositions of the individual waste fractions a certain uncertainty may be associated with the results. However, the compositional data for the residues produced during the experiments (y-axis in Figure 2 and 3) are significantly more accurate as these data are determined by actual measurements on the ashes themselves. The fraction of the input distributed in the bottom or fly ashes is on the other hand based on the previously mentioned mass balance calculations and are therefore less accurate. The use of error bars as described above in the interpretation of the results acknowledges this inherent uncertainty.



Figure 2. Concentrations of As, Cr, Cu, Mo, Ni, and Pb in bottom ashes and fly ashes as a function of the fraction of the elements in the individual ash samples. Error bars indicate two times the standard deviation of samples from the operational experiments (see Astrup et al. I for details).

As, Cr, Cu, Mo, Ni, and Pb

Generally, most of As, Cr, Cu, Mo, and Ni was found in the bottom ashes. Generally, the variations observed in solid contents due to changes in waste input appeared to be "covered" by the variations observed in the OC experiments. This indicated that the effects from changes in waste input were less or equal to what can be expected from changes in operational conditions.

More than half of As was found in the bottom ashes however in the Impregnated Wood experiment only 20 % of As ended here. Data for the OC experiments generally covered the variations observed for As concentrations during experiments with waste input changes, except in case of Impregnated Wood where both bottom ashes and fly ashes had considerable higher concentrations. More than 80 % of Cr was found in the bottom ashes. The operational experiments generally "covered" the variations in ash concentrations in the waste experiments indicating that Impregnated Wood did not have a significant effect on Cr although this was the case for As.

Cu and Mo were mostly found in the bottom ashes and the observed variations appeared to be covered by variations in the OC experiments; however, combustion of Shredder Waste resulted in significantly higher ash concentrations for both elements, and also higher Mo fraction in the fly ashes. This corresponded with high input levels of these elements in Shredder Waste. Ni was almost entirely found in bottom ashes with elevated concentrations in the Shredder Waste experiment, again corresponding well with high input levels and the possibility of metallic Ni in Shredder waste being likely to stay in the bottom ash. The fly ashes also showed increased Ni concentrations in the Shoes and Batteries experiments; however, this could not be related to significantly higher input levels as determined for these experiments, although Ni is a known constituent of batteries. In the Salt experiment, significantly lower concentrations were observed in the fly ashes although this could not be correlated with a lower input of Ni.

For Pb a more diverse pattern was observed. Although Pb mostly appeared to be favored in the bottom ashes a few experiments showed higher fractions in the fly ashes (increasing order: Salt, Shoes, PVC). The experiments indicated that (organic bound) Cl had a mobilizing effect on Pb (Pedersen et al. **II**). This is further discussed in Chapter 3.

Relating the above to waste composition data in Table 3, it is seen that while increased concentrations of As, Cu, and Mo coincided with waste fractions having high contents of the same elements, increased concentrations of Cr, Ni, and Pb could not be fully explained with higher input levels. This indicated that the waste mix was important and that other constituents and mechanisms controlled ash concentrations such as the case with Cl affecting Pb concentrations.



Figure 3. Concentrations of Cd, Cl, Hg, S, Sb, and Zn in bottom ashes and fly ashes as a function of the fraction of the elements in the individual ash samples. Error bars indicate two times the standard deviation of samples from the operational experiments (see Astrup et al. I for details).

Cd, Cl, Hg, S, Sb, and Zn

Cd was mostly found in the fly ashes. The concentrations were generally covered by the variations observed in the OC experiments; however the fly ashes had higher concentrations in the PVC and Shoes experiments and the bottom ashes higher concentrations in the Batteries experiment. This corresponded well with Cd also being present in high levels in PVC, Shoes, and Batteries. Zn concentrations were within the variations found in the OC experiments, except for Shredder waste where both ashes had significantly higher concentrations. This corresponded with Shredder waste being significantly enriched in Zn compared with the base-load waste. Sb concentrations were also within variations in the OC experiments except for the Shredder waste which was enriched with Sb.

Hg was almost entirely found in the fly ashes and the gas phase with relatively small variations (generally covered by the operational experiments); however feeding Batteries caused significantly higher Hg concentrations in the fly ashes/gas phase (Astrup et al. I). This behavior corresponded well with Batteries contributing with high levels of Hg indicating that batteries may be a significant Hg source in combustible waste.

High Cl concentrations in the Salt, PVC, and Shoe experiments (declining order) were observed in the bottom ashes, and high Cl concentrations in the Salt and Shoes experiments (declining, PVC third highest) were observed for fly ashes. This corresponded with increased Cl levels found in these waste fractions (see also Chapter 3, Flue gas/particle partitioning).

S was found primarily in the fly ashes and the gas phase (> 50 %). S concentrations were generally covered by the OC experiments; however Salt and Impregnated Wood caused higher concentrations in the fly ashes, and Batteries and Wood caused higher concentrations in the bottom ashes.

Stack emissions

To evaluate how the flue gas cleaning system reacted to the changes in waste input and operational conditions, standard measurements were carried out on the clean gas prior to the stack. Selected clean gas concentrations (As, Cd, Cr, Hg, Pb, and Sb) are discussed below.

Generally, the gas cleaning system maintained low As emissions but when combusting Impregnated Wood the emissions increased significantly which corresponded with the higher As load in this experiment. Cd emissions appeared to remain below 7 mg/tonne waste, except in the Undefined Conditions experiment and when incinerating Batteries (about 5 times the average of the others). The higher emission in case of Batteries corresponded with Cd being enriched in the Batteries fraction. Batteries also resulted in significantly increased emissions of Hg, likely due to the presence of Hg in coin cell batteries. Hg is generally very volatile and appeared in the Battery experiment to be almost entirely found in the gas phase without being captured in the flue gas cleaning system. Cr showed slightly increased concentrations for Impregnated Wood (about double the average of the others). Although this did not appear significant in the mass balance calculations, Cr should be present in the Impregnated Wood due to the use as fixation agent in the treatment process. In case of PVC, Sb appeared significantly increased in accordance with the potential presence of Sb as stabilizer in PVC (although this was not apparent from the mass balance calculations).

Pb was an example of an element not showing increased emissions despite increased load in the input waste, e.g. the Wood experiment. Emissions of Cu, Ni, and Zn were also not clearly correlated with input levels. Pb emissions for Undefined Conditions were generally found in the high end of the range (in case of Pb the highest value). This may indicate that poor combustion on the grate may lead to increased air emissions.

Overall, the stack emissions appeared to be strongly affected by the waste input composition. Although the flue gas cleaning system could effectively maintain low emission levels over a wide range of operational conditions, the potential influence from the waste input could not be ignored and extreme input levels resulted in increased air emissions. In particular, the results indicated that contents of PVC, impregnated wood, and batteries in the waste input should be limited.

Implications for residue quality

As indicated in the previous sections, it was not possible to significantly determine correlations between specific changes in operational conditions and residue quality. This does not necessarily indicate that such correlations do not exist; it merely indicates that it was not possible to meaningfully ascribe the observed variations in residue quality to the specific operational changes implemented during the full-scale experiments carried out at FASAN. A main reason for this is likely the complex and dynamic properties of a full-scale furnace which makes it difficult to control the combustion conditions throughout the waste bed and ensure that the operational changes have the anticipated effects. The operational changes implemented at FASAN were, however, done as "large as possible" within the capabilities of the furnace, i.e. without significantly hindering combustion on the grate. The operation during these experiments was significantly "outside" normal operation. This means that for the specific furnace, it is unlikely that operation at normal conditions will be "outside" what was tested during the experiments. It was therefore concluded that the operational changes included in the project would most likely bracket normal operation at FASAN, and that the residue qualities observed during these experiments therefore represented a range which would cover most situations.

The fact that significantly changed operational conditions did not result in discernable effects on the residue quality while significant changes in waste input appeared to have much more dramatic effects on the residues is an important observation. This indicate that the residue quality may be relatively robust with respect to furnace operation, or in other words that the changes in furnace set-points do not have such a dramatic effect on the residues.

In the project, the ranges in residue compositions observed for the OC experiments were in accordance with the above used as a basis for comparison with the experiments with changes in waste input (i.e. Figure 2 and 3). As the changes in waste input generally induced considerably larger changes in the residue quality than changes in operational conditions, it was concluded that the potential for optimizing residue quality were also larger when based on changes in the waste input. Consequently, it was concluded that focus should be put on the waste input composition rather than attempting to change furnace operation.

Combining the results for residue quality with the measured stack gas emissions, it appear that in particular the quantities of PVC, impregnated wood, shredder waste and batteries were important for the emission outputs from the incinerator. It is therefore likely that overall emissions can be reduced if these waste materials are excluded from the input to the incinerator. As these materials (in principle) are easily identifiable, a considerable potential for improvement of the environmental performance of waste incinerators may exist if the materials can be properly separated from the mixed waste input.

Implications for environmental assessments

The full-scale experiments at FASAN clearly showed that both air emissions and residue qualities were significantly affected by changes in waste input composition but also to some extent changes in operational conditions, in particular in case of the residues. The experiments, however, also documented that modern flue gas cleaning systems may cope with substantial changes in waste input and still maintain relatively stable emissions levels for heavy metals. Although no clear correlation with changes in operational conditions were apparent, even worst-case conditions (in-complete burnout) did not appear to significantly affect emissions and element flows.

The presented results are important because authority approvals for incineration of specific waste materials generally assume business-as-usual performance of the incinerators. The experiments illustrate that specific waste materials may dramatically affect air emissions and change element flows. These aspects are important for environmental performance evaluations, e.g. life-cycle assessments. Traditionally such assessments have been based on fixed sets of emissions and element flows irrespectively of the waste input. It is suggested that more accurate results can be obtained when changes in air emissions and residue composition are taken into account. It is recommended that results of environmental assessments involving waste incineration should be thoroughly tested with respect to the variations in output data reported in this paper.

3. Flue gas/particle partitioning

Overview

The main objectives relative to the flue gas/particle phase partitioning study was to improve the understanding of the formation of fly ashes in waste-fired boilers. In this context, the experimental data from FASAN on particle and gas phase compositions (metals, salts) in the boiler/flue gas duct system were analyzed thoroughly. Special focus was on the elements: Na, K, S, Cl, Zn and Pb, corresponding to the six major elements by weight found in the aerosols, and the elements considered most important with respect to corrosion and deposition aspects in waste fired boilers. Besides being highly volatile, chlorides of the alkali metals Na and K, as well as the heavy metals Zn and Pb, are the species thought to be the most chemically aggressive and the ones primarily responsible for corrosion. The heavy metals facilitate corrosive reactions by forming low melting-point mixtures (eutectics) with alkali chlorides in the deposits on metal surfaces.

In addition to the evaluation of the experimental data, a model aiming to predict the gas/particle phase partitioning in the flue gas channel as a function of time and temperature was developed: A physical model for heterogeneous condensation (i.e. surface condensation on existing particles) of trace elements on a size distributed fly ash was developed and coupled with an existing global equilibrium analysis model.

Sampling positions and methodology

During the different full-scale experiments at FASAN, samples of flue gas, fly ash, aerosols and deposits were taken at three different locations in the boiler, as indicated in Figure 1. The results from the flue gas measurements are reported in details in Zeuthen et al. (III) (aerosol measurements), Pedersen et al. (II) (fly ash measurements and over-all flue gas/particle partitioning) and Frandsen et al. (V) (deposit measurements).

The major flue gas components (CO, CO₂, O₂, SO₂, NO) were monitored continuously at a sampling location in the top of the furnace (see Figure 1) using a gas extraction probe and standard gas analyzers. The initial deposit formation was also measured at this location. Air-cooled probes were applied to collect deposits on and inside the boiler. In the standard set-up, the deposition probe was operated with a surface temperature of 400 $^{\circ}$ C, in order to simulate the current steam temperature in the final superheater of the FASAN plant.

Fly ash (total dust) and aerosols were sampled in the flue gas duct downstream of the boiler, but upstream of the flue gas cleaning (NID) reactor and bag filter (see Figure 1). A probe for sampling of fly ash (total dust) from the hot flue gas by isokinetic extraction was developed at CHEC for the present campaign. The sampling of fly ash served two main purposes: (1) to find the total dust load during the different experiments, and (2) to collect samples for further analyses. Approximately 20 g of particulate material was collected from each experiment (minimum fly ash sampling time 1.6 h per experiment).

The measurements of mass-based size distributions of aerosols were made using a 10-stage Berner Low-Pressure Cascade Impactor (LPI), and number-based size distributions were obtained using a Scanning Mobility Particle Sizer (SMPS). Both aerosol measurement methods (LPI and SMPS) give information on the particle size distribution of the aerosols (mass based and number based, respectively), and the LPI method also allows for chemical analysis of the deposited aerosols by means of e.g. SEM/EDX. Additional SMPS measurements were made on the cleaned flue gas in the stack.

Dust and aerosol loads

The total dust load, as found by the fly ash sampling method, was varying between 1.8 and 2.8 g/m³ (std. T,P) in most experiments, with an average of 2.2 g/m³ (std. T,P) and no apparent correlations to the operational conditions or waste input composition (Pedersen et al. **II**). The particle size distribution of the total dust samples, as measured by laser diffraction, revealed a bimodal size mode with a peak below 5-10 μ m (fine mode) and a second peak at app. 100 μ m (coarse mode). The particle size distribution tended to change towards finer particles when firing NaCl salt, indicating increased (homogeneous) condensation of alkali salts in this experiment (Pedersen et al. **II**).

The mass-based aerosol size distribution revealed a peak around 0.4 μ m (see Figure 4). The aerosol mass-load was very stable when no changes in fuel composition were introduced (PM_{2.5}: 252 ± 21 mg/m³) and increased significantly during combustion of waste when including automotive shredder waste, CCA-impregnated wood, and NaCl (PM_{2.5}: 313, 320, and 431 mg/m³). The number concentrations of particles varied during combustion and between varying operational conditions and waste mixes (from 43·10⁶ to 87·10⁶ #/cm³). The mass of the PM_{2.5} fractions (252 mg/m³) amounts to approximately 10 % of the total particle mass (Zeuthen et al. **III**).

The mass-based particle size distribution for reference conditions, the PVC experiment, and the salt experiment, respectively, is illustrated in Figure 4.



Figure 4. Mass-based size distributions for aerosols from reference, PVC and Salt experiments, respectively. Meassured with cascade impactor.

The increased mass load of fine particles when adding salt was not found when adding PVC, in spite that in both cases high amounts of chlorine was added. However, the chlorine in PVC was not bound in the particles, probably due to lack of alkali metals or other condensable cations. The chlorine in this case was released as HCl(g).

Characterization of fly ash, aerosols and deposits

Chemical composition of fly ash and aerosols

A complete chemical analysis of the fly ash samples from the FASAN measuring campaign, as analyzed by wet chemical techniques, is presented in Appendix **VII** (Table VII-1).

Detailed results for selected, volatile elements (Na, K, S, Cl, Zn and Pb) is presented and discussed in Pedersen et al. (II) and summarized here. Figure 5 illustrates the concentrations of these elements in the fly ash samples from the experiments adding different waste fractions, relative to the concentration in the fly ash from the reference experiment. For comparison, it was found that for the experiments with changes in grate operation parameters, the variations in concentration of the target elements were less than 50 % relative to the reference run (i.e. $< \pm 0.5$ relative concentration).

Figure 5 reveals a significant increase (> 100 %) in the concentration of an element in the fly ash in the following cases: Pb when firing PVC and shoes, Zn when firing shredder waste, and, Na and Cl when firing salt. For the salt experiment, the Na and Cl actually accounted for more than 80 % (w/w) of the total mass of the fly ash, as compared to less than 40 % in the reference experiment (and most other experiments).



Figure 5. Concentration of target elements (Pb, Zn, K, Na, Cl, S) in fly ash samples from "dedicated waste" experiment relative to reference conditions (Pedersen et al. II).

The following elements were increased slightly (50 - 100 %) in concentration: Pb when firing shredder waste, salt, and CCA-impregnated wood, Zn when firing shoes and CCA-impregnated wood, K when firing salt, and, S when firing CCAimpregnated wood. Relating Figure 5 to Table 3 reveals that the variations found in the fly ash compositions only in some cases coincide with high input concentrations in waste fractions. This is further discussed later on, in the "partitioning of volatile elements" section, and in Pedersen et al. **II**.

The chemical composition of the aerosols from the experiments with different waste fractions added is illustrated in Figure 6. Only the experiments with changes in waste composition were investigated, as the operational conditions were not expected to influence the composition considerably. For each measured element the content is depicted for all 6 experiments with changes in fuel composition together. Details on more metals are provided in Zeuthen et al. (III).



Figure 6. Size dependent elemental content of main components of the combustion aerosols in experiments with addition of special waste fractions (Zeuthen et al. III).

In all experiments the six major elements in the aerosols by weight were Cl, Na, K, S, Zn and Pb. The chlorine content was increasing with the particle diameter. The content of alkali (Na + K) was rather constant for all experiments. The molar Na/K ratio decreased with the particle diameter indicating that the sodium salts condenses at higher temperatures than those of potassium. The ratio was increased significantly when adding salt (+ 67 %) or shredder waste (+ 112 %) to the base-load waste. For shredder waste this increase was also due to very low K content in the particles. The Zn content increased with particle diameter, and for the experiments with shredder waste, shoes, PVC and impregnated wood the content was increased significantly. For the shredder waste experiment the content by weight was increased more than seven times with respect to Zn. The content of Pb was rather constant as a function of particle diameter and throughout the series of experiments. However, for the shredder waste combustion a 40 % increase was found.

Deposit formation

The deposit study (Frandsen et al. V) indicated that the grate operation did indeed affect the chemical composition and extent (i.e., the flux: g deposit/area/time) of deposit formation significantly.

The experiment with minimum addition of primary air, that is, maximum addition of secondary air, showed a significantly higher upstream deposition flux at a surface temperature of 400 $^{\circ}$ C, see Frandsen et al. V for details. The primary air is let in through the grate, from below, while the secondary air is partly blown downward against the grate, through furnace top nozzles. Thus, a minimum of primary air would expect to give a lesser lift of particles from the fuel bed, but then the jets of secondary air may blow up particles from the top of the fuel bed, thereby more than compensating for the lack of lift from below.

The experiments with "undefined conditions" and "minimum O_2 " in the flue gas gave significantly lower Cl in the deposits, compared to the rest of the operational conditions (Frandsen et al. V). This is an important feature from a material and corrosion point of view.

The addition of PVC and shoes to the base-load waste caused significantly increased deposition fluxes, and increased (not surprisingly) the concentration of Cl in the deposit. In contradiction, shredder waste and CCA impregnated wood caused lower Cl in the deposits (Frandsen et al. V).

Partitioning of volatile elements

The partitioning, that is, the distribution of the elements between bottom ash, fly ash, aerosols and gas phase in the boiler, was studied for the target elements Na, K, Cl, S, Zn, and Pb (Pedersen et al. **II**).

The partitioning study suggested that the vaporization/condensation behavior of some elements was influenced by changes in feedstock composition, as there was not necessarily a correlation between the input concentration of an element in the feedstock, and the amount of the element recovered in the fly ash and flue gas fractions.

The study indicated that Cl and S play an important role for the partitioning of Pb, and maybe also Zn. When firing Cl-rich waste fractions (PVC, salt, shoes), the partitioning of Pb seemed to shift towards increased vaporization (and subsequent recovery in fly ash and aerosol fractions), whereas an opposite effect (decreased vaporization) was seen when firing S-rich waste fractions (CCA impregnated wood

and batteries). The effect, which was explained by the formation of volatile Pbchlorides at high Cl concentrations, and stable Pb-sulphates at high S concentrations, is illustrated in Figure 7.

The partitioning study also implied that organically bound Cl (as in PVC, shoes) was preferably released (as HCl(g)) to the gas phase, whereas the inorganically bound Cl (salt) was recovered in the bottom ash and fly ash fractions (indicating alkali-chloride bonding), see Figure 8. This is in consistency with the findings from the aerosol study described earlier in this Chapter.

The results for overall-partitioning of alkali metals, Zn and S were less conclusive, probably because the changes induced by adding specific waste fractions were not big enough to cause significant changes in the over-all picture.



Figure 7. Pb partitioning to fly ash as function of S/(S+Cl) ratio in input waste. Black marks indicate "dedicated waste" experiments, open marks indicate "base-load waste" experiments (Pedersen et al. II).



Figure 8. Cl partitioned to fly ash and flue gas as function of input concentration (mg/kg input waste, dry base). Black marks indicate "dedicated waste" experiments, open marks indicate "base-load waste" experiments. The dotted lines indicate the upper and lower values for base-load conditions (Pedersen et al. II).

Link to laboratory-scale release results

Lab-scale release experiments have been conducted on a number of dedicated waste fractions from the FASAN campaign, in order to understand the release pattern and the link to formation of fly ash and aerosols in full-scale waste incinerators (this study was part of the CHEC ITOB consortia activities and is reported separately elsewhere, and partly in Frandsen et al. **V**). The waste fractions studied comprised PVC, shoes, shredder waste and CCA-impregnated wood, and the release of metals, S and Cl to the gas phase was quantified as a function of temperature in the interval 500 – 1000 °C in a lab-scale fixed-bed reactor. In general, the elements Al, Ca, Cr, Cu, Fe, Mg, Si and Ti were not released, in consistency with the non-volatile nature of these elements. For the target elements Pb, Zn, K, Na, Cl, and S, the lab.-scale release was found to be both temperature dependent and to some extent fuel specific.

Chlorine release was found to be complete already at 500 °C for PVC, whereas for the other laboratory fuels, there was a gradual increase in the Cl release from 500°C to 1000 °C. This is in good consistency with the full-scale observations and confirming that Cl from PVC is released as HCl(g) already at low temperatures.

K, Na, Zn and Pb were all released to a significant extent to the gas phase above 750 °C, under the conditions prevailing in the fixed bed. The release of Pb at 750 °C was found to correlate linearly with the S/(S+Cl) ratio in the fuels ($R^2 = 0.92$), consistent with the full-scale observations and confirming that at least at moderate temperatures (around 750 °C) Cl and S in the fuel may be influencing the volatility of Pb.

All in all, the lab-scale release results were found to correlate quite well with the observations from the full-scale study, when taking into account the different fuel matrix (dedicated waste fractions vs. blended waste) and the generally more complex conditions prevailing during the full-scale incineration experiments. The obtained results suggested that lab-scale release experiments on dedicated waste fractions can be used to gain reliable information on the release of inorganic elements, and subsequent formation of fly ash and aerosols, during full-scale waste incineration. It may not be possible to extrapolate the data quantitatively, as the conditions during full-scale waste incineration are very complex, but the method proves a useful tool when it comes to predicting overall trends and identify possible release mechanisms attributed to specific waste fractions.

Overall link: Aerosols – fly ash – deposits

An increased mass-load of fine particles when adding salt to the feedstock, but not when adding PVC, was suggested to reflect that the chlorine in PVC was released as HCl(g), while the chlorine in salt was particle bound. Addition of salt also changed the chemical composition and particle size distribution of the total fly ash, towards an increased concentration of Na and Cl, and a smaller particle size, indicating increased condensation of alkali-chlorides.

The study of the deposit formation, however, displayed somewhat different results. Still, in the interpretation of these results one should bear in mind that the deposits were sampled at the entrance to the convective section, at a flue gas temperature close to 600 °C and with a surface temperature of the deposition probe of 400 °C, whereas the aerosols were sampled downstream of the convective section, at a flue gas temperature of app. 175 °C (see Figure 1). This gives rise for different ash transformations in the two cases.



Figure 9. Model approach – heterogeneous condensation of trace metals on a size distributed fly ash.

It was found in the deposition study that the rate of deposit build-up $(g/m^2/h)$, as well as the concentration of Cl in the deposits, was increased when firing PVC, while no significant effects on the deposit formation were observed when firing salt. The chemical analysis of the deposits from the PVC run revealed that the deposits were also relatively high in [Ca], while the alkali content was relatively low. Hence we suggest that the increased deposition flux, and high [Cl], in the deposits from the PVC run may be explained by increased condensation of CaCl₂ on the deposition probe in this specific run. Ca-compounds, such as CaCO₃, are added routinely to various PVC-products, as fillers.

Modeling of heterogeneous condensation

A physical model for heterogeneous condensation of trace elements on a size distributed fly ash was to be developed and coupled with an existing equilibrium model. The model is summarized below, and more details are provided in Appendix **VIII**.

Approach

A two-compartment system: (1) Release from the grate: Vaporization of elements. (2) Condensation of elements in the flue gas channel as a function of T. The idea is sketched in Figure 9. The model is one-dimensional in time and consists of (1) specifying the initial conditions in the system (2) computing the target element condensation as a function of T. Gas/solid equilibrium is calculated at each step using the commercial equilibrium model FactSageTM (version 5.2), and interactions between vapor phase and particle surface are calculated by fluxes, with units of moles per second per particle. A flow chart of the model is outlined in Figure 10.

Assumptions and limitations

Heterogeneous condensation (i.e. surface condensation on existing fly ash particles) is assumed to be the only mechanism by which particulate matter becomes enriched with heavy metals (chemical reactions are neglected).

The condensation flux equation, R_{cond} , $(mol/m^3 \cdot s)$, on uniform sized spherical particles of diameter d_p (m) in the continuum (supermicrometer) regime is diffusion limited and given by equation (2).



Figure 10. Flow chart for modeling of heterogeneous condensation on a size distributed fly ash. Target element: Pb, Zn, Cu, Cd etc.

$$R_{cond} = F_i(d_p) = \frac{2 \cdot \pi \cdot d_p \cdot D_{Xm}}{R \cdot T} \cdot P_{LOAD} \cdot (p_{\infty} - p_s) \cdot \left[\frac{1 + Kn}{1 + 1.71 \cdot Kn + 1.333 \cdot (Kn)^2}\right]$$
(2)

where D_{Xm} is the molecular diffusivity of X in a multicomponent gas, m, p_{∞} and p_s is the bulk and surface pressure of X, $P_{LOAD}(1/m^3)$ is the particle loading of the gas, and $Kn = 2\lambda/d_i$ is the Knudsen number. The term in the brackets is the Fuchs-Sutugin correction for the non-continuum behavior of small particles. λ : the mean free path of the gas molecules (~0.5 µm for combustion gases). D_{Xm} can be estimated for any molecular component X in m using *an empiric equation* of Fuller, Schettler and Giddings, which utilizes diffusion volumes.

To solve for a <u>size distributed fly ash</u>, the particle size distribution is <u>discriminated</u> and the flux equation is solved for each size fraction.

Data fitting / validation of the model

The mass fraction of a condensed species, $Wi(d_p)$ is given by the mass condensed (in time, t), on the particle of size d_p divided by the particle mass.

Different condensation mechanisms will yield different dependencies of $F(d_p)$ on d_p , and it can be shown that for Kn<<1 (coarse fraction particles), Wi $(d_p) \sim 1/d_p^2$, while for Kn>>1 (fine fraction particles), Wi $(d_p) \sim 1/d_p$. However, several studies of coal fly ash have shown that experimental data are not always in consistence with this physical condensation theory. Often the trace elements (As, Ni, Cd etc.) better obey a $1/d_p$ dependence, even for larger particles, suggesting that surface condensation alone cannot account for the trace element distribution (chemical reaction mechanisms are also involved).

The Pb concentration in the FASAN ash as a function of particle diameter is illustrated in Figure 11 (measured and calculated values). The calculated value is a "first rough estimate" of the condensation flux, assuming an initial Pb concentration of 0.005 ppm(v) in the gas phase. At the point of sampling (T ~175 °C), all Pb is assumed to be condensed (as PbCl₂(s)). The initial Pb concentration (C₀) in the coarse particles is set to 697 mg/kg (residual concentration), while C₀ is set to 0 in the aerosols.

It is seen that the Pb concentration better obeys a $1/d_p$ dependence as compared to a $1/d_p^2$ dependence (the same trend is found for other trace elements as Hg, Cd, Zn and Cu, results not shown). The heterogeneous condensation model seems to correctly predict the trends in the experimental data, although it is also seen that the model may tend to over-estimate the Pb concentration in the fine particles.

An over-all evaluation of the equilibrium condensation temperatures revealed that most of the trace elements (including As, Zn, Cu) have relative high condensation temperatures (> 500 - 700 °C), and only Hg, Cd and Pb may be expected to be still in the gas phase at temperatures below ~400 °C. This lead us to conclude that the potential for alternative/selective capture of fly ash (at higher temperatures), in order to improve the ash quality, is limited. On the other hand, the quality of the bottom ash might be improved slightly by taking out the boiler ash separately, instead of mixing it with the bottom ash (which is currently the normal procedure at many plants).



Figure 11. Pb concentration in FASAN ash as a function of particle diameter. Measured and calculated values.

4. Residue leaching

Overview of leaching experiments

As mentioned in the Introduction, bottom ash samples from 12 different experiments were collected at FASAN; these will be referred to as "FASAN residues" (elemental composition is given in Appendix **VII**, Table VII-2). In order to evaluate leaching, the FASAN residues were subjected to CEN 12457-1 (L/S 2 L/kg) compliance leaching experiments (Appendix **VII**, Table VII-3,4,5).

Secondly, in order to provide a reference for comparison of bottom ash leaching data, a range of "averaged one-year" samples of bottom ash was collected from Vestforbrænding, Nordforbrænding, KARA, and Århus (furnace 2 and 3). Note that these materials were not subjected to CEN 12457-1 as these data were obtained from regulatory compliance testing supplied by either the incinerators or AFATEK A/S. Instead, these materials were subjected to CEN 14405 column experiments (Appendix **VII**, Table VII-8) thus providing additional information about time-dependent leaching.

Thirdly, in order to provide sort of a reference leaching data for Danish fly ash/APC residues, a range of "averaged one-year" samples of fly ash/APC residues was collected from Amagerforbrænding, Nordforbrænding, KARA, Odense, Vestforbrænding, and Århus. These materials were subjected to CEN 12457-1,4 batch tests (L/S 2 and 10 L/kg, respectively) as well as CEN 14405 column experiments (Appendix **VII**, Table VII-6,7,9).

In the following text primarily the results from the 12 experiments at FASAN are discussed. The experiments carried out using different operational parameters and different waste input compositions are in this chapter referred to as "OC" and "WI", respectively.

Comparison with regulatory compliance leaching data is provided where appropriate.

Cl leaching

Figure 12 shows Cl contents in fresh bottom ash samples and associated leaching of Cl⁻. The highest Cl⁻ leaching was observed in samples from the experiments with road salt (4720 mg kg⁻¹ TS) and PVC (2700 \pm 320 mg kg⁻¹ TS). Correlation between the Cl content and Cl⁻ leaching was not observed for the OC experiments ($r^2 = 0.18$) whereas a clear correlation was found for the WI samples ($r^2 = 0.86$). Thus, positive correlation between the Cl content in bottom ash and Cl⁻ leaching was found. Comparing the results from the OC and WI experiments it appears that although changes in OC may affect the leaching of Cl⁻ to some extent, the presence of Cl in the incinerated material has a more direct influence on Cl⁻ leaching.


Figure 12. Bulk contents of Cl in fresh bottom ash (primary y-scale) and Cl⁻ leaching from fresh bottom ash (secondary y-scale) (Hyks and Astrup IV).

Evolution of pH during ageing

The leaching of metals in particular is strongly pH-dependent and may thus only be discussed when pH during the leaching experiments is considered. Table 4 shows range of pHs for fresh, naturally aged, and lab-scale aged bottom ash samples. Three months of outdoor ageing did not affect pH compared with the fresh bottom ashes while the lab-scale aged samples showed about 1-2 pH units lower values compared with the fresh bottom ashes. Although pH of the lab-scale aged samples was found semi-stable, ageing/carbonation process was likely far from complete as this would typically be characterized by pH 8.5.

	OC expe	riment: refe	rence waste on	ıly		
Pottom ash	Normal	#1	#2	#3	#4	#5
Douoin asii	(Ref.)	(Min O ₂)	(Max O ₂)	(Dec.sec.)	(Inc.sec.)	(Undef.)
Fresh	11.35	11.38	11.34	11.32	11.03	11.19
Naturally aged	11.35	11.09	11.41	11.31	11.22	-
Lab-scale aged	9.44	9.45	9.24	9.21	10.03	9.70
	WI exper	iment: refe	rence waste + a	added fracti	on	
Bottom ash	PVC	Imp.	Shredder	Shoes	Road	Rottorios
Dottoin asii	IVC	wood	waste (ASR)	Silves	salt	Datteries
Fresh	11.69 ± 0	.2 11.57	11.03	11.40	11.49	10.88
Naturally aged	11.29 ± 0	.2 11.31	11.05	11.33	11.32	10.56
Lab-scale aged	9.65 ± 0.1	10 9.29	9.80	9.53	9.83	9.16

Table 4. pH of fresh, naturally aged, and lab-scale aged bottom ash (in leachates) generated during incineration of the reference waste with or without addition of specific waste fractions (Hyks and Astrup IV).



Figure 13. Leaching from bottom ash (BA) generated during incineration of the reference waste at different OC and different WI. Leaching from fresh, naturally aged, and lab-scale aged BA samples is expressed as black, gray, and white diamonds respectively. Mean values and 90 % confidence intervals are shown; note the different scale in case of DOC (Hyks and Astrup IV).

Leaching of metals

Fresh FASAN residues

Figure 13 shows the mean values and 90 % confidence intervals for the leaching of Cu, Pb, Zn, Ba, Ni, DOC, As, Cr, Mo, Sb, Se, and V from fresh bottom ashes generated in the OC/WI experiments. The leaching was notably similar (within a factor of 3) for all element pairs except Ba, which was likely related to different leaching of sulfate (details in Hyks and Astrup **IV**). The WI leaching data showed somewhat larger variations compared with the OC leaching data (e.g. Ni, Cr). Correlations between the bulk content and leaching were not found for any metal (including Ba). Overall, the leaching of metals appeared to be independent of the bulk contents and, in turn, changes in OC as well as WI; note similar pH of the fresh samples (Table 4).

Leaching of Be and Cd in all samples was found below 0.9 and 0.3 μ g L⁻¹, respectively (i.e. below detection limit of used ICP-OES). These elements are therefore neither shown graphically nor discussed further.

Leaching of Co (not shown graphically) was generally found around the detection limit except the residues generated in the experiment with batteries where the leaching from fresh and also lab-scale aged samples increased to about 30 μ g kg⁻¹ TS. Co is currently not regulated (in case of landfilling in EU) and so far Co leaching has received little attention in particular with respect to leaching control mechanisms. The results of the CEN 12457-1 leaching test did not allow concluding on any specific leaching mechanism or calculating correlations between the bulk contents and leaching. Nevertheless, caution is advised in relation to incineration of batteries.

Aged FASAN residues

Figure 13 also shows the mean values and 90 % confidence intervals for leaching from naturally aged and lab-scale aged samples. For many elements (e.g. Pb, Zn, Cr), ageing had greater impact on leaching than the induced changes to OC/WI. At full-scale, ageing of bottom ash is generally accepted to decrease leaching of most metals and is a widely used method to improve bottom ash quality. This was, however, only the case for Pb, Zn, and Cu which were likely controlled by sorption, precipitation and complexation with DOC (Hyks and Astrup **IV**).

Leaching of Ba from the naturally aged OC samples was found lower than from the lab-scale aged OC samples whereas the situation was opposite for the WI samples. Nevertheless, Ba leaching from both lab-scale aged residues was found at the same level: ± 0.2 mg kg⁻¹ TS. Barite (BaSO₄) and to some extent also Ba_{0.5}Ca_{0.5}SO₄ solid solution ($K = 10^{-7.41}$) were found plausible to explain Ba concentrations.

Ni showed similar leaching patterns for both OC and WI samples with leaching around 7-20 μ g kg⁻¹ TS. This could be caused by a combination of Ni(OH)₂(s) precipitation and surface complexation to Fe/Al (hydr)oxides. According to our modeling the solution concentrations of Ni fell between the solubility of Ni(OH)₂(s) and bunsenite (NiO).

The leaching of some oxyanions (Cr, Sb) increased with carbonation possibly due to dissolution of their bearing phases (e.g. Cr-ettringite and calcium antimonite) at pH < 10.5 (Fig. 13c and 13d). Thus, although ageing can generally decrease leaching of cationic metals the same process may also result in significant increase in the leaching of some oxyanions compared with the fresh residues: Cr and Sb seem to be of major concern due to their toxicity and concentration levels. In real-life system the one order of magnitude increase in the leaching of Cr may further increase towards higher L/S ratios (or with continued carbonation) as Cr leaching from the carbonated bottom ash was reported to reach concentrations up to 2 mg kg⁻¹ TS at pH 8.5. In addition, oxidation of Al(0) during carbonation may lower the "reduction capacity" of the ashes thereby allowing Cr to get oxidized and leach from the aged ashes.

Other oxyanions (As, Mo) showed relatively stable leaching from all three material types. For As, fairly flat sorption edges on Fe-oxides were reported at pH > 9 and phosphate levels similar to our experiments. For Mo, virtually the same leaching from all three material types (Figure 13c) could well be caused by a single controlling process. Since the leachates were found undersaturated with respect to CaMoO₄ and low affinity of MoO₄²⁻ for substituting SO₄²⁻ in ettringite is low, weak surface complexation at pH > 9 seemed plausible.

Comparison with other leaching data

Leaching data from regulatory compliance testing

In Figure 14 leaching data for the FASAN residues (F_{OC} , F_{WI}) were compared with leaching from bottom ash generated during 2006 – 2008 at six Danish MSW incinerators which typically incinerate mixes of MSW from households, recycling stations, industry, etc. with varying ratios between household waste and industry waste. Bottom ash from I/S FASAN (referred to as "F" in Figure 14) used in this comparison was not collected as part of the OC/WI experiments at I/S FASAN but collected as part of the required compliance testing. All bottom ashes were aged outside 1-4 months prior to the CEN 12457-1 leaching experiments.



Figure 14. Mean values and 90 % confidence intervals for leaching from bottom ash (BA) generated at different Danish MSW incinerators in a period from January 2006 to September 2008. Abbreviations and sample populations: F (I/S FASAN, n = 11), AF (I/S Amagerforbrænding, n = 37), VF (I/S Vestforbrænding, n = 39), Aa (I/S Århus, n = 13), NF (I/S Nordforbrænding, n = 13), and Ka (I/S KARA, n = 21). Leaching data generated during the experimental campaign at I/S FASAN are referred to as F_{OC} and F_{WI} . Solid black lines in case of Cu, Pb, and Cl^T represent Danish material utilization limits for Category 3 (Hyks and Astrup **IV**).

The leaching of some elements (e.g. Cu, DOC, Cl⁻) was similar for all MSW incinerators whereas other elements (e.g. Cr) showed larger fluctuations between individual plants. However, it should be emphasized that given the differences in incinerator technology, operation, MSW composition, ash pH and sampling (performed by the incinerator operators), the leaching was in fact similar: typically with variations within one order of magnitude between the incinerators. The Danish regulatory limits for material utilization (Category 3) were met in all cases (these limits are indicated by solid lines in Figure 14 where appropriate). In this perspective, incineration of the previously mentioned amounts of additional waste materials (PVC, impregnated wood, ASR, shoes, batteries) during the OC/WI experiments at I/S FASAN did *not* appear to affect the leaching properties of the bottom ash significantly, at least when compared with typical naturally aged ashes from Danish MWS incinerators.



Figure 15. pH, conductivity and leaching of Cl obtained during the CEN 14405 column experiments with "averaged one-year" samples of bottom ash; mean values and STD are shown. Data indicated by a dashed line represents leaching from fresh bottom ash subjected to CEN 14405).

Column leaching data

Bottom ash samples

Results of the CEN 14405 column experiments carried out with five "averaged oneyear" samples of bottom ash collected at Vestforbrænding, Nordforbrænding, KARA, and Århus (oven 2 and oven 3) are shown in Figure 15 and 16 (details in Appendix **VII**, Table VII-8). Note that leaching data determined in dynamic experiments (i.e. column test) cannot be directly compared with results of the batch experiments since different mechanisms are involved. Nevertheless, the columns still provide a valuable piece of information about time-dependent leaching and possible long-term trends.

It could be seen in Figure 15 that mean pH value did not changed significantly over the course of the experiments while the variations likely reflected the fact that different "aged" residues were tested (compare with variations in pH values in Figure 14). Cl as well as conductivity displayed their typical behavior: both decreased rapidly before L/S 3 L/kg was reached as a consequence of initial washout of the readily soluble compounds (e.g., NaCl, KCl, CaCl₂).

Leaching of metals (Figure 16, next page) from the five materials was relatively similar, although variations were found somewhat larger (up to three orders of magnitude in total) than in case of the batch experiments shown in Figure 14. Higher variations between columns could be caused, among other things, by different magnitude of non-equilibrium conditions during the dynamic leaching experiments. Still, it appears from the data shown in Figure 16 that the leaching of metals was controlled by similar mechanisms: mainly solubility control (e.g. Pb, Ba, Zn, Sb) and depletion/complexation (e.g. Cu, Mo). The differences in leaching between aged and fresh residues (represented by dashed lines) reflected the same patterns outlined in Figure 13. For instance, the leaching of Pb from the fresh residues was higher than the leaching from the "averaged one-year" samples, which were aged to some extent.



Figure 16. Leaching of metals (refined) obtained during the CEN 14405 column experiments with "averaged one-year" samples of bottom ash; mean values and STD are shown. Data indicated by a dashed line represent leaching from fresh bottom ash subjected to CEN 14405. Horizontal line in case of Pb and Se symbolizes detection limit.

Fly ash/APC residues samples

Dynamic leaching was also quantified in case of fly ash/APC residues. In contrast with the above discussed bottom ash samples, the leaching from fly ash/APC residues showed significantly greater variations for some pollutants (e.g. Cr, Sb, Pb); for details see Appendix **VII**, Table VII-9.

In Figure 17 an example of observed leaching from "averaged one-year" samples of fly ash/APC residues collected at Århus (Aa), Vestforbrænding (VF), Amagerforbrænding (AF), Odense (Od), Nordforbrænding (NF) and KARA (Ka) is given together with leaching data for fly ash/APC residues collected by simple grab-sampling at AF and VF (data from another study). It could be seen that e.g. the leaching of Ba and Mo from the "averaged one-year" samples (indicated by data points) is to some extent comparable with leaching from fly ash/APC residues collected by simple grab-sampling (indicated by lines). At the same time, however, the leaching of Cu and Pb in particular shows variations of up to five orders of magnitude during the initial leaching (below L/S 1 L/kg).



Figure 17: Leaching of Ba, Pb, Mo, and Cu from "averaged one-year" samples of fly ash/APC residues compared with leaching of two residues sampled by grab-sampling (indicated by lines). Horizontal line in case of Cu symbolizes detection limit.

This reflected the nature of fly ash/APC residues which composition is extremely dependent on the composition of the waste input (Astrup et al. **I**; Pedersen et al. **II**) and performance of the flue gas control system. In addition, great bulks of the total contaminant contents may be present as part of highly soluble compounds (generally metal chlorides). Consequently, in contrast with bottom ash, it may not be possible to define "typical" range for leaching from Danish fly ash/APC residues and residues from different plants should always be tested individually.

5. Project conclusions

The overall objective of the project was to improve the understanding of the formation and characteristics of residues from waste incineration. This was done focusing on the importance of the waste input and the operational conditions of the furnace. Data and results obtained from the project have been discussed in this report according to the following three overall parts: i) mass flows and element distribution, ii) flue gas/particle partitioning and corrosion/deposition aspects, and iii) residue leaching. This has been done with the intent of structuring the discussion while tacitly acknowledging that these aspects are interrelated and cannot be separated.

Overall, it was found that the waste input composition had significant impact of the characteristics of the generated residues. A similar correlation between operational conditions and residue characteristics could not be observed. Consequently, the project recommend that optimization of residue quality should focus on controlling the waste input composition. The project results showed that including specific waste materials (and thereby also excluding the same materials) may have significant effects on the residue composition, residue leaching, aerosol and deposit formation It is specifically recommended to minimize Cl in the input waste. Based on the project results, it was found that a significant potential for optimization of waste incineration exist.

In the following, more detailed conclusions of the various aspects are provided.

Mass flows and element distribution

Based on mass balance calculations, it was found that experiments with different waste materials significantly altered the waste input composition of both heavy metals, organic and inorganic Cl. Salt contributed with considerable quantities of Na and Cl, Shoes were adding Cd and Cl to the waste mix, Shredder Waste contributed with a range of metals. Batteries contributed with high amounts of Cd, Co, Hg, S and Sr relative to the base-load waste, PVC Waste contributed with Cd and Cl, while Impregnated Wood added As, K, Pb, S and Si.

The consequences for the solid residues and air emissions were evaluated by taking into account both the concentration levels in the actual ashes, as well as the amount of an element routed to the individual ashes. While increased concentrations of e.g. As, Cu, and Mo in the solid residues coincided with waste fractions having high contents of the same element, increased concentrations of Cr, Ni, and Pb could not be fully explained with higher input levels. This indicated that the waste mix was important but also that other constituents and mechanisms, such as mobilization of Pb by Cl, controlled ash concentrations: organic-bound Cl (as in PVC and shoes) favored higher concentrations in the fly ashes while inorganic Cl (salt) shifted Pb towards the bottom ashes.

It was not possible to relate the composition of the solid residues to changes in plant operational parameters (the variations induced by changes in operational conditions could not be distinguished from the natural fluctuations). This does not mean, however, that such correlations may not exist only that in view of the complexity and dynamic properties of a full-scale system, such mechanisms could not be observed. The operational changes induced in these experiments were made as large as possible considering the furnace, still the resulting effects on residue characteristics were less pronounced than the effects from changes in waste input. Based on this it was concluded that priority should be placed on optimization with respect to the waste input.

Flue gas/particle partitioning and corrosion/deposition aspects

Cl and S played an important role for the behavior of Pb, and probably also other metals. When incinerating Cl-rich waste fractions, the partitioning of Pb appeared to shift towards increased vaporization, whereas the opposite was observed for S-rich waste fractions. This was explained by formation of volatile Pb-chlorides at high Cl concentrations, and stable Pb-sulphates at high S concentrations. Organic-bound Cl (as in PVC and shoes) was preferably released (as HCl(g)) to the gas phase, while inorganic Cl (salt) was mainly recovered in the bottom ash and fly ash fractions (indicating alkali-chloride binding).

In addition to affecting overall distribution of Pb between the residues, Cl in the waste appeared to: i) enhance the vaporization of heavy metals (primarily Pb, but probably also Zn), ii) cause increased mass-load of aerosols if present as alkalichloride in the waste, and iii) cause increased deposition fluxes and higher concentrations of Cl in the deposits. It was suggested to minimize the Cl/S ratio in the input waste as this may reduce corrosion in the boiler sections by shifting the equilibrium from the highly corrosive species PbCl₂ and ZnCl₂ to less corrosive species such as PbSO₄ and ZnSO₄. Deposit formation was the only aspect which appeared significantly correlated with the operational conditions. "Minimum O₂" and "Undefined Conditions" gave significantly lower Cl in the deposits while "Increased Secondary Air" showed a significantly higher deposition flux.

High input concentrations of heavy metals also appeared to be problematic: e.g. in case of shredder waste and CCA impregnated wood, high loads of Zn and/or Pb resulted in increased concentrations of these two elements in both fly ash and aerosols (and deposits), although the overall partitioning was not altered. Consequently enhanced deposition and corrosion problems may be induced in the boiler.

The potential for alternative/selective capture of fly ashes, in order to improve the ash quality, appeared to be limited as most heavy metals condense at relatively high temperatures. Control of the input waste composition was concluded more relevant for optimization.

Residue leaching

No clear correlation between the bulk contents of the bottom ashes and the leaching was observed for the changes induced in waste input and operational conditions, except in the case of Cl ($r^2 = 0.86$). This indicated that Cl leaching from bottom ashes may be reduced by limiting Cl in the waste received at the incinerator, potentially by improved source separation. Leaching experiments on fly ash samples collected during the experiment could not be carried out as only a few grams could be collected due to practical reasons. It was not possible to observe correlations between changes in operational conditions and leaching.

Differences in metal leaching between samples obtained in this study could primarily be associated with differences in ageing: leaching of cationic metals from aged bottom ashes was generally lower than from fresh ashes whereas ageing had negative effects on leaching of Sb and Cr. The observed variations in leaching data from these experiments generally appeared within the ranges in leaching data observed for typical Danish MSW incineration bottom ash. Although similar mechanisms control the leaching in different bottom ashes, variations in element leaching of about one order of magnitude appear typical for Danish bottom ashes.

6. Suggestions for further research

During the course of the project, two workshops were carried out with participation from major Danish stakeholders. The latter of these workshops was concluded with a discussion of the outcome of the project. A general appreciation with the work performed in the project was expressed from the participants and it was found that the project results would likely form a solid basis for future research and development within the field. Based on discussions in the workshops, among the project partners, feedback on international conferences and other stakeholders a range of important aspects relevant for further research has been identified:

- The interdisciplinary collaboration within the project group has been very advantageous in our opinion. In order to better benefit from the data generated in this project, further develop the understanding of chemical interactions between the different elements, and make the experimental full-scale data from the project more evident, we hope to continue with the PSO-5784 group of participant in future follow-up projects.
- The experiences related to the full-scale experiments carried out in this project clearly showed that the approach of adding specific waste materials to a base-load waste were extremely useful for assessing to which extent these waste materials affected the emissions from the plant. The project also showed that it was possible to complete a range of individual experiments within a rather limited time schedule. Further full-scale experiments with other waste mixes are suggested to supplement the results from this project.
- Although this project included an unprecedented range of measurements on a fullscale plant while at the same time varying key parameters, the dynamics and complexities of the involved full-scale processes is a limiting factor for investigation of the more detailed processes responsible for formation of the residues. More detailed data supplementing the results gained in this project is therefore suggested to be carried out based on lab-scale experiments.
- Lab-scale experiments focused on element release and ash formation of specific waste materials and mixtures of waste materials are suggested to be carried out under controlled conditions. These experiments should be combined with investigation of the leaching properties of the treated waste materials, in order to better understand the relation between combustion temperature, waste composition, combustion atmosphere, and the leaching from the ashes generated.
- Lab-scale experiments are also suggested to improve the understanding of relations between the operational conditions and deposit formation as this project indicated a link in this respect.
- The importance of Cl and S is suggested to be further studied, e.g. by further development of full-scale measurements of the release profile just above the grate as done in PSO-6368.

Appendix I

Astrup, T., Riber, C. & Pedersen, A.J. (2009): MSW Incinerator Emissions and Residues: Influence of Waste Input and Operational Conditions

Appendix II

Pedersen, A.J., Frandsen, F.J., Riber, C., Astrup, T., Thomsen, N.S., Lundtorp, K. & Mortensen, L.F. (2009): A Full-Scale Study on the Partitioning of Trace Elements in Municipal Solid Waste Incineration – Effects of Firing Different Waste Types

Appendix III

Zeuthen, J.H., Pedersen, A.J., Hansen, J., Frandsen, F., Livbjerg, H., Riber, C. & Astrup, T. (2007): Combustion aerosols from municipal waste incineration - effect of fuel feedstock and plant operation

Appendix IV

Hyks, J. & Astrup, T. (2009): Influence of operational conditions, waste input and ageing on leaching of contaminants from waste incineration bottom ashes

Appendix V

Frandsen, F.J.; Pedersen A.J.; Hansen, J.; Madsen, O.H.; Lundtorp, K.; Mortensen, L. (2008): Deposit Formation in the FASAN WtE Boiler as a Function of Feedstock Composition and Boiler Operation

Appendix VI

Operational conditions and flue gas data for the FASAN experiments

Log data and m	easured flu	e gas comp	osition frc	m FASAI	N experim	ents (avera	age values						
			OC exper	<i>iments</i>				WI exper	iments: Ref	erence wast	e + added fr	action.	
			Dec.	Undef.			Incr.			Shredder	Imp.		
Parameter	Units	Reference	Sec.	Cond.	Min 02	Max O2	Sec.	Salt	Batteries	waste	poom	PVC	Shoes
Primary air flow	m ³ /h	20385	25843	24606	22820	26225	17677	23412	25846	23026	21146	20683	20982
Furnace top													
nozzles (sec. 1)	m ³ /h	3682	2713	2781	2503	3146	5476	4243	4249	3029	2413	3261	3420
OFA 1 (sec. 2)	m ³ /h	5231	4084	3640	3086	6428	4993	4265	3511	5722	5754	5490	4825
OFA 2	$m^{3/h}$	1983	508	1500	500	4762	1709	1623	1508	1731	1834	2450	1585
Secondary air flow	m ³ /h	10896	7305	7922	6089	14335	12177	10130	9268	10481	10001	11201	9830
Total air flow	m ³ /h	31282	33147	32528	28909	40560	29854	33542	35113	33507	31147	31884	30812
O ₂ wet	vol. %	7	8	8	L	6	7	8	8	8	8	8	8
Humidity (%)	%	15	15	15	17	14	15	15	14	15	14	12	14
Flue gas flow 1	m ³ /h	78607	77862	75632	72461	95127	77120	80161	83433	78561	81049	79693	77801
Flue gas flow 2													
(actual)	Am^{3}/h	92351	92117	90331	86192	116361	91951	94659	98618	92578	95803	92389	92416
T1, lower IR	°C	1037	1010	985	1018	991	1011	1014	1017	1028	1018	1010	1011
T2, furnace top	°C	865	912	928	912	890	1179	920	963	919	1024	1112	1124
T3, EBK	°C	976	953	934	973	924	961	946	949	965	970	954	958
T4, EBK burner	°C	895	832	824	845	851	910	895	874	924	872	895	884
T4, flue gas duct	°C	175	175	173	176	184	175	175	176	175	176	175	177
HCl before NID	mg/m ³	322	263	342	290	344	598	335	317	331	358	749	519
Measured flue gas co	omposition (ent	trance to superh	leater section.	, T(gas) ~600	C):								
0_2	vol. %	7.81	8.35	8.78	7.69	10.16	7.61	7.99	8.7	7.59	8.31	8.29	7.67
CO_2	vol. %	10.9	10.5	10	11.1	8.9	11	10.9	10.1	11.1	10.7	10.4	10.4
\mathbf{SO}_2	ppmv (dry)	43.4	32.5	47.7	29.5	21.9	53.2	43.4	66.3	56.5	92.9	42.3	23.6
CO	ppmv (dry)	2.1	2.5	0.4	9.6	1.1	2.9	1.4	2.9	1.5	0.3	4.7	2.9
NO	ppmv (dry)	100.1	89.3	90.8	80.1	103.6	89.7	87.3	84.4	102	111.4	100.5	68.6
Total dust load	g/m ³	2.3	2.6	1.9	2.2	2.2		2.2	2.8	1.9	2.1	2.3	1.8

Appendix VII

Reside characteristics: solid composition and leaching

			OC experi	ments: Refe	rence waste	only		M	xperiment	s: Reference	e waste + 8	added frac	tion	One-yea	r Århus
Flomont	11	Dofoundation	Undef.	Min O	More	Dec Con	Inc.	Colt	Dottonios	Shredde • Weste	Imp.		Change	۲.A	
As	mg/kg TS	102	191.0	67.8	160.8	88.7	DEC.	102	111	1 W aste 109.4	2260	187	211.6	333	210
Ba	mg/kg TS	568	571.0	811.2	954.1	613.9		218	594	1502.8	651	895	585.7	1600	446
Be	mg/kg TS	<0,6	0.6	<0,6	0.8	<0,6		<0,6	0.631	<0,6	<0,6	0.637	<0,6	0,695	<0,6
Cd	mg/kg TS	203	408.0	137.7	205.0	159.0		189	428	387.0	192	595	578.1	186	107
Co	mg/kg TS	15	21.3	14.9	23.3	15.9		3.98	66.69	17.8	16.6	15.7	14.4	28,9	7,98
Cr	mg/kg TS	476	557.0	411.2	951.2	510.9		185	581	379.9	914	687	775.0	579	159
Cu	mg/kg TS	1260	2530.0	1562.9	976.0	1308.9		1950	606	3348.5	2340	2730	1930.1	1210	532
Hg	mg/kg TS	2.32	17.1	8.9	4.2	4.2		8.23	53.9	5.8	10.5	8.37	7.4	20,1	14,3
Mo	mg/kg TS	36.4	34.1	29.7	34.1	32.2		26.4	35.6	46.8	29.9	32.5	33.6	21,6	8,47
Ni	mg/kg TS	104	9.99	120.9	122.7	93.3		49.4	144	109.1	95.8	93.9	148.9	89,3	32,4
Pb	mg/kg TS	3680	5500.0	5193.2	3172.9	3900.8		7360	2980	7405.7	6080	16900	12028.1	5600	2270
S	mg/kg TS	37500	52100.0	39656.0	36655.7	37873.0		55500	47800	44163.0	63500	38800	45217.8	53400	36700
Sb	mg/kg TS	726	749.0	736.2	380.7	620.2		549	559	1362.8	855	647	929.9	967	560
Sn	mg/kg TS	356	410.0	197.6	91.2	235.4		573	203	791.7	419	503	898.5	400	120
Sr	mg/kg TS	468	405.0	429.4	549.4	452.7		71.1	441	316.2	426	437	295.2	402	262
٧	mg/kg TS	17.4	23.6	43.5	28.8	21.2		3.89	23.8	22.9	26.2	38.6	35.6	57,4	14,9
Zn	mg/kg TS	24000	33400.0	27169.1	13954.4	21034.7		30000	24700	86405.9	36700	29700	37944.1	24500	12200
SiO_2	%	12.1	10.5	11.9	16.2	12.4		2.38	13.6	10.6	11.3	14.4	9.3	24,3	5,28
A12O ₃	%	6.67	6.0	6.4	10.5	6.8		0.708	8.5	5.1	6.17	8.21	6.2	7,96	2,22
CaO	%	22.4	18.3	20.8	28.0	23.1		2.36	22.3	14.0	20.2	20.2	14.2	21,6	41,5
$Fe2O_3$	%	1.15	1.1	1.6	1.4	1.2		0.259	1.26	3.5	1.11	1.71	1.0	1,93	0,64
K_2O	%	9.13	9.5	8.5	5.5	8.5		15.5	7.54	8.0	8.87	7.26	11.0	5,37	2,56
MgO	%	2.23	2.1	2.4	3.1	2.7		0.285	2.41	1.8	1.98	2.62	1.8	2,22	1,24
MnO	%	0.0887	0.1	0.1	0.1	0.1		0.0177	0.221	0.1	0.105	0.128	0.1	0,109	0,0374
Na_2O	%	9.13	10.1	8.8	6.1	8.6		24.9	8.76	8.8	9.14	8.83	12.6	5,51	2,87
P_2O_5	%	1.96	1.9	2.1	2.3	2.0		0.464	1.96	1.3	1.84	1.86	1.3	1,41	0,45
TiO_2	%	1.15	1.3	1.1	1.8	1.3		0.146	1.55	1.0	1.11	1.62	0.8	1,91	0,194
CI	% TS	19	22.0	20.8	11.9	18.6		39.1	17.3	22.7	18.4	24.4	26.3	6,8	13,1
F	% TS	0.21	0.2	0.5	12.8	0.2		0.1	0.23	0.5	0.27	0.24	0.2	0,23	0,14
LOI	% TS	21.2	21.1	22.1	13.9	20.2		45.5	13.4	29.7	18.3	20.8	21.8	8,9	22,1
TS	%	97	96.6	94.5	97.8	96.2		99.1	97.2	97.1	96.8	94.5	95.5	99,1	95.1
^a No fly a:	sh sample av	vailable fror.	n this expe	vriment											

T ADIC A				nnieni VI	Inninn) en	I abileb l							
		OC experim	nent: refere	nce wast	e only			WI experiment:	: reference wa	ste + added	fraction		
Element	Units	Reference	Min O ₂	Max O ₂	Decsec.	Inc.sec.	Undefined	PVC ^a	Imp. wood	Shredder	Shoes	Salt	Batteries
As	mg/kg TS	26.7	28.8	32.0	22.7	19.4	20.7	28.77 ± 5.92	57.3	27.5	23.2	34.8	23.3
Ba	mg/kg TS	917	3170	1040	1350	1080	1690	1747 ± 298.6	1380	4830	1230	905	1440
Be	mg/kg TS	0.79	0.621	0.902	0.738	0.975	0.838	0.91 ± 0.074	0.804	0.631	0.861	0.876	0.631
Cd	mg/kg TS	2.68	1.67	4.02	2.52	4.41	2.29	3.12 ± 1.713	1.02	1.44	2.55	3.59	7.53
Co	mg/kg TS	19.8	38.9	47.0	41.0	79.4	30.1	33.06 ± 11.16	27.3	49.4	54.7	32.9	222
Cr	mg/kg TS	316	552	727	590	402	387	469.5 ± 60.53	527	487	617	386	378
Cu	mg/kg TS	18700	10600	4240	1840	5620	3930	3473 ± 3448	5500	21400	2200	1950	3090
Hg	mg/kg TS	0.021	0.0185	0.026	0.024	0.018	0.022	0.0402 ± 0.029	0.03	0.029	0.017	0.024	0.0295
Mo	mg/kg TS	6.98	29	7.06	15.0	14.9	11.9	10.58 ± 1.294	9	37	12.3	10.3	13.5
Ni	mg/kg TS	179	290	305	170	731	304	213.7 ± 77.1	515	913	170	200	331
Pb	mg/kg TS	697	1080	1740	9150	1770	2960	1899.8 ± 1273	4470	2230	704	1090	2640
S	mg/kg TS	2190	2640	4930	4360	4500	4900	3016 ± 495	9260	3860	5310	2970	10100
$\mathbf{S}\mathbf{b}$	mg/kg TS	54	126	55.7	75.9	182	57.6	97.87 ± 12.04	72.4	203	59.2	65	74.4
Sn	mg/kg TS	325	117	91.8	173	53.2	143	133.5 ± 132.2	160	462	58.2	92.9	49.4
Sr	mg/kg TS	388	494	333	351	391	348	378.4 ± 30.97	352	422	361	336	833
^	mg/kg TS	34.3	52.8	43.1	48.4	53.3	37.4	46.4 ± 5.429	47.7	56.1	58.4	40	36.2
Zn	mg/kg TS	2480	5240	2540	3660	3250	3750	2872.8 ± 502	3420	10600	2690	4900	3340
SiO_2	% TS	50.5	43.0	45.7	47.5	39.4	47.4	48.79 ± 1.51	46.7	40.2	45.6	47.6	44.9
$A12O_3$	% TS	11.7	12.2	14.2	11.7	20.1	13.0	12.26 ± 0.78	11.6	10.8	14.3	12.6	13.2
CaO	% TS	13.2	13.4	15.7	14.0	16.2	13.7	15.70 ± 0.39	15	11.4	16	14.4	15
$\mathrm{Fe_2O_3}$	% TS	7.76	17.2	9.24	9.42	10.2	7.97	10.76 ± 1.03	10.9	20.9	9.73	8.57	9.97
$ m K_2O$	% TS	1.56	1.26	1.28	1.35	1.21	1.57	1.11 ± 0.096	1.48	1.11	1.16	1.36	1.43
MgO	% TS	1.89	2.11	2.21	1.99	2.11	1.9	2.19 ± 0.113	1.83	1.87	2.21	1.9	1.94
MnO	% TS	0.115	0.178	0.151	0.128	0.13	0.104	0.14 ± 0.021	0.149	0.184	0.122	0.115	0.173
Na_2O	% TS	5.41	4.19	4.63	4.68	3.35	4.53	3.98 ± 0.269	4.33	3.74	4.19	5.83	4.67
P_2O_5	% TS	1.58	1.51	1.47	1.46	1.35	1.6	1.32 ± 0.09	1.34	0.976	1.23	1.46	1.21
TiO_2	% TS	0.995	1.04	1.36	1.14	1.58	1.21	1.34 ± 0.059	0.966	0.972	1.26	1.07	1.1
TOC	% TS	1.3	1.0	0.4	0.7	1.2	2.3	0.61 ± 0.195	0.6	0.4	0.5	0.7	1.2
CI	% TS	0.9	1.3	1.3	1.1	0.9	0.7	1.73 ± 0.345	0.7	1.2	1.5	2.3	1.0
Ь	ST %	0.04	0.07	0.06	0.01	0.07	0.04	0.049 ± 0.007	0.04	0.05	0.06	0.1	0.12
LOI	% TS	2.4	0.6	1.4	3.0	1.8	4.2	0.929 ± 0.795	2.5	1.2	1.5	2.6	2.6
TS	%	98.8	95.5	97.8	97.0	98.3	97.9	96.44 ± 2.319	96.6	98.9	98	98.6	95.7
^a Mean vi	alues and S7	ID obtained	l from sep	arately i	treated bot	tom ash	samples gene	rated in the expe	eriment with	PVC (n =	7).		

Table VII-2. Solid contents of the FASAN residues (hottom ashes)

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		lab	.70	.81	5.5	.23	1.1	0.2	ī	1.0	193	.63	t08	.50	0.1	9.5	385	130	5.1	.22	3.6	8.8
	lefined	atural	- 5	۰ ک	1	-	ı		ı	1		- 1	ч т	-	- 1	- 1		I	1	ı س	1	ı س
	Und	fresh n	1.19	5.73	10.0	0.22	1.1	0.2	1343	1.0	32.5	10.3	666	0.47	27.1	1162	360	80.1	1.9	2.16	4.0	139
		lab 1	10.03 1	6.38	5.5	0.20	1.1	0.2		0.4	414	0.70	93.6	0.66	5.1	6.4	673	94.3	1.2	3.85	5.1	2.7
	nc. sec.	natural	11.22	4.97	5.2	0.09	1.1	0.2	2088	0.4	76.3	0.98	88.8	0.43	4.0	9.8	142	31.0	2.1	1.55	13.6	43.1
	I	fresh ı	11.03	6.52	7.5	0.18	1.1	0.2	2070	0.4	19.9	2.07	220	0.38	7.1	26.3	214	71.6	0.7	2.84	2.0	101
		lab	9.21	6.57	5.0	0.17	1.1	0.2	ı	0.8	358	1.63	124	0.60	24.7	0.1	1097	109	2.3	3.27	2.7	2.7
	bec. sec.	natural	11.31	4.27	3.7	0.04	1.1	0.2	1360	0.4	26.8	1.42	70.8	0.47	8.7	4.7	197	27.9	1.7	0.64	9.3	47.5
	Ω	fresh	11.32	ı	7.9	0.09	I.I	0.2	1655	1.7	48.2	5.96	310	0.38	13.7	22.3	303	46.2	1.5	1.38	2.5	138
		lab	9.24	6.05	3.6	0.16	1.1	0.2	ı	1.2	816	1.40	97.8	0.44	28.2	0.7	1087	90.7	1.0	3.30	2.1	2.7
	$Max O_2$	natural	11.41	4.38	2.7	0.03	1.1	0.2	1380	0.4	79.8	1.68	51.5	0.44	10.8	10.9	192	21.0	2.3	0.77	9.7	47.3
	4	fresh	11.34	6.22	13.3	0.12	1.1	0.2	1632	0.7	108	5.66	257	0.41	15.2	39.0	268	48.9	2.7	1.51	8.0	121
		lab	9.45	5.23	3.3	0.27	1.1	0.2	ı	0.4	65.2	1.43	105	0.64	18.0	3.7	365	131.1	2.1	2.01	2.2	68.0
	Min O ₂	natural	11.09	5.03	3.1	0.10	1.1	0.2	1741	0.4	17.3	1.41	57.8	0.6	8.2	2.1	214	48.7	3.6	0.81	6.4	35.7
		fresh	11.38	7.00	6.3	0.41	I.I	0.2	2112	0.4	9.9	4.64	281	0.18	19.4	119	106	36.0	1.3	1.24	0.4	221
		lab	9.44	4.96	2.7	0.13	1.1	0.2	ı	0.8	353	2.43	226	0.83	12.6	2.8	457	107	5.6	1.40	2.5	175
	eference	natural	11.35	3.71	13.4	0.02	1.1	0.2	1190	0.4	0.2	0.003	46.9	0.001	0.2	0.2	147	0.8	0.5	0.003	0.3	2.7
	R	fresh	11.35	5.11	9.7	0.07	1.1	0.2	1630	0.9	39.5	9.01	437	0.57	26.2	59.4	178	45.3	3.7	0.98	4.1	122
		Units		mS/cm	µg/kg	mg/kg	µg/kg	µg/kg	mg/kg	µg/kg	µg/kg	mg/kg	mg/kg	mg/kg	µg/kg	µg/kg	mg/kg	µg/kg	µg/kg	mg/kg	µg/kg	µg/kg
Ichicci nc		Element	рН	EC	\mathbf{As}	Ba	Be	Cd	CI	Co	Cr	Cu	DOC	Mo	Ni	Pb	S	Sb	Sn	Sr	Λ	Zn

[able VII-4. Leaching data from the FASAN experiments: Cumulative release from "WI" bottom ash samples determined in C. Ukg batch experiments with fresh, naturally aged at AFATEK A/S (natural), and lab-scale aged (lab) bottom ashes. Values in Arotica limits	EN 12457-1 L/S	ndicated in italic	
[able VII-4. Leaching data from the FASAN experiments: Cumulative release from "WI" bottom ash samples de 1/kg batch experiments with fresh, naturally aged at AFATEK A/S (natural), and lab-scale aged (lab) bottom as	termined in C	shes. Values in	
[able VII-4. Leaching data from the FASAN experiments: Cumulative release from "WI" bottom a 1/kg batch experiments with fresh, naturally aged at AFATEK A/S (natural), and lab-scale aged (ish samples de	lab) bottom a	
[able VII-4. Leaching data from the FASAN experiments: Cumulative release from ' 1/kg batch experiments with fresh, naturally aged at AFATEK A/S (natural), and la	"WI" bottom 2	b-scale aged (
Cable VII-4. Leaching data from the FASAN experiments: Cumulative LIkg batch experiments with fresh, naturally aged at AFATEK A/S (notice datation limits)	release from	latural), and la	
Cable VII-4. Leaching data from the FASAN experiment 1/kg batch experiments with fresh, naturally aged at AF	s: Cumulative	ATEK A/S (r	
Cable VII-4. Leaching data from the FASA 1 Jkg batch experiments with fresh, natural	N experiment	ly aged at AF	
Cable VII-4. Leaching data fr 1/kg batch experiments with	om the FASA	fresh, natural	
Table VII-4. Le 1/kg batch exp 1/kg batch exp	eaching data fr	eriments with	limite
` `	Table VII-4. L	1/kg batch exl	aftant datantion

reflect deter	ction limits.									
			PVC			Imp. wood			Shredder	
Element	Units	fresh	natural	lab	fresh	natural	lab	fresh	natural	lab
PH	1	11.69 ± 0.15	11.29 ± 0.16	9.65 ± 0.10	11.57	11.31	9.29	11.03	11.05	9.80
EC	mS/cm	7.82 ± 0.86	5.50 ± 0	8.20 ± 0.41	5.72	3.90	7.40	4.68	4.34	6.37
As	µg/kg TS	5.0 ± 0.2	4.60 ± 0.3	6.10 ± 1.7	3.11	3.90	6.20	9.19	4.30	5.10
Ba	mg/kg TS	6.94 ± 2.47	1.13 ± 0.07	0.21 ± 0.03	0.06	0.05	0.12	0.42	0.19	0.20
Be	µg/kg TS	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	I.I
Cd	µg/kg TS	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2
CI	mg/kg TS	2703 ± 332	2440 ± 683	ı	1407	1240	I	1535	1546	·
Co	µg/kg TS	1.45 ± 0.7	0.40 ± 0.20	4.9 ± 8.6	2.81	0.70	2.50	0.4	0.4	0.4
Cr	µg/kg TS	32.1 ± 6.5	46.2 ± 10.1	609 ± 75.0	410	117	1131	13.5	13.7	16.4
Cu	mg/kg TS	1.66 ± 0.51	0.97 ± 0.7	1.26 ± 0.27	5.44	1.40	1.63	1.72	0.64	0.72
DOC	mg/kg TS	148 ± 17.5	43.0 ± 5.2	60.9 ± 19.7	20.0	75.3	55.3	152	41.6	34.8
Mo	mg/kg TS	0.20 ± 0.04	0.21 ± 0.06	0.36 ± 0.03	0.55	0.38	0.50	1.28	0.71	1.02
Ni	µg/kg TS	12.1 ± 4.6	7.20 ± 1.1	21.9 ± 22.2	61.5	9.90	9.90	5.33	4.60	3.90
Pb	µg/kg TS	574 ± 206	32.5 ± 14.6	6.5 ± 3.1	56.1	10.5	0.50	5.16	6.0	0.2
S	mg/kg TS	15.4 ± 7.3	50.7 ± 3.3	702 ± 322	396	252	1500	218	201	<i>L</i> 66
\mathbf{Sb}	µg/kg TS	21.1 ± 5.7	34.5 ± 3.2	101 ± 16.1	23.9	18.8	123	99.3	48.9	197
Sn	µg/kg TS	39.8 ± 5.1	48.9 ± 9.0	420 ± 46.6	310	104	686	40.2	30.9	33.7
Sr	mg/kg TS	6.05 ± 4.4	4.25 ± 0.31	4.59 ± 0.54	1.27	0.97	4.47	2.03	06.0	3.14
^	µg/kg TS	1.93 ± 0.4	5.50 ± 1.7	3.90 ± 1.4	28.0	13.0	3.20	10.3	13.3	4.40
Zn	µg/kg TS	380 ± 225	87.5 ± 50.0	47.8 ± 72.1	220	96.7	8.10	220	41.5	2.5

Table VII-5. Data from the FASAN experiments: Cumulative release from "WI" bottom ash samples determined in CEN 12457-1 L/S 2 l/kg batch experiments with fresh, naturally aged at AFATEK A/S (natural), and lab-scale aged (lab) bottom ashes. Values indicated in italic reflect

detection II	mits.									
			Shoes			Road salt			Batteries	
Element	Units	fresh	natural	lab	fresh	natural	lab	fresh	natural	lab
Hq	,	11.40	11.33	9.53	11.49	11.32	9.83	10.88	10.56	9.16
EC	mS/cm	6.80	6.27	8.78	13.31	7.46	11.55	6.02	6.15	7.50
\mathbf{As}	μg/kg TS	12.42	6.80	3.80	3.92	5.10	4.0	3.20	6.40	39.4
Ba	mg/kg TS	0.21	0.11	0.20	0.13	0.07	0.22	0.19	0.13	0.18
Be	μg/kg TS	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1
Cd	µg/kg TS	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2	0.2
CI	mg/kg TS	2455	2380	ı	4717	3142	I	2084	1979	ı
Co	µg/kg TS	0.4	0.4	0.4	0.4	0.4	0.4	32.8	3.60	20.9
Cr	μg/kg TS	64.1	54.1	514	26.8	31.0	348	68.2	48.2	116
Cu	mg/kg TS	2.15	1.50	0.56	3.0	2.32	0.72	6.58	1.32	2.05
DOC	mg/kg TS	136	66.9	39.8	223	59.3	75.2	397	59.7	113
Mo	mg/kg TS	0.27	0.52	0.46	0.51	0.43	0.55	0.83	0.46	0.76
Ni	μg/kg TS	9.71	13.3	1.20	1.10	11.4	19.2	58.8	9.50	0.2
Pb	μg/kg TS	16.85	11.4	0.2	113	42.5	0.2	9.14	5.20	0.2
S	mg/kg TS	169	193	1202	614	150	665	524	750	1557
Sb	μg/kg TS	43.2	45.7	76.3	59.0	50.4	97.8	15.6	14.0	147
Sn	μg/kg TS	76.7	61.3	364	111	45.2	259	2.0	0.5	0.5
Sr	mg/kg TS	2.70	1.45	4.42	1.41	0.69	3.63	3.98	3.22	5.37
Λ	μg/kg TS	4.54	15.7	4.30	4.03	7.60	3.0	11.0	9.50	4.60
Zn	ug/kg TS	160	90.4	2.5	120	118	2.50	50.0	31.8	3.20

expressed in mg/.	L. Values indicated	l in italic reflect d	letection limits.				
mg/L	VF (FA)	NF (FA)	Aarhus (APC)	AF (APC)	Aarhus (FA)	Odense (FA)	Kara (FA)
	LS2 S1	LS2 S2	LS2 S3	LS2 S4	LS2 S5	LS2 S6	LS2 S7
Al	5.02E-01	5.97E-01	5.44E-01	9.61E-01	1.52E+00	4.66E-01	8.99E-01
As	3.70E-02	4.00E-04	8.53E-02	9.14E-02	1.39E-01	4.00E-04	1.11E-01
Ba	8.67E-01	5.72E-01	4.99E+00	3.03E+01	9.17E-01	6.97E-01	1.40E+01
Be	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04
Ca	2.32E+03	1.77E+03	1.48E + 04	4.57E+04	1.67E+03	1.39E+03	1.89E+04
Cd	2.18E-01	3.81E-02	1.00E-04	2.09E-01	1.19E+00	2.82E-01	2.52E-02
Co	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04
Cr	1.00E-04	3.25E+00	2.67E-01	2.60E-02	5.39E+00	3.08E+00	1.09E-02
Cu	5.49E-01	5.80E-04	5.80E-04	5.99E-01	2.00E-01	5.80E-04	1.16E+00
Fe	6.97E-02	2.80E-04	2.80E-04	6.59E-01	2.52E-01	2.80E-04	1.92E-01
K	2.88E+04	2.44E+04	1.42E + 04	1.78E+04	4.27E+04	3.71E+04	2.37E+04
Li	3.55E+00	6.96E-01	1.44E + 00	4.82E-01	5.00E+00	2.69E+00	2.26E+00
Mg	1.09E-01	3.13E-01	2.26E-01	1.26E+00	6.33E-01	1.82E-01	3.27E-01
Mn	7.00E-05	7.00E-05	7.00E-05	8.30E-03	1.95E-02	7.00E-05	3.00E-02
Mo	3.17E+00	2.29E+00	2.89E-01	4.58E-01	6.41E+00	4.21E+00	8.07E-01
Na	2.10E+04	1.81E+04	1.57E+04	1.34E+04	3.22E+04	2.71E+04	1.81E+04
Ni	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05
Ρ	4.16E-01	4.52E-01	4.56E-01	8.37E-01	7.93E-01	3.43E-01	4.53E-01
Pb	1.09E+01	2.22E-01	2.11E+02	1.03E+03	1.54E+00	9.26E-01	4.26E+02
S	1.40E+03	1.49E+03	3.98E+02	1.45E+02	1.68E+03	2.11E+03	3.49E+02
Sb	4.00E-04	4.00E-04	4.86E-02	3.71E-01	4.47E-01	5.62E-02	1.83E-01
Se	3.32E-01	2.13E-01	1.54E-01	2.98E-01	7.04E-01	4.11E-01	1.35E-03
Si	5.08E-01	8.29E-01	2.01E-01	2.78E-01	1.44E+00	2.95E-01	2.21E-01
Sn	2.74E-02	2.70E-04	2.70E-04	2.70E-04	1.91E-01	4.02E-02	6.70E-02
Sr	1.92E+01	2.13E+01	7.13E+01	5.06E+01	1.21E+01	1.64E+01	5.00E+01
Ti	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05
Λ	1.50E-04	1.50E-04	1.98E-02	2.28E-02	1.50E-04	1.50E-04	1.97E-02
Zn	3.57E+00	7.89E-01	6.45E+00	4.23E+01	3.99E+00	1.43E+00	1.76E+01
LS [L/kg]	2.18	2.22	2.13	2.13	2.15	2.08	2.13
[-] Hd	12.14	11.08	11.81	11.29	10.353	11.07	11.669
c [mS/cm]	180.5	154	139.3	213.6	246.2	216.6	196.8
CI	69801	71273	62853	125755	102116	86933	91227
DOC	7.4	15.3	22.6	1.5	0.4	9.9	160.5

Table VII-6. Raw leaching data from different Danish fly ash/APC residues obtained in CEN 12457-1 batch test (L/S 2 L/kg). Leaching

Table VII- expressed in	7. Raw leaching 1 mg/L. Values i	g data from d ndicated in ital	ifferent Danish fly lic reflect detection	y ash/APC resi n limits.	dues obtained in	CEN 12457-4 ba	tch test (L/S	[0 L/kg). Leaching
mg/L	VF (FA)	NF (FA)	Aarhus (APC)	AF (APC)	Aarhus (FA)	Odense (FA)	Kara (FA)	Aarhus (FA) Dupl.
	LS10 S1	LS10 S2	LS10 S3	LS10 S4	LS10 S5	LS10 S6	LS10 S7	LS10 S8
AI	1.48E-01	1.47E-01	6.70E-01	2.23E-01	2.26E+00	4.29E-01	2.32E-01	2.20E+00
\mathbf{As}	8.91E-03	8.18E-03	7.55E-03	7.77E-03	4.70E-02	1.29E-02	7.65E-03	5.17E-02
\mathbf{Ba}	8.97E-03	2.33E-01	8.39E-01	2.37E+00	3.18E-01	2.91E-01	7.44E-01	4.83E-01
Be	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04
Ca	1.76E+03	1.24E+03	3.65E+03	7.45E+03	1.30E+03	1.25E+03	3.83E+03	1.30E+03
Cd	1.42E-02	3.19E-03	1.00E-04	5.68E-03	1.90E-02	9.59E-03	1.37E-03	1.82E-02
Co	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04
Cr	3.57E-01	6.06E-01	1.02E-01	4.50E-01	1.71E+00	7.65E-01	2.39E-01	1.63E+00
Cu	3.03E-02	5.80E-04	4.05E-02	1.44E-01	5.80E-04	5.80E-04	1.21E-01	5.80E-04
Fe	2.80E-04	2.80E-04	2.80E-04	2.80E-04	2.80E-04	2.80E-04	2.80E-04	2.80E-04
K	4.00E+03	3.53E+03	2.04E+03	2.51E+03	6.72E+03	5.19E+03	3.14E+03	7.35E+03
Li	7.21E-01	2.49E-01	2.36E-01	1.70E-01	1.13E+00	6.71E-01	3.80E-01	1.25E+00
Mg	5.00E-02	2.49E-01	3.10E-02	7.91E-02	3.89E-01	2.10E-01	4.53E-02	4.58E-01
Mn	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05
Mo	5.17E-01	4.07E-01	1.22E-01	2.61E-01	1.13E+00	5.92E-01	3.81E-01	1.27E+00
Na	2.89E+03	2.52E+03	2.21E+03	1.84E+03	4.47E+03	3.41E+03	2.33E+03	4.95E+03
Ni	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05
Р	2.54E-01	1.14E-01	1.04E-01	1.96E-01	1.32E-01	9.04E-02	1.10E-01	1.63E-01
Pb	2.17E+01	1.86E-01	3.64E + 01	5.27E+01	9.84E-02	9.15E-01	5.15E+01	9.70E-02
S	8.72E+02	8.31E+02	4.40E+02	2.97E+02	1.12E+03	1.14E+03	3.87E+02	1.15E+03
\mathbf{Sb}	8.56E-03	4.05E-03	7.71E-03	8.51E-03	7.33E-02	2.01E-02	8.51E-03	8.33E-02
Se	3.71E-02	3.33E-02	2.36E-02	4.62E-02	1.91E-01	6.38E-02	5.11E-02	2.12E-01
Si	1.89E-01	1.16E+00	1.38E-02	6.93E-02	1.46E+00	4.05E-01	1.40E-02	1.49E+00
Sn	4.38E-03	2.70E-04	2.70E-04	4.66E-03	2.84E-03	2.70E-04	2.70E-04	4.34E-03
Sr	4.46E+00	6.62E+00	1.27E+01	7.41E+00	6.42E+00	6.11E+00	7.02E+00	6.46E + 00
Ti	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05
v	1.57E-03	3.74E-03	2.90E-03	3.01E-03	5.74E-03	1.84E-03	2.01E-03	6.04E-03
Zn	2.36E+00	2.62E-01	7.69E-01	5.00E+00	1.28E-01	4.95E-01	7.31E-01	1.35E-01
LS [L/kg]	9.82	10.02	10.01	10.06	10.01	10.03	10.36	9.11
[-] Hq	12.08	11.05	12.15	11.96	10.67	11.08	12.15	10.63
c [mS/cm]	33.4	25.9	30.2	44.2	43.2	34	37.4	46
CI	9603	8595	9658	18179	14862	10430	12735	16208
DOC	0.3	2.2	3.8	1.3	0.1	0.3	26.8	0.2

nish bottom ashes obtained in CEN 14405 column tests. Leaching expressed in mg/L. Values	
Table VII-8. Raw leaching data from different Danish bo	indicated in italic reflect detection limits.

indicated in it	alic reflect dete	ction limits.							
KARA (BA)	C1-S1	C1-S2	C1-S3	C1-S4	C1-S5	C1-S6	C1-S7	C1-S8	C1-S9
Al	0.7394	1.36275	2.1587	3.05809	4.1677	4.82269	5.02677	4.93194	4.79328
$A_{\rm S}$	0.00674	0.00692	0.0004	0.0004	0.00489	0.00485	0.0004	0.0004	0.0004
Ba	0.04587	0.05612	0.05098	0.0396	0.04876	0.01945	0.0233	0.01988	0.02096
Be	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04
Ca	365.5989	444.8793	231.7298	144.5619	99.35198	74.54667	70.02644	60.74382	54.5463
Cd	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04
Co	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04
Cr	0.01666	0.01826	0.01499	0.01295	0.0127	0.01217	0.01229	0.01166	0.01036
Cu	1.04421	1.27337	0.64785	0.31324	0.19686	0.11123	0.08186	0.05312	0.04132
Fe	2.70E-04	2.70E-04	2.70E-04	2.70E-04	2.70E-04	2.70E-04	2.70E-04	2.70E-04	2.70E-04
К	352.2596	446.2251	239.5325	128.8043	73.94989	41.10255	29.7282	18.1255	13.28347
Li	0.00823	0.01399	0.01364	0.00846	0.00567	0.00409	0.00252	0.00096	0.00009
Mg	0.1361	0.1667	0.08631	0.05276	0.03269	0.02769	0.02214	0.01987	0.01706
Mn	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05
Mo	0.86713	1.05694	0.52868	0.28498	0.18707	0.11454	0.08807	0.06103	0.04675
Na	1202.249	1577.644	700.35	301.0309	172.7418	59.2519	31.99986	19.55715	11.454
Ni	0.00424	0.00458	8.00E-05						
Ρ	0.0758	0.10095	0.00958	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027
Pb	0.00361	0.00533	0.00588	0.00182	0.00409	0.00116	0.00335	2.00E-04	0.00581
S	453.1789	563.9011	324.0156	174.0742	116.6051	74.15961	54.05805	34.04923	25.08063
Sb	0.00575	0.00573	0.00931	0.01107	0.01187	0.01614	0.01378	0.01212	0.0157
Se	0.14316	0.1201	0.06825	0.05704	0.02188	0.00135	0.01786	0.02354	0.00135
Si	3.2877	4.51575	6.10885	5.67438	4.99187	4.9555	4.65268	4.60294	4.50579
Sn	0.01199	0.00412	0.008	0.00027	0.00388	0.00621	0.00478	0.00027	0.00482
Sr	1.56785	1.91069	0.9537	0.57892	0.44999	0.30137	0.27301	0.26828	0.24084
Ti	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05
Λ	0.01115	0.01561	0.01903	0.0176	0.01583	0.0143	0.01325	0.01181	0.01113
Zn	7.00E-05	7.00E-05	7.00E-05	0.00663	0.00246	0.13596	0.0485	0.03129	0.02578
LS [L/kg]	0.1	0.24	0.58	1.04	2.06	3.47	4.86	7.45	9.85
[-] Hq	10.6	10.4	10.63	10.58	10.52	10.34	10.42	10.54	10.4
c [mS/cm]	11.3	10.33	4.25	2.25	1.29	0.74	0.597	0.452	0.4
CI	1725.6	2098.3	978.2	403.4	139.7	20.5	9.9	5.5	5.535
DOC	70.775	82.188	60.726	28.437	16.378	8.716	6.766	4.514	4.069

VF (BA)	C2-S1	C2-S2	C2-S3	C2-S4	C2-S5	C2-S6	C2-S7	C2-S8	C2-S9
AI	52.18394	54.60409	116.051	118.2116	59.94591	37.46442	32.4057	25.06284	21.70185
\mathbf{As}	0.0186	0.0004	0.0004	0.00665	0.0004	0.0004	0.0004	0.00436	0.0004
Ba	0.04889	0.01617	0.01287	0.00283	0.01208	0.0077	0.01019	0.00549	0.00511
Be	5.50E-04								
Ca	89.25114	86.11063	50.59206	14.77722	27.20501	51.72954	61.77832	64.66293	64.90796
Cd	1.00E-04								
Co	2.20E-04								
Cr	0.18616	0.19366	0.24313	0.10392	0.03135	0.02223	0.01854	0.01317	0.01172
Cu	1.55253	1.59948	1.5711	0.50751	0.16055	0.07118	0.06242	0.04193	0.03498
Fe	2.70E-04								
K	281.3648	295.2246	233.0488	86.42543	52.68639	35.69509	27.39443	18.38049	15.3447
Li	0.00009	0.00009	0.00009	0.00009	0.0000	0.00009	0.0000	0.0000	0.00009
Mg	0.00027	0.00027	0.00027	0.00027	0.00027	0.00525	0.00027	0.0064I	0.00027
Mn	7.00E-05								
Mo	0.84637	0.89058	0.97715	0.27478	0.07428	0.04277	0.03152	0.02114	0.01686
Na	1560.001	1641.642	1185.754	309.211	135.789	40.71384	23.36858	15.323	10.27809
Ni	0.00218	0.00253	0.00232	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05
Р	1.43206	0.1864	0.24189	0.05525	0.00027	0.00027	0.00027	0.00027	0.00027
Pb	0.03514	0.00764	0.01394	0.00992	0.00993	0.00704	0.00691	0.00221	0.00284
S	159.8906	167.3342	200.8363	84.27351	29.14246	22.14479	21.42365	18.5781	17.51892
Sb	0.01919	0.02415	0.0382	0.04569	0.02667	0.02227	0.01918	0.01838	0.01957
Se	0.07674	0.04845	0.08064	0.04052	0.01471	0.02689	0.02066	0.00135	0.00135
Si	0.89605	0.89054	1.29188	1.55522	1.31967	0.97828	1.00302	1.09374	1.2696
Sn	0.00027	0.00027	0.00027	0.00645	0.00701	0.00027	0.00326	0.00027	0.00027
Sr	0.11716	0.11905	0.06953	0.01913	0.04146	0.08401	0.10158	0.10305	0.10399
Ti	5.00E-05								
Λ	0.00446	0.00445	0.00929	0.01139	0.00617	0.00426	0.00399	0.00387	0.00379
Zn	0.02346	0.01031	0.02972	0.04014	0.02128	0.11001	0.02114	0.11884	0.0126
LS [L/kg]	0.13	0.28	0.63	1.1	2.13	3.53	4.92	7.52	10.02
[-] Hq	11.21	11.11	11.27	11.13	10.86	10.55	10.53	10.62	10.52
c [mS/cm]	10.76	10.6	6.35	2.15	1.01	0.63	0.576	0.505	0.484
CI	2452.9	2533.7	1538.6	132.3	11.8	11.8	ı	ı	ı
DOC	84.144	86.258	70.229	19.825	7.518	4.822	4.195	2.878	3.187

NF (BA)	C3-S1	C3-S2	C3-S3	C3-S4	C3-S5	C3-S6	C3-S7	C3-S8	C3-S9
Al	29.63286	32.02439	67.29694	50.51087	3.9081	27.31062	20.5691	15.62171	13.35506
\mathbf{As}	0.0081	0.0004	0.0004	0.00491	0.0004	0.0004	0.0004	0.0004	0.0004
Ba	0.01003	0.00925	0.00547	0.00027	0.02285	0.0111	0.00414	0.00346	0.00377
Be	5.50E-04								
Ca	63.48091	64.80472	27.33345	18.98434	453.4785	38.94226	46.66599	49.39505	51.2814
Cd	1.00E-04								
Co	2.20E-04								
Cr	0.214	0.2261	0.16257	0.065	1.00E-04	0.02296	0.01655	0.01287	0.01089
Cu	1.12494	1.2007	0.76204	0.25918	0.13248	0.06474	0.04281	0.02272	0.01638
Fe	2.70E-04								
K	288.1251	290.6512	192.0727	92.08999	34.68568	37.73905	26.56477	18.3684	15.02591
Li	0.00009	0.00009	0.0000	0.00009	0.15075	0.0000	0.0000	0.00009	0.00009
Mg	0.00027	0.00027	0.00027	0.00027	0.9933	0.00582	0.00796	0.00467	0.00027
Mn	7.00E-05	7.00E-05	7.00E-05	7.00E-05	0.00339	7.00E-05	7.00E-05	7.00E-05	7.00E-05
Mo	0.68468	0.73618	0.41639	0.13694	0.03822	0.02267	0.01543	0.01189	0.00928
Na	1398.765	1404.388	753.8094	262.7055	188.1946	66.81817	35.02582	24.74465	15.33893
Ni	0.00149	0.00119	8.00E-05						
Ρ	0.17797	0.19068	0.09536	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027
Pb	0.00758	0.00381	0.0096	0.00938	0.00221	0.00287	0.00487	0.00579	0.00176
S	246.6186	247.9519	200.9798	89.44139	433.7441	32.98371	25.36746	19.34042	17.95148
Sb	0.02007	0.02545	0.03676	0.03661	0.02606	0.02259	0.01911	0.01626	0.01842
Se	0.05401	0.05896	0.05567	0.03149	0.00135	0.01817	0.00135	0.01527	0.00135
Si	1.0527	1.17072	1.80654	2.18779	0.38745	1.75787	1.67839	1.79265	1.96142
Sn	0.00794	0.00027	0.00027	0.0044	0.00403	0.0028	0.00027	0.00027	0.00371
Sr	0.18536	0.19827	0.09247	0.06975	0.74081	0.1581	0.20693	0.25527	0.28593
Ti	5.00E-05								
Λ	0.00837	0.00982	0.01796	0.01352	0.00321	0.00678	0.00565	0.00576	0.00591
Zn	0.01317	0.00947	0.02235	0.02897	0.30055	0.18516	0.04634	0.17549	0.01007
LS [L/kg]	0.12	0.27	0.62	1.09	2.13	3.56	4.96	7.56	10.02
[-] Hq	11.19	11.18	11.27	11	10.85	10.64	10.57	10.59	10.32
c [mS/cm]	9.73	9.24	4.4	1.95	1.21	0.728	0.581	0.476	0.438
CI	1959.3	2113.5	881.9	185.7	45.8	8.5	6.2	5.2	ı
DOC	68.956	72.321	40.166	14.564	8.646	4.482	3.39	2.857	2.566

AI 19E+00 24E+00 3.5E+00 3.5E+00 3.5E+00 3.5E+00 1.0E+00 1.0E Ba 3.5E+02 3.5E+00 3.5E+00 3.5E+00 3.5E+00 4.0E 4.0E Ba 3.5E+02 3.5E+02 3.5E+02 3.5E+02 3.5E+02 3.5E+02 3.5E+01 3.5E+01 <th>Aa (BA-ovn3)</th> <th>C4-S1</th> <th>C4-S2</th> <th>C4-S3</th> <th>C4-S4</th> <th>C4-S5</th> <th>C4-S6</th> <th>C4-S7</th> <th>C4-S8</th> <th>C4-S9</th>	Aa (BA-ovn3)	C4-S1	C4-S2	C4-S3	C4-S4	C4-S5	C4-S6	C4-S7	C4-S8	C4-S9
As 5.1E-03 1.5E-02 8.6E-03 5.7E-03 2.7E-04 4.0E-04 4.0E-01 4.0E-04 4.	AI	1.9E+00	2.4E+00	3.5E+00	3.5E+00	4.0E+01	7.2E+00	1.0E+01	1.1E+01	1.1E+01
Bat 3.32E-02 5.3E-04	As	5.1E-03	1.5E-02	8.6E-03	5.7E-03	4.0E-04	4.0E-04	4.0E-04	4.0E-04	4.0E-04
Be 5.5E-04 5.5	Ba	3.2E-02	4.7E-02	3.3E-02	2.5E-02	2.7E-04	2.5E-02	3.8E-02	2.0E-02	1.9E-02
Ca 5.2E+02 7.5E+02 7.5E+02 7.5E+02 7.5E+02 0.5E+01 2.1E+02 1.1E+02 1.1E+02 1.1E+02 1.1E+02 1.1E+02 1.1E+02 1.1E+01 1.0E-04 1.	Be	5.5E-04	5.5E-04							
Cd 1.0E-04 1.	Ca	5.2E+02	7.5E+02	7.5E+02	6.5E+02	2.6E+01	2.1E+02	1.5E+02	9.8E + 01	7.5E+01
Co $22E04$ 2	Cd	1.0E-04	1.0E-04							
Cr 1.0E-04 1.	Co	2.2E-04	2.2E-04							
Cu 3.3E-01 5.0E-01 4.2E-01 2.4E-01 1.7E-01 7.6E-02 7.2E-04 2.7E-04 2.	Cr	1.0E-04	1.0E-04	1.0E-04	1.0E-04	4.4E-02	1.0E-04	1.0E-04	1.0E-04	1.0E-04
Fe $2.7E-04$ 2.7	Cu	3.3E-01	5.0E-01	4.2E-01	2.4E-01	1.7E-01	7.6E-02	7.2E-02	2.5E-02	2.0E-02
K $1.7E+02$ $2.3E+02$ $1.9E+02$ $7.5E+01$ $6.3E+01$ $1.8E+01$ $1.5E+01$ Mi $1.2E-01$ $1.6E-01$ $2.1E-01$ $1.9E-01$ $9.6E-02$ $7.9E-02$ $7.9E-02$ $7.9E-03$ $8.9E-05$	Fe	2.7E-04	2.7E-04							
Li 1.2E-01 1.6E-01 2.1E-01 1.9E-01 1.9E-01 9.6E-05 9.6E-02 7.9E-03 Mg 2.2E+00 3.2E+00 3.2E+00 3.2E+00 3.9E-01 2.4E-0 Mn 5.0E-03 8.0E-03 7.6E-03 7.6E-03 7.0E-05 7.0E-05 7.0E-03 Mo 1.1E-01 1.7E-01 1.7E-01 1.5E-03 7.0E-05 7.0E-05 7.0E-05 7.0E-05 Na 2.0E+03 2.2E+03 1.9E+03 4.6E+02 7.2E-03 2.9E+01 2.9E+01 2.9E+01 Ni 2.8E+03 4.5E+03 1.6E-01 1.1E-01 1.1E-02 2.3E+02 3.9E-01 2.9E+03 Ni 2.8E+03 9.0E-04 9.0E-05 9.0E-05 8.0E-05 8.0E-05 8.0E-05 So 1.8E+02 3.1E+02 3.1E+02 7.3E+02 7.3E+02 2.3E+02 1.7E+02 So 1.8E+02 3.6E+01 2.7E+04 2.7E+02 2.3E+02 1.2E+02 2.3E+02 So 1.	K	1.7E+02	2.3E+02	1.9E+02	7.5E+01	6.3E+01	1.8E+01	1.5E+01	1.1E+01	9.7E+00
Mg $2.2E+00$ $3.2E+00$ $3.0E+00$ $3.0E+00$ $3.0E-03$ 3.0	Li	1.2E-01	1.6E-01	2.1E-01	1.9E-01	9.0E-05	9.6E-02	7.9E-02	5.7E-02	4.9E-02
Min 5.0E-03 8.0E-03 7.6E-03 5.2E-03 7.0E-05 7	Mg	2.2E+00	3.2E+00	3.0E+00	2.2E+00	6.0E-03	3.9E-01	2.4E-01	1.3E-01	9.1E-02
Mo 1.1E-01 1.7E-01 1.5E-01 6.8E-02 7.2E-02 2.2E-02 1.7F-01 Na 2.0E+03 2.2E+03 1.9E+03 4.6E+02 1.9E+02 5.6E+01 2.9E+03 P 1.8E-01 2.2E+03 1.9E+03 4.6E+02 1.9E+02 5.6E+01 2.9E+03 P 1.8E-01 2.2E+03 1.9E+03 4.6E+02 1.9E+02 5.6E+01 2.9E+03 P 1.8E-01 2.2E+03 1.6E-01 1.1E-01 7.7E-02 7.3E-02 8.0E-05 8.0E-05 <td< td=""><td>Mn</td><td>5.0E-03</td><td>8.0E-03</td><td>7.6E-03</td><td>5.2E-03</td><td>7.0E-05</td><td>7.0E-05</td><td>7.0E-05</td><td>7.0E-05</td><td>7.0E-05</td></td<>	Mn	5.0E-03	8.0E-03	7.6E-03	5.2E-03	7.0E-05	7.0E-05	7.0E-05	7.0E-05	7.0E-05
Na $2.0E+03$ $2.2E+03$ $1.9E+03$ $4.6E+02$ $1.9E+02$ $5.6E+01$ $2.9E+$ P $1.8E-01$ $2.2E-01$ $1.6E-01$ $1.1E-01$ $7.7E-02$ $7.3E-05$ $8.0E-05$	Mo	1.1E-01	1.7E-01	1.5E-01	6.8E-02	7.2E-02	2.2E-02	1.7E-02	1.3E-02	1.1E-02
Ni $2.8E-03$ $4.5E-03$ $4.2E-03$ $4.2E-03$ $8.0E-05$ $5.0E-05$ 5	Na	2.0E+03	2.2E+03	1.9E+03	4.6E+02	1.9E+02	5.6E+01	2.9E+01	1.9E+01	1.2E+01
P 1.8E-01 2.2E-01 1.6E-01 1.1E-01 7.7E-02 7.3E-02 8.0E-05 Pb 1.8E-03 9.0E-04 9.0E-05 5.0E-05	Ni	2.8E-03	4.5E-03	4.2E-03	8.0E-05	8.0E-05	8.0E-05	8.0E-05	8.0E-05	8.0E-05
Pb 1.8E-03 9.0E-04 $9.0E-05$ $9.0E-02$ $2.3E-02$ $1.4E-03$	Ρ	1.8E-01	2.2E-01	1.6E-01	1.1E-01	7.7E-02	7.3E-02	8.0E-02	6.4E-02	5.7E-02
S $4.3E+02$ $6.4E+02$ $7.5E+02$ $5.8E+02$ $6.3E+01$ $1.8E+02$ $1.2E+02$ $2.3E+01$ $3.3E+01$ $3.3E+01$ $3.3E+01$ $3.3E+02$ $3.3E$	Pb	1.8E-03	9.0E-04	9.0E-05	9.0E-05	1.2E-03	9.0E-05	9.0E-05	9.0E-05	9.0E-05
Sb 1.8E-02 $3.1E-02$ $3.6E-02$ $2.9E-02$ $2.4E-02$ $1.6E-02$ $2.3E-02$ Se $3.0E-02$ $4.2E-02$ $1.4E-03$ </td <td>S</td> <td>4.3E+02</td> <td>6.4E+02</td> <td>7.5E+02</td> <td>5.8E+02</td> <td>6.3E+01</td> <td>1.8E+02</td> <td>1.2E+02</td> <td>7.2E+01</td> <td>5.1E+01</td>	S	4.3E+02	6.4E+02	7.5E+02	5.8E+02	6.3E+01	1.8E+02	1.2E+02	7.2E+01	5.1E+01
Se $3.0E-02$ $4.2E-02$ $2.1E-02$ $1.4E-03$ $3.7E-04$ $2.7E-04$ 2	Sb	1.8E-02	3.1E-02	3.6E-02	2.9E-02	2.4E-02	1.6E-02	2.3E-02	2.1E-02	2.4E-02
Si $3.4E-01$ $4.9E-01$ $5.6E-01$ $4.1E-01$ $2.0E+00$ $3.3E-01$ $3.7E-04$ Sn $2.7E-04$ $4.1E-03$ $2.7E-04$ $2.7E-04$ $2.7E-04$ $2.7E-04$ $2.7E-04$ Sr $1.5E+00$ $2.3E+00$ $2.0E+00$ $1.4E+00$ $1.0E-01$ $4.0E-01$ $2.9E-05$ Ti $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ Ti $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ V $2.3E-03$ $3.3E-03$ $3.3E-03$ $3.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ Zn $7.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ Zn $7.0E-05$ $3.3E-03$ $3.3E-03$ $3.3E-03$ $3.2E-03$ $3.3E-03$ $3.2E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ $5.0E-05$ Zn $7.0E-05$ $3.8E-03$ $3.3E-03$ $3.2E-05$ $5.0E-05$ $5.0E$	Se	3.0E-02	4.2E-02	2.1E-02	1.4E-03	1.4E-03	1.4E-03	1.4E-03	1.4E-03	1.4E-03
Sn $2.7E-04$ $4.1E-03$ $2.7E-04$ 2.7	Si	3.4E-01	4.9E-01	5.6E-01	4.1E-01	2.0E+00	3.3E-01	3.7E-01	4.0E-01	4.9E-01
Sr 1.5E+00 2.3E+00 2.0E+00 1.4E+00 1.0E-01 4.0E-01 2.9E-03 Ti $5.0E-05$ <t< td=""><td>Sn</td><td>2.7E-04</td><td>4.1E-03</td><td>2.7E-04</td><td>2.7E-04</td><td>2.7E-04</td><td>2.7E-04</td><td>2.7E-04</td><td>2.7E-04</td><td>2.7E-04</td></t<>	Sn	2.7E-04	4.1E-03	2.7E-04	2.7E-04	2.7E-04	2.7E-04	2.7E-04	2.7E-04	2.7E-04
Ti $5.0E-05$	Sr	1.5E+00	2.3E+00	2.0E+00	1.4E+00	1.0E-01	4.0E-01	2.9E-01	2.4E-01	1.8E-01
V $2.3E-03$ $2.9E-03$ $3.3E-03$ $3.0E-03$ $1.0E-02$ $2.7E-03$ $2.5E-03$ $2.4E-02$ $3.4E-02$ $3.4E$	Ti	5.0E-05	5.0E-05							
Zn7.0E-053.8E-037.0E-051.6E-013.4E-023.4E-02LS [L/kg] 0.08 0.22 0.56 1.01 2.02 3.4 4.77 pH [-] 9.35 9.12 9.3 8.79 8.5 8.87 8.88 c [mS/cm] 17.88 17.8 10.23 4.41 2.16 1.03 $0.72t$ Cl 3051.7 4515.8 2532.3 619.1 92.5 13.2 6.8 PAC 27.45 10700 5.174 2.025 $3.03t$	Λ	2.3E-03	2.9E-03	3.3E-03	3.0E-03	1.0E-02	2.7E-03	2.5E-03	2.4E-03	2.5E-03
LS [L/kg] 0.08 0.22 0.56 1.01 2.02 3.4 4.77 pH[-] 9.35 9.12 9.3 8.79 8.5 8.87 8.88 c [mS/cm] 17.88 17.8 10.23 4.41 2.16 1.03 0.724 CI 3.051.7 4515.8 2532.3 619.1 92.5 13.2 6.8 PMC 27.157 3.7157 3.8 27107 10.700 5.174 7.035 3.034	Zn	7.0E-05	3.8E-03	7.0E-05	7.0E-05	1.6E-01	3.4E-02	3.4E-02	3.0E-02	6.8E-03
pH[-] 9.35 9.12 9.3 8.79 8.5 8.87 8.88 c [mS/cm] 17.88 17.8 10.23 4.41 2.16 1.03 0.720 CI 3.051.7 4515.8 2532.3 619.1 92.5 13.2 6.8 DOC 27.157 3.7157 3.8 7.107 107 5.170 2.35 3.030	LS [L/kg]	0.08	0.22	0.56	1.01	2.02	3.4	4.77	7.3	9.64
c [mS/cm] 17.88 17.8 10.23 4.41 2.16 1.03 0.726 Cl 3051.7 4515.8 2532.3 619.1 92.5 13.2 6.8 DOC 27.157 38 24.107 10.700 5.174 2.035 3.034	[-] Hq	9.35	9.12	9.3	8.79	8.5	8.87	8.88	9.06	8.98
Cl 3051.7 4515.8 2532.3 619.1 92.5 13.2 6.8 DOC 27.152 38 24.107 10.700 5.174 2.035 3.034	c [mS/cm]	17.88	17.8	10.23	4.41	2.16	1.03	0.726	0.488	0.372
DAG 27 153 28 24 10700 5174 2025 37034	CI	3051.7	4515.8	2532.3	619.1	92.5	13.2	6.8	6.4	I
UUC 2/1.12 JO 24.17/ 10.177 J.1.14 2.7.1 JUUC	DOC	27.152	38	24.197	10.799	5.174	2.935	3.036	2.399	2.323

Aa (BA-ovn2)	C5-S1	C5-S2	C5-S3	C5-S4	C5-S5	C5-S6	C5-S7	C5-S8	C5-S9
Al	7.71576	6.10729	11.18449	18.95443	21.64917	23.49007	23.83395	22.77618	21.73415
As	0.00585	0.00625	0.0004	0.0004	0.00454	0.0004	0.0004	0.0004	0.0004
Ba	0.05586	0.04395	0.03107	0.02475	0.02995	0.07365	0.06117	0.04116	0.04292
Be	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04	5.50E-04
Ca	478.8422	392.9027	237.6397	96.33159	74.47054	74.94353	71.89002	62.01084	58.5088
Cd	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04	1.00E-04
C_0	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04	2.20E-04
Cr	0.03938	0.0281	0.02304	0.00692	0.00398	0.0067	0.00953	0.0089	0.008
Cu	0.61981	0.49469	0.40639	0.19191	0.09511	0.1252	0.06099	0.0399	0.031
Fe	0.00027	0.00027	0.00027	0.00027	0.00027	0.12591	0.00027	0.00027	0.00027
K	259.7314	222.9158	159.2837	57.26992	30.62681	26.31322	18.23674	15.81949	14.8755
Li	0.00174	0.00009	0.00009	0.00009	0.0000	0.0000	0.00009	0.00009	0.0000
Mg	0.21481	0.1576	0.0925	0.03178	0.01778	0.02385	0.01746	0.01155	0.00568
Mn	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05	7.00E-05
Mo	0.58082	0.4709	0.3995	0.1473	0.04738	0.02716	0.01959	0.01473	0.01176
Na	1771.786	1374.915	852.0794	211.3245	74.19478	26.9867	19.99106	10.20948	6.55019
Ni	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05	8.00E-05
Ρ	0.24957	0.15378	0.10753	0.05609	0.04598	0.04523	0.03728	0.03582	0.02034
Pb	0.00009	0.00009	0.00009	0.00009	0.0000	0.00197	0.00009	0.00009	0.00009
S	479.9063	392.3902	366.5054	182.3561	86.51232	55.80879	46.50207	31.82519	27.35056
Sb	0.01024	0.00785	0.01326	0.01328	0.01237	0.01378	0.01243	0.01404	0.02226
Se	0.02589	0.03702	0.02662	0.00135	0.00135	0.00135	0.00135	0.00135	0.00135
Si	0.99608	0.76821	0.88432	0.75289	0.65873	0.61133	0.63073	0.71339	0.85783
Sn	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027	0.00027
Sr	1.13805	0.92253	0.54493	0.28313	0.23537	0.24002	0.24049	0.22392	0.21609
Ti	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05	5.00E-05
Λ	0.0071	0.00576	0.0081	0.00833	0.00643	0.00553	0.00567	0.00532	0.0056
Zn	0.0008	0.00079	7.00E-05	7.00E-05	7.00E-05	0.06857	0.09759	0.02536	0.00987
LS [L/kg]	0.16	0.3	0.64	1.11	2.13	3.53	4.92	7.49	9.85
[-] Hq	9.84	9.7	10.1	10.11	9.98	9.83	9.87	9.98	9.89
c [mS/cm]	9.61	9.6	5.23	1.62	0.69	0.47	0.424	0.359	0.333
CI	2798.4	2189.5	1147.7	115.8	11.9	11.9	I	ı	ı
DOC	35.526	33.177	24.797	9.744	4.592	3.628	3.429	2.982	2.791

Table VII-9. Raw leaching data from different Danish fly ash/APC residues obtained in CEN 14405 column tests. Leaching expressed in mg/L. Values indicated in italic reflect detection limits.

Aarhus (FA)												
mg/L	C1-S1	C1-S2	C1-S3	C1-S4	C1-S5	C1-S6	C1-S7	C1-S8	C1-S9	C1-S10	C1-S11	C1-S12
AI	0.274535	0.0003	0.0003	0.103021	0.088347	0.053656	0.034016	0.035167	0.033417	0.034471	0.035768	0.028511
$\mathbf{A}_{\mathbf{S}}$	0.05127	0.09494	0.0004	0.027368	0.013916	0.0004	0.008103	0.007548	0.008983	0.00654	0.0004	0.00768
Ba	10.70868	5.855589	3.796037	3.894054	2.497295	1.402275	0.966662	0.942235	0.849392	0.828741	0.835289	0.735582
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	2245.431	3561.727	3154.401	1423.465	845.9958	796.7089	852.2352	865.1537	854.5496	880.7329	790.2519	782.9487
Cd	0.122736	0.2433	0.060906	0.003166	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Co	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.011839	0.0002
Cr	0.017122	0.01955	0.010019	0.008438	0.004107	0.001976	0.002364	0.001616	0.001675	0.001962	0.008614	0.001528
Cu	0.0006	0.0006	0.0006	0.0006	0.0006	0.027875	0.0006	0.0006	0.0006	0.0006	0.011175	0.0006
Fe	1.60924	0.0003	0.0003	0.398475	0.018448	0.046492	0.022481	0.020773	0.022105	0.029708	0.125481	0.026329
K	28441.23	56189.77	45811.6	24135.29	7428.239	1122.977	94.48516	23.44906	16.55786	12.61954	9.45268	8.58043
Li	2.66728	4.10405	3.46297	1.86802	0.87232	0.356362	0.167814	0.129717	0.10216	0.080958	0.07307	0.076796
Mg	0.345614	0.0003	0.0003	0.025811	0.018545	0.030768	0.01477	0.012193	0.014859	0.021643	0.030577	0.043144
Mn	0.044764	0.006967	0.0001	0.01332	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Mo	1.350303	2.98687	3.258154	2.601859	0.976784	0.290888	0.144038	0.096988	0.075131	0.057869	0.046669	0.040798
Na	27151.43	52844.8	37342.27	20483.07	3420.402	335.1666	44.78441	15.03045	8.442512	5.496361	3.797036	3.420918
N:	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Ρ	0.454163	0.840751	0.259379	0.226467	0.039835	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014
Pb	0.82082	1.55272	0.8991	0.7726	0.45948	0.31221	0.292624	0.451218	0.459762	0.361656	0.233771	0.192632
S	273.6229	922.091	1722.207	3445.965	1494.731	680.9294	510.9445	500.6567	501.1247	541.9797	500.5068	502.7322
\mathbf{Sb}	0.03905	0.04698	0.06281	0.01337	0.006594	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Se	1.02059	1.35639	1.39733	0.84987	0.48283	0.14318	0.092262	0.074242	0.056907	0.066979	0.040737	0.063192
Si	0.75126	0.58505	0.72446	1.20137	0.92905	0.40096	0.265248	0.144206	0.104098	0.077998	0.056734	0.065588
\mathbf{Sn}	0.12279	0.12669	0.09035	0.05285	0.00519	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
Sr	8.96323	16.0788	14.12928	9.94331	5.69485	4.32342	3.51008	2.85992	2.50146	2.21126	1.92549	1.94331
Ti	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
^	0.01857	0.01457	0.01568	0.00484	0.00455	0.002765	0.002896	0.00276	0.003008	0.003113	0.004423	0.003062
Zn	0.15414	0.55615	1.18674	1.52046	1.58098	0.833218	0.838836	0.955363	1.32537	1.442871	1.245925	1.416175
DOC	12.147	12.535	7.518	5.943	3.363	1.943	1.185	1.032	1.714	1.689	2.044	2.879
CI	84000	173500	123820	45910	7918	938	92.2	27.5	25.7	10	10	IO
[-] Hd	11.12	11.23	11.58	11.68	11.74	11.57	11.46	11.61	11.61	11.50	11.50	11.30
LS [L/kg]	0.10	0.27	0.44	0.86	1.23	1.85	3.20	5.27	7.59	10.70	13.46	16.18

Vestforbrær ma/I	nding (FA) C2-S1	C2-C2	53-CJ	C7.54	C7-55	77.SK	C2-S7	C7.58	C7_50	C2-S10	C2-S11	C7-S12
Al	0.0003	0.047955	0.0003	0.80925	0.45647	0.375633	0.217852	0.119638	0.089519	0.065213	0.053345	0.088349
\mathbf{As}	0.047474	0.034752	0.0004	0.019125	0.008446	0.004641	0.004851	0.0004	0.0004	0.0004	0.0004	0.005423
Ba	3.904136	2.473766	2.799065	3.168534	2.58652	0.624123	0.580963	0.53613	0.558272	0.702832	0.533468	0.479112
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	795.3121	2113.869	2541.634	1842.545	1054.215	805.0188	755.4229	745.5506	751.9051	754.1114	743.7279	736.5637
Cd	0.659592	1.906476	1.202374	0.072102	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Co	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.004279
Cr	1.642237	3.36726	4.043467	3.563539	1.390803	0.792383	0.671946	0.522703	0.451521	0.368833	0.303959	0.275182
Cu	0.0006	0.0006	0.0006	0.0006	0.010476	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.006715
Fe	0.0003	0.0003	0.0003	0.020691	0.021134	0.022949	0.023608	0.024979	0.029107	0.023887	0.023267	0.048157
K	30783.2	65262.09	68055.56	31516.38	4483.251	413.3488	44.62932	21.5223	17.57719	13.29115	10.44853	10.65261
Li	0.04449	0.18381	0.34511	0.8606	0.86303	0.477883	0.304549	0.208438	0.165989	0.168077	0.128254	0.126012
Mg	0.0003	0.0003	0.0003	0.065072	0.090141	0.064352	0.080487	0.099763	0.111161	0.120391	0.13081	0.13186
Mn	0.017997	0.009083	0.009041	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Mo	2.34679	3.897401	4.44337	3.160038	1.287196	0.695959	0.206576	0.112278	0.080547	0.057967	0.047597	0.043853
Na	40652.47	73225	51737.73	20880.52	2079.946	193.3141	39.42932	18.72147	13.22851	9.225006	7.690669	8.183828
Ni	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Р	0.0014	0.0014	0.0014	0.049567	0.015473	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014
Pb	0.13303	0.64972	0.49857	0.14416	0.04009	0.042017	0.020198	0.018566	0.018141	0.031966	0.037475	0.044274
S	505.8604	884.3904	1035.832	1339.626	650.8582	523.3971	531.4449	527.8207	526.055	527.1148	520.049	513.6312
\mathbf{Sb}	0.08938	0.05591	0.1071	0.013	0.00991	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Se	1.03164	1.09938	1.20009	0.73688	0.43577	0.128493	0.072264	0.070602	0.04922	0.063327	0.042454	0.055074
Si	2.3635	1.84585	1.56722	3.13592	2.00313	1.672587	1.37302	1.416606	1.509174	1.718325	1.896629	2.029403
Sn	0.0731	0.054	0.04991	0.0539	0.01354	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
\mathbf{Sr}	8.95492	19.00274	18.19559	16.95733	11.30175	7.70446	4.78886	3.10015	2.56469	1.83622	1.99682	1.99834
Ti	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
^	0.0002	0.0002	0.0002	0.0018	0.00298	0.00197	0.002478	0.003059	0.004217	0.004594	0.004736	0.004796
Zn	0.15176	0.18506	0.14663	0.06463	0.03104	0.00776	0.0003	0.004735	0.005622	0.0059	0.009696	0.018658
DOC	18.413	20.672	17.892	10.118	4.48	3.177	1.96	1.439	1.762	1.6634	0.458	1.615
CI	112390	230000	193000	57200	6113.6	786	69.7	44.8	29.4	30.8	24	29
[-] Hq	8.92	10.58	10.76	11.12	11.10	10.98	10.84	11.08	11.14	11.11	11.16	10.88
LS [L/kg]	0.07	0.20	0.35	0.68	0.96	1.44	2.49	4.10	5.90	8.32	10.50	12.67

Amagerforb	rænding (APt	C)										
mg/L	C3-S1	C3-S2	C3-S3	C3-S4	C3-S5	C3-S6	C3-S7	C3-S8	C3-S9	C3-S10	C3-S11	C3-S12
Al	0.42603	0.49132	0.39436	0.01786	0.0003	0.024287	0.031092	0.05066	0.0003	0.0003	0.007389	0.0003
\mathbf{As}	0.0597	0.10271	0.09991	0.04189	0.01726	0.00652	0.00427	0.0004	0.00457	0.00412	0.00466	0.0004
Ba	9.118985	12.82241	13.87141	8.859854	2.983392	1.058942	0.939793	0.788749	0.625444	0.560341	0.546644	0.478337
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	15336.99	26748.19	29740.3	26170.97	8668.483	3001.585	1787.658	1608.977	1537.454	1522.99	1505.249	1509.997
Cd	0.006828	0.019172	0.0001	0.003644	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Co	0.046156	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002
Cr	0.164078	0.21066	0.216526	0.191139	0.110549	0.045456	0.015035	0.002778	0.00168	0.001659	0.0001	0.00114
Cu	0.0006	0.0006	0.0006	0.066116	0.027875	0.009694	0.007349	0.006029	0.005948	0.0006	0.0006	0.0006
Fe	1.324405	0.890927	1.086815	0.100572	0.016207	0.023045	0.030861	0.0003	0.0003	0.0003	0.0003	0.0003
K	28123.32	46152.23	35494.11	14588.28	2971.604	374.272	144.4728	93.92492	81.08266	41.733	41.95912	32.52794
Li	1.82695	2.76514	2.6594	1.88138	1.0027	0.454054	0.247615	0.146816	0.162639	0.11582	0.108299	0.092841
Mg	0.060431	0.190515	0.296172	0.402751	0.061753	0.010898	0.003955	0.0003	0.002974	0.0003	0.0003	0.002989
Mn	0.0001	0.015138	0.018481	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Mo	0.156817	0.259191	0.32852	0.298327	0.182706	0.114688	0.085876	0.06146	0.038113	0.025688	0.023873	0.024518
Na	25458.01	38340.3	28492.08	15369.02	1903.254	197.9345	43.85388	17.22151	8.917762	3.913758	3.862758	3.178941
Ni	0.0001	0.0001	0.0001	0.002841	0.001186	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Р	0.143235	0.0014	0.0014	1.412964	0.033564	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014
Pb	705.4586	1225.307	1345.397	1028.729	190.7799	38.19748	15.11802	11.3232	9.3664	7.837	64.4111	5.7882
S	96.26787	173.4357	193.3035	253.4114	330.3488	361.7341	407.0159	409.6492	405.0829	400.4234	402.8724	401.894
\mathbf{Sb}	0.21702	0.37826	0.47508	0.40747	0.07548	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Se	0.87223	0.55188	0.79746	0.34959	0.29124	0.106964	0.099275	0.047482	0.054458	0.051174	0.032784	0.035403
Si	0.18987	0.79867	0.25758	0.41985	0.13704	0.104986	0.074369	0.0014	0.026997	0.0014	0.0014	0.0014
Sn	0.14366	0.09912	0.14585	0.02894	0.01796	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
\mathbf{Sr}	125.8277	207.2795	168.9215	81.3968	21.19054	8.07105	3.35106	2.2009	1.99018	2.10692	1.68748	1.55183
Ti	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Λ	0.04859	0.04575	0.03795	0.00864	0.00509	0.003007	0.002504	0.002338	0.002822	0.002571	0.002192	0.002668
Zn	2.63336	6.69565	9.99494	12.71761	5.99982	2.839592	2.501677	2.786475	2.951128	2.862912	2.320256	2.643774
DOC	26.993	44.443	41.663	26.371	11.497	7.269	3.335	2.046	1.397	1.048	0.7786	2.792
CI	105200	194920	163260	84060	18130	3290	543	153	68.8	35.4	20.4	19.7
[-] Hq	11.24	11.33	11.41	11.65	11.88	12.03	12.05	12.11	12.17	12.2	12.37	12.00
LS [L/kg]	0.08	0.21	0.35	0.70	1.01	1.50	2.57	4.18	6.07	8.57	10.68	12.69
Aarhus (Al	PC)											
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mg/L	C4-S1	C4-S2	C4-S3	C4-S4	C4-S5	C4-S6	C4-S7	C4-S8	C4-S9	C4-S10	C4-S11	C4-S12
Al	83.81275	60.60045	35.9691	3.726449	0.0003	0.041603	0.0003	0.0003	0.0003	0.0003	0.125492	0.007119
$\mathbf{A}_{\mathbf{S}}$	0.10361	0.09819	0.1116	0.0228	0.02294	0.005822	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Ba	58.0868	70.66846	55.03336	28.35961	6.628893	0.964423	0.478318	0.414993	0.304586	0.305357	4.736383	0.242574
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	48532.23	70968.83	63590.56	40840.67	12327.46	2597.159	1450.56	1370.514	1332.156	1330.321	1247.432	1064.087
Cd	7.747518	8.316516	5.325048	1.126263	0.057166	0.005338	0.002846	0.001463	0.0001	0.001767	0.001397	0.002057
Co	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002
Cr	0.762475	0.717975	0.531243	0.106674	0.023906	0.010172	0.005099	0.002598	0.001515	0.001035	0.003092	0.0001
Cu	3.60989	3.630902	2.205035	0.301044	0.106211	0.031686	0.019406	0.01447	0.007158	0.011266	0.015124	0.01072
Fe	51.80614	41.51334	26.13124	3.296981	0.0003	0.0003	0.0003	0.005548	0.029783	0.0003	0.524745	0.026086
K	20785.13	32123.64	27186.07	15161.4	4156.222	349.3796	22.26183	15.26744	12.87964	10.65988	0.007	0.007
Li	0.34722	0.61273	0.80047	0.96452	1.02339	0.422316	0.174187	0.145606	0.116845	0.08819	0.060654	0.054233
Mg	1.498897	1.454441	1.10692	0.715611	0.088733	0.015583	0.004965	0.00713	0.0003	0.006381	0.09158	0.010442
Mn	0.223813	0.149762	0.07555	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.034976	0.0001
Mo	0.118915	0.250409	0.332312	0.399726	0.419357	0.407709	0.198485	0.094804	0.076211	0.066073	0.054349	0.041992
Na	22341.23	25213.03	21172.75	16224.26	3101.276	256.6375	16.50227	9.195182	7.732763	4.906322	1.747641	0.007
Ż	0.0001	0.0001	0.0001	0.003876	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Ρ	0.130937	0.130423	0.190365	0.108137	0.038237	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014
Pb	499.2827	547.576	361.9202	147.0014	18.13998	2.090542	1.019292	0.587883	0.411102	0.506265	0.482198	0.412339
S	67.59722	102.2021	155.7144	212.0053	322.3964	396.6193	447.6047	418.3793	410.6177	397.9925	350.4824	356.7184
\mathbf{Sb}	0.10374	0.09389	0.12816	0.07555	0.03078	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Se	0.53504	0.35247	0.54137	0.4426	0.45979	0.18495	0.086856	0.045959	0.052658	0.051392	0.021789	0.03895
Si	1.09715	0.87943	0.36914	0.22873	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.085957	0.0014
\mathbf{Sn}	0.14383	0.15443	0.15733	0.09049	0.02585	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
\mathbf{Sr}	52.62001	80.12356	74.61988	51.51312	16.99698	4.79914	3.59429	2.88856	2.50476	2.10692	1.68748	1.55183
Ti	0.01937	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
^	0.04377	0.04027	0.04198	0.02768	0.00618	0.001794	0.002039	0.002503	0.002469	0.002496	0.002602	0.002714
Zn	202.9247	238.8751	173.9101	112.1973	38.51196	6.92317	3.3161	2.32407	1.05743	2.02697	1.83968	1.59202
DOC	25.269	25.388	28.414	20.109	6.86	3.356	3.464	1.041	1.029	1.502	0.211	0.7464
CI	156430	240116	217531	115900	ı	ı	ı	ı	ı	ı	ı	ı
[-] Hq	10.55	10.62	10.80	11.19	11.83	12.00	12.04	12.03	12.05	12.11	12.16	11.95
LS [L/kg]	0.11	0.30	0.49	0.98	1.38	2.00	3.39	5.48	7.81	10.93	13.89	16.74

Odense (FA												
mg/L	C5-S1	C5-S2	C5-S3	C5-S4	C5-S5	C5-S6	C5-S7	C5-S8	C5-S9	C5-S10	C5-S11	C5-S12
AI	1.22842	0.60779	2.0335	31.48327	40.26382	12.25814	4.863865	1.876452	0.924345	0.509087	0.371396	0.338854
\mathbf{As}	0.064337	0.039793	0.047353	0.048769	0.049806	0.016291	0.011729	0.01246	0.018773	0.019647	0.01616	0.027684
Ba	47.53912	2.560714	2.851877	1.7931	0.994984	0.435878	0.251453	0.308157	0.320296	0.266768	0.257888	0.233315
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	867.8185	1998.786	3133.658	1069.416	586.3426	642.3358	685.1082	676.1734	686.6039	678.3978	673.3376	673.8416
Cd	49.95202	53.77328	11.46049	2.225348	0.012616	0.002229	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Co	0.0002	0.0002	0.0002	0.0002	0.006679	0.0002	0.003246	0.0002	0.0002	0.0002	0.0002	0.0002
Cr	0.325196	0.279866	0.034209	0.096075	0.086826	0.021898	0.009346	0.00312	0.001237	0.001657	0.013833	0.035695
Cu	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Fe	0.0003	0.036046	0.0003	0.0003	0.085252	0.0003	0.027156	0.0003	0.0003	0.0003	0.0003	0.0003
K	29000.18	59019.14	76884.69	45008.87	13962.69	1326.479	54.12999	19.71715	10.51383	6.439125	2.884364	9.917589
Li	3.0604	2.59413	0.68774	0.55918	0.50477	0.688122	0.484465	0.307448	0.22339	0.156921	0.172188	0.16004
Mg	70.8115	68.64971	3.435646	1.454845	0.12137	0.076571	0.074817	0.117982	0.171145	0.229283	0.258897	0.265971
Mn	0.237819	0.10591	0.008835	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Mo	0.69547	3.046465	4.751899	9.980648	3.536725	0.813197	0.280561	0.158691	0.118252	0.093785	0.070236	0.066548
Na	37655.26	80834.72	59951.78	23191.1	1417.871	184.5339	28.79993	12.55379	8.08059	5.787441	4.009126	5.122024
Ni	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Р	0.215737	0.448767	0.460002	0.255994	0.100067	0.0014	0.0014	0.0014	0.0014	0.514225	0.0014	0.0014
Pb	10.55356	10.94411	2.13203	0.07364	0.00887	0.0001	0.005378	0.000932	0.0001	0.0001	0.0001	0.0001
S	333.9031	652.5656	656.4822	3913.284	5677.603	823.9665	461.8747	446.7313	448.4204	441.7253	438.9664	434.579
\mathbf{Sb}	1.31292	3.14779	2.42901	1.09397	0.25788	0.067416	0.020073	0.012264	0.0004	0.005782	0.0004	0.00599
Se	0.9855	1.45043	1.60752	1.39442	0.67423	0.131711	0.114198	0.07576	0.068443	0.048013	0.075438	0.051577
Si	1.52295	3.51178	0.56349	3.93631	0.48961	0.509779	1.18445	2.50539	3.369449	4.146848	4.4332	4.594174
Sn	0.08419	0.11245	0.11759	0.06327	0.0154	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
Sr	6.65253	9.66038	12.05542	3.80232	5.064	4.87065	4.65405	3.07524	2.40449	2.00751	1.78126	1.7933
Ti	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
^	0.0149	0.0002	0.0002	0.00317	0.01108	0.004899	0.006776	0.008953	0.011499	0.012624	0.01353	0.014721
Zn	40.23399	37.78993	1.32515	0.10231	0.02661	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003	0.0003
DOC	2.6412	2.897	1.158	2.989	7.32	0.028	0.601	0.3406	4.597	2.454	2.525	1.16
CI	I	I	I	I	I	I	I	I	I	ı	I	I
[-] Hq	4.8	7.2	8.59	9.79	10.17	10.70	10.62	10.55	10.69	10.69	10.85	10.70
LS [L/kg]	0.09	0.28	0.48	0.97	1.38	2.06	3.61	5.87	8.42	11.86	14.99	17.93

	C6-S12	0.012318	0.008106	0.401663	0.0006	1396.034	0.0001	0.0002	0.001355	0.015963	0.0003	11.62636	0.077805	0.0003	0.0001	0.057946	6.285215	0.0001	0.0014	0.375924	417.4879	0.0004	0.034143	0.041647	0.005736	1.48338	0.0001	0.003885	3.46099	2.201	57.6	12.06	JJ 66
	C6-S11	0.008934	0.006363	0.436901	0.0006	1395.587	0.0001	0.0002	0.0001	0.010269	0.0003	15.3039	0.125135	0.0003	0.0001	0.062368	10.82215	0.0001	0.0014	0.454159	415.4495	0.0004	0.030611	0.014887	0.006387	1.82807	0.0001	0.002836	2.92337	3.462	102.0	12.13	
	C6-S10	0.012756	0.005159	0.463394	0.0006	1391.233	0.0001	0.0002	0.001326	0.00913	0.0003	18.22608	0.118479	0.0003	0.0001	0.06829	7.843266	0.0001	0.0014	0.607145	429.0303	0.0004	0.064896	0.025301	0.005744	1.74641	0.0001	0.002993	2.94696	3.846	107.0	12.1	14.00
	C6-S9	0.0003	0.009072	0.571488	0.0006	1495.028	0.0001	0.0002	0.001774	0.028008	0.0003	15.48371	0.145683	0.0003	0.0001	0.105425	8.107261	0.0001	0.0014	0.779371	450.2135	0.0004	0.070534	0.08422	0.005766	2.11536	0.0001	0.003118	3.36808	3.984	51.8	12.06	0 83
	C6-S8	0.0003	0.0004	0.578703	0.0006	1507.516	0.0001	0.0002	0.001428	0.024182	0.0003	15.1069	0.14919	0.0003	0.0001	0.128453	9.3822	0.0001	0.0014	0.784498	442.3859	0.0004	0.062528	0.0014	0.0003	2.22287	0.0001	0.001929	1.99105	4.583	68.3	12.02	CVL
	C6-S7	0.0003	0.00467	0.60813	0.0006	1600.518	0.0001	0.0002	0.00267	0.063826	0.0003	18.51031	0.175912	0.0003	0.0001	0.261667	13.28187	0.0001	0.0014	1.244698	424.5736	0.0004	0.077093	0.024521	0.0003	2.41172	0.0001	0.002454	3.04194	7.287	111.0	12.03	LVV
	C6-S6	0.0003	0.0004	0.716301	0.0006	1668.585	0.0001	0.0002	0.003443	0.082792	0.0003	86.12913	0.237941	0.003582	0.0001	0.225033	83.99576	0.0001	0.0014	3.212982	323.617	0.0004	0.151982	0.023215	0.0003	3.41909	0.0001	0.002519	5.55817	19.976	626.0	12.12	7 56
	C6-S5	0.0003	0.0004	1.701976	0.0006	3026.472	0.0001	0.0002	0.006099	0.086477	0.0003	829.3007	0.56976	0.003606	0.0001	0.253582	600.2462	0.0001	0.0014	8.034069	305.8555	0.0004	0.37024	0.0014	0.08921	6.68044	0.0001	0.01311	4.77283	44.673	4124.0	12.12	1 60
	C6-S4	0.0003	0.012881	7.332757	0.0006	10704.24	0.011243	0.0002	0.023647	0.530372	0.100108	11037.4	1.53009	0.11041	0.020448	0.400659	10528.55	0.0001	2.447447	80.99246	254.0118	0.03053	0.43279	0.10898	0.01174	22.06697	0.0001	0.01639	4.80627	151	39616.0	12.00	1 18
	C6-S3	1.00936	0.06483	20.92387	0.0006	25662.69	0.179588	0.0002	0.220834	1.515775	3.230256	33225.56	3.01964	0.215358	0.0001	0.468574	25913.87	0.0001	0.0014	350.082	169.9963	0.07683	0.55025	0.0014	0.0984	58.72405	0.0001	0.04755	66.67332	343	130054.0	11.46	0.60
	C6-S2	4.0586	0.11038	32.15697	0.0006	34356.57	0.5422	0.0002	0.481735	3.00514	9.239515	45499.92	3.39236	0.445645	0.007302	0.479381	38254.52	0.0001	0.243417	687.2322	115.1859	0.25391	0.62335	0.18148	0.10956	85.1641	0.0001	0.04013	86.1956	532	196813.0	11.16	036
nding (FA)	C6-S1	5.09603	0.14011	23.15929	0.0006	22436.93	0.606342	0.0002	0.62178	2.593542	15.13971	29478.82	2.24767	0.282238	0.010785	0.221026	24545.05	0.0001	0.342766	609.4585	46.59778	0.19572	0.6228	0.0014	0.0987	55.4382	0.0001	0.03903	80.54898	331.5	117154.0	10.99	0 12
Nordforbrær	mg/L	Al	\mathbf{As}	Ba	Be	Ca	Cd	Co	Cr	Cu	Fe	K	Li	Mg	Mn	Mo	Na	Ni	Ρ	Pb	S	Sb	Se	Si	Sn	Sr	Ti	v	Zn	DOC	CI	[-] Hq	$1 S \Pi / k\sigma$

KARA (FA	(
mg/L	C7-S1	C7-S2	C7-S3	C7-S4	C7-S5	C7-S6	C7-S7	C7-S8	C7-S9	C7-S10	C7-S11	C7-S12
Al	1.76637	1.07301	0.3645	2.88555	2.83127	1.109602	0.431055	0.151336	0.08147	0.073538	0.049307	0.03529
\mathbf{As}	0.0004	0.07577	0.0004	0.028608	0.029369	0.011014	0.005762	0.0004	0.008148	0.006658	0.006599	0.007067
Ba	1.893152	2.211494	2.259597	1.350647	0.0014	0.321405	0.265911	0.293008	0.256311	0.272063	0.266067	0.378946
Be	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006
Ca	1251.243	2264.57	2303.452	1116.012	640.8901	702.7749	761.3362	769.068	780.4855	792.3329	774.3679	767.4794
Cd	0.646459	1.624015	1.030356	0.031111	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Co	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.0002	0.003363	0.0002	0.0002	0.0002	0.0002
Cr	1.976437	1.566421	0.974186	0.235169	0.038163	0.009013	0.002472	0.003126	0.0001	0.001194	0.001519	0.0001
Cu	0.0006	0.0006	0.0006	0.0006	0.006105	0.0006	0.0006	0.0006	0.0006	0.0006	0.0006	0.007104
Fe	0.467505	2.841441	0.0003	0.0003	0.0003	0.0003	0.0003	0.013534	0.0003	0.0003	0.0003	0.003883
K	36361.91	68141.46	63967.93	29204.42	10091.89	1278.4	38.8548	18.26802	12.57197	12.10647	9.12464	10.50978
Li	0.77268	1.81582	2.50004	1.98115	1.20373	0.766798	0.440896	0.259402	0.19488	0.143795	0.167323	0.118522
Mg	1.261366	0.56691	0.0003	0.026587	0.0439	0.088246	0.093283	0.087543	0.087265	0.085573	0.079939	0.109231
Mn	0.027732	0.226738	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Mo	1.821015	3.523372	4.822087	5.410548	2.429059	0.814849	0.288994	0.147655	0.10088	0.078973	0.073385	0.056024
Na	46052.97	70175.01	48216.6	18651.48	1651.129	198.9119	23.16855	10.86753	7.013543	5.312918	4.466827	5.704903
Ni	0.751513	0.017114	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001	0.0001
Р	0.495847	0.826657	0.272681	0.18385	0.044062	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014	0.0014
Pb	3.57847	11.6306	8.90746	3.9878	1.56403	0.77674	0.531734	0.751564	0.596735	0.433884	0.39156	0.395084
S	530.2005	977.9917	1568.961	5445.905	4551.615	968.6442	571.4159	568.2394	566.6977	570.1573	560.4871	555.4936
\mathbf{Sb}	0.06796	0.09832	0.076	0.03654	0.03109	0.016633	0.0004	0.0004	0.0004	0.0004	0.0004	0.0004
Se	0.63396	0.89852	1.11303	0.7204	0.3935	0.103753	0.059351	0.058199	0.052477	0.04844	0.039698	0.034883
Si	2.99001	2.09274	0.69289	1.280491	1.319578	1.205152	1.14979	1.143218	1.065867	1.098238	1.108944	1.016782
Sn	0.08299	0.0682	0.08801	0.01578	0.00387	0.0003	0.005983	0.0003	0.005773	0.0003	0.0003	0.0003
\mathbf{Sr}	8.17308	12.74248	11.56303	7.18879	7.27548	6.76801	4.01292	2.89722	2.52241	2.2978	2.24716	2.1083
Ti	1.41198	0.03467	0.0001	0.0001	0.0001	0.0001	0.0001	0.077777	0.0001	0.0001	0.0001	0.0001
Λ	0.0002	0.01271	0.01999	0.00201	0.00218	0.002825	0.003889	0.005849	0.005721	0.006172	0.005785	0.006131
Zn	0.41451	0.77825	0.96381	0.82285	0.38451	0.137604	0.057269	0.068359	0.062832	0.047464	0.049701	0.071433
DOC		12.017	7.554	6.297	3.695	2.102	1.717	1.665	1.3001	1.98	1.533	1.876
CI	120590	214607	158727	42242	4310	466.9	65.1	44.9	24.1	99.5	24	15.8
[-] Hq	10.33	10.66	10.98	11.36	11.11	11.06	11.05	11.00	11.03	11.03	11.21	11.05
LS [L/kg]	0.07	0.27	0.46	0.95	1.37	2.06	3.52	5.80	8.37	11.42	13.65	15.93

$ \begin{array}{ $				200KEGB9F	R (ALS)	Side 12 (42)				200KEGB9F	
PSO Vettor Ref cence ample Ref cence ample Ref cence ample Ref cence Bart Ref cence Bart<											" (ALS
Montechnic Montech	Se P.	SO Vestion Re	ference sa	mple BA:	Vestforbrænding	Deres betegnelse	PSO Kara		Ref	erence sa	ample BA: KARA
Raturation Material formation Tennutor Resultant Material formation Materian formation Materian formation </th <th>à</th> <th>10085424</th> <th></th> <th></th> <th></th> <th>Labrummer</th> <th>D100854</th> <th>25</th> <th></th> <th></th> <th></th>	à	10085424				Labrummer	D100854	25			
433 103 <th>Result</th> <th>at Måleusikkerhed (±)</th> <th>Enhed</th> <th>Metode</th> <th>Teknik/Udf. Enhed</th> <th>Parameter</th> <th>Resultat</th> <th>Måleusikkerhed (±)</th> <th>Enhed</th> <th>Metode</th> <th>Teknik/Udf. Enhed</th>	Result	at Måleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed	Parameter	Resultat	Måleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed
1410 0.36 mg/gris 1 <	43.3	12.3	mg/kg TS		H	As	24.1	8.3	mg/kg TS		н
	1460	109	mg/kg TS	*	ш	Ba	1930	143	mg/kg TS	-	ц
1,17 0,36 mg/kg13 1 H 0.17 0,36 mg/kg13 1 H 0.30 407 0,1 0,0 </td <td>1.31</td> <td>0.34</td> <td>mg/kg TS</td> <td></td> <td>ш</td> <td>Be</td> <td>2.21</td> <td>0.48</td> <td>mg/kg TS</td> <td>F</td> <td>Е</td>	1.31	0.34	mg/kg TS		ш	Be	2.21	0.48	mg/kg TS	F	Е
931 6.5 mg/kg13 1 H Co 25.0 6.3 mg/kg13 1 H 931 8.0 mg/kg13 1 E Mu S1.0 1190 mg/kg13 1 H 930 4.4 mg/kg13 1 E Mu 9.0 0.00 mg/kg13 1 H 9.0 9.1 mg/kg13 1 E Mu 9.0 0.00 mg/kg13 1 E 9.0 9.1 mg/kg13 1 E Mu 9.0 0.00	1.77	0.36	mg/kg TS		I	Cd	3.09	0.56	mg/kg TS		H
980 870 980 870 980 870 980 870 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 7 980 980 7 980 980 7 980 9	31.9	6.5	mg/kg TS	1	I	Co	26.0	5.3	mg/kg TS	+	Т
3060 457 $m0675$ 1 E 100 0000 000013 1 E 100 0000 1100 000013 1 E 307 666 $m00013$ 1 E 000013 1 E 000013 1 E 112 $m00013$ 1 E 112 1100 112 110013 11 112 110013 11 110013 11 110013 11 110013 11 110013 110013 11 110013 11 110013 11 110013 11 110013 11 110013 110103 110103 110103 110103 110103 110103 1101033 110133 110013	598	80	mg/kg TS	1	ш	Cr	1020	136	mg/kg TS	+	ш
	3050	437	mg/kg TS	1	Ē	Cu	8310	1190	mg/kg TS	1	E
3.28 6.44 mg/dg13 1 E 3.30 67 mg/dg13 1 E Mo 950 6.12 mg/g13 1 E 3.30 67 mg/dg13 1 E Mo 950 6.12 mg/g13 1 E 3.30 543 mg/g13 1 E Mo 950 6.12 mg/g13 1 E 4.30 mg/g13 1 E Mo 950 6.23 1.43 mg/g13 1 E 1.30 mg/g13 1 E S 2.44 1.43 mg/g13 1 E 2.30 mg/g13 1 E S 2.45 1.43 mg/g13 1 E 3.10 mg/g13 1 E S 3.7 mg/g13 1 E E 3.11 mg/g13 1 E Mo 1.43 mg/g13 1 E E	0.037	3 0.0085	mg/kg TS	1	ш	Hg	0.0400	0.0090	mg/kg TS	1	ш
948 665 mg/lg15 1 E 330 57 mg/lg15 1 E 330 53 mg/lg15 1 E 434 131 mg/lg15 1 E 136 131 mg/lg15 1 E 233 30 mg/lg15 1 E 331 30 mg/lg15 1 E 461 1 E 1 E 1 461 1 E 1 E 1 E 333 30 mg/lg15 1 E 1 E 1 461 1 1 1 1 1 1 1 E 1 E 312 1 1 1 1	12.8	4.4	mg/kg TS	+	ц	Mo	59.4	11.2	mg/kg TS	1	ш
300 647 mg/gars 1 H No 960 12 mg/gars 1 \mathbb{R}^{1} 4340 131 mg/gars 1 1 mg/gars 1 mg/gars 1 \mathbb{R}^{1} $$	9.48	5.65	mg/kg TS	1	ш	Nb	9.00	6.32	mg/kg TS	-	ш
3200 548 mg/kg/s 1 E 520 173 mg/kg/s 1 E E 1.96 1.31 mg/kg/s 1 <	307	29	mg/kg TS	1	T	Ni	850	120	mg/kg TS	-	ш
4340 1 \mathbb{R} 520 \mathbb{R} 620 \mathbb{R} 620 \mathbb{R} 620 \mathbb{R}	3280	548	mg/kg TS	1	ш	Pb	1040	173	mg/kg TS	-	ш;
136 131 mg/g/15 1 E 2,49 1,48 mg/g/15 1 E 4.37 30 mg/g/15 1 E 2,49 1,48 mg/g/15 1 E 4.37 30 mg/g/15 1 E 37 37 mg/g/15 1 E 4.00 mg/g/15 1 E 37 37 mg/g/15 1 E 4.00 mg/g/15 1 E 37 37 mg/g/15 1 E 4.00 mg/g/15 1 E 37 3250 410 37 mg/g/15 1 E 3.240 40 mg/g/15 1 E 27 3250 411 1 E 10.3 %/15 1 E 27 3250 413 1 E 27 3.34 0.10 %/15 1 E 27 3250 413 1 1 1 <	4340		mg/kg TS	1	A	S*	6220		mg/kg TS		A
227 327 327 937 967 167 167 167 167 167 167 167 1 </td <td>1.96</td> <td>1.31</td> <td>mg/kg TS</td> <td>1</td> <td>ш</td> <td>Sc</td> <td>2,49</td> <td>1.48</td> <td>mg/kg TS</td> <td>+</td> <td>w,</td>	1.96	1.31	mg/kg TS	1	ш	Sc	2,49	1.48	mg/kg TS	+	w,
433 30 mg/kg1s 1 e 4.3 37 mg/kg1s 1 e e 60 9.0 mg/kg1s 1 1 mg/kg1s 1 mg/kg1s 1 e	227		mg/kg TS	-	w	Sn*	167		mg/kg TS	-	n on
510 90 mg/sg 13 1 E W 108 18 mg/sg 13 1 E E Mg/sg 13 1 E	433	30	mg/kg TS	-	ш	Sr	537	37	mg/kg TS	1	ш
	51.0	9.0	mg/kg TS	F	ш	>	109	18	mg/kg TS	1	ш
12.2 1.7 mg/kg TS 1 E 32.40 469 mg/kg TS 1 E 471 mg/kg TS 1 E 32.40 mg/kg TS 1 E 3250 471 mg/kg TS 1 E 47.1 3.7 $\% TS$ 1 E 3260 471 1 E 10.3 $\% TS$ 1 E 326 0.9 $\% TS$ 1 E 11.9 $\% TS$ 1 E 360 1.1 1.1 E 15.6 0.10 $\% TS$ 1 E 1.1 $\% TS$ 1 E 2.03 16.3 0.11 E 0.13 $\% TS$ 1 E 2.03 0.11 0.11 1 E 2.03 0.13 0.11 1 E 2.03 0.13 0.11 <t< td=""><td><60</td><td></td><td>mg/kg TS</td><td>1</td><td>ш</td><td>M</td><td><60</td><td></td><td>mg/kg TS</td><td>-</td><td>ш</td></t<>	<60		mg/kg TS	1	ш	M	<60		mg/kg TS	-	ш
3240 469 mg/kg TS 1 E 236 40 mg/kg TS 1 E 47 3.3 9.7 9.7 1 E 47 3.3 9.7 9.7 9.7 1 E 47 3.3 9.7 9.7 1 E 1 2 47 1.0 9.7 1 E 1 2 9.7 1 E 15.6 1.2 9.7 1 E 3.42 5.8 9.7 1 E 15.6 1.2 9.7 9.7 1.3 1 E 1 1 E 1 1 1 E 1 1 1 E 1 1 1 1 1 1 E 1	12.2	1.7	mg/kg TS	1	ш	X	17.1	2.1	mg/kg TS		ш
236 40 mg/kgTS 1 E 47.1 3.7 $\%/TS$ 1 E 47.1 0.3 $\%/TS$ 1 E 47.1 0.3 $\%/TS$ 1 E 10.3 $\%/TS$ 1 E 3.7 $\%/TS$ 1 E 10.3 $\%/TS$ 1 E 3.7 $\%/TS$ 1 E 11.9 1.0 $\%/TS$ 1 E 3.7 $\%/TS$ 1 E 11.9 0.10 $\%/TS$ 1 E 3.7 $\%/TS$ 1 E $1.4.9$ 0.15 $\%/TS$ 1 E 3.6 0.17 0.13 $\%/TS$ 1 E 1.34 0.013 $\%/TS$ 1 E 0.174 0.017 $\%/TS$ 1 E 3.96 0.130 $\%/TS$ 1 0.16 0.13 $\%/TS$ 1 E	3240	469	mg/kg TS	1	ш	Zu	3250	471	mg/kg TS	-	ш
47.1 3.7 $\%1S$ 1 E $\%1S$ $\%1S$ $\%1S$ $\%1S$	236	40	mg/kg TS	1	ш	Zr	342	58	mg/kg TS	-	ш
10.3 0.9 %TS 1 E AI203 10.4 0.9 %TS 1 E 15.6 1.2 %TS 1 E 0.0 %TS 1 E E 15.6 1.0 %TS 1 E 0.1 %TS 1 E 1.3 %TS 1 E 0.1 %TS 1 E E 1.34 0.10 %TS 1 E 0.13 %TS 1 E 1.35 0.013 %TS 1 E 0.1 %TS 1 E 0.130 %TS 1 E 0.174 0.17 0.17 1 E 0.130 %TS 1 E 0.14 0.16 1 E 0.130 %TS 1 E 0.174 0.178 0.17 1 E 0.130 %TS 1 E 0.13 0.13 0.13 1	47.1	3.7	%TS	÷	ш	Si02	48.0	3.8	%TS	+	ш
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135 0.15 %TS 1 E 0.14 0.013 %TS 1 E 0.14 0.013 %TS 1 E 0.14 0.017 %TS 1 E 3.04 0.013 %TS 1 E 0.10 %TS 1 E National 1 E 0.130 %TS 1 E National %TS 1 E 0.10 %TS 1 E National National National 1 E 0.10 %TS 1 E National National National 1 E 0.10 %TS 1 1 1 E E National National 1 E 0.10 %TS 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	1.24	0.10	%TS	+	ш	K20	1.64	0.13	%TS	-	ш
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0.904 0.130 %TS 1 E 0.134 0.134 %TS 1 E 1.24 0.10 %TS 1 E 1.24 0.10 %TS 1 E 3.3 %TS 1 1 1 3.4.3 %TS 1 1 1 %TS 1 1 1 1 1 %TS 1 1 1 1 1 1 %T %T 1 1 1 1 1 1 %T %T 98.3 98.3 98.3 1 1 1 1	3.96	0.31	%TS	-	ш	Na2O	2.83	0.33	%TS	1	Е
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	96.7		%	1	>	1S*	98.3		%	F	~

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Hg 6.04 mg/kg13 1 E Hg 0.103 0.02 0.02 0.03 <t< td=""><td>380. 770</td><td>mg/kg TS 1</td><td>E</td><td>Cu</td><td>7890</td><td>1140</td><td>mg/kg TS</td><td>1</td><td>а</td></t<>	380. 770	mg/kg TS 1	E	Cu	7890	1140	mg/kg TS	1	а
Mo e6 mg/601S 1 E NB 56 54 mg/601S 1 6 NB 56 54 mg/601S 1 6 P 5800 467 mg/601S 1 6 P 5800 467 mg/601S 1 6 Sc 51 91 1 6 379 141 P 583 53 38 mg/601S 1 6 379 141 Sc 531 38 mg/601S 1 6 379 141 V 561 mg/601S 1 6 379 141 V 562 mg/601S 1 6 70 7 V 563 36 56 7 7 7 22 V 563 36 56 7 7 7 23 60 7 V 560 96 56 </td <td>0.01</td> <td>mg/kg TS 1</td> <td></td> <td>Hg</td> <td>0.103</td> <td>0.022</td> <td>mg/kg TS</td> <td>-</td> <td>u.</td>	0.01	mg/kg TS 1		Hg	0.103	0.022	mg/kg TS	-	u.
NB <6 mg/gris 1 E NI 56 964 $mg/gris$ 1 164 164 164 164 164 164 617 Se 5130 467 $mg/gris$ 1 8 5530 304 Se 5130 467 $mg/gris$ 1 8 55230 304 Se 5130 467 $mg/gris$ 1 8 5520 304 Sr 513 381 $mg/gris$ 1 8 552 304 Sr 533 3810 562 $mg/gris$ 1 8 872 306 4140 562 996 106 W 600 166 166 166 872 862 906 106 V 660 166 166 166 166 166 166 166 166 166 166 <td< td=""><td><6</td><td>mg/kg TS 1</td><td>В</td><td>Mo</td><td>12.5</td><td>5.3</td><td>mg/kg TS</td><td>4</td><td>ш</td></td<>	<6	mg/kg TS 1	В	Mo	12.5	5.3	mg/kg TS	4	ш
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Pb 2800 467 mg/kgTS 1 E Pb 1820 467 mg/kgTS 1 E 141 1420 353 353 373 </td <td>50 54</td> <td>mg/kg TS 1</td> <td>H</td> <td>Ni</td> <td>198</td> <td>43</td> <td>mg/kg TS</td> <td>4</td> <td>н</td>	50 54	mg/kg TS 1	H	Ni	198	43	mg/kg TS	4	н
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m^* 252 $mg/kgTS$ 1 S 543 38 $mg/kgTS$ 1 228 236 $mg/kgTS$ 1 512 51	4	mg/kg TS 1	Е	Sc	3.79	1.47	mg/kg TS	1	ш
ir 543 38 molegers 1 E 512 362 362 362 362 362 362 362 362 362 362 362 362 362 362 362 362 366 362 366 362 366 362 366 362 366 362 366 367 37	252	mg/kg TS 1	S	Sn*	228		mg/kg TS	1	s
/ 51.4 3.1 mg/kgTS 1 E $/$ 6.2 7 7	343 38	mg/kg TS 1	Ш	Sr	512	36	mg/kg TS	1	ш
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r 3310 552 mg/kg TS 1 E Zn 4140 562 r -45 mg/kg TS 1 E 238 40 50 r -45 mg/kg TS 1 E 238 40 37 r 160 1.1 $%$ TS 1 E 238 40 r 0.87 9.13 1.0 $%$ TS 1 E 0.37 r 0.10 $%$ TS 1 E 1.2 0.117 $%$ TS 1.2 0.12 r 0.33 0.76 1 E 0.20 1.17 0.01 r 0.139 0.14 $%$ TS 1 E 0.02 r 0.136 0.76 1.17 0.01 0.117 0.02 r 0.36 0.76 1.17 0.02 0.01 0.02 r 0.36 0	0.9 1.6	mg/kg TS 1	ιu	X	12.2	1.7	mg/kg TS	t	ш
rr 266 45 mg/gTS 1 E 27 238 40 5002 47.5 3.8 $\% TS$ 1 E 237 246 37 5102 8.75 1.1 $\% TS$ 1 E 238 0.0 3100 15.0 1.1 $\% TS$ 1 E 233 3.7 3100 15.0 1.1 $\% TS$ 1 E 0.0 300 15.0 1.1 $\% TS$ 1 E 0.0 200 12.4 1.0 $\% TS$ 1 E 0.0 200 1.33 0.013 $\% TS$ 1 E 0.0 200 0.16 $\% TS$ 1 E 0.0 1.7 0.0 200 0.16 $\% TS$ 1 E 0.0 1.7 0.0 200 0.16 $\% TS$ 1 E 0.0 0.0	810 552	mg/kg TS 1	ш	Zn	4140	599	mg/kg TS	1	ш
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P205 0.808 0.116 %TS 1 E P205 1.20 0.11 TiO2 1.16 0.09 %TS 1 E 1.12 0.01 Sum 93.8 %aTTS 1 1 1 1 20 1.12 0.00 Sum 93.8 %aTTS 1 1 1 20 1.12 0.00 CI 3.0 %aTTS 1 1 1 2.0 2.0 1.12 0.00 Sum 93.8 1 1 1 2.0 </td <td>0.31</td> <td>%TS 1</td> <td>ш</td> <td>Na2O</td> <td>3.60</td> <td>0.29</td> <td>%TS</td> <td>+</td> <td>ш</td>	0.31	%TS 1	ш	Na2O	3.60	0.29	%TS	+	ш
TiO2 1.16 0.09 %TS 1 E 1102 1.12 0.06 Sum 93.8 %afTS 1 1 1 200	808 0.116	%TS 1	ш	P205	1.20	0.17	%TS		ш
Sum 93.8 % afTS 1 1 20 C01 3.0 % fS 1 1 20 4.3 L01 3.0 % fS 1 1 1 4.3 FS* 97.9 % 1 W 98.4 fS* 97.9 moleTS 1 S	.16 0.09	1 1%	ш.	Ti02	1.12	0.09	%TS	F	ш.
LOI 3.0 % TS 1 1 LOI 4.3 TS* 97.9 % 1 W 98.4 TS* 91.9 molerTS 1 SN* 72.8	13.8	% af TS 1	1	Sum	92.0		% af TS		
TS* 97.9 W TS* 98.4 Sh+ 96.4 Sh+ 96.4 Sh+ 96.4 Sh+ 72.8 Sh+ 97.8 S	3.0	1 1S %	1	rol	4.3		% TS	1	4
Sh ⁺ 96.1 molka TS 1 S	6'2	% 1	M	TS*	98.4		0/0	1	M
	6.1	mg/kg TS 1	S	Sb*	72.8		mg/kg TS	-	ŝ

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Rapport				080006	8	Rappo	t	
Side 25 (42)				200KEGB9F	R (ALS)	Side 26 (42)		
Deres betegnelse	609A	BA	Time seri	es: Amag	erforbrænding	Deres betegnels	se 611A	
Labnummer	D10085	438				Labnummer	D1008	5439
Parameter	Resultat	Måleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed	Parameter	Resultat	Måleusikke
As	25.5	8.5	mg/kg TS	1	H	As	17.4	7.0
Ba	1730	129	mg/kg TS	1	E	Ba	1400	104
Be	1.68	0.39	mg/kg TS	1	ш	Be	1.71	0.38
Cd	1.94	0.37	mg/kg TS	+	H	Cd	3.17	0.58
Co	19.3	4.0	mg/kg TS	1	H	Co	18.0	3.7
c	447	60	mg/kg TS	1	9	Cr	463	62
Cu	3190	457	mg/kg TS	1	ш	Cu	3580	513
Hg	0.0252	0.0062	mg/kg TS	+	F	Hg	0.0450	0.009
Mo	10.8	4.3	mg/kg TS	1	E	Mo	7.97	4.31
Nb	13.0	5.8	mg/kg TS	4	ш	Nb	10.4	6.0
N	181	40	mg/kg TS	1	H	Ni	170	37
Pb	1420	237	mg/kg TS		ш	Pb	1420	237
S*	5260		mg/kg TS		A	S*	5400	
Sc	3.16	1.38	mg/kg TS	1	E	Sc	2.93	1,47
Sn*	190		mg/kg TS	-	S	Sn*	216	
Sr	451	31	mg/kg TS	E I	B	Sr	395	27
٨	55.8	9.6	mg/kg TS	1	8	~	51.3	9.1
M	<60		mg/kg TS	1	U	M	<60	
Y	11.7	1.6	mg/kg TS	4	ш	Y	12.7	1.8
Zn	3950	572	mg/kg TS	1	E	Zn	3250	471
Zr	258	44	mg/kg TS	1	ш	Zr	240	41
Si02	45.4	3.6	%TS	1		Si02	45.9	3.6
AI203	9.11	0.81	%TS	1	B	A1203	9.03	0.80
CaO	16.0	1.2	%TS	-	a	CaO	14.7	1.1
Fe203	13.3	1.1	%TS	1	ш Ш	Fe203	13.0	1.0
K20	1.09	0.09	%TS		ш	K20	1.17	0.10
MgO	1.93	0.15	%TS	1	ш	MgO	1.75	0.14
MnO	0.141	0.014	%TS	1	μ	MnO	0.142	0.01
Na2O	3.98	0.32	%TS	-	ш	Na2O	3.47	0.28
P205	1.39	0.20	%TS	1	Ш	P205	1.31	0.15
Ti02	1.07	0.08	%TS	1	a	TiO2	1.10	0.0
Sum	93.4		% af TS	-	1	Sum	91.6	
FOI	3.2		% TS	1	1	LOI	4.4	
TS*	98.3		%		M	TS*	98.5	
Sb*	68.4		mg/kg TS	1	ŝ	Sb*	51.8	

0.10 0.14 0.014 0.28 0.19 0.09 1.47 27 1.8 41 3.6 0.80 10 91.6 4.4 98.5 51.8 LOI TS* Sb*

3

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D0800068

ALS)

200KEGB9FR

Amagerforbrænding

series:

Time

BA

Teknik/Udf. Enhed

Metode

eusikkerhed (±)

шшт

ZOOKEGB9FR	Time series: Amagerf	Enhed Metode T	mg/kg TS 1	mg/kg 15	ma/ka TS 1	mg/kg TS 1	mg/kg TS 1	mg/kg TS 1	mg/kg 15 1 mg/kg TS 1	mg/kg TS 1	mg/kg TS 1	mg/kg TS 1	mg/kg TS 1	ma/ka TS 1	mg/kg TS 1	%TS 1	%IS 1	7015 1 2015 4	ATS 4	%TS 1	%TS 1	%TS 1	%TS 1	%TS 1	% af TS 1						
	BA	Måleusikkerhed (±)	9.1	106	0.41	0.40	4.7	94	0.0074	5.1	2.09	71	368	1 66	2011	27	9.4	40	487	34	3,4	0.9	EF.	010	0.14	0.013	0.29	0.26	0.08		
	se 613A D100854	Resultat	28.3	1420	1.41	2.09	23.0	697	4780	10.7	9.15	499	2200	9 4500	200	383	50.0	<60	3360	203	43.3	10.6	4.47	1.13	1.81	0.138	3.61	1.78	1.04	92.0	44
Side 28 (42)	Deres betegnel Labnummer	Parameter	As	Ba	Be	Cd	Co	ů	Cu	Bu	Nb	Ni	Pb	22	Sn*	Sr	>	M >	Zu	Zr	Si02	AIZO3	CaU	K2D	MaO	Oum	Na2O	P205	Ti02	Sum	101
		_												_													_				-
(SIR)	erforbrænding	Teknik/Udf. Enhed	н	ш	ш	н	T	ш,		- 14	, u	т	ш <	A	100	μ.	ш			Ш	ш	ш і	U U		ιш	1 111	ш	ш			+
ZOOKEGBBFR	eries: Amagerforbrænding	Metode Teknik/Udf. Enhed	5 1 H	ш Т	5 T	I	T	ш I	ш н		5 I	H H	ш×	A II	100	E	т П			с П	- E			5 UI			ω.	ш -			-

D10085440 Resultat Maleustikkenheid (2) 11.6 11.6 11.6 11.6 11.6 11.6 11.6 11.6 11.6 11.3 11.12 1

612A

Deres betegnelse

Rapport

Side 27 (42)

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Deres betegnelse

Labnummer

Parameter

Rapport

Side 29 (42)

As Ba Ba Co So¹ So¹ So¹ So² So

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Rappor	4		ב			Kappor	-			oonnoon	
Side 31 (42)				200KEGB9FI	(ALS)	Side 32 (42)				ZOOKEGB9FR	ALS
Deres betegnelse	616A	BA	Time seri	es: Amag	erforbrænding	Deres betegnels	617A	BA	Time ser	ies: Amage	rforbrænding
Labnummer	D10085444					Labnummer	D100854	45			
Parameter	Resultat	Maleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed	Parameter	Resultat	Maleusikkerhed (±)	Enhed	Metode	Teknik/Udf, Enhed
AS	0.62	0.4	CI BY/BU	,	C I	AS	50.3	0.3	CI BX/Bu		EI
Ra	1540	114	mg/kg 1S	-	ш I	Ba	1490	111	mg/kg 1S	-	u) I
Be	1.64	0.37	mg/kg 15	-	ц::	Be	1.74	0.39	mg/kg 1S	-	بر
Cd	2.23	0.41	mg/kg TS	-	I	Cd	2.44	0.46	mg/kg TS	-	I.
Co	23.3	4.8	mg/kg TS	-	Ŧ	Co	19.9	4.2	mg/kg TS	-	I
c	539	72	mg/kg TS	1	u u	Cr	434	58	mg/kg TS	1	E
Cu	3070	440	mg/kg TS	1	E	Cu	3180	456	mg/kg TS	-	ш
Ha	<0.01		ma/ka TS	+	12.	Ha	0.0473	0.0103	ma/ka TS	-	u
Mo	9.33	4.15	ma/ka TS	-	ш	Mo	99		ma/ka TS		ш
Nb	11.4	5.8	ma/ka TS	+	W	Nb	9>		ma/kg TS	-	ш
IN	324	46	ma/kg TS	*-	ш	N	232	33	ma/ka TS	-	ш
Pb	645	114	ma/kg TS	+	T	Pb	1210	203	ma/kg TS	1	ш
S*	4030		mg/kg TS	1	A	S.	4510		mg/kg TS	-	A
Sc	2.72	1.36	mg/kg TS	÷	ш	Sc	1.42	1.39	mg/kg TS		Ш
Sn*	222		mg/kg TS	*	s	Sn*	237		mg/kg TS	-	S
Sr	322	22	mg/kg TS	1	ш	Sr	461	32	mg/kg TS	1	ш
>	47.8	8.5	mg/kg TS	F	ш	N	43.5	8.1	mg/kg TS	1	ш
×	<60		mg/kg TS		ш	M	<60		mg/kg TS	1	ш
٨	9.36	1.48	mg/kg TS	1	ш	X	11.2	1.6	mg/kg TS	-	ш
Zn	3360	487	mg/kg TS		U	Zu	3100	449	ma/ka TS	F	ш
Zr	246	42	mg/kg TS	1	ш	Zr	236	40	mg/kg TS	+	ш
Si02	43.8	3.5	%TS	ł	ш	SI02	44.2	3.5	%TS	-	ш
AI203	10.8	1.0	%TS	1	ш	AI203	9.47	0.84	%TS	1	ш
CaO	15.4	12	%TS		ш	CaO	15.4	1.2	%TS	-	ш
Fe203	12.5	1,0	%TS	1	ш	Fe2O3	12.0	1.0	%TS	1	ш
K2O	1.21	0.10	%TS		ш	K20	1.15	0.09	%TS	+	ш
OBM	1.73	0.14	%TS	,	ш	OBM	1.80	0.14	%TS	-	ш
MnO	0.138	0.013	%TS		ш	MnO	0.131	0.013	%TS	-	ш
Na2O	4.02	0.32	%TS	•	ш	Na2O	3.77	0.30	%TS	-	ш
P205	2.62	0.38	%TS	-	а	P205	2.30	0.33	%TS	1	ш
TiO2	1.06	0.08	%TS		ц.	Ti02	0.983	0.077	%TS	1	ш
Sum	93.3		% af TS	-	1	Sum	91.2	10 mm	% af TS	1	
LOI	3.6		% TS		10	TOI	4.9		% TS	1	1.
TS*	98.8		%	1	M	TS*	98.4		%	1	W
sh*	P. 4.0		CH TOTAL	,	0	*10	000				

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	4	erforbrænding		Teknik/Udf. Enhe	H	ш	ш	E	T	ш	ш	4	ш	ū	ш	ш	A	ш	S	Ш	ш	ш	ш	ш	ш	ш	ш	ш	ш	Ш	ш	ш	ш	щ	щ	+	4	M	ŝ
800068	ZOOKEGB9FR	s: Amage		Metode	-	-	-		-	1	1	Ŧ	1	+	1	÷	4	+		1	+	1	4	-	1	1	1	1	+		÷	t	1	*	1 - 10	*	1	41 - 1	+
ă		Time serie		Enhed	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	% af TS	% TS	%	ma/ka TS
		BA		Måleusikkerhed (±)	7.7	66	0.39	0.39	3.8	122	426	0.0071			50	208		1.44		31	8.0		1.6	523	39	3.5	0.82	1.2	1.0	0.09	0.14	0.012	0.31	0.33	0.08				
		702A	D10085447	Resultat	21.6	1340	1.52	2.09	18.8	906	2970	0.0302	99	99	355	1240	4180	1.33	230	439	43.1	<60	10.3	3610	230	44.0	9.22	15.4	12.6	1.13	1.80	0.130	3.84	2.27	1.02	91.4	4.3	98.5	616
tapport	le 34 (42)	tres betegnelse	bnummer	rameter															*							02	203	0	203	0	0	0	20	05	32	E			
											_		_					_										_			_								_
	(ALS)	corbrænding		reknik/Udf. Enhed	T	ш	ш.:	I	H	E	ш	ш.	ш	u	I	8	A	u	s	ш	ш	ш	ш	E	ш	ш	ш	ш	ш	ш	ш	ш	ш	ш	Е		1	M	U
800068	200KEGB9FR	s: Amagerforbrænding		Metode Teknik/Udf. Enhed	H	ш.	ш.:	H	H	- E	1 E	- E	E		T T	E	A A	E	s	1	E	1	1	1 E	1 E	E	1 E	1 E	1 E	1 E	1 6		E	E	- E	1		1 W	
DUSUUU68	200KEGB9FR	Time series: Amagerforbrænding		Enhed Metode Teknik/Udf. Enhed	mg/kg TS 1 H	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 H	mg/kg TS 1 H	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 F	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 H	mg/kg TS 1 E	mg/kg TS 1 A	mg/kg TS 1 E	mg/kg TS 1 S	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 E E	mg/kg TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	%TS 1 E	B 1 81%	%afTS 1 1	%TS 1 1	% 1 W	and to the test of
DUSUUU68	200KEGBBFR	BA Time series: Amagerforbrænding	99	Måleusikkerhed (±) Enhed Metode Teknik/Udf. Enhed	8.9 mg/kg TS 1 H	92 mg/kg TS 1 E	0.41 mg/kg TS 1 E	0.56 mg/kg TS 1 H	3.7 mg/kg TS 1 H	63 mg/kg TS 1 E	394 mg/kgTS 1 E	0.029 mg/kg TS 1 F	4.3 mg/kgTS 1 E	5.8 mg/kgTS 1 E	51 mg/kg TS 1 H	202 mg/kg TS 1 E	mg/kg TS 1 A	1.40 mg/kg TS 1 E	mg/kgTS 1 S	23 mg/kgTS 1 E	8.2 mg/kg TS 1 E	mg/kg TS 1 E	1.6 mg/kgTS 1 1 E	631 mg/kg TS 1 E	33 mg/kgTS 1 E	3.5 %TS 1 E	0.81 %TS 1 E	1.1 %TS 1 E	0.9 %TS 1 E	0.10 %TS 1 E	0.13 %TS 1 E	0.021 %TS 1 E	0.33 %TS 1 E	0.34 %TS 1 E	0.074 %TS 1 E	% af TS 1 1	%TS 1 1 1	% 1 W	matter TO A
t D0800068	200KEGB9FR	701A BA Time series: Amagerforbrænding	D10085446	Resultat Måleusikkerhed (±) Enhed Metode Teknik/Udf. Enhed	27.8 8.9 mg/kg TS 1 H	1240 92 mg/kgTS 1 E	1.93 0.41 mg/kg TS 1 E	3.11 0.56 mg/kg TS 1 H	17.3 3.7 mg/kg TS 1 H	467 63 mg/kgTS 1 E	2750 394 mg/kgTS 1 E	0.137 0.029 mg/kgTS 1 F	10.7 4.3 mg/kgTS 1 E	10.8 5.8 mg/kgTS 1 E	235 51 mg/kg TS 1 H	1200 202 mg/kgTS 1 E	3950 mg/kgTS 1 A	3.08 1.40 mg/kgTS 1 E	205 mg/kgTS 1 S	326 23 mg/kg TS 1 E	45.2 8.2 mg/kg TS 1 E	<60 mg/kgTS 1 E	11.5 1.6 mg/kgTS 1 E	4360 631 mg/kgTS 1 E	198 33 mg/kgTS 1 E	44.1 3.5 %TS 1 E	9.09 0.81 %TS 1 E	14.7 1.1 %TS 1 E	11.7 0.9 %TS 1 E	1.17 0.10 %TS 1 E	1.69 0.13 %TS 1 E	0.216 0.021 %TS 1 E	4.15 0.33 %TS 1 E	2.35 0.34 %TS 1 E	0.947 0.074 %TS 1 E	90.1 % afTS 1 1	5.8 %TS 1 1	98.2 W W	Ed 7 C

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Deres belegnelse 76A BA Time Serries: Amager for Jahnumner D10086449 A BA Time Serries: Amager for Jahnumner D10086449 A BA Time Serries: Amager for As 33.3 0.2 mg/kg TS 1 Amager for Amager for As 3.12 0.03 0.03 mg/kg TS 1 Amager for Col 3.12 0.03 mg/kg TS 1 Amager for Amager for Col 1.340 0.9 B Mg/kg TS 1 Amager for Col 1.340 0.03 0.0389 0.0389 Mg/kg TS 1 Amage/kg TS 1 Col 46.4 52 mg/kg TS 1 Mg/kg TS 1 Amage/kg TS 1	Side 36 (42)				COUNEGERI	×
J10085449 Labhummer D10085449 Rasultat Måleusikkerhed (±) Enhed Metode Tekni As 3120 92 mg/kg TS 1 1 As 1320 0.0.33 mg/kg TS 1 1 Be 3.1.20 0.0.33 mg/kg TS 1 1 Co 1230 1739 mg/kg TS 1 1 Co 1230 0.0387 0.0089 mg/kg TS 1 1 Co 1230 0.0089 mg/kg TS 1	Deres belegnels	e 706A	ВА	Time seri	es: Amag	lerfor
Parameter Resultat Måleusikkerhed (±) Enhed Metode Terkni As 13.3 10.2 mg/kg TS 1 Ba 1.30 0.33 mg/kg TS 1 Cid 3.12 0.58 mg/kg TS 1 Cid 3.12 0.033 mg/kg TS 1 Cid 3.12 0.038 mg/kg TS 1 Cid 3.12 0.038 mg/kg TS 1 Ci 5650 811 mg/kg TS 1 Mb <6 811 mg/kg TS 1 Nb <6 0.0089 mg/kg TS 1 Nb <6 0.0089 mg/kg TS 1 Nb <6 52 mg/kg TS 1 Nb <6 52 mg/kg TS 1 Sr 240 51 7.9 mg/kg TS 1 Sr 230 7.9 mg/kg TS 1 1 Sr 3	Labnummer	D10085	449			
As 33.8 10.2 mg/kg TS 1 Ba 1.240 92 mg/kg TS 1 Ba 1.240 92 mg/kg TS 1 Ba 3.12 0.58 mg/kg TS 1 Co 1230 0.33 mg/kg TS 1 Co 1230 0.33 mg/kg TS 1 Co 1230 0.0387 0.0089 mg/kg TS 1 Co 1230 0.0387 0.0089 mg/kg TS 1 Mo 6 0.0089 mg/kg TS 1 1 Mo 6 0.0089 mg/kg TS 1 1 Mo 6 0.0089 mg/kg TS 1 1 Mo 240 52 mg/kg TS 1 1 1 Sec 1 1 mg/kg TS 1 1 1 Sec 1 0.013 0.014 0.014	Parameter	Resultat	Måleusikkerhed (±)	Enhed	Metode	Teknik
Ba 1240 92 mg/kg TS 1 Be 1.30 0.33 mg/kg TS 1 Cd 3.12 0.56 mg/kg TS 1 Cd 1.30 0.33 mg/kg TS 1 Cd 1.30 0.33 mg/kg TS 1 Cd 1.30 0.0089 mg/kg TS 1 Cu 6560 179 mg/kg TS 1 Mb <6	As	33.8	10.2	mg/kg TS	+	
Be 1.30 0.33 mg/kg1S 1 Cd 1.31 0.58 mg/kg1S 1 Cd 1.31 0.58 mg/kg1S 1 Co 1.31 0.58 mg/kg1S 1 Cu 5560 811 mg/kg1S 1 Cu 5560 811 mg/kg1S 1 Mo c6 3 0.0069 mg/kg1S 1 Mo c6 5 mg/kg1S 1 1 No c6 2.11 mg/kg1S 1 1 No c6 2.11 mg/kg1S 1 1 Sc c1 2.11 mg/kg1S 1 1 Sc c1 2.11 mg/kg1S 1 1 Sc c1 7.9 mg/kg1S 1 1 Sc c1 7.9 mg/kg1S 1 1 Sc c1 3.0 0.00 0.01 1	Ba	1240	92	mg/kg TS	+	
Cd 3.12 0.68 mg/kg TS 1 Co 1230 179 mg/kg TS 1 Cu 464 6.2 mg/kg TS 1 Cu 5660 811 mg/kg TS 1 Hg 0.0397 0.0089 mg/kg TS 1 Mo <6	Be	1.30	0.33	mg/kg TS	-	
Co 1230 179 mg/kg TS 1 Cr 464 62 mg/kg TS 1 Hg 0.0387 0.0089 mg/kg TS 1 Hg 0.0387 0.0089 mg/kg TS 1 Mo <6	Cd	3.12	0.58	mg/kg TS	1	
Cr 464 62 mg/kg TS 1 Cu 5560 811 mg/kg TS 1 Hg 0.0337 0.0089 mg/kg TS 1 Mb -6 811 mg/kg TS 1 Nb -6 52 mg/kg TS 1 Nb -6 240 52 mg/kg TS 1 Nb -6 211 mg/kg TS 1 1 Sr 550 211 mg/kg TS 1 1 Sr 55 mg/kg TS 1 <	Co	1230	179	mg/kg TS	ł	
$ \begin{array}{c ccccccccccccccccccccccccccccccccccc$	c	464	62	mg/kg TS	1	
Hg 0.0387 0.0089 mg/kg TS 1 Mo <6	cu	5560	811	mg/kg TS	1	
Mo e6 mg/kg TS 1 Nb -6 52 mg/kg TS 1 Nb 240 52 mg/kg TS 1 Pb 1270 211 mg/kg TS 1 S* 5650 211 mg/kg TS 1 S* 5650 211 mg/kg TS 1 S* 5650 211 mg/kg TS 1 S* 57 7.9 mg/kg TS 1 Sr 200 mg/kg TS 1 1 V 42.8 30 mg/kg TS 1 V 435 30 mg/kg TS 1 V 43 mg/kg TS 1 1 V 10.3 1.6 mg/kg TS 1 Zr 236 496 mg/kg TS 1 Zr 3450 0.14 16 1 Zr 256 4.43 %rTS 1 1 Zr 234 <td>Hg</td> <td>0.0397</td> <td>0,0089</td> <td>mg/kg TS</td> <td>1</td> <td></td>	Hg	0.0397	0,0089	mg/kg TS	1	
Nb 66 mg/kgTS 1 Ni 240 52 mg/kgTS 1 Pb 1270 211 mg/kgTS 1 Sr 5650 211 mg/kgTS 1 Sr 435 30 mg/kgTS 1 Sr 435 7.9 mg/kgTS 1 V 42.8 7.9 mg/kgTS 1 V 436 mg/kgTS 1 1 V 446 0.1 9/KTS 1 Zr 256 3.7 %TS 1 1 1	Mo	99		mg/kg TS	1.	
Ni 240 52 mg/kgTS 1 Pb 1270 211 mg/kgTS 1 Sc <1	Nb	9>		mg/kg TS	1	
Pb 1270 211 mg/kgTS 1 Sr 41 mg/kgTS 1 mg/kgTS 1 Sr 435 30 mg/kgTS 1 mg/kgTS 1 Sr 435 30 mg/kgTS 1 1 1 1 Sr 435 30 mg/kgTS 1 1 1 1 1 V 42.8 7.9 mg/kgTS 1	Ni	240	52	mg/kg TS	Ŧ	
S* 5650 mg/kg TS 1 Sc <1	Pb	1270	211	mg/kg TS	*	
Sc <1 mg/kg TS 1 Sn* 209 30 mg/kg TS 1 Sn* 235 30 mg/kg TS 1 N 42.8 30 mg/kg TS 1 V 42.8 7.9 mg/kg TS 1 V 42.8 7.9 mg/kg TS 1 V 40.3 1.6 mg/kg TS 1 Zr 3430 496 mg/kg TS 1 Zr 256 4.3 mg/kg TS 1 Aloo 0.846 0.846 7 1 Aloo 1.14 0.04 %TS 1 Mg0 1.178 0.017 %TS 1 Mg0 1.178 0.017 %TS 1 Mg0 0.177	S*	5650		mg/kg TS	1	
Sn* 209 mg/kg TS 1 Sr 435 30 mg/kg TS 1 V 423 343 496 mg/kg TS 1 Zn 256 43 mg/kg TS 1 1 Zn 266 0.04 %TS 1 1 AlzO3 14.4 0.09 %TS 1 1 AlzO3 14.4 0.01 %TS 1 1 MnO 0.17 %TS 1 1 1	Sc	۲		mg/kg TS	1	
Sr 435 30 mg/kg TS 1 V < 42.8	Sn*	209		mg/kg TS	1	
V 42.8 7.9 mg/kg TS 1 W <60	Sr	435	30	mg/kg TS	1	
W <60 mg/kg TS 1 Y 10.3 1.6 mg/kg TS 1 Zr 3430 496 mg/kg TS 1 Zr 3430 496 mg/kg TS 1 Zr 356 43 mg/kg TS 1 Zr 256 3.7 %TS 1 Aloca 9.46 0.84 %TS 1 Aloca 1.1 1 %TS 1 Aloca 1.1 %TS 1 1 Cao 1.41 0.9 %TS 1 1 Mg0 1.78 0.14 %TS 1 1 Mg0 1.78 0.014 %TS 1 1 Mg0 0.177 %TS 1 1 1 Mg0 1.73 0.017 %TS 1 1 Mg0 1.73 0.017 %TS 1 1 Mg0 1.73 0.717 <td< td=""><td>></td><td>42.8</td><td>7.9</td><td>mg/kg TS</td><td>1</td><td></td></td<>	>	42.8	7.9	mg/kg TS	1	
Y 10.3 1.6 mg/kgTS 1 Zn 3430 496 mg/kgTS 1 Zn 3430 496 mg/kgTS 1 Zio2 46.3 3.7 mg/kgTS 1 SiO2 46.3 3.7 mg/kgTS 1 Al203 9.46 0.84 %TS 1 Al203 9.46 0.84 %TS 1 Al203 9.46 0.84 %TS 1 Al203 1.14 0.09 %TS 1 K200 1.14 0.09 %TS 1 Mg0 1.78 0.17 %TS 1 Ma20 3.94 0.017 %TS 1 P205 1.30 0.017 %TS 1 P205 0.366 0.077 %TS 1 Sum 91.1 %TS 1 1 Sum 91.1 %TS 1 1 T02	N	<60		mg/kg TS	1	
Zh 3430 496 mg/kg TS 1 Zr 256 43 mg/kg TS 1 Siz 46.3 3.7 mg/kg TS 1 Siz 46.3 3.7 mg/kg TS 1 Siz 46.3 0.84 %TS 1 Al203 9.46 0.84 %TS 1 Al203 9.46 0.84 %TS 1 Al203 9.46 0.84 %TS 1 Cao 14.7 1.14 0.09 %TS 1 Mg0 1.14 0.014 %TS 1 1 Mg0 1.78 0.14 %TS 1 1 Mg0 1.78 0.14 %TS 1 1 Mg0 0.17 %TS 1 1 1 1 Na2O 3.94 0.017 %TS 1 1 1 1 Na2O 1.20 0.17 %TS 1	٨	10.3	1.6	mg/kg TS	1	
Zr 256 43 mg/kgTS 1 SiO2 46.3 3.7 %15 1 SiO2 94.6 0.8 %15 1 Al203 94.6 0.8 %15 1 Al203 94.6 0.8 %15 1 CaO 14.7 1.1 %15 1 CaO 14.7 0.9 %15 1 K2O 1.14 0.09 %15 1 MnO 0.173 0.017 %15 1 MnO 0.173 0.017 %15 1 Na2O 3.94 0.31 %15 1 Pazos 0.931 %17 1 1 Sum 9.1 0.077 %15 1 Sum 9.1 %15 1 1 Sum 9.1 %15 1 1 Sum 9.1 %15 1 1 Sum 9.1 %15<	Zn	3430	496	mg/kg TS	1	
SiO2 46.3 3.7 %TS 1 Al203 9.46 0.84 %TS 1 Al203 9.46 0.84 %TS 1 CaO 14.7 1.1 %TS 1 Fe203 11.4 0.9 %TS 1 MgO 1.78 0.09 %TS 1 MnO 0.178 0.014 %TS 1 MnO 0.178 0.017 %TS 1 Na2O 3.94 0.017 %TS 1 P205 1.20 0.31 %TS 1 TI20 0.31 %TS 1 1 MnO 0.17 %TS 1 1 Mazo 39.4 0.077 %TS 1 1 Sum 91.4 %TS 1 1 1 Sum 91.4 5.2 93.7 1 1 Sum 97.1 97.5 1 1 <t< td=""><td>Zr</td><td>256</td><td>43</td><td>mg/kg TS</td><td>1</td><td></td></t<>	Zr	256	43	mg/kg TS	1	
Al203 9.46 0.84 %TS 1 CaO 14.7 1.1 %TS 1 Fa203 11.4 0.9 %TS 1 K20 1.14 0.9 %TS 1 Mg0 1.78 0.14 %TS 1 Ma20 1.78 0.17 %TS 1 P205 1.30 0.077 %TS 1 P205 0.985 0.077 %TS 1 P205 0.985 0.077 %TS 1 VIIO2 0.985 0.077 %TS 1 P205 1.0 0.17 %TS 1 Sum 91.1 %TS 1 1 TIO2 0.985 0.077 %TS 1 1 Sum 91.1 %TS 1 1 1 TS* 7.3 0.077 %TS 1 1	Si02	46.3	3.7	%TS	1	
Ca0 14.7 1.1 %TS 1 Fe203 11.4 0.9 %TS 1 KF00 1.74 0.9 %TS 1 Mg0 1.78 0.14 %TS 1 Ma2O 3.94 0.017 %TS 1 Pa2OS 3.94 0.017 %TS 1 Pa2OS 3.94 0.017 %TS 1 Pa2OS 1.20 0.17 %TS 1 Pa2OS 0.017 %TS 1 1 Pa2OS 0.956 0.017 %TS 1 1 Pa2OS 0.966 0.077 %TS 1 1 Sum 91.1 0.77 %TS 1 1 Sum 91.3 0.77 %TS 1 1 Sum 97.1 9.78 1 1 1 Str 73.9 0.077 %TS 1 1 Sum 97.3	AI203	9.46	0.84	%TS	1	
Fe2O3 11.4 0.9 %TS 1 K2O 1.1.4 0.09 %TS 1 MgO 1.78 0.01 %TS 1 MaO 0.173 0.017 %TS 1 MaO 0.173 0.017 %TS 1 Na2O 3.94 0.31 %TS 1 P205 1.20 0.017 %TS 1 TiO2 0.365 0.077 %TS 1 Sum 91.1 %TS 1 1 Sum 91.3 0.077 %TS 1 TiO2 9.865 0.077 %TS 1 Sum 91.4 7 %TS 1 TiO2 9.865 0.077 %TS 1 Sum 97.1 5.2 7 1 TS* 97.1 97.1 1 1 Str 73.9 0.077 7.8 1	CaO	14.7	1.1	%TS	1	
K20 1.14 0.09 %TS 1 Mg0 1.78 0.14 %TS 1 Mn0 0.173 0.017 %TS 1 Nazo 3.94 0.017 %TS 1 Nazo 3.94 0.17 %TS 1 TiO2 0.956 0.077 %TS 1 Sum 91.1 %TS 1 1 LOI 5.2 0.077 %TS 1 1 Sum 91.1 9.71 %TS 1 1 1 Sum 91.1 6.2 0.077 %TS 1	Fe203	11.4	0.9	%TS	1	
MgO 1.78 0.14 %TS 1 MnO 0.173 0.017 %TS 1 MnO 0.173 0.017 %TS 1 Na2O 3.94 0.017 %TS 1 P205 1.20 0.17 %TS 1 TiO2 0.985 0.077 %TS 1 Sum 91.1 %TS 1 1 LOI 5.2 0.077 %TS 1 1 TiO2 91.1 %TS 1	K20	1.14	0.09	%TS	1	
MnO 0.173 0.017 %TS 1 Na2O 3.94 0.31 %TS 1 P2O5 1.20 0.17 %TS 1 TiO2 0.985 0.077 %TS 1 Sum 91.1 %TS 1 1 LOI 5.2 9.86 0.077 % TS 1 TiO2 9.936 0.077 % TS 1 1 Sum 91.1 0.077 % TS 1 1 TiO2 9.935 0.077 % ATS 1 1 Sum 91.1 % TS 1	MgO	1.78	0.14	%TS	1	
Na2O 3.94 0.31 %TS 1 P2O5 1.20 0.17 %TS 1 P2O5 1.20 0.17 %TS 1 TIO2 0.985 0.077 %TS 1 Sum 91.1 %afTS 1 1 LOI 5.2 %TS 1 %TS 1 TLO1 5.2 %TS 1 %TS 1 Sum 97.1 %TS 1 %TS 1 Sum 97.3 73.9 mgkgTS 1 1	MnO	0.173	0.017	%TS		
P205 1.20 0.17 %TS 1 TiO2 0.985 0.077 %TS 1 Sum 91.4 %TS 1 7 LOI 5.2 %TS 1 7 TS* 97.1 %TS 1 7	Na20	3.94	0.31	%TS	-	
TiO2 0.985 0.077 %TS 1 Sum 91.1 %TS 1 %TS 1 LOI 5.2 %TS 1 %TS 1 TS* 97.1 %T %TS 1 %TS Sb* 73.8 mg/kTS 1 %TS 1	P205	1.20	0.17	%TS	F	
Sum 91.1 % afTS 1 LOI 5.2 % TS 1 LOI 5.2 % TS 1 Sb* 73.8 mg/kdTS 1	TiO2	0.985	0.077	%TS	1	
LOI 5.2 %TS 1 TS* 97.1 % 1 Sb* 73.9 mg/kg TS 1	Sum	91.1		% af TS		
TS* 97.1 % 1 Sb* 73.9 mg/kg TS 1	rol	5.2		% TS	1	
Sb* 73.9 mg/kg TS 1	TS*	97.1		%	-	
	Sb*	73.9		mg/kg TS	-	

ALS BA Time series: Amagerforbrænding Metode Teknik/Udf. Enhed D0800068 200KEGB9FR mg/kg TS mg/ mg/kg TS Enhed %TS %TS %TS %TS %TS %TS %TS %TS %TS SP 8 D10085448 10.7 96 0.37 0.37 0.37 4.0 5.8 392 368 5.8 5.8 5.8 5.8 5.8 5.8 5.8 281 281 61 3.5 0.9 0.9 0.09 0.09 0.015 1.33 0.30 1.6 26 703A Rapport Deres belegnelse Side 35 (42) abnummer Parameter Zr Si02 Si02 Al203 Fc203 R20 Mg0 Mn0 Mn0 Na20 P205 P205 Fri02 Sum Sum Sum Sum Sum Sum Si02 Si0 Zn

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	ALS)	rhus		nhed														1																	
	~	series: Å		Teknik/Udf. E		u	I	H	ш	ш	UL.	ш	ш I	ш	ш<	< u	u oo	ш	ш	ш	ш	ii i	иш	ш	ш	ш	ш	ш	ш	ш	ш	ш	F	-	N
0800080	200KEGB9FR	BA Time		Metode						1	+	1	÷-	-	+ +			1	1	÷					1	÷	1	1	1		+		1	4	
2				Enhed	CI fiving	ma/ka TS	ma/ka TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg 13	ma/ka TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg TS	mg/kg 15	ST%	%TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	%TS	% af TS	% TS	%a
		94a	151	Måleusikkerhed (±)	3,4	0.48	0.83	6.9	73	1160	0.0040	5.9	7.32	43	271			29	10.1		1.7	616	36	0.88	1.1	1.1	0.080	0.15	0.020	0.31	0.124	0,10			
		Aarhus	D100854	Resultat	1.15	1 89	4.66	33.9	537	8110	0.0111	11.0	6.10	303	1620	3800	396	415	55.8	<60	11.6	4250	320	9.92	14.8	14.2	0.933	1.90	0.212	3.87	0.862	1.21	92.2	4.5	98.9
2		sgnelse	5	I																			0	03		03		0	0	0	55	2	-		
карро	Side 38 (42)	Deres bete	Labnumme	Paramete	A	Ba	20	3	Շ	5	ΒH	Mo	qN	ī	92	0	Su*	ŝ	>	3	>	UN I	21	AIZ	CaC	Fe2	K2O	Mgl	Mno	Na2	P2C	TIO	Sun	PO	TS*
rappo	Side 38 (42)	Deres bet	Labnumue	Paramete	AS	Ba a	2d	00	ŗ	Cu	Hg	Mo	qN	N	Ph 10	0	Sn*	Sr	>	N	>	Zu	SIO	AIZ	CaC	Fe2	K2C	Mg(Mnd	Na2	P2C	TIO	Sun	FOI	TS*
	R (42) Side 38 (42)	erforbrænding	Labnumme	Teknik/Udf. Enhed	AS L	E	H	H	ш	E	Hg	Mo	UP I	II I	Pb		Sol Sol	E	E	E		2U	20	E AI2	E	E Fe2	E K2C	E	E	E Na2	E P2C	E TIO:	Sun	1	W TS*
vappo	200KEGB9FR (42) Side 38 (42)	es: Amagerforbrænding	Labrumme	Metode Teknik/Udf. Enhed	AS	E E	T	H H	ш С	1 E	H	1 E Mo	m :	I	A A		30 Soft	1 E	1 E	т М				1 E AI2	1 E CaC	T E Fe2	1 E K20	Mg(A E Mu	1 E Na2	P2C	TIO	1 3un	1 1	1 W 15*
	200KEGB9FR	Time series: Amagerforbrænding	Labnumm	Enhed Metode Teknik/Udf. Enhed Paramete		mo/kaTS 1 E	marka TS 1 H	mg/kg TS 1 H H	mg/kg TS 1 E	mg/kg TS 1 E E Cu	mg/kg TS 1 F F Hg	mg/kg TS 1 E E	mg/kg TS 1 E	mg/kg TS 1 H			mg/kg TS 1 S	mg/kg TS 1 E E Sr	mg/kg TS 1 E	mg/kg TS 1 E	mg/kg TS 1 E			%TS 1 E AI2	%TS 1 E CaC	%TS 1 E E	%TS 1 E E K20	%TS 1 E Mg(%TS 1 E E	%TS 1 E E Na2	%TS 1 E P2C	%TS 1 E E	%afTS 1 1 Sun	%TS 1 1 100	% 1 W
Kappo	200KEGB9FR	BA Time series: Amagerforbrænding	t50 Labnumme	Måleusikkerhed (±) Enhed Metode TeknikUdf. Enhed Paramete	101 III00910 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	0.55 muka 12 1 E Ra	0.50 mo/kg TS 1 H	4.6 mg/kgTS 1 H	78 mg/kg TS 1 E	377 mg/kg TS 1 E E	0.0178 mg/kg TS 1 F	6.7 mg/kgTS 1 E	8.2 mg/kg TS 1 E Nb	54 mg/kg 1S 1 H	164 mg/kg IS 1 E	183 multer 1 A	1.00 mg/kg TS 1 SS SS	26 mg/kg TS 1 E E Sr	9.7 mg/kgTS 1 E	mg/kg TS 1 E E	1.6 mg/kg TS 1 E		34 1100 12 1 E	0.86 %TS 1 E	1.2 %TS 1 E	1.0 %TS 1 E	0.09 %TS 1 E	0.14 %TS 1 E MgG	0.018 %TS 1 E Mrd	0.32 %TS 1 E Na2	0.27 %TS 1 E P2C	0.08 %TS 1 E TO	%afTS 1 1 1 Sun	%TS 1 1 1	% 1 W TS*

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Rapport Side 37 (42)

707A Deres belegnelse

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00068	Rappo	t		Δ	0800068	
OKEGBBFR	Side 40 (42)				200KEGB9FF	AL AL
A Time series: Århus	Deres belegnel	se Aarhus 71			BA Time	: series: Århu
	Labnummer	D1008545	8			
etode Teknik/Udf. Enhed	Parameter	Resultat	Måleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed
H	As	31.7	9.7	mg/kg TS	F	н
- E	Ba	1370	102	mg/kg TS		ш
1 E	Be	1.52	0.37	mg/kg TS	-	ш
H	Cd	3.29	0.58	mg/kg TS	1	н
H H	Co	21.6	4.4	mg/kg TS	-	H
ш •-	Cr	513	69	mg/kg TS		ш
1	Cu	7150	1070	mg/kg TS	+	ш
1 1	Hg	0.0727	0.0154	mg/kg TS	1	ı.
П	Mo	99		mg/kg TS	1	ш
E	Nb	99		mg/kg TS		ш
- -	iz	420	91	mg/kg TS		H
	Pb	835	140	mg/kg TS	1	ш
1 A	S*	3550		mg/kg TS	-	A
	Sc	4		mg/kg TS	1	ш
1 S	Sn*	220		mg/kg TS	1	S
ш Ш	Sr	345	24	mg/kg TS	1	ш
т П	Λ	50.4	9.2	mg/kg TS	1	ш
m	N	<60		mg/kg TS	1	ш
μ	Y	13.4	1.8	mg/kg TS	-	ш
	Zn	2820	411	mg/kg TS	•	ш
E	Zr	275	47	mg/kg TS	-	ш
- E	Si02	48.8	3.9	%TS	1	ш
1	AI203	10.0	0.9	%TS	F.	ш
-1 E	CaO	13.5	1.0	STa%	1	ш
1	Fe203	11.7	0.9	%TS	4	ш
1	K20	1.09	0.09	%TS	1	ш
т. Т.	MgO	1.83	0.14	%TS	-	ш
m	MnO	0.160	0.015	%TS	-	ш
1	Na2O	3.71	0.30	%TS	1	ш
E	P205	0.876	0.126	%TS	6	ш
1	TiO2	1.11	0.09	%TS	1	ш
-	Sum	92.8		% af TS	+	÷
1	LOI	3.9		% TS		+
1 W	TS*	98.7		%		M
s	Sb*	88.6		mg/kg TS	1	s

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Rapport Side 39 (42)

D08006 200KEGB

Enhed

D10085452 Resultat Måleusikkerhed (±)

Labnummer

Parameter

Aarhus 91a Deres belegnelse

 $\begin{array}{c} 4.1.4\\ 4.1.63\\ 1.53\\ 2.1.6\\ 2.1.5\\ 2.1.6\\ 2.1.5\\ 2.1.6\\ 2.1.6\\ 2.1.6\\ 2.1.6\\ 1.55\\ 1.2.2\\ 1.2.2\\ 1.2.2\\ 1.3.3\\ 1.2.3\\ 1.3.3\\ 1.3.3\\ 1.0.7\\ 1.0$ Zh Zr Zr Zr Al203 Al203 Ca0 Mg0 Mg0 Mg0 Mg0 Na205 Fr205 Fr205 Ti(02 Sum L01 TS* Sum

11.8 139 0.38 0.38 0.40 4.4 77 372 372 5.82 5.82 5.82 24 226

1.36 25 1.7 1.7 38 38 3.7 3.7 3.7

91.02/02 92.02/02/02 92.02/02 92.02/02/02 92.02/02 92.02/02 92.02/02

mg/kg TS

%TS %TS %TS %TS %TS %TS % TS %

0.09 0.15 0.014 0.26 0.128 0.11

6TS

1.0

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÷	
2	
0	
0	
0	
3	
20	

D0800068

Side 41 (42)

200KEGB9FR

ALS

series: Århus

Time BA Aarhus 89a Deres betegnelse

arameter	Resultat	Måleusikkerhed (±)	Enhed	Metode	Teknik/Udf. Enhed
4s	81.9	21.2	mg/kg TS	1	H
3a	2290	171	mg/kg TS	1	ш
3e	1.69	0.54	mg/kg TS	1	ш
p	1.86	0.35	mg/kg TS	1	н
00	20.4	4.2	mg/kg TS	1	Н
5	630	86	mg/kg TS	1	E
2	2550	366	mg/kg TS	L	ш
B	0.0113	0.0041	mg/kg TS	ł	e e
No	22.3	7.3	mg/kg TS	F	ш
q	14.6	7.7	mg/kg TS	*	ίΩ.
	377	82	mg/kg TS	1	Ŧ
q	1070	621	mg/kg TS	1	ш
1	5530		mg/kg TS	-	A
0	2.31	1.52	mg/kg TS	÷	ш
"u"	292		mg/kg TS		s
	431	30	mg/kg TS	ł	ш
	71.2	12.3	mg/kg TS	1	E
>	<60		mg/kg TS	1	u
	14.8	1.9	mg/kg TS	1	u.
u	3110	451	mg/kg TS	1	ш
	236	40	mg/kg TS	+	ш
102	44.9	3.6	%TS	+	ш
1203	9.89	0,88	%TS	1	ш
ao	16.1	1.2	%TS	÷	u.
e203	12.8	1.0	%TS	1	ш
20	1.02	0.09	%TS	F	ш
lgo	2.07	0.16	%TS	e	ш
Ino	0.198	0.019	%TS	-	μ.
a20	3.71	0.30	%TS	1	ш
205	0.870	0.126	%TS	1	ш
102	1.31	0.10	%TS	+	ш
um	92.9		% af TS	1	1
0	3.9		% TS	1	F
S*	96.7		%	1	W
••	85.0		mg/kg TS	-	s

Rapport

Side 42 (42)

D0800068

200KEGB9FR

(SIN)

efter parameternavn indikerer ikke akkrediteret analyse.

Bestemmelse af sporstoffer samt oxider inkl. oplukning. Oplukning: For bestemmelse af As, Cal, Co, Cu, Hag, NI, Pb, S og Zn tørres prøven ved 50°C og opløses i sallpetersyren/yrdrogenperoxid i lukkede teflonbeholdere i mikroovn. (indeholdet korrigeres til TS 105°C.) svinge ammer bestemmels after snelle med tilhummelaborat efterfulgt af opløsning i fortyndet. saltpetesyre, Metode

Analysen er gennmeført iht. EPA metoder 200.7 (ICP-AES) og 200.8 (ICP-QMS/SFMS).

	A	ш.	LL.	т	s	>	3	-
Teknik / Udforende enhed ¹	Mäling udføres med ICP-AES Akk. Lab Reg.nr. 1087.	Måling udføres med ICP-AES Akk. Lab Reg.nr. 1087.	Måling udføres med AFS Akk. Lab. Reg.nr. 1087.	Maiing udføres med ICP-SFMS Akk. Lab. Reg.nr. 1087.	Måling udføres med ICP-SFMS Akk. Lab. Reg.nr. 1087.	Vátkemi	Vādkemi	Akk. Lab. Reg.nr. 1087

Følgende gælder for tekniske enheder i ALS Scandinavia AB:

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Appendix VIII

Modeling of heterogeneous condensation

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Model approach

Task: A physical model for heterogeneous condensation on a <u>size distributed</u> fly ash is developed and coupled with an existing equilibrium model.

Approach: A two-compartment system (as described by Sørum et al., 2003): (1) Release from the grate: Vaporization of elements. (2) Condensation of elements in the flue gas channel as a function of T. The idea is sketched in Figure 1.



The model is one-dimensional in time and consists of (1) specifying the initial conditions in the system (2) computing the target element condensation as a function of T.

Gas/solid equilibrium is calculated at each step using the commercial equilibrium model FactSageTM (version 5.2), and interactions between vapor phase and particle surface are calculated by <u>fluxes</u>, with units of moles per second per particle.

While most other modeling studies have considered <u>uniform sized spherical particles</u> and <u>coal fly</u> <u>ash</u>, our study consider <u>size distributed</u>, <u>non-spherical</u>, <u>waste incineration fly ash</u>.

A flow chart of our model is outlined in Figure 2. The model is inspired by a physical model derived by Senior et al. (2006), considering arsenic partitioning in coal-fired power plants.

Assumptions, limitations and uncertainties are addressed and discussed below (with references to relevant literature).

Model assumptions and limitations

Heterogeneous condensation is defined as condensation on the surface of existing particles.

In the present model heterogeneous condensation is assumed to be the <u>only</u> mechanism by which particulate matter becomes enriched with heavy metals. Surface reactions are not included, neither are homogeneous condensation of trace metals (formation of new aerosol particles by nucleation).



Figure 2. Flow chart for modeling of heterogeneous condensation on a size distributed fly ash. Target element: Pb, Zn, Cu, Cd etc.

Particle growth:

The net source of particles of size v, at position x, and time t $[1/m^3s]$, can be described by *the particle growth equation* (a) (Linak & Wendt, 1994):

 $R_v = R_{nucl} + R_{coag} + R_{cond}$ (nucleation + coagulation + condensation) (a)

The volatile trace metals condense only in <u>the cooler zones</u> of the furnace, therefore nucleation and coagulation can be ignored.

Vapor phase – particle surface interactions. The condensation flux equation:

Interactions between vapor phase and particle surface are calculated by <u>fluxes</u>. The mechanisms depend on the gas-phase transport regime: free molecular (submicrometer) or continuum (supermicrometer).

The condensation flux equation, R_{cond} , $[mol/m^3 s]$, on uniform sized spherical particles of diameter d_p [m] in the continuum (supermicrometer) regime is diffusion limited and given by equation (b) (Frandsen, 1995; Neville & Sarofim, 1985, Senior et al., 2006). To solve for a <u>size distributed fly</u> ash, the particle size distribution is <u>discriminated</u> and the equation is solved for each size fraction.

$$R_{cond} = F_i(d_p) = \frac{2 \cdot \pi \cdot d_p \cdot D_{Xm}}{R \cdot T} \cdot P_{LOAD} \cdot (p_{\infty} - p_s) \cdot \left[\frac{1 + Kn}{1 + 1.71 \cdot Kn + 1.333 \cdot (Kn)^2}\right]$$
(b)

where D_{Xm} is the molecular diffusivity of X in a multicomponent gas, m, p_{∞} and p_s is the bulk and surface pressure of X, $P_{LOAD}[1/m^3]$ is the particle loading of the gas, and $Kn = 2\lambda/d_i$ is the Knudsen number. The mean free path, λ (computed by (c)) is ~0.5 µm for combustion gases (Senior et al., 2006; Neville & Sarofim, 1985):

$$\lambda = \frac{\mu}{\rho} \left(\frac{\pi MW}{2RT}\right)^{1/2} \tag{c}$$

where the molecular weight, viscosity, and density of the gas are calculated using the overall gas composition

The term in the brackets in equation (b) is the Fuchs-Sutugin correction for the non-continuum behavior of small particles. D_{Xm} can be estimated for any molecular component X in m using *an empiric equation* (d) of Fuller, Schettler and Giddings (Fuller et al., 1969). The empirical equation of Fuller, Schettler, and Giddings utilizes empirical constants derived from experimental data:

$$D_{AB} = D_{BA} = \frac{0.00143T^{1.75}}{PM_{AB}^{1/2} \left[\left(\sum_{V} \right)_{A}^{1/3} + \left(\sum_{V} \right)_{B}^{1/3} \right]^{2}},$$
 (d)

Where D_{AB} is in cm²/s, P is in atm, T is in K. M are the molecular weights (g/mol). V are the atomic diffusion volumes.

The condensation flux in the free molecular regime (submicrometer particles) is given by (e) (Senior et al., 2006; Neville & Sarofim, 1985):

$$R_{cond} = \frac{\psi \pi d_p^2 (p_{\infty} - p_s)}{(2\pi M W R T)^{1/2}} \cdot P_{LOAD} , \text{ MW} = \text{molecular weight (g/mol)}$$
(e)

where Ψ (psi) is the accommodation coefficient, or the fraction of successful collision with the particle surface, taken as unity (if spherical particles. Corrections (> 1) must be made for non-spherical particles).

For the submicrometer particles an allowance must be made for the Kelvin effect (increase in equilibrium pressure with decrease in droplet diameter) (Neville & Sarofim, 1985).

Condensation vs. particle size:

A simple geometric model for volume percentage of condensed surface layer on a spherical fly ash particle is outlined on Figure 3 (Nutalapati et al., 2007).



Figure3: Geometric model for volume percentage of condensed surface layer on a spherical fly ash particle

The thickness of the condensed surface layer is in the <u>nm range (0.01 - 50 nm)</u> for coal combustion fly ash (Davison et al., 1974), but in the <u>µm range (0.1 - 0.5 µm)</u> for biomass combustion fly ash (Nutalapati et al., 2007), due to a high percentage of ash vaporization in biomass combustion. <u>Waste incineration fly ash</u> may be similar to biomass combustion (high degree of ash vaporization)

Dependence of concentration on particle size – the physical condensation theory:

The mass fraction of a condensed species, $Wi(d_p)$ is given by the mass condensed (in time, t), on the particle of size d_p divided by the particle mass.

Different condensation mechanisms will yield different dependencies of $F(d_p)$ on d_p , and it can be shown that for Kn<<1 (coarse fraction particles), Wi(d_p) ~ $1/d_p^2$, while for Kn>>1 (fine fraction particles), Wi(d_p) ~ $1/d_p$ (neglecting the Kelvin effect) (Neville & Sarofim, 1985, Davison et al., 1974).

Several studies of coal fly ash have, however, shown that experimental data are not always in consistence with this physical condensation theory (Neville and Sarofim, 1985; Davison et al., 1974). Often the trace elements (As, Ni, Cd etc.) better obey a $1/d_p$ dependence, even for larger particles, suggesting that surface condensation alone can not account for the trace element distribution (chemical reaction mechanisms are also involved).

Fitting the FASAN data / validation of the model

The FASAN fly ash was divided (experimentally) into 10 + 3 discrete size fractions: 10 aerosol fractions from LPI measurements and 3 fly ash sub samples (filter, filter house, undivided). Assuming that that the 10 aerosol fractions can be treated as one fraction with an average element concentration and an average geometric mean diameter (GMD) of 0.393 µm, these discrete fractions were used for data fitting. The condensation flux was calculated using equation (c) (condensation in the continuum regime, with correction for non-continuum behavior of small particles).

The Pb concentration in the FASAN ash as a function of particle diameter is illustrated in Figure 11 (measured and calculated values). The calculated value is a "first rough estimate" of the condensation flux, assuming an initial Pb concentration of 0.005 ppm(v) in the gas phase (based on gas data from Bøjer et al., 2008). At the point of sampling (T ~175 °C), all Pb is assumed to be

condensed (as $PbCl_2(s)$). The initial Pb concentration (C₀) in the coarse particles is set to 697 mg/kg (residual concentration), while C₀ is set to 0 in the aerosols.



Figure 11: Pb concentration in FASAN ash as a function of particle diameter. Measured and calculated values

It is seen that the Pb concentration better obeys a $1/d_p$ dependence as compared to a $1/d_p^2$ dependence (the same trend is found for other trace elements as Hg, Cd, Zn and Cu, results not shown). The heterogeneous condensation model seems to correctly predict the trends in the experimental data, although it is also seen that the model may tend to over-estimate the Pb concentration in the fine particles.

Model uncertainties

Especially the quality of the input data - specifying the initial condition - is a very critical factor, which need further attention. In particular, there is a lack on reliable data for the degree of vaporization of elements from the grate during waste incineration. This could be a topic for future research projects.

The present model does not allow for chemical surface reactions. Including chemical surface reactions may improve the model, as our (and others) results indicate that surface condensation alone can not account for the trace element distribution.

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