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MARS-RELEVANT PHOSPHATE MINERALS AND IMPLICATIONS FOR

MARTIAN HABITABILITY

by

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Bachelor of Science in Earth and Planetary Sciences University of New Mexico 1995

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A dissertation submitted in partial fulfillment of the requirements for the

Doctor of Philosophy - Geoscience

Department of Geoscience College of Sciences The Graduate College

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THE GRADUATE COLLEGE

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Christopher Adcock

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August 2014

ABSTRACT

Mars-Relevant Phosphate Minerals and Implications for Martian Habitability

by

Christopher T. Adcock

Dr. Elisabeth Hausrath, Examination Committee Chair Assistant Professor of Geoscience University of Nevada, Las Vegas

This dissertation is comprised of three studies focused on martian phosphate availability, with an introductory chapter introducing and linking the three studies. Chapter two is on the subject of merrillite synthesis. Merrillite is an extraterrestrial Caphosphate mineral similar to the mineral whitlockite and is found as a dominant primary phosphate mineral in martian meteorites. The chapter includes methods of whitlockite and merrillite synthesis as well as a detailed characterization of the produced minerals and a mechanism by which charge balance can be maintained when merrillite is synthesized through dehydrogenation of whitlockite.

Chapter three presents the results of kinetic and thermodynamic studies on the Mars-relevant minerals chlorapatite and merrillite, as well as the more terrestriallyrelevant minerals whitlockite and fluorapatite. The results of these studies indicate that the dominant primary Ca-phosphate minerals on Mars possess higher solubilities that could lead to more than twice the phosphate concentration in solution. Dissolution rates for the Mars-relevant minerals derived in the study, when combined with the higher martian phosphorus abundance, could result in phosphate release rates of up to 45x faster for a given set of aqueous conditions on Mars when compared to Earth. The implications of the results for past or present martian habitability are discussed.

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In Chapter four, reactive transport modeling was applied to investigate the transport and mobility of phosphate under martian conditions. The kinetic and thermodynamic data derived in Chapter three were combined with Mars mission data, results from an investigation of Mars analog basalts at Craters of the Moon National Monument in Idaho, and previously published data to inform a reactive transport code and model dissolution profiles measured by Mars Exploration Rover (MER) Spirit in Wishstone class rocks. The modeling results suggest phosphate release into near-neutral waters occurred at some point in the past at Gusev Crater on Mars with positive implications for past martian habitability.

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A Ph.D. is a tremendous undertaking for an individual and a task I do not believe a single individual can accomplish alone. Thus, it is unfortunate that this dissertation allows for only a single author. In fact, it is somewhat misleading. While the work contained herein is my own, it certainly would never have been completed without the help, support, and contributions of many individuals. The list of those I need to acknowledge is a long one, and I apologize that I will fail to mention everyone - I am sure.

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Elisabeth Hausrath has been my committee chair. I don't really have the room here to express my thanks to her. She is always positive, supportive, and understanding.

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In doing this research I had an office. And, in that office, I had office-mates. Office-mates are like siblings after awhile. We argued and fussed, we had fun together, we joked with each other and commiserated - we competed with each other often. We also helped each other often, and not simply with school or research work. Seth Gainey, Mike Steiner, Fritz Freudenberger, Valerie Tu, Britney Downs, Julie Baumeister - I thank you all for the great conversations, debates, technical support, and a list of other things too long to include. I also want to thank Brenda Buck, who helped me get my mind back into research science after 5 years of consulting and applied science. And she did that in a single class on soils. Scott Nowicki deserves my thanks as well. His views on what education should be and how I should approach my own career were invaluable. He also helped me keep a finger on my broader interest and I regret we didn't have more time for thought provoking discussions.

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And then there is Miranda Jane Tillery Adcock (a.k.a. Pandora or Panda), my daughter. When I started this she couldn't walk. Now she has finished kindergarten. She is no small part of the reason I attempted a Ph.D. I do not know if one dedicates a

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dissertation, but if they do, this one is for Miranda, because it has been a true adventure. I do not know what you will be when you "grow up" Miranda, but let it be adventurous. Remember the words of the crew of the Galaxy Quest, "Never give up, never surrender". Just try to plan a little better than they did.

Lastly I have to thank all of the sources of funding that made this research even possible. This material is based upon work supported by the National Aeronautics and Space Administration (NASA) Mars Fundamental Research Program grant NNX10AP58G and NASA grant no. NNX10AN23H issued through the NASA training grant National Space Grant College and Fellowship Program to E. M. Hausrath, as well as Nevada Space Grant Consortium fellowships to C. T. Adcock, a GSA research grant to C. T. Adcock, and by a cooperative agreement through the National Nuclear Security Administration under the Stewardship Science Academic Alliances program through DOE Cooperative Agreement #DE-NA0001982 to P. M. Forster and O. Tschauner. The Advanced Light Source was used as part of this work and is supported by the Director, Office of Science, Office of Basic Energy Sciences, of the U.S. Department of Energy under Contract No. DE-AC02-05CH11231. C. T. Adcock also received funding from the UNLV Graduate and Professional Student Association and the UNLV Department of Geoscience.

On to the science...

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CHAPTER ONE

INTRODUCTION AND OVERVIEW

The possibility of the development and persistence of life on Mars has been a topic of scientific interest for centuries (e.g. Huygens, 1698). In the last five decades, dozens of unmanned missions to the planet have returned data that have significantly reshaped the general understanding of Mars and its potential to have harbored life, past or present. The present day martian surface has been revealed as a generally inhospitable, cold, dry, and radiation-bombarded environment. However, Mars also possesses abundant geomorphic and mineralogical evidence of significant surface and near-surface waters in the past (Carr, 1996; Squyres et al., 2004; Baker, 2006; Milliken et al., 2008; Treiman, 2008; Murchie et al., 2009; Milliken et al., 2010; Grotzinger et al., 2013; Williams et al., 2013; Gainey et al., 2014). Provided the availability of bio-essential chemistry, these aqueous environments may have been habitable places conducive to the origin and development of life. Further, present day water in the form of ice currently exists on Mars (Squyres et al., 1992) and liquid water may exist in the subsurface (Malin and Edgett, 2000). If these more localized environments also posses the necessary bioessential chemistry, habitable niches for martian life might still persist.

Phosphate is among the nutrients considered critical for all known life (Johnston et al., 1934; Miller and Urey, 1959; Wald, 1962, 1964; Madigan et al., 2000). The ion is a component in ATP, DNA, RNA, phospholipid cell membranes and required in a number of fundamental biochemical reactions (Madigan et al., 2000; Pasek and Kee, 2011). Phosphorus, either as phosphate or a more reduced species such as phosphite,

may have also been crucial in pre-biotic reactions that led to the origin of life on Earth (Schwartz, 2006; Powner et al., 2009; Pasek and Kee, 2011). Thus, the availability of phosphate may be a determining factor for an environment to be suitable to maintain life (i.e. habitable). In fact, in long term ecosystems on Earth, phosphate availability is often a biologically limiting factor (Chadwick et al., 1999). If life ever arose on Mars, the availability of phosphorus likely played a significant role.

Though Mars is rich in phosphate (5-10x Earth) (Wanke and Dreibus, 1988; Brückner et al., 2008; Ming et al., 2008; Adcock et al., 2013), the presence of phosphate alone is not equivalent to phosphate availability. Unlike other bioessential nutrients, phosphate has no significant volatile phase and remains locked in phosphate-bearing minerals within rocks until water/rock interactions release it into the environment (Filippelli, 2002). The goal of the research presented here, therefore, is to investigate the mobility and transport of phosphate on Mars and assess martian phosphate availability and the implications for martian habitability in the past, present and, if we choose to make extended visits to Mars, the future. The bioessential nature of phosphate requires that any long term visits to the planet have a secure source of the nutrient. Further, because phosphate is released primarily through aqueous processes, phosphate mobility and transport have implications for past aqueous conditions on Mars. The chapters that follow document three studies focused on the study of martian phosphate mobility and transport.

Chapter two focuses on the synthesis and characterization of the Mars-relevant phosphate mineral merrillite [Ca₉(Na,Fe,Mg)(PO₄)₇], as well as the similar mineral whitlockite [Ca₉(Fe,Mg)(PO₃OH)(PO₄)₆]. Merrillite is an extraterrestrial Ca-phosphate

mineral similar to whitlockite and is found as a dominant primary phosphate mineral in martian meteorites (McSween and Treiman, 1998). Though merrillite has been previously synthesized (e.g. Gopal and Calvo, 1972; Gopal et al., 1974; Hughes et al., 2008), this study developed methods to produce relatively large quantities of end-member Fe-, Mg- and mixed Fe/Mg-merrillite in an economical manner. The synthesized minerals were also characterized in detail by electron microprobe, single-crystal X-ray diffraction, and synchrotron X-ray diffraction, which allowed for the development of a potential mechanism by which merrillite maintains charge balance during the dehydrogenation of whitlockite. The developed methods are documented in chapter two along with the characterization results of the synthesized materials, including atomic structures of the minerals. This chapter in the form of a manuscript, was accepted and published in American Mineralogist in a July 2014 issue (Adcock et al., 2014).

Chapter three documents the results of thermodynamic and kinetic experiments performed on fluorapatite, synthetic chlorapatite, and the synthetic whitlockite and merrillite produced by methods documented in chapter two. On Earth, the most common primary phosphate mineral is fluorapatite $[Ca_5(PO_4)_3F]$. The dissolution of fluorapatite in igneous rocks is a major source of inorganic phosphate available for biological reactions, and because it is biologically important, it has been well studied (Valsami-Jones et al., 1998; Welch et al., 2002; Guidry and Mackenzie, 2003; Köhler et al., 2005; Chaïrat et al., 2007; Harouiya et al., 2007). In contrast, meteorite evidence suggests that the most abundant primary phosphate minerals on Mars are Cl-rich apatite $[Ca_5(PO_4)_3Cl]$ and the mineral merrillite $[Ca_9(Na,Fe,Mg)(PO_4)_7]$, an extraterrestrial mineral similar to whitlockite $[Ca_9(Fe,Mg)(PO_3OH)(PO_4)_6]$ (McSween and Treiman, 1998). Few kinetic

(e.g. dissolution rates) and thermodynamic (e.g. solubility) data exist for these Marsrelevant minerals. The experiments conducted in chapter three allowed for the derivation of dissolution rates laws and solubilities for these Mars-relevant minerals. The resulting data indicate faster dissolution rates for the Mars-relevant minerals that, when the overall higher phosphorus abundance on Mars is taken into consideration, could lead to phosphate release rates as much as 45 times faster on Mars for a given set of aqueous conditions. Solubility measurements of the Mars-relevant minerals also suggest final concentrations of phosphate in aqueous systems on Mars could be as much as twice that of Earth for a given set of conditions. The potential implications for phosphate availability and martian abiogenesis are discussed. This chapter, in the form of a manuscript, was accepted and published in the journal Nature Geoscience (Adcock et al., 2013).

Finally, in chapter four the measured thermodynamic and kinetic data derived in chapter three are applied to Mars-specific reactive transport modeling of Wishstone class martian rocks to model phosphate transport under martian conditions. The Mars Exploration Rover (MER) Spirit encountered Wishstone class rocks in the Columbia Hills of Gusev Crater. Alpha particle X-ray spectrometry (APXS) was performed on two rocks of this class (i.e. Wishstone and Champagne) and Mini-TES data indicate that the rock class is common among the float rocks of Columbia Hills. Comparisons of APXS brushed surface analyses and analyses taken on fresh surfaces of the rocks, after treatment with Spirit's Rock Abrasion Tool (RAT), indicate the loss of Ca and P from the rocks in ratios consistent with a Ca-phosphate mineral. Reactive transport modeling was performed to investigate the conditions under which these somewhat unique profiles

might have formed. The findings of this study indicate that a period of near-neutral aqueous conditions may have persisted at Gusev Crater at some point in the past. This has positive implications for the potential origin and development of life on Mars. This chapter has been submitted for publication to a peer reviewed planetary science journal.

CHAPTER TWO

SYNTHESIS AND CHARACTERIZATION OF THE MARS-RELEVANT PHOSPHATE MINERALS FE- AND MG-WHITLOCKITE AND MERRILLITE AND A POSSIBLE MECHANISM THAT MAINTAINS CHARGE BALANCE DURING WHITLOCKITE TO MERRILLITE TRANSFORMATION

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Abstract

Merrillite $[Ca_9NaMg(PO_4)_7]$ occurs as a dominant primary Ca-phosphate mineral in martian meteorites and therefore presumably also on Mars. The mineral is an important phase in exploring differences in geologic processes between Earth and Mars, and also has astrobiological implications due to its potential role as a significant source of the bio-essential nutrient phosphate. Merrillite does not occur terrestrially as a discrete mineral phase, making it difficult to obtain for Mars-relevant studies. It can, however, be synthesized from a similar terrestrial mineral, whitlockite (natural or synthetic), through dehydrogenation. Here we present methods for synthesizing relatively large quantities (0.5 g or greater per batch) of coarse crystalline (75 μ m+) Mg-whitlockite, Fewhitlockite, mixed Fe/Mg-whitlockites, and from these synthesized minerals produce Mg-merrillite, ferrous and ferric Fe-merrillite and ferrous and ferric mixed Fe/Mgmerrillite. Chemistry and atomic structures of synthesized Fe- and mixed Fe/Mgwhitlockite and ferrous and ferric Fe- and mixed Fe/Mg-merrillite resulting from singlecrystal X-ray diffraction, infra-red spectroscopy, and electron microprobe analyses are presented. We also present a mechanism for maintaining charge balance during the formation of merrillite from whitlockite. Our results shed light on these mineral structures for future martian studies, and provide methods for creating coarse crystalline merrillite for use in Mars-relevant thermodynamic, kinetic, soil/dust simulant, crystallographic, astrobiological, and other studies.

Introduction

Merrillite is an anhydrous calcium phosphate mineral that occurs in lunar samples and meteorites, including martian meteorites (Jolliff et al., 1993; McSween et al., 1996; Xie et al., 2002; Terada et al., 2003). The IMA formula for merrillite is Ca₉NaMg(PO₄)₇, though merrillite can also be iron-bearing (Jolliff et al., 2006). Synthetic merrillite also lacks Na⁺ and charge balance is maintained by additional Ca²⁺ [i.e. Ca_{9.5}Mg (PO₄)₇]. Merrillite is of particular interest to martian studies because it is the dominant primary phosphate mineral in martian meteorites (McSween et al., 1996; McSween and Treiman, 1998), and thus presumably also on Mars. This is in contrast to Earth where fluorapatite [Ca₅(PO₄)₃F] is the dominant terrestrial primary phosphate mineral and the nearest analog to merrillite, the mineral whitlockite [Ca₉Mg(PO₃OH)(PO₄)₆], occurs as a component of teeth and bones in vertebrates, but outside of biological systems is a rare alteration product of phosphate pegmatites. Merrillite is also the major carrier of Rare Earth Elements (REE) in martian meteorites (McSween et al., 1996; McSween and Treiman,

1998; Shearer et al., 2011a) and likely one of the major sources of the phosphorus observed in phosphorus rich martian soils and dust (Goetz et al., 2005; Greenwood and Blake, 2006; Brückner et al., 2008). Phosphate minerals such as merrillite and apatite are thought to react in late stage martian magmas (McCubbin and Nekvasil, 2008; Filiberto and Treiman, 2009a; Gross et al., 2013b), and their presence and intergrowth relationships with each other within martian rocks (or meteorites) can be useful indicators of crystallization and melt evolution, including the evolution of both water and halogen budgets over time (Patiño Douce and Roden, 2006; Patiño Douce et al., 2011; Gross et al., 2013a; Gross et al., 2013b). Such use of merrillite/apatite intergrowth relationships may also be applicable to other differentiated bodies outside of Mars (Shearer et al., 2011b). Phosphate minerals have also been shown to be altered in acid vapor environments, potentially recording important characteristics of the reacting environment (Lane et al., 2008; Hausrath et al., 2013; Hausrath and Tschauner, 2013). In addition, phosphate is a required component in fundamental biologic reactions as well as considered critical to reactions that may have led to life on Earth (Gulick, 1955; Westheimer, 1987; Schwartz, 2006). The alteration of phosphate minerals such as merrillite has important implications for phosphate availability and the potential for martian abiogenesis and development of life on that planet (Adcock et al., 2013). Thus merrillite is an important phase in exploring the differences in petrogenesis, mantle evolution, surface processes and other geologic processes between Earth, Mars, and possibly other bodies (McSween et al., 1996; Treiman, 2003; Shearer et al., 2011a).

The importance of merrillite in martian processes makes a readily available source of the mineral desirable for laboratory study. However, single-phase natural

merrillite only occurs in extraterrestrial materials and thus studies requiring merrillite must obtain the mineral directly from meteorites or lunar samples, or through synthesis. Here we present, methods for producing fairly large quantities of crystalline Fe/Mgbearing merrillite for use in Mars-relevant studies, the chemistry and atomic arrangements of the synthetic minerals, and a proposed mechanism for maintaining charge balance during the formation of merrillite from whitlockite.

Background

Merrillite was first described in meteorites by Merrill (1915; 1917). Wherry (1917) was the first to suggest applying "merrillite" as the name of the extraterrestrial mineral. Merrillite is chemically and structurally similar to the terrestrially occurring mineral whitlockite; in fact, though merrillite is not found terrestrially as a discrete mineral phase outside of biological systems, it does occur as a minor component within whitlockite, with which it forms a solid solution (Hughes et al., 2008). In biological systems, merrillite forms in dental calculi and urinary stones (Dorozhkin and Epple, 2002). The similarities of merrillite and whitlockite have resulted in overlapping use of the mineral names within the literature, though they are not strictly the same mineral. Whitlockite was first described by Frondel (1941) as a late stage calcium phosphate mineral associated with pegmatites. Frondel and Prien (1946) later applied the term "whitlockite" to synthetic tricalcium phosphate $[\beta$ -Ca₃(PO₄)₂ or commonly β -TCP] because of chemical and structural similarities of that phase. The similarity of merrillite, whitlockite, and β -TCP led Fuchs (1962) to argue, based primarily on powder X-ray diffraction data, that the three phases were the same and that the term "merrillite" should

be abandoned in favor of "whitlockite". Subsequent research conclusively demonstrated that merrillite and β -TCP, while isostructural and very similar to whitlockite, are distinct, and lack the required hydrogen of whitlockite (Gopal and Calvo, 1972; Dowty, 1977; Rubin, 1997; Hughes et al., 2008). Dowty (1977) suggested "merrillite" as the more appropriate name for the hydrogen-free form found in meteorites. However, because of the historical usage of "whitlockite" and the difficulty of confirming the hydrous or anhydrous nature of the phases, the term "whitlockite" continues to be occasionally used synonymously or interchangeably when describing merrillite (e.g. Ruszala and Kostiner, 1980; Jolliff et al., 1993; McSween et al., 1996; Terada et al., 2003; Orlova et al., 2009). This is especially the case in biomedical fields where synthetic merrillite compounds, such as β -TCP, are important in bio-ceramics and are often referred to as whitlockite or β -whitlockite (e.g. Jarcho et al., 1979; Lagier and Baud, 2003). In this paper, as in some other studies that discuss both merrillite and whitlockite (e.g. Hughes et al., 2006; Jolliff et al., 2006) the term "merrillite" is used to describe the anhydrous extraterrestrial (and synthetic) Ca-phosphate mineral, and the term "whitlockite" to refer to the hydrogenated terrestrial mineral form.

The most common method of synthesizing merrillite powder is solid state sintering, which involves heating relevant powdered calcium phosphate chemistry to >1025 °C. The method typically requires several cycles of regrinding the material during the synthesis process (e.g. Lazoryak et al., 1996; Belik et al., 2002; Orlova et al., 2009), and therefore produces a very fine powder. This general method is used to produce synthetic Ca-endmember merrillite materials including commercially available β -TCP powders and ceramics. The same methods have been used to synthesize other types of

merrillite powders (Teterskii et al., 2007; Orlova et al., 2009), including Mg-merrillite and ferric Fe-merrillite (Lazoryak et al., 1996). A drawback to this general method is that it produces only very fine powders, which may not be preferred for some studies (e.g. dissolution kinetics, crystallography, soil simulants). The commercial version is also an endmember lacking in Fe or Mg and thus chemically dissimilar to the merrillite commonly found in lunar samples or martian meteorites (Jolliff et al., 2006).

Another method of synthesizing merrillite is through the heating of whitlockite to >1000 °C for at least 24 hours to drive off the "essential hydrogen" (Gopal and Calvo, 1972; Hughes et al., 2008). This method has the advantage of being simple to perform and has been applied to both synthetic Mg-whitlockite and natural whitlockite crystals to produce crystalline merrillite rather than powders. However, natural terrestrial whitlockite is itself uncommon and generally restricted to small quantities associated with pegmatites. Thus, in order to synthesize quantities of high purity crystalline Mg- and Fe-merrillite for use in Mars-related kinetic, thermodynamic, astrobiological, or other studies, the synthesis of Mg- and Fe-whitlockite in large quantity is required.

Whitlockite can be synthesized by the Rowles method (Rowles, 1968) which involves hydrolysis of brushite in magnesium chloride solution to create Mg-whitlockite. The method can create pure Mg-whitlockite crystalline material and large output masses (500+ g). The crystalline material generated by this method is very fine with specific surface areas (SSA) of 3-7 m²/g (geometrically equivalent to 0.3-0.6 µm diameter average particle size)(Hamad and Heughebaert, 1986) making it of limited usefulness in kinetic, simulant, crystallographic, or other studies where coarser crystalline material is preferred. Hydrothermal methods, however, can create coarser crystalline material.

Gopal et al. (1974) synthesized coarse, up to 500 µm per side, crystalline Mg-whitlockite using such a method. The method required the pre-synthesis and processing of a calcium phosphate gel and a large (6mm x 100mm) single-use gold tube to contain the reagents during heating to 300 °C, making it less practical for synthesizing in quantity. Hughes et al. (2008) inadvertently produced a small amount (i.e. "a few crystals") of Mg-whitlockite by similar methods while hydrothermally recrystallizing hydroxyapatite $[Ca_5(PO_4)_3OH]$. Their method involved first synthesizing hydroxyapatite powder from a calcium phosphate solution and suspending it in 10ml of distilled water with the pH adjusted to 2.79 using phosphoric acid. The resulting suspension was sealed in a reusable 23ml Teflon lined acid digestion vessel and heated to 240 °C for 5 days. The output mass from the hydrothermal treatment consisted mainly of elongate, recrystallized hydroxyapatite along with a small number of large, well-formed, Mg-whitlockite crystals. The apparent cause of whitlockite synthesis was Mg contamination from an undetermined source. Here we present methods based on Hughes et al. (2008) to generate large quantities of single phase Mg-whitlockite and, from that whitlockite, Mg-merrillite by intentionally supplying Mg^{2+} to the synthesis. We also developed methods to generate large amounts of crystalline Fe-whitlockite, as well as mixed Fe/Mg-whitlockites, and ferrous, ferric, and mixed Fe/Mg-merrillites. Atomic parameters are presented for ferrous, ferric, and mixed merrillite and Fe- and mixed whitlockite, as well as a proposed mechanism for maintaining charge balance during the formation of merrillite from whitlockite.

Methods

Whitlockite synthesis

Based on preliminary experiments (Table S1 in Appendix I), the following synthesis methods were developed for Mg- and Fe-whitlockite production: a solution containing 90ml of high purity (18.2 M Ω) water, laboratory grade hydroxyapatite (Spectrum, reagent grade), and either magnesium nitrate hexahydrate for Mg-whitlockite (J.T. Baker, ACS grade) or troilite for Fe-whitlockite (Alfa Aesar, Fe[II]S 99%) were combined in a 125ml Parr acid digestion vessel (Parr #4748) with an acid washed Teflon liner. Specific volumes and masses are contained in Table 1. Once the solution was mixed, it was acidified to pH <2.8 with concentrated phosphoric acid (Alfa Aesar, ACS grade). The vessel was then sealed and incubated in an oven at 240 °C for 7 days. At the end of 7 days, the vessel was removed from the oven and quenched in a water bath in an effort to prevent any further reaction. After cooling, the vessel was opened and the solution decanted leaving the solids. Solid material was rinsed from the vessel using ethanol, allowed to air dry for 24 hours, weighed, and inspected by optical microscopy for preliminary phase identification. Impurities, first determined by powder XRD and then optically by morphology thereafter, were typically hydroxyapatite and monetite (CaHPO₄). In the case of Fe-bearing whitlockite, additional opaque materials presumed to be Fe-phases occurred in quantities too minor for powder XRD analysis. These impurities were primarily confined to the $<75 \mu m$ fraction and output masses were brush sieved on 200 mesh screen to remove that size fraction. Crystals were confirmed as Mgor Fe-whitlockite by Scanning Electron Microscopy (SEM) with Energy Dispersive

Spectroscopy (EDS), Electron Microprobe Analysis (EMP), Powder X-ray Diffraction (XRD), and Single Crystal X-ray Diffraction (SC-XRD).

Based on results from endmember experiments described above, mixed Fe/Mgwhitlockites were also successfully synthesized by combining both troilite and magnesium nitrate hexahydrate (J.T. Baker ACS grade, and Alfa Aesar Fe[II]S 99%, respectively) in the solution used within a single synthesis experiment. Table 1 contains the specific masses and volumes used to produce the mixed Fe/Mg-whitlockite that was further characterized in this study and it includes sodium (as NaCl) as part of the solution. Subsequent experiments indicated that the addition of NaCl in this method has no observable effect and can be omitted (see Results and Discussion, and Appendix I Supplementary Discussion and Table S1), but it is included in Table 1 as this was the material further characterized in this study. Outside of solution chemistry, all other parameters used in mixed Fe/Mg-whitlockite synthesis were the same as for the endmember whitlockite synthesis (e.g. temperature, acidity, incubation time). Impurities found in mixed Fe/Mg synthesis batches (typically large polycrystalline aggregates, hydroxyapatite and opaque phases as determined optically) were primarily confined to the $<75 \ \mu m$ and $>150 \ \mu m$ fractions and output masses were brush sieved to isolate the 75-150 µm size fraction. Remaining crystalline material was confirmed as Fe/Mgwhitlockite by EMP and SC-XRD.

Merrillite synthesis

Merrillite can be synthesized from whitlockite (natural or synthetic) by dehydrogenation through heating to >1000 °C for at least 24 hours (Gopal et al., 1974;

Hughes et al., 2008). In this study, crystalline Mg-merrillite was synthesized from synthetic Mg-whitlockite by heating to 1055 °C (+/-5 °C) for a minimum of 24 hours in a loosely covered 10ml Pt crucible nested within a larger (100ml) alumina crucible for easier handling. After heating, the crucibles were removed and allowed to air cool. After cooling, resulting materials, typically white to clear in color, were removed from the crucible using a synthetic brush.

Crystalline ferric Fe-merrillite was synthesized in an identical manner as the Mgmerrillite described above, in a Pt crucible in ambient atmosphere. The process produces an orange/red colored material. Ferrous Fe-merrillite was synthesized in an evacuated (10^{-3} Torr) sealed SiO₂ tube in place of the Pt crucible to inhibit oxidation of Fe from ferrous to ferric in the Fe-whitlockite. Tubes were purged with argon three times to remove oxygen before final evacuation and sealing. SiO₂ glass wool (average 4 µm diameter) was used to prevent the loss of material from the tubes during purge and evacuation cycles. Whitlockite heat treated in this manner ranged in color from white/translucent to black and nearly opaque.

Mixed Fe/Mg-merrillite was synthesized both in a 10 ml Pt crucible in ambient atmosphere, and also in a triple argon purged evacuated (10⁻³ Torr) sealed SiO₂ tube to inhibit oxidation of the Fe in the Mg-Fe whitlockite, both as described above. Resulting materials from all merrillite syntheses were analyzed by optical microscopy, EMP and SC-XRD to confirm the phases. Oxidation of Fe during merrillite synthesis can mask detection of dehydrogenation by SC-XRD, so Fe-containing phases were additionally analyzed by Infrared Spectroscopy (IR). As with the Fe-endmember whitlockite treated

in the same manner, the output material ranged in color from white/translucent to black and nearly opaque.

Analytical Methods

Powder X-ray Diffraction (XRD) analyses were performed in the UNLV Geoscience XXL facility. Samples of synthesized Mg- and Fe-whitlockite, as well as impurities from synthesis, were powdered by hand in an agate mortar and pestle and subjected to powder XRD in a Panalytical X'pert Pro diffractometer using Cu K α radiation. Scans were run at 45 Kv and 40 mA from 5-75° 2 θ in continuous mode with step-size of 0.0084° and scan rate of 0.06° sec⁻¹. Patterns were identified and confirmed using Panalytical's X'pert Pro High Score Plus software.

Optical microscopy was used to identify synthesized phases based on crystal habit/morphology and previous XRD analyses. These identifications were carried out with a Barska tri-ocular stereomicroscope with a magnification range of 7x to 45x. An adaptor mounted Canon XTi camera was used for all optical photomicrographs.

Scanning Electron Microscopy (SEM) in secondary electron imaging mode (SEI) and Energy Dispersive Spectrometry (EDS) was carried out on a JEOL 5600 SEM at the UNLV Electron Microanalysis and Imaging Laboratory (EMiL). Operating conditions were 20 KeV and a 20 mm working distance. Analyses by Electron Microprobe (EMP) wavelength dispersive spectroscopy (WDS) were carried out in a JEOL JXA-8900 at the UNLV EMiL facility on polished epoxy mounts. Analysis conditions were 20 KeV and 10 nA using a 10 µm beam. Analyses were standardized using Smithsonian mineral

standards of olivine (Mg, Fe) and apatite (Ca, P)(Jarosewich, 2002), and Harvard (Amelia) albite (Na)(McGuire et al., 1992).

Single Crystal X-ray Diffraction (SC-XRD) analyses were carried out using a Bruker APEXII single crystal diffractometer. An appropriate crystal was selected under paratone oil on a polarizing microscope and attached to a Kapton mount. A full sphere of data was collected with the sample cooled to 100 K using an Oxford nitrogen cryostream. An absorption correction utilized the Bruker software SADABS, structure solution was performed using a beta-version of SHELXT, and refinement was carried out using a combination of SHELX (Sheldrick, 2007) and OLEXII (Dolomanov et al., 2009). Bond valence calculations from SC-XRD data were conducted using VALIST (Wills, 1999).

Infrared (IR) spectra were obtained with a Digilab FTS 7000 series FT-infrared spectrometer on powdered samples using a photo-acoustic chamber attachment. Analyses were performed under ambient conditions at 2.5 KHz and a resolution of 8 cm⁻¹ for 64 total scans. Data were collected over the range of 400 to 4000 cm⁻¹. Carbon black was used to obtain a background versus subsequent spectra.

Synchrotron X-ray diffraction experiments were conducted on two samples (a red ferric Fe/Mg-merrillite crystal and a dark colored ferrous Fe/Mg-merrillite crystal) at the superconducting bending magnet beamline 12.2.2 at the Advanced Light Source (ALS), Lawrence Berkeley National Laboratory, with the goal of identifying any trace fine-grained phases within the merrillite. Primary beam energy was 20.000 keV and a MAR345 image plate detector was utilized. Detector parameters were calibrated and corrected for geometric distortions based on a LaB₆ NIST powder diffraction standard using the Fit2D software (Hammersley et al., 1996). The X-ray beam at beamline 12.2.2

of the ALS is focused by Kirkpatrick-Baesz mirrors vertically and horizontally to 10 x 15 μ m². Crystal samples were mounted on the tips of quartz glass fibers of 30 μ m diameter and centered on the phi rotation axis. Exposures were taken over 160 s while the samples were oscillated by +/-40 degrees around the phi axis. Recorded images were integrated using Fit2D (Hammersley et al., 1996). We first used Fit2D to mask strong reflections from merrillite and any remaining minor untransformed whitlockite in the diffraction image frames and integrated the remaining patterns. Integrated patterns were still dominated by merrillite reflections and less intense whitlockite reflections. The contributions from both these phases were fitted by a Le Bail refinement (Le Bail, 2005). Remaining reflections were then identifiable as fine-grained inclusions. We used the first strong diffraction peaks of these fine-grained inclusions for a search in the American Mineralogist database (Downs and Hall-Wallace, 2003) and commercial data bases (ICSD) for identifying possible inclusion phases (Bergerhoff and Brown, 1987).

Results and Discussion

Optimization of Whitlockite Synthesis

Results of preliminary Mg-whitlockite experiments indicated that the concentration of solids in solution (g/L) was a strong control on Mg-whitlockite formation (Appendix I Table S1). Concentrations of greater than approximately 22 g/L produced primarily monetite (CaHPO₄) and hydroxyapatite (HAP) crystalline material rather than whitlockite. Concentrations of less than 22 g/L, including the 14.5 g/L used in the finalized method, produced yields consisting primarily of well formed crystals of Mg-whitlockite confirmed by morphology, SEM EDS, powder XRD, and SC-XRD

(Figures 1A and 2A). Solution concentrations of significantly less than 14.5 g/L (e.g. 11.1 g/L) produced whitlockite, but in smaller amounts (Appendix I Table S1). Variations in incubation time (5-14 days) indicated that most crystal growth occurred in the first 7 to 8 days. Longer incubation times (up to 14 days) did not produce significantly larger crystals. Variations in pH from 2.1 to 2.8 did not appear to have any significant effect. Scaling up from 23ml vessels to 125ml vessels had no effect as long as solution concentrations were maintained. Temperature (i.e. 240 °C) was not varied and represented the highest safe temperature rated for the vessels used. The ratio of HAP to magnesium nitrate hexahydrate (3.33: 1) was not varied significantly, but minor changes (2.93:1 to 3.52:1) had no noticeable effect on syntheses. Overall yields from multiple batches of Mg-whitlockite synthesis using the final method had an average yield of >700 mg (Appendix I Table S1). The $<75\mu$ m fraction was removed to minimize impurities (i.e. monetite and hydroxyapatite); this fraction averaged $\leq 2\%$ of the total output mass. Imperfections within some crystals, mainly in the larger size fraction, were apparent optically, but centralized clusters of inclusions similar to those seen by Hughes at al. (2008), while present, were uncommon. Most crystals (>80%) were over 150 μ m and the largest crystals were 2mm. Polycrystalline aggregates of whitlockite occurred in the >75 µm fraction but the majority of the material were single crystals. EMP analyses of synthesized crystals (Table 2) confirmed Mg-whitlockite stoichiometry as $Ca_{9.0}Mg_{0.9}(PO_3OH)(PO_4)_6.$

Attempts to synthesize Fe-whitlockite using $FeCl_2$ as the source of Fe failed to produce the mineral in amounts detectable by optical microscopy, even at total solid in solution concentrations as low as 12.0 g/L. Instead, crystal morphologies consistent with
monetite and recrystallized HAP were produced as well as minor opaques (presumably Fe phases). The subsequent use of Fe(II)S as the source of Fe did produce Fewhitlockite. Aggregate and hydroxyapatite production were minimized by decreasing the concentration of solids in solution from an initial 16.1 g/L to < 12.0 g/L. To minimize iron sulfide phases, the ratio of hydroxyapatite to Fe(II)S in solution was increased from \sim 9:1 (near stoichiometric ideal) to between 16:1 and 20:1 in an effort to "starve" the system of Fe. Yields from Fe-whitlockite syntheses using the final method described in the methods section averaged >500 mg of largely well formed whitlockite crystals (Figs. 1B and 2B, Appendix I Table S1). Impurities (i.e. iron sulfides, hydroxyapatite, and monetite) mainly occurred in the $<75 \,\mu\text{m}$ fraction which could be removed through sieving. The $<75 \mu m$ fraction accounted for 33% of the total output mass on average. Inter-crystalline Fe-films were apparent optically in some aggregates (Fig. 1C). Some Fe-whitlockite crystals themselves contained minor intra-crystalline Fe-film deposits (determined optically) on earlier growth surfaces (Fig. 1B), with the crystals separable by picking. Minor amounts of Mg (0.14 wt. % as MgO) were also detected in Fewhitlockites by EMP (Table 2). Like Hughes et al. (2008), the source of Mg is unknown, but is most likely from trace impurities in the reagents used. EMP analyses of synthesized crystals (Table 2) confirmed synthetic Fe-whitlockite stoichiometry as $Ca_{9.0}Fe_{1.0}(P_{1.0}O_3OH)(PO_4)_6.$

While Mg- and Fe-endmember whitlockite (and the merrillite synthesized from them) represent closer analogs to naturally occurring minerals than a phase like β -Ca₃(PO₄)₂, mixed Fe/Mg phases are closer analogs still. Two early experiments to produce mixed Fe/Mg-whitlockite failed to produce any significant whitlockite

(Appendix I Table S1). With understanding gained from further endmember experiments, another set of experimental conditions was tested with a lower solution concentration (13.0 g/L as opposed to >17 g/L for the earlier attempts). This experiment included a source of Na⁺ as part of the solution (included in the solution concentration noted above). Though Na⁺ does not generally occur in whitlockite, the ion does occur in a specific Ca²⁺ site in natural merrillite, including merrillite found in martian meteorites where Na⁺ content can be as high as 2 wt.% (Jolliff et al., 2006). The batch produced Fe/Mg mixed whitlockite (Figs. 1D and 2C) along with impurities of hydroxyapatite and monetite, and large (>150 μ m) polycrystalline aggregates. Most impurities were in the >150 μ m and <75 μ m size fractions, leaving the 75-150 μ m size fraction the purest. This fraction comprised 39% of the total output mass. EMP analyses showed no incorporation of Na⁺ into any of the synthesized material and an average stoichiometry of $Ca_{9.0}Fe_{0.7}Mg_{0.3}(PO_3OH)(PO_4)_6$ (Table 2). The whitlockite seems to preferentially incorporate Mg, as the whitlockite from this batch had a Fe to Mg molar ratio of ~7:3 but the molar ratio of Fe to Mg in the reactants added was ~9:3. EMP analyses of results from additional mixed Fe/Mg-whitlockite synthesis experiments support this observation (Appendix I Table S2). It is also of note that the mixed Fe/Mg-whitlockite synthesis experiments had a higher failure rate than those of the endmembers, even after finding a generally successful set of conditions. Further investigation might yield the source of the failures or better refine the method. The lack of Na⁺ incorporated within the whitlockite synthesized in these experiments suggests it may not be possible to incorporate the ion into the structure under hydrothermal conditions. Thus, mixed Fe/Mg-merrillite subsequently synthesized from the mixed Fe/Mg-whitlockite would also lack the sodium

generally found in martian and lunar merrillite. Further investigation to determine conditions that favor Na⁺ substitution in merrillite are warranted as they could potentially reveal important information about the specific martian environments in which merrillite forms. Nevertheless, sodium-free mixed Fe/Mg-bearing merrillite represents a better analog to martian merrillite than pure endmember phases. EMP analyses revealed standard deviations of Fe and Mg contents were slightly higher than those of endmember synthesized material (Table 2). However, the minor increase in inhomogeneity should not limit the usefulness of the material for most studies.

Merrillite Synthesis

Merrillite and whitlockite structures are too similar to be easily distinguished by powder XRD, thus heat-treated materials were analyzed by SC-XRD and EMP to determine if the whitlockite to merrillite transformation had occurred. In the case of Mgmerrillite, SC-XRD (discussed in more detail below under Crystallography) confirmed synthesis had occurred and EMP analyses of the resulting Mg-merrillite (Table 2) indicated a stoichiometry of Ca_{9.4}Mg_{1.1}(PO₄)_{7.0}, compared to Ca_{9.0}Mg_{0.9}(PO₃OH)(PO₄)₆ of the starting Mg-whitlockite. The higher stoichiometric value for Ca²⁺ results from Ca substitution for charge balance due to dehydrogenation. Dehydrogenation is also the likely cause of higher observed totals for EMP analyses of Mg-merrillite versus the Mgwhitlockite used to synthesize it (99.35% versus 98.02% respectively). The synthesized Mg-merrillite maintained the crystallinity and morphology of the starting whitlockite material (Fig. 1E). Fe-merrillite synthesis proved more complex than Mg-merrillite synthesis due to the apparent oxidation of ferrous Fe^{2+} to ferric Fe^{3+} when heated in open air. Fewhitlockite samples heat-treated in open air (i.e. in the Pt crucible) changed color from white to orange or yellow. Like the Mg-merrillite, EMP analyses of the heat treated material showed overall totals rise from 98.15% for the original whitlockite to 98.80% for the heat treated material (with Fe calculated as FeO, see Table 2 and associated table notes), likely due to the loss of H⁺ during heating. However, EMP analyses also indicated a stoichiometry that did not significantly change from the Fe-whitlockite of $Ca_{9,0}Fe_{1,0}(PO_4)_{7,0}$. This EMP result, and the color change, suggested oxidation of Fe^{2+} to Fe^{3+} to produce a ferric Fe-merrillite phase balancing the change in charge due to the loss of H⁺ as shown below:

$$Ca_9Fe^{2+}(PO_3OH)(PO_4)_6 \rightarrow Ca_9Fe^{3+}(PO_4)_7.$$
 (1)

Bond valence calculations (Skowron and Brown, 1990) and Fe – O bond lengths (Table 3) based on SC-XRD analyses of the heat treated material were also consistent with Fe³⁺. To confirm dehydrogenation, IR analyses were performed on both the unreacted Fewhitlockite and the heat-treated Fe-merrillite to determine the presence or absence of O– H bonds (Fig. 3A). The IR spectra show a large reduction in absorption between the starting material and the heat-treated material over the range of 3330 to 2570 cm⁻¹ where a broad feature is associated with an O–H stretch (White, 1990; Belik et al., 2002). There is also an O–H absorption peak at 2380 cm⁻¹ (Belik et al., 2002) which disappears between the starting whitlockite material and the samples heated treated in the Pt crucible open to the atmosphere (Fig. 3A), further supporting the conclusion that the material is ferric (Fe³⁺) Fe-merrillite. Previous research using solid state methods has also produced

a ferric Fe-merrillite (e.g. Lazoryak et al., 1996; Belik et al., 2002), although the material in those experiments was a fine powder.

In order to prevent Fe oxidation and promote the formation of ferrous Femerrillite, Fe-whitlockite was heated to 1055 °C in triple argon purged evacuated fused SiO₂ tubes instead of the open air Pt crucible. After heat treatment, EMP analyses indicated a stoichiometry of $Ca_{9,2}Fe_{1,0}Mg_{0,1}(P_{1,0}O_4)_7$ and overall totals rose compared to the starting material (Table 2). Bond valence calculations performed using VALIST (Wills, 1999) and Fe–O bond lengths (Table 3) from SC-XRD data are consistent with an Fe oxidation state of 2+, indicating that the phase is ferrous. However, IR results showed a remaining peak at 2380 cm⁻¹ indicating that full dehydrogenation did not occur during the heat treatment (Fig. 3A). Thus the resulting phase consists of ferrous Fe-merrillite containing some whitlockite. Extending the incubation time at 1055 °C to three days had no effect on the IR results. The result of heat treating the Fe-whitlockite in a sealed SiO_2 appears to be ~80-90% transformation to merrillite based on SC-XRD and is discussed in more detail in the Crystallography section. The incomplete dehydrogenation of the ferrous merrillite may be the result of changing partial pressure due to H⁺ loss in the sealed system and warrants further exploration. Many of the heat treated crystals were darker in color suggesting the formation of other, possibly Fe-containing phases, although BSE and EMP analyses were unable to detect a separate phase within crystals or a color dependent difference in chemistry.

Formation of mixed Fe/Mg-merrillite followed similar trends to that of endmember Fe-merrillite synthesis, with samples heated in open air oxidizing to Fe³⁺/Mg-merrillite and samples heated in an evacuated SiO₂ tube forming Fe²⁺/Mg-

merrillite. Fe/Mg-whitlockite heat treated in a Pt crucible in open air showed a rise in EMP totals over the starting material beyond that attributable only to total calculations based on Fe₂O₃ rather than FeO (Table 2 and associated table notes). IR analyses, as described above (Fig. 3B), indicated dehydrogenation had occurred. Bond valence calculations based on SC-XRD data showed that Fe had oxidized to Fe³⁺, and EMP analyses of the reacted material showed a stoichiometry of Ca_{9.2}Fe_{0.7}Mg_{0.4}(P_{1.0}O₄)₇ compared to the unreacted whitlockite of Ca_{9.0}Fe_{0.7}Mg_{0.3}(P_{1.0}O₃OH)(PO₄)₆ indicating Ca mobility to balance the change in charge due to the loss of H⁺ had occurred. Heat treatment also caused a color change in the material from white/translucent to red/orange.

Dehydrogenation of mixed Fe/Mg-whitlockite by heating in an evacuated fused SiO₂ tube resulted in crystalline material that varied in color from clear translucent to nearly opaque dark black. SC-XRD analysis performed on a clear crystal indicated it to be Fe^{2+}/Mg -merrillite that contained some whitlockite. Fe – O distances determined by SC-XRD indicate that the Fe had not significantly oxidized. A smaller, sharp O-H absorption peak at 2380 cm⁻¹ remained (Fig. 3B) indicating that the phase contained some whitlockite. Similar to the ferrous endmember Fe-merrillite extending the incubation time to three days had no effect on the IR results. Estimates of the merrillite/whitlockite content are discussed in more detail in the Crystallography section, but indicate that most of the mixed Fe/Mg-reacted material is merrillite (~70%). The color change observed in post heat treatment crystals, as with the Fe-merrillite treated in a SiO₂ tube, suggested the possible formation of another phase. However, as with the similarly treated endmember Fe-whitlockite, BSE and EMP analyses were unable to detect a separate phase within the crystals or a difference in chemistry between lighter or darker colored material.

Surface deposits or films on crystals of synthesized endmember and mixed merrillite crystals are apparent under SEM imaging (Figs. 2D to 2F) and by optical microscopy (e.g. Fig. 1E). Large (> 150 µm) Mg-merrillite crystals were completely covered with the film, which is generally sub-micron in thickness. The amount of coverage appeared to decrease with decreasing grain size (i.e. smaller crystals exhibited less complete or thinner coverage). In contrast, deposits on Fe²⁺-bearing crystals rarely covered the entire crystal surface and in the case of ferric Fe-merrillite, most crystals showed no coating (Fig. 1F). SC-XRD analyses did not detect a second phase, although SC-XRD analysis would be unlikely to detect such small amounts of a second phase. The surface textures also lack any morphology that would suggest crystallinity and thus may be amorphous. The coatings were also not detectable in EMP prepared samples due to their sub-micron thickness and similar chemistry. EDS in SEM of the reacted surfaces indicated that the surface deposits were mainly Ca and P with no evidence of Mg or Fe, and therefore suggest that they may be the result of some form of diffusion to the surface during heating.

In an attempt to determine what additional phases might be present in the merrillite samples, two samples were analyzed by synchrotron XRD; a translucent redcolored crystal of the Fe/Mg-merrillite heat treated in open air (ferric Fe/Mg-merrillite), and a crystal containing a hazy grey to black core within a clear crystal matrix of Fe/Mgmerrillite heat treated in a SiO₂ tube (ferrous Fe/Mg merrillite). Specimens were 100 to 150 μ m in diameter. In both cases, diffraction patterns exhibited three distinct sets of features: reflections that belong to multiple domains of merrillite, reflections that belong

to domains from whitlockite, and Debye fringes from very fined grained (grain size $< 1 \mu m$) inclusions. Thus the samples were not true single crystals.

Modeled and observed patterns of the red-colored ferric Fe/Mg-merrillite sample along with the residual of the fit appear in Figure 4A. In addition to the expected merrillite and whitlockite, monetite and a minor amount of holtedahlite [Mg₂(PO₄)(OH)] were also identified. Holtedahlite occurs as a fine grained phase which generated a smooth powder diffraction pattern. With the given experimental settings and based on the observed peak profiles the average grain size of holtedahlite is less than 1 μ m and more than 100 nm. Since the diffraction signals from whitlockite, merrillite, and monetite were not powderous, a Reitveld analysis for phase quantification could not be conducted. However, the strength of the diffraction signal of holtedahlite indicates that it is significantly less than merrillite, whitlockite, and monetite.

The ferrous Fe/Mg-merrillite specimen with the grey-black core region was composed of a solid solution whitlockite/merrillite matrix (>3 micron grain size) containing small amounts of monetite and an additional unidentified fine grained (< 1 μ m) phase. Limited diffraction data are consistent with a Na-bearing Ca-ferrate, although a confident identification is not possible and oxidized Fe would be surprising in the ferrous Fe/Mg-merrillite. The observed pattern with the whitlockite/merrillite contributions masked and modeled patterns of monetite, a Na-bearing Ca-Ferrate, and the residual of the fit appear in Figure 4B.

Presumably, some of these additional phases may also occur in the endmember merrillite materials, though we were not able to confirm that as part of the current study. Synchrotron diffraction patterns of monetite might explain the calcium and phosphorus

phase observed as films on crystals. SC-XRD analyses did not detect monetite, holtedahlite or another unidentified phase, but the technique and sample selection criteria are designed to refine single structures from single crystals and are not applicable to detecting minor phases that exhibit reflections from multiple domains or powder fringe patterns.

The mechanism by which calcium balances the charge imbalance due to the loss of H^+ from whitlockite and increases the stoichiometric Ca in non-ferric merrillite is unclear. Gopal and Calvo (1972) and Gopal et al. (1974), who synthesized merrillite from both natural and synthetic Mg-whitlockite, present no mechanism. Hughes et al. (2008), who synthesized merrillite from synthetic Mg-whitlockite, suggest Ca-rich reagents entrained in the synthetic whitlockite crystals yielded Ca sufficient to counter the charge imbalance due to H^+ loss, with the Ca presumably diffusing through the crystal. However, this mechanism fails to explain enhanced Ca concentrations in the whitlockite to merrillite dehydrogenation experiments of Gopal and Calvo (1972) who used natural whitlockite, which would contain no entrained reagents.

The apparent enrichment of Ca relative to phosphate during the whitlockite to merrillite transformation could take place through the formation of additional phases that either diffuse to the surface or are dispersed and difficult to detect within the bulk crystals. This would be consistent with the synchrotron diffraction data that indicate additional phases were present within Fe/Mg-merrillites. An additional phase or phases, like the holtedahlite or Fe-bearing phase indicated by synchrotron diffraction, would be required in order to take up extra Mg or Fe made available during the transformation from whitlockite to merrillite. Synchrotron diffraction data from the ferric Fe/Mg-

merrillite heat treated open to atmosphere indicated holtedahlite as the only additional crystalline phase beyond monetite, which would take up remaining Mg. It is plausible that if the Fe is oxidizing during transformation as per Equation 1, an additional Febearing phase may not necessarily form. However, in the case of the ferrous Fe/Mgmerrillite treated in the SiO₂ sealed tube, both Fe- and Mg-bearing additional phases would be expected and synchrotron diffraction data only indicate a possible Fe-bearing crystalline phase. It is possible that an undetected amorphous Mg-phase is taking up the excess Mg, and that a similar Fe-bearing phase may also form to take up any excess Fe in the Ferric merrillite that is not accounted for by oxidation (Equation 1). Candidates for such a phase include Fe- and Mg- pyrophosphates $[X_2(P_2O_7)]$ where X is Mg or Fe], which can be produced in high temperature metal/phosphate systems and can be amorphous (Bensalem et al., 1997; Lee et al., 2012). These Fe- and Mg-pyrophosphate phases may also act as transitional phases – for example, a hypothetical Mgpyrophosphate may be a transitional phase to the holtedahlite observed by synchrotron XRD (Equation 2), which would explain why holtedahlite was detected in one sample, but no Mg-containing phase in the other:

$$Mg_2(P_2O_7) + 2 H_2O \rightarrow Mg_2(PO_4)(OH) + H_3PO_4.$$
 (2)

A general reaction reflecting the transformation from whitlockite to merrillite may therefore include either holtedahlite or an amorphous metal pyrophosphate, and be represented by:

Whitlockite + Heat \rightarrow Merrillite + Monetite + Additional Phase(s) + H₃PO₄ + H₂O, where non-merrillite phases take up any remaining chemistry resulting from the whitlockite-to-merrillite transformation. A balanced example that includes both monetite and pyrophosphate as the metal phosphate is:

$$24 \operatorname{Ca}_{9}X(PO_{3}OH)(PO_{4})_{6} \rightarrow 22 \operatorname{Ca}_{9.5}X(PO_{4})_{7} + 7 \operatorname{Ca}(HPO_{4}) + X_{2}(P_{2}O_{7}) + 5 \operatorname{H}_{3}PO_{4} + H_{2}O_{4}$$
(3)

Equations similar to 1, 2, or 3, or combinations thereof, are consistent with the observations of this study and allow for a mechanism of charge balance that does not require entrained synthesis reagents. Such a mechanism would better explain the transformation of natural whitlockite to merrillite confirmed by Gopal and Calvo (1972), where no entrained reagents from synthesis were present.

<u>Crystallography</u>

The atomic parameters of Mg-whitlockite and Mg-merrillite synthesized by conceptually similar methods to those used here have been previously measured by SC-XRD and discussed (Gopal et al., 1974; Hughes et al., 2008), and therefore we only focus on differences between the Fe- and Fe/Mg-whitlockite and merrillite structures determined in this work. The structures of Fe-whitlockite and ferric Fe-merrillite have been previously refined using powder XRD diffraction (Corlett and Keppler, 1966; Keppler, 1968; Lazoryak et al., 1996; Belik et al., 2002). Belik et al. (2002) utilized neutron powder XRD and located two deuterium sites for Fe–whitlockite; hydrogen cannot be located by SC-XRD methods. Since significantly more accuracy is possible with single crystal XRD, we report the structures of Fe-whitlockite and ferric Femerrillite here (Table 4). Atomic parameters of synthetic ferrous Fe-merrillite, and mixed Fe/Mg-merrillite do not appear to be in the literature, and we present atomic coordinates and full listing of bond distances as Tables S3 and S4 in Appendix I. A .cif file containing all six crystal structures is also in Appendix I. Broadly speaking, the six crystal structures determined (Fe- and Fe/Mgwhitlockite, as well as ferrous and ferric Fe- and Fe/Mg-merrillite)(Table 4) are similar with differences related mainly to the chemistry. The position of the proton has previously been established as protonating the phosphate group located on the 3-fold axis (labeled P3 in this work). Protonated phosphates point in the opposite direction compared to non-protonated phosphates. Modeling of this disorder for the synthetic Fewhitlockite found 89.0% of P3 on the inverted site; 100% should be inverted if the structure were fully protonated. It is possible that some protons occupy other sites in the structure, some Fe³⁺ is present, minor merrillite is present, or a combination of the three. Small amounts of merrillite within natural whitlockite and synthetic Mg-whitlockite have been previously noted (Hughes et al., 2008).

Heat treating the Fe-whitlockite in a Pt crucible open to the atmosphere yields an Fe-merrillite sample where 97.1% of the P3 are not on the inverted site, in agreement with essentially full conversion to merrillite. This does not require 2.9% protonation – it is possible that a fraction of the P3 phosphate remains inverted after deprotonation. The observed Fe – O distances (Table 3) are highly sensitive to the oxidation state of the iron. We find distances in the Fe-whitlockite sample (Fe – O bond lengths of ~2.03 Å), and in the ferric merrillite sample (Fe – O bond length of ~2.10 Å) that match expected metal – oxygen distances for Fe²⁺ and Fe³⁺, respectively. Note that the approximate difference of 0.1 Å between Fe for Fe²⁺ and Fe³⁺ is larger by roughly a factor of 50 than standard deviations on these bond distances (Table 3). Bond valence calculations are normally only an approximate method of identifying oxidation state as bond distances can be influenced by a variety of structural features in a specific compound. In this particular

instance, bond valence calculations are more conclusive as the crystal structures and compositions are very similar to each other. While the observed bond distances as well as the bond valence calculations clearly suggest primarily Fe^{2+} in the Fe-whitlockite and ferrous merrillite, and primarily Fe^{3+} in ferric merrillite, they only indicate which oxidation state is predominant in each structure, not the relative amounts of Fe^{2+} and Fe^{3+} .

Heating the Fe-whitlockite samples in evacuated SiO₂ tubes apparently prevented some oxidation of Fe^{2+} to Fe^{3+} , as evidenced by little change in the measured Fe - Obond distances (Table 3). Charge balance in the merrillite is maintained by the migration of Ca^{2+} cation sites to around where protons are found in the whitlockite form. The Ca is distributed over two sites, Ca4A and Ca4B, with refined occupancies of 14.8 and 29.6%, respectively. As it is not possible from crystallography to assign elements to partially occupied sites such as these, we infer that they are likely Ca based on its increase in microprobe data (Table 2) and the bonding environments for both sites. Table 3 includes bond distances for these sites, all of which are longer than expected for Fe^{2+} . In addition, the environments are quite irregular, which would be unfavorable for d⁶ Fe but not unusual for Ca^{2+} . As 50% occupancy of a divalent cation site would provide sufficient charge balance to replace that lost by full dehydrogenation, these occupancies are consistent with 88.8% conversion to merrillite. Of the P3 phosphates, 23.9% are inverted, which is consistent with a modestly higher degree of protonation than suggested by the Ca²⁺ occupancies, though some of the P3 phosphate may have remained inverted after deprotonation. The Ca^{2+} occupancies of the Fe-whitlockite heat treated in a sealed SiO₂ tube, together with P3 observations, indicate ~80-90% conversion to merrillite.

There was no evidence of significant electron density on or around these sites in the Fewhitlockite and Fe-merrillite heat treated in a Pt crucible.

The mixed Fe/Mg-whitlockite sample refined to a similar overall structure as the pure Fe-whitlockite. Refinement of the Fe/Mg ratio yielded a 63.7% Fe occupancy, with the balance Mg, in general agreement with the microprobe results (Table 2). The Fe/Mg ratios refined for the two heated samples refined to 75.4% (Pt crucible) and 64.6% (SiO₂ tube). Inter-crystal standard deviations on this ratio determined by microprobe were high, so it is quite possible that this value varies some from crystal to crystal. P3 phosphates are inverted 84.0% of the time, suggesting the whitlockite is 16% merrillite already, and the Fe – O distances are consistent with Fe²⁺. The sample heated in air ended up with 6.5% of the P3 site inverted, suggesting near complete conversion to merrillite structure, and the Fe – O distances are consistent with Fe³⁺.

Results of heating mixed Fe/Mg samples in sealed SiO₂ tubes were also quite similar to the pure Fe case. Fe – O distances are consistent with Fe²⁺, and 30.9% of the P3 phosphate group are inverted, indicating that the sample is largely merrillite. Two Ca sites with similar environments and occupancies (9.8% and 26.4%) were also found.

Implications for Martian Studies

Merrillite is a potentially important indicator of martian geologic processes, including the evolution of halogen and water content over time in martian magmas. A deeper understanding of the relationship of merrillite formation to halogen and water content in melts may hold clues to parts of the martian igneous and hydrologic cycles. The relationship of merrillite/whitlockite to apatite in melts may also hold important clues to the evolution of not only martian magmas, but those of other differentiated bodies including Earth and asteroids. The phosphate in martian soils and dust may also be present as merrillite, making the mineral of interest to martian surface process studies and, as a major source of bio-essential phosphate, studies pertaining to martian habitability. Access to a supply of synthetic, coarse crystalline whitlockite and merrillite of various chemistries and an understanding of the conditions under which these phases can be formed, as presented here, will likely aid in facilitating a number of these studies as well as further investigations into more fundamental thermodynamic and kinetic parameters of the minerals. Complete dehydrogenation of ferrous merrillite in this study may have been hindered by the sealed tube used. A flow through apparatus may eventually prove more successful, and while Na⁺ incorporation into merrillite may not be possible under hydrothermal conditions, a high temperature approach may yield success. Future efforts in producing completely dehydrogenated ferrous merrillite as well as incorporating Na⁺ into synthetic merrillite are warranted as the conditions under which these are achieved may reveal further insights into the natural environments, martian or otherwise, in which these minerals form.

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Synthetic Mineral	MgNO₃ (g)	FeS (g)	NaCl (g) ^a	$HAP\left(g ight)^{b}$	Total solid Mass (g)	H₂O (ml)	Solids in Solution (g/L)
Mg-whitlockite	0.300 (±0.003)	-	-	1.000 (±0.003)	1.300	90	14.4
Fe-whitlockite	-	0.055 (±0.005)		1.000 (±0.003)	1.055	90	11.7
Fe/Mg-whitlockite	0.050 (±0.003)	0.050 (±0.005)	0.067 (±0.005)	1.000 (±0.003)	1.167	90	13.0

Table 1. Masses and volumes used in finalized whitlockite synthesis methods

^{*a*}Later experiments showed NaCl to have no effect on synthesis. ^{*b*}HAP = hydroxyapatite.

	Mg-v	vhitlockite	Mg-merrillite				
CaO	47.61	(0.20)	48.89	(0.21)			
P_2O_5	46.85	(0.27)	46.38	(0.32)			
MgO	3.54	(0.14)	4.06	(0.12)			
FeO	0.02	(0.03)	0.02	(0.03)			
Na₂O	B.D.		B.D.				
^ª H₂O	0.86		-				
Total	98.02	(0.34)	99.35	(0.36)			
n		17		16			
Stoic.	Ca _{9.0} Mg _{0.9}	9(PO3OH)(PO4)6	Ca _{9.4} M	g _{1.1} (PO ₄) _{7.0}			
			F	erric	Ferrous Fe-merrillite		
	Fe-n	hitlockite/	Fe-n	nerrillite			
			(Pt	cruc.)	(SiO₂ tube)		
CaO	46.00	(0.45)	46.37	(0.27)	47.00	(0.19)	
P_2O_5	45.26	(0.47)	46.02	(0.50)	45.42	(0.46)	
MgO	0.14	(0.03)	0.34	(0.07)	0.39	(0.10)	
^b Fe ₂ O ₃	-		6.74	-	-		
FeO	6.75	(0.27)	-		6.85	(0.18)	
Na₂O	B.D		B.D.		B.D.		
^ª H₂O	0.86		-		-		
Total	98.15	(0.63)	99.47	(0.45)	99.65	(0.53)	
n		18	17		20		
Stoic.	Ca _{9.0} Mg _{0.9}	9(PO3OH)(PO4)6	Ca _{9.0} Fe _{0.9}	$Ca_{9.0}Fe_{0.9}Mg_{0.1}(PO_4)_{7.0}$		Ca _{9.2} Fe _{1.0} Mg _{0.1} (PO ₄) _{7.0}	
		Ferric		erric	Ferrous		
	Fe/Mg-w	hitlockite	Fe/Mg	-merrillite	Fe/Mg	-merrillite	
			(Pt	(Pt cruc.)		2 tube)	
CaO	46.54	(0.59)	46.79	(0.33)	47.09	(0.22)	
P ₂ O ₅	45.83	(0.36)	45.96	(0.39)	45.40	(0.38)	
MgO	1.00	(0.20)	1.29	(0.09)	1.20	(0.16)	
^⁰ Fe₂O₃	-		4.98	-	-		
FeO	4.62	(0.71)	-		5.68	(0.23)	
Na ₂ O	B.D.		B.D.		B.D.		
^ª H₂O	0.86		-		-		
Total	97.99	(0.49)	99.02	(0.60)	99.36	(0.48)	
n		15		13	19		
Stoic.	Ca _{9.0} Fe _{0.7} M ₈	g _{0.3} (PO ₃ OH)(PO ₄) ₆	Ca _{9.1} Fe _{0.7}	Mg _{0.3} (PO ₄) _{7.0}	Ca _{9.2} Fe _{0.9} Mg _{0.3} (PO ₄) _{7.0}		

Table 2. Chemical analyses of whitlockite and merrillite by electron microprobe in wt%.

Parenthetical values are 1 standard deviation. n = number of analyses averaged chemistry is based on. B.D. = below detection.

^{*a*} H_2O is based on ideal whitlockite (Hughes et al., 2008) and is not included in the EMP totals. ^{*b*}Fe was measured assuming Fe^{2+} and recalculated to Fe^{3+} with total and resulting stoichiometry adjusted accordingly. For ferric Fe-merrillite, the original FeO wt. % = 6.07 with a standard deviation of 0.13, analysis total of 98.80, and calculated stoichiometry of $Ca_{9.1}Fe_{0.9}Mg_{0.1}(PO_4)_{7.0}$. For Ferric Fe/Mgmerrillite the original FeO wt. % = 4.48 with a standard deviation of 0.08, analysis total of 98.53, and calculated stoichiometry of $Ca_{9.2}Fe_{0.7}Mg_{0.4}(PO_4)_{7.0}$. NOTE: It is not possible by microprobe alone to actually determine Fe^{2+}/Fe^{3+} content and these values are estimates.

				·	Ferrous		Ferrous
					Fe-	Ferric	Fe/Mg-
				Ferric Fe-	merrillite	Fe/Mg-	merrillite
		Fe-	Fe/Mg-	merrillite	(SiO ₂	merrillite	(SiO ₂
		whitlockite	whitlockite	(Pt cruc.)	tube)	(Pt cruc.)	tube)
Fe(1)-	O(4)	2.1131(16)	2.1067(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(4)#1	2.1130(16)	2.1068(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(4)#2	2.1130(16)	2.1068(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(5)	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
	O(5)#1	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
	O(5)#2	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
Ca(4A)-	O(6)#18	NA	NA	NA	2.736(5)	NA	2.758(6)
	O(6)#19	NA	NA	NA	2.736(5)	NA	2.758(6)
	O(7)#18	NA	NA	NA	2.541(4)	NA	2.541(4)
	O(7)#19	NA	NA	NA	2.541(4)	NA	2.541(4)
Ca(4B)-	O(6)#18	NA	NA	NA	2.516(3)	NA	2.518(2)
	O(6)#19	NA	NA	NA	2.516(3)	NA	2.518(2)
	O(7)#18	NA	NA	NA	2.846(5)	NA	2.851(4)
	O(7)#19	NA	NA	NA	2.846(5)	NA	2.851(4)

Table 3. Select bond lengths discussed in text for synthetic minerals Å.

Parenthetical values are standard deviations. "Pt cruc." denotes merrillite produced in a platinum crucible in open air. "SiO₂ Tube" denotes merrillite produced using a triple argon purged and seal glass tube. Only Fe-O and Ca-O bonds discussed in text appear in table. A full table of bond lengths appears in deposit item Table S4.

	-		Ferric Fe-	Ferrous Fe-		
	Fe-		merrillite (Pt	merrillite (SiO ₂	Ferric Fe/Mg-	Ferrous Fe/Mg-merrillite
Mineral	whitlockite	Fe/Mg-whitlockite	cruc.)	tube)	merrillite (Pt cruc.)	(SiO ₂ tube)
Emp. formula	Ca ₉ FeHO ₂₈ P ₇	$Ca_9Fe_{0.64}HMg_{0.36}O_{28}P_7$	$Ca_9FeH_{0.02}O_{28}P_7$	Ca _{9.44} FeH _{0.12} O ₂₈ P ₇	$Ca_9Fe_{0.75}Mg_{0.25}O_{28}P_7$	$Ca_{9.36}Fe_{0.65}H_{0.28}Mg_{0.35}O_{28}P_7$
Formula wt.	1082.37	1070.93	1081.38	1099.12	1073.59	1084.98
a (Å)	10.3510(7)	10.3499(4)	10.3278(11)	10.3453(6)	10.3301(13)	10.3392(7)
c (Å)	37.059(2)	37.0715(16)	37.050(4)	37.118(2)	37.062(5)	37.081(2)
Volume (ų)	3438.6(4)	3439.1(2)	3422.4(8)	3440.3(5)	3425.1(10)	3432.8(5)
Density (calc.) (Mg/m³)	3.136	3.103	3.148	3.183	3.123	3.149
Reflections collected	17516	17609	17589	11475	17501	17513
Independent reflections	2352	2355	2334	2238	2338	2342
R(int)	0.0221	0.0202	0.0248	0.0236	0.0249	0.0213
parameters	140	142	140	146	143	153
Goodness-of-fit on F2	1.148	1.085	1.095	1.078	1.053	1.085
Final R indices						
[I>2sigma(I)]						
R1	0.0185	0.0154	0.0172	0.0204	0.0171	0.0157
wR2	0.0552	0.041	0.0454	0.0517	0.0483	0.0411
R indices (all data)						
R1	0.0185	0.0155	0.0175	0.0211	0.0178	0.016
wR2	0.0553	0.0411	0.0455	0.0521	0.0489	0.0414
Largest diff. e Å ⁻³						
peak	0.361	0.315	0.372	0.590	0.787	0.472
hole	-1.913	-0.956	-1.092	-1.032	-0.366	-0.425

Table 4. Single crystal X-ray diffraction refinement statistics

Note: 5.1° to 61.0° 2 Θ range, temperature 100 K, space group R3c. Parenthetical values are standard deviations.

Figures



Figure 1. Optical images of synthesized phases. A) Mg-whitlockite single crystals with a polycrystalline aggregate in the upper left corner. B) Fe-whitlockite single crystal. Dark color inside crystal suggests some intra-crystalline iron film on growth faces. C) Fe-whitlockite polycrystalline aggregates. Darker colors in some aggregates from inter-crystalline iron films. D) Mixed Fe/Mg-whitlockite single crystals. E) Mg-merrillite showing the hazy surface texture due to a thin film deposit post heat treatment. The extent of coverage appeared to decrease with decreasing grain size (i.e. smaller crystals exhibited less complete or thinner coverage). F) Ferric Fe-merrillite crystals showing the incomplete hazy surface texture post heat treatment on some crystals (arrows). Deposits on the minerals rarely covered the entire crystalline surface. "White crystals" in this image are polycrystalline aggregates.



Figure 2. Scanning Electron Microscope images of synthesized materials. A) SEM SEI image of synthesized Mg-whitlockite. B) SEM SEI image of synthetic twinned Fe-whitlockite. C) SEM SEI image of synthesized Fe/Mg-whitlockite. D) SEM SEI with inset (taken at 3000x) showing coating developed on Mg-merrillite crystal during synthesis. Most Mg-merrillite crystals were completely covered, although smaller grains exhibited less complete or thinner coverage. E) SEM SEI image of Ferric Fe-merrillite showing coating developed during synthesis. In contrast to Mg-merrillite, the deposit on the mineral rarely covered the entire crystalline surface and in some cases crystal showed no coating, consistent with discussion of our proposed mechanisms. F) SEM SEI image of ferrous Fe-merrillite showing coating developed during synthesis.



Figure 3. IR Spectra of whitlockite and merrillite A) Fe-whitlockite, ferrous merrillite treated in a SiO2 tube, and ferric merrillite treated open to atmosphere. **B)** Fe/Mg-whitlockite, ferrous Fe/Mg-merrillite heat treated in a SiO2 tube, and ferric Fe/Mg-merrillite heat treated open to atmosphere. Whitlockite data show an apparent absorption from 3330 to 2570 cm-1 where a broad O-H stretch feature exists. The absorbance is mainly absent in material treated in a Pt crucible exposed to air (ferric Fe- and Fe/Mg-merrillite), with some absorption in the O-H stretching band present in material treated in a sealed SiO2 tube (ferrous Fe- and Fe/Mg-merrillite) including a peak at 2380 cm-1. Spectra offset for clarity.



Figure 4. Synchrotron X-ray diffraction. A) Observed and modeled patterns of the red-colored ferric Fe/Mg-merrillite sample along with the residual of the fit. Beyond the expected whitlockite/merrillite observed in the crystal, monetite and holtedahlite also appear to be present. B) Similar data for a dark-colored ferrous merrillite crystal with whitlockite/merrillite pattern data masked out. This crystal also contained monetite and an additional unconfirmed phase with structure similar to a ferrate.

CHAPTER THREE

READILY AVAILABLE PHOSPHATE FROM MINERALS IN EARLY AQUEOUS ENVIRONMENTS ON MARS.

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Abstract

Given the presence of essential chemistry, the processes that led to life on Earth may have also occurred on Mars. Phosphate is among the chemical nutrients thought to be essential for life and low prebiotic availability of phosphate may have been a complicating factor in terrestrial abiogenesis. A similar hurdle may have potentially confronted the development of life on Mars. Here we present dissolution rates, phosphate release rates and solubilities of phosphate minerals found in martian rocks as determined from laboratory measurements. Our findings indicate phosphate release rates and phosphate concentrations in solution are higher on Mars than Earth for a given set of conditions.

Introduction

If the chemistry essential to life was present in water containing environments on Mars, the processes that led to life on Earth may have also occurred on the red planet (Miller and Urey, 1959). Phosphate is one of the chemical nutrients thought to be

essential for life and is also considered critical to reactions that may have led to life on Earth (Westheimer, 1987; Schwartz, 2006). However, low prebiotic availability of phosphate may have been a complicating factor in terrestrial abiogenesis (Gulick, 1955; Schwartz, 2006), suggesting that a similar hurdle may have confronted the development of life on Mars. Phosphate available for biological reactions can be introduced into aqueous environments through dissolution of primary phosphate minerals during waterrock interactions, but little is known about the dissolution of the dominant phosphate minerals found in martian meteorites and presumably on Mars (McSween and Treiman, 1998; McCubbin and Nekvasil, 2008; Filiberto and Treiman, 2009b; McCubbin et al., 2012). Here we present dissolution rates, phosphate release rates and solubilities of phosphate minerals found in martian rocks as determined from laboratory measurements. Our experimental findings predict phosphate release rates during water-rock interactions on Mars that are as much as 45 times higher than on Earth and phosphate concentrations of early wet martian environments more than twice those of Earth. We suggest that available phosphate may have mitigated one of the hurdles to abiogenesis on Mars.

Phosphate possesses many attributes that exclusively suit the ion for biotic and prebiotic roles, including stabilizing adenosinetriphosphate (ATP), deoxyribonucleic acid (DNA), ribonucleic acid (RNA) and contributing to the amphiphilic character of phospholipids that make them ideal components of cell membranes (Westheimer, 1987). In DNA, for example, the phosphate bond configuration and resulting net negative charge act to ionize and protect the macromolecule from hydrolysis, qualities required for cell containment and the persistence of genetic material (Westheimer, 1987). The hydrolysis half-life of phosphate-based DNA is of the order of millions of years, its stability making

it ideal for retaining genetic information (Schroeder et al., 2006). In contrast, DNA-type diesters based on anions similar to phosphate, such as arsenate, sulfate, silicate, or orthovanadate, possess half-lives of the order of days at most, which is not sufficient for the faithful reproduction of genetic material (Westheimer, 1987; Schroeder et al., 2006). No terrestrial organism has been observed to persist in an environment completely devoid of phosphate. Phosphate availability is also considered essential for prebiotic reactions that eventually led to terrestrial life (Miller and Urey, 1959). Phospholipid polymers have been observed to abiotically form in water and self-assemble into bilayers that may have acted as precursors to present biological cell membranes (Hargreaves et al., 1977). Phosphate has also been shown to be key in reactions leading to abiotic formation of RNA, both in the RNA itself and acting as a catalyst and pH buffer, which may have been a step towards a so-called RNA world and a possible precursor to present DNA-using life (Powner et al., 2009). Phosphate availability for prebiotic reactions is considered critical enough that the low solubility of phosphate minerals resulting in low concentrations of phosphate in prebiotic terrestrial waters and the relatively poor reactivity of phosphate towards organic compounds, may have been a hurdle for the origin of life (that is, the socalled phosphate problem)(Keefe and Miller, 1995; Schwartz, 2006). Numerous studies have proposed plausible mechanisms for the synthesis of organophosphates using agents such as cyanamide, salts of phosphoric acid, or heating in the presence of aluminum phosphate (Schwartz, 2006), thus addressing the poor reactivity. However, all of these mechanisms require phosphate availability. The fundamental necessity of aqueous phosphate availability has led to a number of studies suggesting mechanisms to enhance global or local P concentrations, including the introduction of more soluble reduced P

species by volcanism, meteorites, or lightning (Yamagata et al., 1991; Schwartz, 2006; Pasek and Lauretta, 2008; Pasek and Block, 2009). Whatever mechanism allowed for terrestrial abiogenesis, the phosphate problem underlines the crucial nature of phosphate in biotic and probably prebiotic reactions. Given the possible similarity of past and perhaps present environments on Mars and the chemical attributes of phosphate that make it essential for terrestrial life, if life arose on Mars, phosphate availability probably played a critical role (Miller and Urey, 1959).

Mars is phosphate rich (five to ten times more than Earth) as evidenced from martian meteorites and alpha particle X-ray spectrometry rock analyses by the Mars Exploration Rovers (Appendix II, Table S5). However, the presence of phosphate is not equivalent to phosphate availability. Phosphate has no significant volatile phase under terrestrial or martian conditions. Phosphate is first introduced and made bioavailable in an environment through the dissolution of primary phosphate minerals in igneous rocks during rock-water interactions.

Dissolution of phosphate-containing minerals has probably occurred on Mars. Alpha particle X-ray spectrometry analyses from Mars Exploration Rover Spirit indicate a loss of Ca and P from the surfaces of high-P rocks such as Wishstone at Gusev Crater (Gellert et al., 2006), suggesting dissolution of a Ca- and P-containing mineral. The dissolution rate of primary phosphate minerals from rocks such as Wishstone controls the rate of introduction of phosphate into aqueous and therefore potentially habitable environments and the solubility of these minerals controls equilibrium phosphate concentrations. Thus, fundamental thermodynamic and kinetic data (for example, solubility and dissolution rates) from Mars-relevant phosphate minerals can be applied to

interpret phosphate release as part of a conceptual model of martian phosphate mobility. Such application of fundamental kinetic and thermodynamic data is extensively used to interpret natural systems (Drever, 2005).

The dissolution rates and solubility of fluorapatite, $Ca_5(PO_4)_3F$, the most common primary phosphate mineral on Earth, are well studied (Valsami-Jones et al., 1998; Welch et al., 2002; Guidry and Mackenzie, 2003; Chaïrat et al., 2007; Harouiya et al., 2007). In contrast, few kinetic and thermodynamic data exist for merrillite, $Ca_9NaMg(PO_4)_7$ [an extraterrestrial variety of whitlockite, $Ca_9Mg(PO_3OH)(PO_4)_6$] and Cl-rich apatite, $Ca_5(PO_4)_3(Cl; F)$, the dominant primary phosphate minerals in martian meteorites (McSween and Treiman, 1998). Therefore, to better understand martian phosphate release and availability, we measured chlorapatite, merrillite and whitlockite dissolution rates over a range of pH values using the initial rate method and carried out solubility experiments (see Appendix II Supplementary Methods).

Fluorapatite dissolution rates and solubility were also measured to determine the relative differences in dissolution rates and equilibrium concentrations between the Mars-relevant primary phosphate minerals and fluorapatite. Although mineral dissolution rates measured in the laboratory often cannot be directly extrapolated to the field, relative differences in dissolution rates between minerals can, allowing for comparison of the rate of phosphate release during water-rock interactions in terrestrial versus martian environments.

Mineral dissolution was measured in duplicate or triplicate experiments for each mineral over a pH range of 2 to 5 in 35 individual batch experiments. Results indicate higher Ca and P release rates for chlorapatite, whitlockite and merrillite than for

fluorapatite (Fig. 5 and Supplementary Sheets S1-S35 of Appendix II). Ca: P release ratios are generally stoichiometric or close to stoichiometric in experiments (Appendix II Supplementary Sheets S1-S35). Mineral dissolution rates for individual experiments were determined through linear regression analysis of cumulative calcium release over time (Appendix II Supplementary Methods) and are compiled in Supplementary Table S6 of Appendix II.

Rate laws for each mineral were developed by applying linear regression to log dissolution rates of averaged batch experiments versus pH (Table 5 and Appendix II Supplementary Sheets S1-S35). The resulting pH dependence of mineral dissolution was fit to:

 $\log R = \log k_{H+} - n_{H+} p H,$

where *R* is the mineral dissolution rate (in mol m⁻² s⁻¹), k_{H^+} is the intrinsic rate constant in units of mol m⁻² s⁻¹ and n_{H^+} is the reaction order with respect to H+ (Table 6). To compare our results for well-studied fluorapatite with dissolution rates from the literature, our mineral dissolution rates of fluorapatite were plotted against results from previous studies (Valsami-Jones et al., 1998; Welch et al., 2002; Guidry and Mackenzie, 2003; Chaïrat et al., 2007; Harouiya et al., 2007). Our fluorapatite dissolution rates all fall within half an order of magnitude of previously published data. Our derived value of k_{H^+} for the fluorapatite rate law is within error of a previously published rate law, and our value of n_{H^+} is within 10% of a previously published rate law (Appendix II, Fig. S2)(Chaïrat et al., 2007).

Comparison of chlorapatite, merrillite, whitlockite and fluorapatite dissolution rates indicates that chlorapatite dissolves more rapidly than fluorapatite over the entire pH range of the experiments, and merrillite and whitlockite exhibit faster dissolution than

fluorapatite over the pH range of 2 to 4.5 (Fig. 6a), although, owing to mineral stoichiometry, P release is greater over the entire pH range for these minerals (Fig. 6b). The rapid dissolution of whitlockite and merrillite relative to fluorapatite is probably the result of significant structural differences between the two mineral types. In contrast, similar dissolution rates of whitlockite and merrillite are probably owing to the very similar mineral structures that differ mainly by the higher calcium in the merrillite used in these experiments and slight structural differences resulting from dehydrogenation of whitlockite (Appendix II Supplementary Methods). The faster dissolution of chlorapatite relative to fluorapatite may be the result of somewhat weaker bonds in chlorapatite from incorporating the larger Cl⁻ ion into the anion column sites of the mineral. The approximately equal dependence on pH of fluorapatite and chlorapatite dissolution is similar to that of other minerals that form solid solution series such as celestite/anglesite/barite and the olivine series (Wogelius and Walther, 1992; Dove and Czank, 1995). However, a change in structure from P63=m hexagonal structure (fluorapatite) to P21=b monoclinic structure (chlorapatite) occurs at the Cl endmember (Hughes et al., 1989) and enthalpy data suggest nonlinearity may exist along the Cl-F join (Hovis and Harlov, 2010). Nonetheless, dissolution and solubility data presented here should represent upper and lower bounds and be useful in approximating dissolution of mixed F and Cl apatites that have been observed in martian meteorites (McCubbin and Nekvasil, 2008; McCubbin et al., 2012).

Mineral dissolution rates are generally expressed in terms of moles of mineral released per unit of time (Fig. 6a); however, differences in mineral stoichiometry result in more moles of phosphate released per mole of merrillite or whitlockite than per mole of

apatite. The mineral dissolution rates and stoichiometry measured here indicate phosphate release from whitlockite and merrillite that is 1.9 to 4.8 times greater than fluorapatite over the measured pH range. Over the same pH range, phosphate release from chlorapatite ranges from 2.2 to 4.1 times higher than fluorapatite (Fig. 6b). Factoring in the approximate order of magnitude greater presence of phosphate minerals inferred from the P content of bulk Mars versus Earth (Appendix II Table S5), rates of phosphate introduction into martian environments during rock-water interactions could be >45 times that of Earth.

Solubility experiments carried out on the minerals (Table 6 and Appendix II Supplementary Methods) indicate that equilibrium concentrations of Ca and P from chlorapatite, whitlockite, and merrillite are significantly greater than from fluorapatite (Fig. 7). Solubility product constant (K_{sp}) values for fluorapatite from our experiments generally agree with previous studies (Appendix II Table S7) lending confidence to the method. The solubility experiments also allowed for calculation of dissolution rates under near-neutral conditions (pH ~7; Appendix II Supplementary Methods and Fig. S3), which indicate that phosphate release from chlorapatite, merrillite and whitlockite dissolution remain higher than from fluorapatite under neutral conditions as well as acidic (Appendix II Fig. S4). Although Earth and Mars have probably possessed both alkaline and acidic aqueous environments (Ehlmann et al., 2008), and the pH necessary for crucial prebiotic reactions is not known for either Earth or potentially Mars, it is not unlikely that prebiotic reactions that lead to abiogenesis occur in a specific pH range (for example, the abiotic synthesis of RNA requires pH buffering) (Powner et al., 2009). Therefore, under such expected similar pH environments, the greater phosphate release from the Mars-

relevant minerals relative to fluorapatite would be important in alleviating any shortage of phosphate.

The higher solubilities of Mars-relevant primary phosphate minerals and faster dissolution rates measured here suggest higher phosphate concentrations in solution and that phosphate removed from solution by prebiotic, biotic, or abiotic reactions would be replaced rapidly, effectively increasing phosphate availability in situations of rapid uptake or removal. Phosphate is but one bioessential component and the fundamental data presented here do not specifically replicate conditions in a particular martian environment. However, the faster dissolution rates of Mars-relevant dominant primary phosphate minerals, higher concentrations of phosphate minerals in the martian crust and higher final phosphate concentrations in solution from Mars-relevant phosphate minerals all suggest higher phosphate availability on Mars relative to Earth. The results suggest that the phosphate problem may be much less significant on Mars than on Earth, possibly mitigating one of the hurdles faced by potential martian abiogenesis.

Methods

Dissolution rates of synthetic chlorapatite, whitlockite, merrillite and natural fluorapatite were measured by the initial rate method in batch dissolution experiments (Rimstidt and Newcomb, 1993). This method, where a rate is determined under conditions at the onset of far-from-equilibrium batch experiments, has been demonstrated to yield results comparable to other experimental methods (Rimstidt and Newcomb, 1993). A graphical representation of the method is included as Appendix II Fig. S1. Solubility experiments were carried out in pure water using previously published

methods (Zhu et al., 2009). Detailed materials, methods, data analysis and any associated references are provided in Appendix II Supplementary Methods.

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Tables

Mineral	Initial pH	n*	Ave. log <i>R</i> (mol m ⁻² s ⁻¹)
Whitlockite	2	2	-6.43
Whitlockite	3	2	-7.20
Whitlockite	4	2	-8.46
Whitlockite	5	2	-9.17
Merrillite	2	2	-6.49
Merrillite	3	2	-7.22
Merrillite	4	2	-8.20
Merrillite	5	2	-9.25
Chlorapatite	2	2	-6.27
Chlorapatite	3	2	-6.66
Chlorapatite	4	2	-8.15
Chlorapatite	5	2	-8.81
Fluorapatite	2	3	-6.64
Fluorapatite	3	3	-7.58
Fluorapatite	4	3	-8.39
Fluorapatite	5	2	-9.09

Table 5. Log mineral dissolution rates and accompanying pH values.

*n = number of experiments average is based on.

Mineral	Dissolution rate experiments					
	log k _{H+} (mol m ⁻² s	¹) n _{H+}				
Merrillite	-4.56 ±0.192	-0.92 ±0.052	2			
Whitlockite	-4.50 ±0.274	-0.95 ±0.075	5			
Chlorapatite	-4.28 ±0.524 -0.91 ±0.143		3			
Fluorapatite	-5.07 ±0.138	±0.138 -0.82 ±0.038				
	Solubility experiments ^a					
	Ca mmol*	P mmol*	рΗ	\mathbf{K}_{sp}		
Merrillite	0.0358 ±0.0005	0.0254 ±0.0010	6.6	-39		
Whitlockite	0.0490 ±0.0008	0.0368 ±0.0014	6.8	-39		
Chlorapatite	0.0466 ±0.0012	0.0226 ±0.0009	6.9	-22		
Fluorapatite	0.0224 ±0.0007	0.0109 ±0.0012	6.5	-27		

Table 6. Dissolution rate constants (log k_{H^+}), pH dependence (n_{H^+}), and final concentrations from solubility experiments with calculated K_{sp} values.

Fluorapatite 0.0224 ±0.0007 0.0109 ±0.0012 6.5 -27 *Range values are standard errors (dissolution) or standard deviations (solubility). aReactions: Merrillite;* $Ca_{9.5}Mg(PO_4)_7 + 7H^+ = 9.5Ca^{2+} + Mg^{2+} + 7HPO_4^{2-}$, *Whitlockite;* $Ca_9Mg(PO_3OH)(PO_4)_6 + 6H^+ = 9Ca^{2+} + Mg^{2+} + 7HPO_4^{2-}$, *Chlorapatite;* $Ca_5(PO_4)_3Cl + 3H^+ = 5Ca^{2+} + 3HPO_4^{2-} + Cl$, *Fluorapatite;* $Ca_5(PO_4)_3F + 3H^+ = 5Ca^{2+} + 3HPO_4^{2-} + F$. *average of 3 analyses




Figure 5. Calcium release over time. Initial calcium release over time (average of linear fits of duplicate or triplicate experiments) at pH 2 for chlorapatite, whitlockite, merrillite and fluorapatite. Relative release rates are representative of all pH conditions tested here, where chlorapatite, whitlockite, and merrillite all exhibit faster Ca and P release rates than fluorapatite.



Figure 6. Mineral dissolution rates and phosphate release rates. **a**, Log mineral dissolution rates versus pH for whitlockite, merrillite, chlorapatite and fluorapatite. Plotted points are the average of two or more dissolution experiments (Appendix II Table S6). Lines are derived from regression analyses and represent measured rate laws. **b**, Log rates of phosphate release versus pH based on measured mineral dissolution rates and stoichiometric release. Plots show that Mars-relevant minerals chlorapatite, whitlockite and merrillite all release phosphate at significantly higher rates than fluorapatite during dissolution under these conditions.



Figure 7. Calcium concentrations over time in solubility experiments. Values are averages of duplicate experiments for each mineral. Plot shows that Ca concentrations over time are higher for chlorapatite, merrillite and whitlockite, versus fluorapatite. Labeled dashed lines represent final concentrations based on three point averages. Phosphate release follows the same trends. Experiment conducted at 25 °C under neutral conditions. End pH values were 6.9, 6.6, 6.8 and 6.5 for chlorapatite, merrillite, whitlockite and fluorapatite, respectively.

CHAPTER FOUR

WEATHERING PROFILES IN HIGH-P ROCKS AT GUSEV CRATER, MARS, SUGGEST DISSOLUTION OF PHOSPHATE MINERALS INTO NEAR-NEUTRAL WATERS

Abstract

Abundant evidence indicates that significant surface and near-surface liquid water has existed on Mars in the past. Evaluating the potential for habitable environments on Mars requires an understanding of the chemical and physical conditions that prevailed in such aqueous environments. Among the geologic features that may hold evidence of past environmental conditions on Mars are dissolution profiles, such as those in the high-P bearing Wishstone class rocks. The dissolution profiles of these rocks indicate the loss of a Ca-phosphate mineral occurred during past aqueous interactions, and the high-P content of these rocks makes them of astrobiological interest because the phosphorus released from them during aqueous interactions is among the elements thought to be required for life. In this work, we used Mars mission data, laboratory-derived kinetic and thermodynamic data, and data from terrestrial analogs for Mars rocks, including our own observations of potential martian high-P analog basalts from Craters of the Moon National Monument (COTM), to model a conceptualized Wishstone class rock using the reactive transport code CrunchFlow. Our results suggest that the weathering profiles in Wishstone class rocks are consistent with dissolution of minerals from a rock containing

multiple Ca-phosphate minerals into near-neutral environments that may have been conducive to life on Mars.

Introduction

Although abundant geologic and mineralogical data indicate that Mars experienced a period of significant and regionally widespread surface and near-surface liquid water (Carr, 1996; Christensen et al., 2000; Bibring et al., 2005; Baker, 2006; Bibring et al., 2006; Clark et al., 2007; Michalski and Dobrea, 2007; Ehlmann et al., 2008; Milliken et al., 2008; Mustard et al., 2008; Murchie et al., 2009; Wray et al., 2009; Milliken et al., 2010; Grotzinger et al., 2013; Williams et al., 2013; Gainey et al., 2014), the specific characteristics of these past waters, such as pH, remain largely unknown. There is evidence of acidic aqueous interactions on Mars, such as the discovery of jarosite at Meridiani Planum (Elwood Madden et al., 2004; Klingelhöfer et al., 2004) and the stratigraphic relationship of Fe-sulfates with opaline silica in young martian deposits in the same area (Milliken et al., 2008), both indicators of a period of acidic aqueous conditions. However, at the Phoenix Lander site, soil/mineral experiments suggest more neutral or slightly alkaline conditions may have prevailed (Hecht et al., 2009) and mineralogical findings at Gale Crater by MSL also suggest a neutral pH environment at that location (Grotzinger et al., 2013). Recent work focused on Comanche rock class outcrops has suggested that the area of Columbia Hills in Gusev Crater on Mars may have experience periodic flooding of mildly acidic to near neutral waters (Ruff et al., 2014), while other studies suggest more acidic aqueous interactions in Gusev (Gellert et

al., 2004; Arvidson et al., 2006; Hurowitz et al., 2006a). It is likely that Mars possessed a diversity of aqueous environmental conditions that evolved over time.

Weathering profiles, persistence of primary minerals, and mineral alteration products of martian rocks can be indicators of the chemical and physical aqueous conditions under which they were formed, such as pH and duration of water/rock interaction (Elwood Madden et al., 2004; Klingelhöfer et al., 2004; Papike et al., 2006; Stopar et al., 2006; Olsen and Rimstidt, 2007; Hausrath et al., 2008a; Hausrath et al., 2008b; Elwood Madden et al., 2009; Hecht et al., 2009; Hausrath et al., 2011; Grotzinger et al., 2013; Hausrath and Olsen, 2013; Gainey et al., 2014). Numerical modeling is a technique which has been extensively applied to such indicators in order to quantitatively interpret past aqueous conditions which may have prevailed on Mars (Elwood Madden et al., 2004; Tosca et al., 2005; Chevrier, 2006; Zolotov and Mironenko, 2007; Hausrath et al., 2008a; McAdam et al., 2008; Schwenzer and Kring, 2008; Hausrath and Olsen, 2013). Here we apply reactive transport modeling to model dissolution profiles in Wishstone class float rocks on Mars.

Wishstone class rocks were encountered by the Mars Exploration Rover (MER) Spirit in the Columbia Hills of Gusev Crater on Mars and two rocks of the class, Wishstone and Champagne, were analyzed with the full MER Spirit science package, including Alpha Particle X-ray Spectroscopy (APXS). APXS analyses by MER Spirit of this rock class indicate a decrease in Ca and P at the rock surfaces, suggesting alteration of the rocks by mineral dissolution. Ca and P are the only elements to show appreciable loss, and Ca/ P ratios at the rock surfaces compared to Ca/ P ratios in the interior suggest that a Ca-phosphate mineral, likely apatite or merrillite, though the ratios also allow for a

secondary mineral like brushite (Hurowitz et al., 2006a; Ming et al., 2006), is the only mineral that has seen significant dissolution from the rocks during past aqueous interactions. This somewhat unusual profile may be indicative of specific aqueous conditions, shedding light on past martian weathering conditions. This rock class is of further astrobiological interest because P is a bioessential nutrient and the release of P from martian rocks has implications for past martian habitability.

Previous work has suggested Wishstone class dissolution profiles could be the result of highly acidic weathering. Hurowitz et al. (2006a), based on fluorapatite dissolution kinetics, suggested leaching driven by repeated episodic exposure to volcanically derived acidic fogs might result in the profiles measured by APXS. Gellert et al. (2004), also suggested acidic conditions based on laboratory observations of meteorites interacting with dilute acid solutions. Here, in order to investigate the aqueous conditions under which Wishstone class rocks may have weathered and released phosphate into the environment, with implications both for potentially widespread weathering conditions on Mars and for mobility of the important nutrient P on Mars, we used reactive transport modeling to quantitatively interpret Wishstone weathering profiles. We used Mars mission data, laboratory derived kinetic and thermodynamic data, and data from terrestrial analogs, including the high-P basalts at Craters of the Moon National Monument (COTM), to inform our model. Our results suggest the weathering profiles in Wishstone class rocks may indicate dissolution of phosphate minerals by near neutral waters, with important implications for past weathering conditions on Mars, as well as a better understanding of P release into aqueous martian environments.

Methods

High-P Wishstone Class Rocks

Wishstone class rocks comprise the dominant population of float rocks encountered by Spirit on the northwest flank of Husband Hill and are common among the float rocks of Cumberland Ridge (Arvidson et al., 2006; Squyres et al., 2006). Miniature Thermal Emission Spectrometer (MiniTES) classified over 95 Wishstone class rocks at Columbia Hills (Ruff et al., 2006), although only two rocks, Wishstone and Champagne, were subjected to the full MER Spirit analysis package. No source or outcrop for the rocks has been identified and, though a number of studies suggest plausible mineralogies for the class based on differing assumptions (Hurowitz et al., 2006b; McSween, 2006; McSween et al., 2006; Ming et al., 2006; McSween et al., 2008; Ming et al., 2008; Usui et al., 2008), the MER science package allowed for only limited direct mineralogy measurements by Mini-TES or by Mossbauer of Fe-containing phases. Alpha Particle Xray Spectroscopy (APXS) of two Wishstone class rocks (i.e. Wishstone and Champagne) showed them to be typified by high P (\sim 5% as P₂O₅), and low Cr (below detection limit) (Gellert et al., 2006; Ming et al., 2006). The P in the rocks is considered high for common basaltic rock (Ming et al., 2006; Squyres et al., 2006), though terrestrially rare phoscorite-carbonatite complex rocks are high in P, and the high-P basalt flows at Craters of the Moon National Monument in Idaho, USA, have also been suggested as plausible analogs for the martian rocks (Ming et al., 2008; Usui et al., 2008). Images from Spirit's Microscopic Imager (MI) show some angular grains present in the rocks, thus, the rocks have also been suggested as pyroclastic in origin or possibly a breccia resulting from

impact processes (Squyres et al., 2006). With these uncertainties in mind, whenever possible we chose modeling inputs that were appropriate for a plausible range of Wishstone mineralogies and formation mechanisms.

To provide additional constraints for the model inputs of a Wishstone class rock and inform other parameters of the modeling, we examined petrographic thin sections of basalts from Craters of the Moon National Monument in Idaho (COTM) by Scanning Electron Microscopy (SEM) (Appendix III supplementary information and accompanying Figs. S5-S12, and Appendix IV). These high-P-bearing basalts (up to 2.9 wt% as P₂O₅) (Kuntz et al., 1992) have been suggested as analogs to high-P rocks on Mars such as Wishstone class (Usui et al., 2008). Insights gained from investigating COTM rocks were used in conjunction with other data to inform the model on secondary minerals, porosity, and specific surface areas of minerals as discussed below (detailed methods and results of COTM observations are contained in Appendix III supplementary information, and Appendix IV).

Model inputs for mineralogy of a conceptualized Wishstone class rock were based on CIPW norm calculations of McSween et al. (2006) from MER Spirit APXS which included information from Mossbauer measurements (Table 7). To simplify the model, minerals present in the CIPW norm calculations at less than 5 wt% were not incorporated into the model, ilmenite and magnetite were combined into a single mineral, and feldspars were also input as a single mineral with an intermediate composition (An₂₅).

Analyses by APXS of an "as is" surface, brushed surfaces, and Rock Abrasion Tool (RAT) treated surfaces of Wishstone class rocks indicated less Ca and P in the surface of the rocks than the interior (Gellert et al., 2006; Brückner et al., 2008) (Table

8). Ratios of Ca and P between rock surfaces and interiors are consistent with the loss of a Ca-phosphate mineral and suggest a Ca-phosphate mineral as the only mineral substantially lost from the rocks due to alteration (Hurowitz et al., 2006a; Ming et al., 2006). Martian meteorite data indicate that the dominant primary phosphate minerals on Mars are Cl-apatite and merrillite (McSween and Treiman, 1998), often found coexisting in martian meteorites, although fluorapatite (the dominate primary phosphate mineral on Earth) also occurs in some martian meteorites with chlorapatite (McCubbin and Nekvasil, 2008). Ca/ P ratio differences in APSX analyses are also reasonably consistent with apatite and merrillite. We therefore configured models with merrillite, chlorapatite, and fluorapatite each as the only Ca-P-bearing mineral, as well as models that included both chlorapatite and merrillite within the model.

The Ca/ P ratios in APXS data of Wishstone class rocks are also reasonably consistent with brushite, a secondary phosphate mineral (Hurowitz et al., 2006a). Because brushite is a secondary mineral and observations of Wishstone class rocks suggests them be less altered than other Columbia Hills materials, brushite is not favored as the source of P in Wishstone class (Ming et al., 2006). Thus, brushite has not been considered in CIPW mineralogy norms of Wishstone class rocks or other petrologic interpretations of the rocks (Gellert et al., 2004; McSween et al., 2006; Ming et al., 2006; McSween et al., 2008; Ming et al., 2008; Usui et al., 2008). The mineral is also rare on Earth and has not been detected in either martian meteorites or on Mars. The presence of brushite in Wishstone class rocks as more than an alteration product in the near surface, would limit Wishstone class petrogenesis to sedimentary or impact breccias derived from a very brushite rich (10 wt %+) source area, or a secondary process that generated

brushite within the rocks while preserving other primary minerals, scenarios that seem unlikely. For this reason, brushite dissolution is not considered further in this paper.

A fit of MiniTES data indicate up to 10% basalt glass may be present in Wishstone class rocks along with 10% phosphate minerals (Ruff et al., 2006). Phosphate is incompatible and can be concentrated in basalt glasses. Thus, a high-P basalt glass could be a Ca-P-containing phase within Wishstone class rocks along with another Caphosphate mineral (or multiple Ca-phosphate minerals). Glass kinetics and thermodynamics vary significantly as a function of glass composition, which itself is highly variable (Hamilton, 1999; Wolff-Boenisch et al., 2004). Thus, a Wishstone glass could be the phase dissolving from the rocks, or be a resistant phase which accounts for the P concentration on brushed surfaces analyzed by APXS. However, the 10% phosphate minerals in the MiniTES fit would account for nearly all of the P measured by APXS in Wishstone class. Further, if a glass does play a significant role in the measured profiles of Wishstone class, the Ca/ P ratio in the glass would have to roughly match that of a Ca-phosphate mineral and the P concentrations would have to be very high (as much as 25 wt% P_2O_5 in the glass to account for the dissolving phase or a resistant phase contributing to the higher P concentrations within the rock). In addition, if the glass is the dissolving component, any remaining chemistry outside of the Ca and P in the glass would have to be consistent with the bulk rock to produce the profiles measured in Wishstone class by APXS. For these reasons we did not include a glass component in finalized models. The possibility of a glass within Wishstone class as one of the Pbearing phases is, however, still plausible.

Since Wishstone class rocks have been previously suggested to be impact breccias, pyroclastics, or extrusive igneous rocks (Arvidson et al., 2006; Squyres et al., 2006), initial porosity values were set to 5% based on image analysis of COTM basalt (Appendix III supplementary information), and data for volcanic breccias (Rejeki et al., 2005; Benning, 2008), tuffs (Keller, 1960; Rejeki et al., 2005), terrestrial basalts (Davis, 1969; Freeze and Cherry, 1977; Sato et al., 1997; Rejeki et al., 2005), lunar basalts (Kiefer et al., 2012; Macke et al., 2012), and brecciated basalt or fractured basalt (Freeze and Cherry, 1977) (Table 9). We varied porosity from 3% to 10% in early modeling and found that porosity variations alone did not significantly change the relative dissolution depths of minerals.

Tortuosity (T) is a measure of the complexity of permeable paths within rocks. Though it can be calculated in a number of ways, in the modeling code used in this study (Crunchflow, discussed below) the value is equivalent to:

$$T=\frac{D_e}{D_w},$$

where D_e is the effective diffusion coefficient and D_w is the coefficient in pure water (Steefel, 2008; Steefel, 2009). Values derived from experimental data on both basalt and volcanic breccias range from 5.76 x 10⁻³ to 3.56 x 10⁻⁶ (Sato et al., 1997; Benning, 2008). We chose an intermediate value of 1 x 10⁻⁴. Varying the tortuosity over 2 orders of magnitude in preliminary models changed the absolute depths of the reaction fronts but did not change the relative positions of the reaction fronts.

Mineral dissolution rates and solubilities were sourced from the literature (Table 10). In order to account for Al immobility, a generalized Al-bearing amorphous-like phase based on kaolinite/halloysite kinetics and thermodynamics was allowed to

precipitate in the models (Table 10). This phase is not listed in Table 7 because it is a precipitate rather than an initial component in the rock. Phosphate was also allowed to sorb to mineral surfaces in the model, after Dzombak and Morel, (1990). Solubility values for Ca-phosphate minerals were varied by +/- 2 orders of magnitude, based on variations in measured solubilities for fluorapatite in the literature (+/- 5 orders of magnitude, Oelkers et al., 2009). This variation showed some change in the overall depth of dissolution of Ca-phosphate minerals in preliminary modeling profiles, suggesting a component of diffusion-limited dissolution exists in the models (see Appendix III Figs. S13 and S14 for a general description of model output profiles and the effects of solubility on Ca-phosphate minerals respectively). Finalized models were run with measured values (Adcock et al., 2013).

Mineral dissolution rates are generally normalized to mineral surface areas. To estimate mineral surface areas for this study, grain size measurements taken from the COTM Kimama lava flow thin sections were used to derive grain diameters for the minerals ilmenite, plagioclase, and fluorapatite (Appendix III supplementary information). Kimama does not contain appreciable pyroxene, however, pyroxenes and plagioclase in both Blue Dragon and Minidoka COTM basalts are roughly the same size, so we assumed pyroxene grain sizes to be similar to plagioclase for modeling. We also assumed merrillite and chlorapatite grain sizes to be similar to fluorapatite in Kimama thin sections. The relative grain sizes followed feldspar ~ pyroxene > ilmenite > Caphosphate mineral. The geometric (A_{geo}) specific surface area (SSA) was then determined for the minerals using:

$$A_{geo} = \frac{6V_m}{W_m D}$$

where $V_{\rm m}$ is the molar volume in m³ mol⁻¹, $W_{\rm m}$ is the molar mass in g mol⁻¹, and *D* is the grain diameter in meters (Rimstidt et al., 2012). Most laboratory-derived dissolution rates (including those used as inputs in this study) normalize dissolution rates to surface areas determined by gas sorption using the Brunauer, Emmett, and Teller method (BET) (Brunauer et al., 1938). The relationship between gas sorption-measured surface areas and geometrically calculated surface areas (the roughness) is expressed as:

$$\lambda = \frac{A_{BET}}{A_{geo}},$$

where λ is a unit-less roughness coefficient, A_{BET} is the specific surface area as measured by BET in m² g⁻¹, and A_{geo} is the geometrically calculated specific surface area also in units of m² g⁻¹. Geometric SSA's calculated for minerals were multiplied by roughness coefficient values from the literature (White et al., 1994; Brantley and Mellott, 2000; Adcock et al., 2013) to approximate BET SSAs. The mineral surface area that interacts with aqueous solutions (i.e. the total reactive surface area) is considered the largest source of uncertainty in field calculations of mineral dissolution rates (White, 2002). Previous research has shown that the mineral surface actually in contact with weathering fluids during rock/water interactions may only be 0.1 to 10% of the measurable mineral surface area (Velbel, 1993). In order to approximate this effect in our modeling, we chose an intermediate value of 5% and applied it to our mineral surface area, shown in Table 10. Varying the surface areas over 2 orders of magnitude did not change the relative positions of reaction fronts in model output.

During chemical weathering, some component of surface retreat is expected as porosity develops to a point where the rock is no longer competent. CrunchFlow is capable of incorporating surface retreat or erosion at a specific rate, however there is no

current option for surface retreat based on porosity due to chemical weathering. Based on observed maximum total porosities for altered or fractured basalt (Freeze and Cherry, 1977; Sak et al., 2004; Navarre-Sitchler et al., 2009) and typical porosities of unconsolidated sediments (Davis, 1969; Freeze and Cherry, 1977), we considered 50% porosity as a reasonable threshold for surface retreat in our models, and applied the surface retreat component into results after modeling. We also considered physical erosion beyond that associated with surface retreat from chemical weathering based on the observation that many of the rocks at Gusev Crater are ventifacts and Wishstone itself has rounded edges, some faceting, and streaking of RAT surfaces (Greeley et al., 2006)(Appendix III Figs. S15 and S16).

Reactive Transport Modeling

The dissolution profiles that form in rocks, including Wishstone class rocks, are the result of a combination of chemical (e.g. mineral kinetics, thermodynamics, solution pH) and physical (e.g. reactive surface area, porosity, tortuosity, mineralogy) conditions at work during water/rock interactions. Reactive transport modeling is a tool that can incorporate the effects of multiple chemical and physical parameters during aqueous interactions, produce dissolution profiles in model space (see Appendix III Fig. S13 as an example), and constrain the conditions under which specific dissolution profiles may form, such as those observed in Wishstone class rocks.

Modeling in this study was conducted using the reactive transport code CrunchFlow written by Carl Steefel (Steefel, 2010). The code allows for a global implicit reactive transport modeling approach (GIMRT) in which transport and

multicomponent reactions are simultaneously solved (Steefel and Lasaga, 1994; Steefel and MacQuarrie, 1996; Steefel, 2009). CrunchFlow has been used in a number of previous water/rock interaction studies, including those applied to Mars (Giambalvo et al., 2002; Knauss et al., 2005; Maher et al., 2006; Hausrath et al., 2008a; Dontsova, 2009; Maher et al., 2009; Navarre-Sitchler et al., 2011; Shen et al., 2011; Beisman et al., 2013; Hausrath and Olsen, 2013).

In order to investigate the solution which interacted with high-P rocks on Mars, we constructed a model using CrunchFlow with the conceptualized Wishstone class rock in contact with waters of variable pH on the martian surface. The water contacting the rock during rock-water interactions was modeled as a low ionic strength solution with the initial pH adjusted for each corresponding model over a range of 2 to 8. Partial pressures for oxygen and CO_2 were set to present day martian atmospheric pressures. Solution temperature was set to 1 °C.

Models were conceptualized as a column of 150 cells of 50 µm depth representing the rock Wishstone from surface to interior, and were run for up to 100,000 years of water-rock interaction. Models included a Dirichlet boundary for aqueous and gaseous species at the interaction surface. Total surface area is calculated by CrunchFlow using a shrinking sphere model (Levenspiel, 1972; Lasaga, 1998), and all parameters (surface area, mineral volumes, surface area, porosity and others) are updated by CrunchFlow at the end of each time iteration.

Results and Discussion

Understanding the persistence and characteristics of liquid water is important in assessing the habitability of Mars. Weathering profiles that develop within rocks during rock/water-interactions can be indicators of the aqueous conditions under which they were formed. Profiles in martian Wishstone class rocks, which indicate dissolution mainly restricted to Ca-phosphate mineralogy (Hurowitz et al., 2006a; Ming et al., 2006) are of particular interest. Not only are these somewhat unusual profiles likely indicative of specific past martian aqueous conditions, but phosphate is a required nutrient for all known life and phosphorus (as phosphate or a more reduced species such as phosphite) was likely required for the origin of terrestrial life (Westheimer, 1987; Schwartz, 2006; Powner et al., 2009; Pasek and Kee, 2011). Thus the profiles in Wishstone rocks could not only yield insight into alteration under martian conditions, but also the characteristics of release of a nutrient essential for both the origin and persistence of life.

In previous studies the profiles of Wishstone rocks have been suggested to be the result of acidic chemical weathering, possibly by volcanically derived acid fogs, based either on kinetic values for fluorapatite dissolution or the observation of apatite dissolution from meteorites by dilute acids in the laboratory (Gellert et al., 2004; Hurowitz et al., 2006a). The use of reactive transport modeling allows the consideration of multiple factors affecting mineral dissolution, including kinetics, thermodynamics, porosity, tortuosity, and specific surface area, both simultaneously and quantitatively.

Model output from CrunchFlow is generated at user-defined modeling times (e.g. at 100,000 years of modeling) and includes modeled values correlated with cells (i.e. depths in this study) of bulk chemistry, mineralogy, surface area, and porosity within the

conceptualized rock, as well as gas concentrations, pore-water chemistry, pH, and the saturation state of solution. Because the output data are correlated with cell values (i.e. depths into the modeled rock), they can be easy plotted as a function of depth. Mineral volume output data plotted against cell depth, for example, represents modeled mineral dissolution profiles within a rock from the original "surface" of the rock down to unaltered (or less altered) "corestone". These model generated profiles can be compared to APXS data of Wishstone on Mars, as well as analyzed using knowledge obtained from terrestrial weathering profiles. In such plots mineral dissolution front depths are represented by curves in mineral volume fraction lines (see Appendix III Fig. S13).

Model outputs in this study showed a number of overall trends consistent with the kinetic and thermodynamic properties of the modeled mineralogy of Wishstone class. For instance, in all modeled profiles, the trend of overall deeper dissolution of minerals with decreasing pH from neutral to acidic is present (Fig. 8 and Appendix III Figs. S17 to S20). This is consistent with the kinetics and thermodynamics of the minerals used in the models which all tend toward faster dissolution rates and higher solubilities with decreasing pH. The depth of 50% developed porosity in models, our threshold for surface retreat due to chemical weathering, follows the same trend of increasing depth with more acidic pH values (Fig. 8).

Similarly, comparing models containing different Ca-phosphate minerals (merrillite, fluorapatite, and chlorapatite), the relative dissolution depth results between the three primary Ca-phosphate phases are in agreement with laboratory-measured dissolution rates and solubilities of the minerals (Table 10)(Adcock et al., 2013). In all pH cases modeled, fluorapatite shows the shallowest dissolution of the three Ca-

phosphate minerals (Fig. 9 and Appendix III Figs. S17-S19). Chlorapatite shows the deepest dissolution fronts of the Ca-phosphate minerals at a given pH, and merrillite is intermediate between chlorapatite and fluorapatite (Fig. 9), although in pH 7 and 8 models, merrillite and fluorapatite dissolution fronts are somewhat similar (Appendix III Figs. S17 and S19). For chlorapatite- and merrillite-bearing models, the Ca-phosphate mineral dissolution depth exceeded the depth of the chemical weathering surface retreat in all but the pH 2 models. In the pH = 2 models, the surface retreat due to enhanced porosity (> 50%) is deeper than the loss of merrillite and matched the depth of chlorapatite dissolution. Fluorapatite-bearing models showed the same trend, however, surface retreat from chemical weathering also exceeded fluorapatite mineral dissolution depth at pH 3. Models containing two Ca-phosphate minerals (i.e. chlorapatite + merrillite) showed the same overall trends as single Ca-phosphate mineral models, with dissolution depth of a particular Ca-phosphate mineral, at a given pH and modeling time, generally matching that of the same Ca-phosphate in the single phosphate mineral models.

All models (including models with two Ca-phosphate minerals) showed significant plagioclase dissolution, beyond the surface retreat from chemical weathering, over the entire pH range with fronts being deep but also relatively gradual in shape (Appendix III Figs. S17- S20). Comparatively sharp ilmenite dissolution fronts were also apparent over the entire pH range, but generally shallow in depth at pH 5 or higher. With the exception of pH 2, the depth of estimated surface retreat from chemical weathering exceeded or coincided with the dissolution front depth for illmenite, effectively removing the fronts from the overall dissolution profiles in the rock. Hypersthene showed no

significant dissolution except in models with pH <5, and in all models the surface retreat due to developed porosity exceeded any significant hypersthene dissolution (Appendix III Figs. S17-S20).

In chlorapatite models run at pH > 3 and merrillite at pH 5-6, the Ca-phosphate minerals show the most significant dissolution with depth and were completely dissolved from the rock before any other mineral. The significant plagioclase dissolution is accompanied by precipitation of the secondary Al-bearing phase at all but pH 2 and 8. This plagioclase dissolution behavior is not inconsistent with terrestrial observations of natural systems where plagioclase has been observed to be the first mineral to dissolve from rocks or exhibit relatively rapid dissolution suggesting solubility as well as kinetics may play an important role in controlling mineral weathering (White, 2008; Brantley and White, 2009; Maher et al., 2009; Navarre-Sitchler et al., 2009; Navarre-Sitchler et al., 2011). Therefore, in low-porosity rocks such as these, and others on Mars, significant dissolution of plagioclase is consistent with what we would expect from interaction of rock surfaces with liquid water.

In addition to surface retreat assumed to occur due to porosity developed from chemical weathering, additional surface erosion is also likely. Many profiles on Earth and Mars show a significant component of physical surface erosion has occurred to modify profiles (Hausrath et al., 2008a). A number of rocks at Gusev crater are fluted or faceted ventifacts or show signs of physical weathering (Greeley et al., 2006; Greeley et al., 2008), including Wishstone which shows some evidence of both faceting and rounded edges (Appendix III Fig. S15). Further, RATed surfaces on Wishstone also showed "streaking" (Greeley et al., 2008) which is an indicator of present day physical abrasion

or weathering (Appendix III Fig. S16). Thus ongoing physical erosion, in addition to surface retreat due to chemical weathering, is expected for Wishstone class rocks.

Incorporating an additional component of physical erosion could yield modeled profiles consistent with the partial loss of Ca and P from the profiles measured by APXS for models containing chlorapatite or merrillite at pH 4-7 (see Fig. 10). Though some dissolution of plagioclase exceeds or matches that of the merrillite and chlorapatite at these pH values, the presence of the secondary phase could potentially mask plagioclase dissolution from elemental-only analytical techniques such as APXS. Figure 10 includes a τ plot of Al content normalized to Ti that shows Al concentrations remain relatively stable despite the plagioclase dissolution. This is also the case for other elements including Si, indicating that even in profiles with significant plagioclase dissolution, APXS data might detect only dissolution of a Ca- and P-containing mineral. However, because of the very similar brushed concentrations of P measured by APXS at the surfaces of both Wishstone and Champagne (Wishstone = 2.64 wt% and Champagne = 2.63 wt% as P_2O_5 both of which represent ~ 50 % of the P_2O_5 measured at depth) (Table 8), in models that contained a single Ca-phosphate mineral, physical erosion fronts would have to precisely intercept the Ca-phosphate reaction fronts in both Wishstone and Champagne rocks at the same point, a scenario that seems unlikely (Fig. 10). In contrast, in models with multiple Ca-phosphate minerals, as would be expected from martian meteorite observations which commonly have a mixture of apatite and merrillite or possibly mixture of chlorapatite and fluorapatite (McSween and Treiman, 1998; McCubbin and Nekvasil, 2008), a range of physical erosion depths could yield similar concentrations measured by APXS if only one Ca-phosphate mineral were fully dissolved

from the rock at the brushed/eroded surface, but both minerals were present in the interior at the RATed depth 3mm+ deeper. This scenario is diagramed in Figure 11 with merrillite and chlorapatite and the accompanying τ plot. Chlorapatite + merrillite models run at initial pH values of 4 to 7 produce profiles that fit this scenario. However, the greatest separation between the two phosphate mineral fronts, and thus the most available leeway for variation in physical erosion depth between rocks while still possessing profiles consistent with APXS measurements, occurs at pH 6 (Appendix III Table S9). Thus, for chlorapatite + merrillite bearing models, profiles produced under pH 6 conditions are the most plausibly consistent with Wishstone as measured by APXS.

While reactive transport models such as those used in this study are numerical approximations that by necessity contain assumptions and simplification, and different assumptions or input might yield different results, the models presented here reasonably approximate dissolution in Wishstone-type rocks on Mars with the available data. The models of this study suggest the dissolution profiles in Wishstone class rocks are best explained if Wishstone class rocks contain more than one Ca-phosphate mineral, a likely scenario based on martian meteorites (McSween and Treiman, 1998; McCubbin and Nekvasil, 2008). Further, the modeled profiles here suggest that the profiles measured by APXS indicate alteration of Wishstone class rocks by mildly acidic to neutral waters, with pH 6 models allowing the best fit. These conditions are consistent with resent research that indicates mildly acidic to near neutral (pH 5-6) waters may have been episodically present in Gusev Crater at Columbia hills in the form of an ephemeral lake, based on observations and modeling applied to Comanche rocks in the Columbia hills (Ruff et al., 2014).

Though the pH conditions most favorable to life are not conclusively known, certain potentially prebiotic reactions and the stability of some bio-molecules are favored under more neutral rather than acidic conditions (Lindahl, 1993; Madigan et al., 2000; Knoll et al., 2005; Powner et al., 2009). The modeling results in this study suggest that more neutral aqueous conditions may have prevailed at some point in the past at Gusev Crater, rather than more acidic conditions such as wide-spread acid vapor alteration (Settle, 1979; Banin et al., 1997; Tosca et al., 2004; Golden et al., 2005; Hurowitz et al., 2006a; Yen et al., 2008; Hausrath et al., 2013). These results are in agreement with recent observation and modeling of other rocks in Gusev Crater (Ruff et al., 2014), with positive implications for past martian habitability.

Conclusions

Weathering profiles can be used as indicators of past aqueous conditions, preserving characteristics such as the duration of liquid water, pH, temperature, and other factors. In Wishstone class rocks on Mars, profiles measured by MER Spirit APXS suggest the loss of a Ca-phosphate mineral has occurred during past aqueous interactions. These profiles have previously been suggested as an indicator of chemical weathering by highly acidic fogs or possibly dilute acidic waters. In this study, models which contained multiple Ca-phosphate minerals, initial pH values of 4-7, and a component of physical erosion produced profiles similar to those apparent in Wishstone class rocks on Mars, with pH 6 models being the most consistent with the APSX-measured data. These modeling results suggest that past environmental conditions at Gusev Crater may have included aqueous dissolution into mildly acidic to neutral conditions.

Though the exact range of required conditions under which life can originate are unknown, certain potentially prebiotic reactions and the stability of some bio-molecules are favored under more neutral conditions (Lindahl, 1993; Madigan et al., 2000; Knoll et al., 2005; Powner et al., 2009). In addition, phosphate is a bioessential nutrient and phosphorus, either as phosphate or a more reduced species such as phosphite, is considered crucial in pre-biotic reactions that may have led to the origin of life on Earth. The results of this study, which indicate the possible dissolution of phosphate minerals by near neutral waters on Mars, therefore have positive implications for potentially habitable environments on Mars.

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Tables

Mineral	McSween et al. 2006 Wt. %	This study Wt%**	Mineral Volume Fraction	Mineral Volume Fraction with 5% porosity	
Feldspars*	57.6	59.3	0.679	0.646	
Albite	42.9				
Anorthite	10.4				
Orthoclase	3.4				
Illmenite	5.0	14.0	0.089	0.084	
Corundum	2.4				
Magnetite	7.6				
Hypersthene (OPX)	13.5	13.9	0.109	0.103	
Ca-phosphate mineral	12.5	12.9	0.123	0.117	
Olivine	1.9				
Porosity				0.05	
Totals	99.6	100	1	1	

Table 7. Mineralogy and volume fraction used in model.

*Total Feldspars (Feldspars were combined into a single mineral of representative composition)

**Minerals comprising <5 wt.% were not included in the model. Ilmenite and magnetite were combined into a single mineral "Ilmenite". Wt% values were then renormalized to 100% before mineral volume fractions were calculated.

	Champagne				Wishstone					
	Depth	Oxide wt. %		Elemental wt. %		Depth Oxide		e wt. %	Elemental wt. %	
	(mm)	CaO	P ₂ O ₅	Са	Р	(mm)	CaO	P ₂ O ₅	Са	Р
Surface	0.00	6.67	1.79	4.77	0.78	-	-	-	-	-
Brushed	0.00	6.59	2.64	4.71	1.15	0.00	6.86	2.63	4.90	1.15
RAT 1*	-	8.78	5.07	6.28	2.21	3.18	8.89	5.19	6.35	2.26
RAT 2	3.95	8.75	5.05	6.25	2.20	-	-	-	-	-
Loss		2.16	2.41	1.54	1.05		2.03	2.56	1.45	1.12

Table 8. APXS Ca and P analyses of Wishstone and Champagne on Mars.

*Champagne received two RAT operations and analyses (included in table). No depth has been reported for Champagne RAT 1. Only one RAT operation was performed on Wishstone.

Rock Type	Porosity (%)	Reference
Terrestrial Welded Tuff	5-14	Keller, 1960; Rejeki et al., 2005
Terrestrial Basalt	0.1-11.4	Davis, 1969; Freeze and Cherry, 1977; Sato et al., 1990; Rejeki et al., 2005
Craters of The Moon Basalts	0.1 -7.5	This study.
Lunar Basalt	2-11	Macke et al., 2012
Brecciated Basalt	10	Freeze and Cherry, 1977
Volcanic Breccia	4-19	Rejeki et al., 2005; Benning, 2008
Lunar Impact/Regolith Breccia	5-12	Macke et al., 2012

Table 9. Porosities of basalts, tuff and breccias relevant to Wishstone.

								Surface
Mineral	Reaction Stoichiometry	Log K (298 K)	Log k _{H+}	n _{H+}	Log _{OH-}	n _{on-}	Ea (kcal/mol)	Area (m ² g ⁻¹)
Plagioclase (An ₂₅)	$Ca_{0.25}Na_{0.75}AI_{1.25}Si_{2.75}O_8 + 5H^+ =$ 1.25 $AI^{3+} + 0.75Na^+ + 0.25Ca^{2+} + 2.75SiO_2 + 2.5H_2O$	8.17 ^ª	-9.22 ^f	0.408 ^f	-9.24 ^f	0.38 ^f	14.8425 ^d	7.66 x 10 ⁻³
Hypersthene	$FeMgSi_2O_6 + 4 H^+ = + Mg^{2+} + Fe^{2+} + 2SiO_2 + 2H_2O$	11.3269 ^b	-9.02 ^d	0.6 ^d	-12.72 ^d	0.00 ^d	19.11132 ^d	6.07 x 10 ⁻³
Ilmenite	$FeTiO_3 + 2H^+ + H_2O = Fe^{2+} + Ti(OH)_4$	0.9046 ^b	-8.35 ^d	0.421 ^d	-11.16 ^d	0.00 ^d	9.053989 ^d	9.44 x 10 ⁻³
Phosphate mineral								
Merrillite	$Ca_{9.5}Mg(PO_4)_7 + 7H^+ = 9.5Ca^{2+} + Mg^{2+} + 7HPO4^{2-}$	-39.1 ^c	-4.56 ^c	0.924 ^c				3.10 x 10 ⁻³
Chlorapatite	$Ca_{5}(PO_{4})_{3}CI + 3H^{+} = 5Ca^{2+} + 3HPO_{4}^{2-} + CI^{-}$	-21.3 ^c	-4.28 ^c	0.912 ^c			11 ^e	3.85 x 10 ⁻³
Fluorapatite	$Ca_5(PO_4)_3F + 3H^+ = 5Ca^{2+} + 3HPO_4^{2-} + F^-$	-26.5 ^c	-5.07 ^c	0.817 ^c			11 ^e	4.89 x 10 ⁻³
Brushite	$CaHPO_4 \cdot 2(H_2O) = Ca2 + HPO_4^{2} + 2H_2O$	6.55 ^b	-4.14	0.541 ^g			3.8 ^h	3.10 x 10 ⁻³
Secondary phase	$AI_2Si_2O_5(OH)_4 + 6H+ = 2AI^{3+} + 2SiO_2 + 5H_2O$	12.55 ⁱ	-13.2 ^j	0.55 ^j	-15.4 ^j	0.00 ^j	15.0	7.66 x 10 ⁻³

Table 10. Thermodynamic data, kinetic data, and specific surface areas used in modeling

^{*a*} Based on work by Maher et al. (2009) for An_{25} .

^b Lawrence Livermore National Laboratory thermochemical modeling database (Thermo.com.V8.R6.230) (Delany and Lundeen, 1990; Johnson et al., 2000).

^c Adcock et al., (2013).

^d Palandri and Kharaka (2004) from compiled mineral data by Schott and Berner (1985) and White et al. (1994).

^e Harouiya et al. (2007).

^f Plagioclase kinetics derived based on data compiled by Bandstra and Brantley (2008) from multiple sources of dissolution rates for albite and anorthite (i.e. Chou and Wollast, 1984; Knauss and Wolery, 1986; Holdren and Speyer, 1987; Casey et al., 1991; Rose, 1991; Amrhein and Suarez, 1992; Stillings and Brantley, 1995; Hodson, 2003) and data on the non-linear relationship between Ca and Na content and dissolution rates from Blum and Stillings (1995).

^g Estimated value based on the kaolinite Ksp from Maher et al. (2009) increased 4 orders of magnitude to approximate a precursor phases like halloysite (Steefel and Van Cappellen, 1990; Yang and Steefel, 2008)

^h Based on kaolinite dissolution kinetics of Huertas et al. (1999) for proton promoted and neutral range dissolution slowed 1 order of magnitude to account for slower precipitation kinetics based on Yang and Steefel (2008)



Figure 8. Modeled dissolution profiles (100 k.y. models at pH values of 3, 5 and 7) of a Wishstone-type rock containing chlorapatite as the Caphosphate mineral. Dashed horizontal lines represent estimated surface retreat due to chemical and physical weathering of the rocks based on 50% porosity. The overall trend of deeper dissolution of minerals with decreasing pH is consistent with the thermodynamics and kinetics of the minerals in the modeled rock. Dissolution profiles of the non-phosphate minerals at a given modeled pH were similar in all models regardless of the Ca-phosphate mineral used in the modeled rock.



Figure 9. Comparison of different modeled Ca-phosphate minerals weathering at pH 6 with 100 k.y. of model time. Dashed horizontal line represents surface retreat of the rocks based on 50% porosity Dissolution profiles for minerals other than the Ca-phosphate and depth of surface retreat do not significantly change between models, thus only those from the chlorapatite model are shown. Figure shows chlorapatite dissolution to the greatest depth and fluorapatite to the shallowest, with merrillite in between. This is consistent with thermodynamic and kinetic data for these minerals (Adcock et al., 2013)



Figure 10. Output dissolution profile for chlorapatite-bearing model at pH 5 for 100 k.y. A) 1. Surface retreat from the original rock surface based on 50% developed porosity. Note that the surface retreat overtakes the dissolution front of ilmenite, effectively removing it from the rock dissolution profile. 2. A component of additional erosion lowers the rock surface even more and generates a profile consistent with APXS brushed surface data for both Wishstone and Champagne, however, the depth of the physical erosion must intercept the Ca-phosphate dissolution front in exactly the same place in both rocks to remain consistent with the APXS data. 3. At an additional depth (3 mm+) reached after use of the RAT, another APXS analysis is taken in relatively unweathered rock. B) τ plot of Al normalized to Ti concentration where 0 = parent concentration for Al and -1 = complete depletion. Plot shows that despite plagioclase dissolution Al remains relatively immobile. This is due to the formation of the Al-bearing secondary phase.



Figure 11. Two mineral dissolution model scenario from a chlorapatite/merrillite bearing model at pH 6 for 100 k.y. A) 1. Surface retreat from the original rock surface based on 50% developed porosity. Note that the surface retreat overtakes the dissolution front of ilmenite, effectively removing it from the rock dissolution profile. 2) A component of additional erosion lowers the rock surface even more (shaded area) overtaking the merrillite dissolution front but leaving the chlorapatite front, thus an APXS brushed surface analysis would indicate some but not all Ca-phosphate mineral loss. 3) At an additional depth (3 mm+) reached after use of the RAT, another APXS analysis is taken in relatively unweathered rock which would show the full phosphorus content of the rock. B) τ plot of Al normalized to Ti concentration where 0 = parent concentration for Al and -1 = complete depletion. Plot shows that despite plagioclase dissolution Al remains relatively immobile. This is due to the formation of the Al-bearing secondary phase.

APPENDIX I

SUPPLEMENTARY INFORMATION FOR CHAPTER TWO

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Supplementary Discussion

Effects of parameter variation during whitlockite synthesis.

In order to synthesize Mg-whitlockite, the Hughes et al. (2008) experiment was replicated, but Mg²⁺ was intentionally supplied to the system in the form of magnesium nitrate hexahydrate (Mg(NO₃)₂·6H₂O) (J. T. Baker, ACS grade). To simplify the method, 40 mesh reagent grade calcium hydroxyapatite (HAP) powder (Spectrum, reagent grade) was used rather than pre-synthesizing this phase. The ratio of HAP to magnesium nitrate hexahydrate by mass was generally 3.33: 1 to provide sufficient Mg²⁺ based on whitlockite stoichiometry, but not excess Mg^{2+} that might lead to the formation of additional Mg phases. Variables during the preliminary experiments included incubation time at 240 °C (5-14 days), adjusted pH (2.1 to 2.8), and the total concentration of solids used in the experiments (6.7 to 90 g/L). Early experiments were run in 23 ml Teflon lined acid digestion vessels (Parr 4749) and later scaled up to 125 ml vessels (Parr 4748). Preliminary experiments were carried out with the goal of first producing Mg-whitlockite crystals, and then refining the synthesis method for phase purity, larger crystal size, and sufficient output mass to carry out crystallographic and chemical characterization as well as supply Mg-whitlockite for use in the synthesis of Mg-merrillite.

Insight gained during synthesis of Mg-whitlockite was used to develop a procedure for Fe-whitlockite synthesis. Initially, $FeCl_2$ was used as a source of Fe^{2+} in

place of the Mg(NO₃)₂·6H₂O used in Mg-whitlockite synthesis. After experiments using FeCl₂ failed to produce significant whitlockite, Fe(II)S was utilized as the Fe²⁺ source and whitlockite crystals were produced along with hydroxyapatite and opaque phases (presumably Fe phases). Variables during the preliminary experiments included incubation time at 240 °C (5-9 days), and total concentration of solids (9.0 to 17.8 g/L). In addition, the production of iron sulfide phases prompted varying the ratio of HAP to Fe(II)S from 9.6:1 (near stoichiometric ideal) to 28.6:1 by mass to limit their formation. The pH value was not significantly varied from 2.8 as results from Mg-whitlockite synthesis indicated it had no effect.

Although endmember whitlockite and merrillite are useful for different studies, natural whitlockite and merrillite generally contain both Mg²⁺ and Fe²⁺. We therefore applied the results of endmember synthesis experiments to synthesize non-endmember whitlockites and from them mixed Fe/Mg-merrillite. Variables included the ratio of Mg and Fe, as well as the concentration of total solids in solution (10.2 to 17.8 g/L). Sodium (as NaCl) was also included in some experiments. Though sodium is not typical within terrestrial whitlockite, it is present in most natural merrillite, particularly martian merrillite, in what is a partially filled calcium site in synthetic merrillite (Jolliff et al., 2006). The methods documented in the accompanying paper are those which successfully produced the material characterized and used for mixed Fe/Mg-merrillite synthesis in this study and include sodium (as NaCl), however, sodium was not incorporated into the whitlockite structure in any of multiple attempts. Additional experiments showed sodium to neither inhibit nor aid the production of mixed Mg/Fe-

whitlockite. Appendix I Table S1 contains parameter values for all whitlockite synthesis experiments in this study, including those used to make mixed Fe/Mg-whitlockite and subsequent Fe/Mg-merrillite.
Tuble 51.	Voscol	MgNQ.,6H.Q	EaCl.	Fos	NaCl	нопто.	Mass	וח		nH		nH	Mass	
ID	Size	(g)	(g)	(g)	(g)	(g)	Tot (g)	(mi.)	σ/I *	start	Davs	end	Out (g)	Note
 Mg-1	23 ml	0.0153	(8/	(8/	(8/	0.0518	0.0671	10	67	2 70	5	-	-	Whit
Mg-2	23 ml	0.2210				0.7716	0.9926	11	90.2	2.78	5	-	-	Monetite
Mg-3	23 ml	0.0159				0.0519	0.0678	10	6.8	2.66	10	2.80	0.016	Whit.
Mg-4	23 ml	0.0461				0.1541	0.2002	10	20.0	2.69	11	2.80	0.113	Whit.
Mg-5	23 ml	0.1093				0.3630	0.4723	13.5	35.0	2.81	14	2.16	0.189	Monetite
Mg-6	23 ml	0.2190				0.7280	0.9470	13.5	70.1	2.80	14	1.92	0.607	Monetite
Mg-7	125 ml	0.4599				1.5432	2.0031	90	22.3	2.80	7	2.09	1.150	Monetite
Mg-8	125 ml	0.4588				1.5423	2.0011	90	22.2	2.78	6	1.85	1.320	Monetite
Mg-9	125 ml	0.4595				1.5452	2.0047	90	22.3	2.80	5	1.86	1.370	Monetite
Mg-10	125 ml	0.4601				1.5426	2.0027	90	22.3	2.77	5	1.85	1.300	Monetite
Mg-11	125 ml	0.4595				1.5423	2.0018	90	22.2	2.80	5	1.90	1.220	Monetite
Mg-12	125 ml	0.2298				0.7709	1.0007	90	11.1	2.70	7	2.10	0.240	Whit.
Mg-13	125 ml	0.3890				1.2813	1.6703	90	18.6	2.79	5	2.01	0.834	Crypto-xtal mix
Mg-14	23 ml	0.0158				0.0512	0.0670	10	6.7	2.10	7	1.85	0.030	Whit.
Mg-15	23 ml	0.2006				0.0000	0.2006	10	20.1	2.70	5	2.12	0.125	Whit.
Mg-16	125 ml	0.3004				1.0014	1.3018	90	14.5	2.77	7	1.70	1.131	Whit.
Mg-17	125 ml	0.3000				1.0000	1.3000	90	14.4	2.76	7	1.84	0.899	Whit.
Mg-18	125 ml	0.3000				1.0000	1.3000	90	14.4	2.74	7	-	-	Whit.
Mg-19	125 ml	0.3000				1.0000	1.3000	90	14.4	2.78	7	-	-	Whit.
Mg-22	125 ml	0.3004				1.0000	1.3004	90	14.4	2.72	7	1.78	0.800	Whit.
Mg-23	125 ml	0.3000				1.0070	1.3070	90	14.5	2.79	7	2.10	0.555	Whit.
Mg-24	125 ml	0.3000				1.0000	1.3000	80	16.3	2.74	7	2.11	0.588	Leaked/Failed
Mg-25	125 ml	0.3004				1.0007	1.3011	90	14.5	2.73	7	1.97	0.510	Whit.
Mg-26	125 ml	0.3005				1.0011	1.3016	90	14.5	2.70	7	1.98	0.480	Whit.
Mg-27	125 ml	0.3003				1.0010	1.3013	90	14.5	2.79	7	1.81	0.720	Whit.
Mg-28	125 ml	0.3000				1.0000	1.3000	90	14.4	2.78	7	1.88	0.699	Whit.
Mg-29	125 ml	0.3000				1.0000	1.3000	90	14.4	2.65	7	1.74	0.702	Whit.
Mg-31	125 ml	0.3026				1.0004	1.3030	90	14.5	2.65	12	1.76	0.810	Whit.
Mg-32	125 ml	0.3016				1.0006	1.3022	90	14.5	2.69	12	1.77	0.710	Whit.
Mg-33	125 ml	0.2997				1.0012	1.3009	90	14.5	2.71	12	1.72	0.730	Whit.
Mg-34	125 ml	0.2994				0.9996	1.299	90	14.4	2.74	6	1.78	0.770	Whit.
Mg-35	125 ml	0.3013				0.9998	1.3011	90	14.5	2.80	6	1.75	0.720	Whit.
Mg-36	125 ml	0.2997				1.0009	1.3006	90	14.5	2.75	6	1.74	0.810	Whit.
Mg-37	125 ml	0.3004				1.0005	1.3009	90	14.5	2.68	7	1.89	0.730	Whit.
Mg-38	125 ml	0.3002				1.0001	1.3003	90	14.4	2.72	7	1.82	0.700	Whit.
Mg-39	125 ml	0.3003				1.0001	1.3004	90	14.4	2.40	7	1.78	0.760	Whit.
Mg-40	125 ml	0.3004				1.0004	1.3008	90	14.5	2.74	8	1.89	0.665	Whit.
Mg-41	125 ml	0.2994				0.9999	1.2993	90	14.4	2.78	8	1.82	0.706	Whit.
Mg-42	125 ml	0.3008				1.0002	1.301	90	14.5	2.75	8	1.78	0.825	Whit.
Mg-44	125 ml	0.3002				1.0012	1.3014	90	14.5	2.73	7	-	0.632	Whit.

Table S1. Batch parameters for whitlockite synthesis experiments.

Supplementary Tables

	Vessel	MgNO ₂ ·6H ₂ O	FeCl	FeS	NaCl	HAP	Mass	DI		рH		рН	Mass	
ID	Size	(g)	(g)	(g)	(g)	(g)	Tot (g)	(mL)	g/L*	start	Davs	end	Out (g)	Note
Mg-45	125 ml	0.3003	(0)	(0)	(0)	1.0008	1.3011	90	14.5	2.76	7	1.83	0.676	Whit.
Mg-46	125 ml	0.3005				1.0005	1.3010	90	14.5	2.74	7	1.77	0.775	Whit.
Mg-47	125 ml	0.2998				1.0003	1.3001	90	14.4	2.74	7	1.77	0.723	Whit.
Mg-48	125 ml	0.2998				1.0005	1.3003	90	14.4	2.76	7	1.76	0.726	Whit.
Mg-49	125 ml	0.2990				1.0000	1.2990	90	14.4	2.75	7	1.69	0.813	Whit.
Mg-50	125 ml	0.3002				1.0001	1.3003	90	14.4	2.66	8	1.80	0.627	Whit.
Mg-51	125 ml	0.3003				1.0001	1.3004	90	14.4	2.73	8	1.73	0.734	Whit.
Mg-52	125 ml	0.3001				1.0000	1.3001	90	14.4	2.75	8	1.68	0.776	Whit.
Mg-53	125 ml	0.3008				1.0000	1.3008	90	14.5	2.66	6	1.84	0.594	Whit.
Mg-54	125 ml	0.3009				1.0003	1.3012	90	14.5	2.79	6	1.75	0.690	Whit.
Mg-55	125 ml	0.3412				1.0006	1.3418	90	14.9	2.76	6	1.74	0.715	Whit.
Mg-56	125 ml	0.3003				0.9994	1.2997	90	14.4	2.69	7	1.85	0.702	Whit.
Mg-57	125 ml	0.3002				0.9995	1.2997	90	14.4	2.76	7	1.74	0.715	Whit.
Mg-58	125 ml	0.2999				1.0003	1.3002	90	14.4	2.74	7	1.70	0.734	Whit.
Mg-59	125 ml	0.3002				0.9998	1.3000	90	14.4	2.77	7	1.83	0.460	Whit.
Mg-60	125 ml	0.2998				0.9998	1.2996	90	14.4	2.74	7	1.85	0.437	Whit.
Mg-61	125 ml	0.3003				1.0012	1.3015	90	14.5	2.74	7	1.83	0.566	Whit.
Fe-1	23 ml		0.0287			0.2307	0.2594	16	16.2	2.68	9	2.26	0.148	No Whit.
Fe-2	23 ml		0.0290			0.2311	0.2601	16	16.3	2.77	9	2.20	0.143	No Whit.
Fe-3	23 ml	0.0356	0.0173			0.2312	0.2841	16	17.8	2.74	8	2.11	0.114	No Whit.
Fe-4	23 ml	0.0357		0.0129		0.2311	0.2797	16	17.5	2.77	8	2.00	0.173	minor Whit.
Fe-5	23 ml			0.0249		0.2334	0.2583	16	16.1	2.75	8	1.80	-	aggregates Whit.
Fe-6	23 ml			0.0183		0.1739	0.1922	16	12.0	2.74	8	1.80	-	Cleaner Whit
Fe-7	125 ml			0.0763		0.7300	0.8063	90	9.0	2.76	7	2.06	0.427	Fe-Whit.
Fe-8	125 ml			0.0769		1.0012	1.0781	90	12.0	2.77	7	2.10	0.520	Fe-Whit.
Fe-9	125 ml			0.0600		0.9994	1.0594	90	11.8	2.78	7	2.05	0.630	Fe-Whit.
Fe-10	125 ml			0.0500		1.0001	1.0501	90	11.7	2.77	7	2.17	0.520	Fe-whit+agg.
Fe-11	125 ml			0.0347		1.0009	1.0356	90	11.5	2.78	7	1.87	0.500	
Fe-12	125 ml			0.0392		1.0000	1.0392	90	11.5	2.69	7	1.96	0.471	Lg xtals Fe-whit.
Fe-13	125 ml			0.0450		1.0000	1.0450	90	11.6	2.74	7	1.87	0.785	Lg xtals Fe-whit.
Fe-14	125 ml			0.0767		1.0008	1.0775	90	12.0	2.75	7	1.90	0.731	Fe-whit+agg.
Fe-15	125 ml			0.1003		1.0004	1.1007	90	12.2	2.72	7	1.92	0.846	Fe-Whit.
Fe-16	125 ml			0.0349		0.9999	1.0348	90	11.5	2.69	7	1.89	0.609	Fe-Whit.
Fe-17	125 ml		0.0800			1.0003	1.0803	90	12.0	2.75	7	1.94	0.451	No-Whit.
Fe-18	125 ml			0.0849		1.1003	1.1852	90	13.2	2.75	7	1.89	0.726	Fe-Whit.
Fe-19	125 ml			0.1101		1.2004	1.3105	90	14.6	2.68	7	1.97	0.845	Fe-whit+agg.
Fe-20	125 ml			0.0549		1.0004	1.0553	90	11.7	2.76	7	1.95	0.419	Fe-Whit.
Fe-21	125 ml			0.0558		1.0001	1.0559	90	11.7	2.69	7	1.92	0.452	Fe-Whit.
Fe-22	125 ml			0.0544		1.0004	1.0548	90	11.7	2.71	7	1.96	0.444	Fe-Whit.

Table S1 continued. Batch parameters for whitlockite synthesis experiments.

	Vessel	MgNO ₃ ·6H₂O	FeCl ₂	FeS	NaCl	HAP	Mass	DI		рН		рН	Mass	
ID	Size	(g)	(g)	(g)	(g)	(g)	Tot (g)	(mL)	g/L*	start	Days	end	Out (g)	Note
Mix-1	125ml	0.0500		0.0500	0.0667	1.0000	1.1667	90	13.0	2.64	7	2.05	0.761	Whit.
Mix-2	125ml	0.1000		0.0500		1.0000	1.1500	90	12.8	2.73	7	2.15	0.694	Whit.
Mix-3	125ml	0.1000			0.1000	1.0000	1.2000	90	13.3	2.78	7	-	-	No Whit.
Mix-4	125ml	0.1000		0.0495		0.9997	1.1492	90	12.8	2.78	7	2.2	0.688	No Whit. (HAP)
Mix-5	125ml	0.1103		0.0403		1.0000	1.1506	90	12.8	2.70	7	2.01	0.821	No Whit. (HAP)
Mix-6	125ml	0.1207		0.0305		1.0005	1.1517	90	12.8	2.78	7	2.27	0.591	Rare Whit.
Mix-7	125 ml	0.0505		0.0499	0.0668	1.0000	1.1672	90	13.0	2.75	7	1.8	0.811	Whit.
Mix-8	125 ml	0.0505		0.0499		1.0005	1.1009	90	12.2	2.74	7	1.84	0.915	Whit.
Mix-9	125 ml	0.1505		0.0500		1.2000	1.4005	90	15.6	2.78	7	2.1	0.974	No Whit.
Mix-10	125 ml	0.1000		0.0500		1.0000	1.1500	90	12.8	2.72	7	2.05	0.700	Rare Whit.
Mix-11	125 ml	0.0800		0.0400		0.8000	0.9200	90	10.2	2.78	7	2.04	0.550	Rare Whit.
Mix-12	125 ml	0.0900		0.0300		0.8000	0.9200	90	10.2	2.78	7	2.02	-	Minor Whit.
Mix-13	125 ml	0.0500		0.0501		1.0002	1.1003	90	12.2	2.75	7	1.89	0.799	Whit. (lg)
Mix-14	125 ml	0.0399		0.0601		0.9999	1.0999	90	12.2	2.78	7	-	0.765	Whit.
Mix-15	125 ml	0.0600		0.0400		1.0000	1.1000	90	12.2	2.77	7	2.04	0.707	Whit.
Mix-16	125 ml	0.0501		0.0500	0.0601	1.0001	1.1603	90	12.9	2.77	7	1.99	0.785	Whit. well fm.
Mix-17	125 ml	0.0251		0.0501		0.9000	0.9752	90	10.8	2.76	7	-	0.556	Whit.
Mix-18	125 ml	0.0300		0.0700		1.0000	1.1000	90	12.2	2.77	7	2.01	0.644	Whit.
Mix-19	125 ml	0.0351		0.0501		1.0001	1.0853	90	12.1	2.77	7	1.98	0.689	Whit.
Mix-20	125 ml	0.0350		0.0502		1.0000	1.0852	90	12.1	2.78	7	1.96	0.674	A.W./Whit.
Mix-21	125 ml	0.0901		0.0302		0.9002	1.0205	90	11.3	2.78	7	2.2	0.470	Rare Whit.

Table S1 continued. Batch parameters for whitlockite synthesis experiments.

Vessel Size = the volume of the Teflon lines Parr reaction vessel used. All chemistry weights are in grams. Mass Tot = Total mass of solids in grams used in the experiment. DI = volume of 18 M Ω water used in experiment. g/L = ratio of total mass of solids (g) in solution (L). pH Start = the adjusted pH at the start of the experiment. Days = number of days of incubation. pH end = measured pH at end of experiment. Mass Out = total mass (g) of material recovered after experiment including impurities. Abbreviations in notes: Whit.=whitlockite, agg. = polycrystalline aggregates, lg = large, HAP = hydroxyapatite, well fm. = well formed crystals. Crypto-xtal mix = mix of microcrystalline phases. A.W. = reaction vessel underwent additional cleaning including additional acid washing to investigate any affect on nucleation.

		Mg/Fe-Whitlo	ckite Synthes	sis Experiment ID			
	Mix-6		Mix-21		Mix-1		
CaO	48.40	(0.26)	47.73	(1.31)	46.54	(0.59)	
P ₂ O ₅	45.83	(0.28)	45.59	(0.92)	45.83	(0.36)	
MgO	3.43	(0.32)	2.63	(0.05)	1	(0.20)	
FeO	0.18	(0.24)	1.92	(2.51)	4.62	(0.71)	
Na₂O	B.D.	-	B.D.	-	B.D.	-	
H₂O ^ª	0.86	-	0.86	-	0.86	-	
Total	98.69	(0.42)	98.73	(0.79)	98.85	(0.49)	
n		11	7		18		
Stoic.	Ca _{9.2} Fe _{0.0} Mg _{0.9} (PO ₃ OH)(PO ₄) _{5.9}		Ca _{9.1} Fe _{0.3} Mg _{0.7} (PO ₃ OH)(PO ₄) _{5.9}		Ca _{9.0} Fe _{0.7} Mg _{0.3} (PO ₃ OH)(PO		
Fe:Mg Soln. ^b	0.	74	0	.98	2.91		
Fe:Mg Prod. ^c	0.00		0	.43	2.33		

Table S2. Microprobe analyses of Mg/Fe-whitlockite synthesized with varied Fe: Mg ratios.

Parenthetical values are 1 standard deviation. n = number of analyses averaged chemistry is based on. B.D. = below detection. ^{*a*} H_2O is based on ideal whitlockite (Hughes et al., 2008) and is included in these EMP totals. ^{*b*}The molar ratio of Fe: Mg added to the solution of the experiment.

^cThe molar ratio of Fe: Mg in the stoichiometry of the resulting whitlockite.

					Ferric	Ferrous
			Ferric Fe-	Ferrous Fe-	Fe/Mg-	Fe/Mg-
	Fe-	Fe/Mg-	merrillite	merrillite	merrillite	merrillite
Atom	whitlockite	whitlockite	(Pt cruc.)	(SiO ₂ tube)	(Pt cruc.)	(SiO ₂ tube)
Fe(1)	Fe1 00	Feacar	Fe1 00	Fe1 00	Feo 754	Feo 646
Y	0.6667	0 6667	0 6667	0.6667	0 6667	0 6667
X	0.3333	0.3333	0.3333	0.3333	0.3333	0.0007
y Z	0.3333	0.3333	0.3333	0.3333	0.3333	0.3333
2	0.1137(1)	0.1130(1)	0.1131(1)	0.1142(1)	0.1149(1)	0.1130(1)
O(eq)	0.008(1)	0.003(1)	0.003(1)	0.000(1)	0.002(1)	0.004(1)
ivig(1)		NIB0.363			NIB0.246	NIB0.354
X	NA	0.0007	NA	NA	0.0007	0.0007
У	NA	0.3333	NA	NA	0.3333	0.3333
Z	NA	0.1138(1)	NA	NA	0.1149(1)	0.1150(1)
U(eq)	NA	0.002	NA	NA	0.002	0.004
Ca(1)	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}
x	0.7191(1)	0.7192(1)	0.7224(1)	0.7245(1)	0.7226(1)	0.7239(1)
У	0.8534(1)	0.8534(1)	0.8547(1)	0.8571(1)	0.8548(1)	0.8566(1)
Ζ	0.2150(1)	0.2150(1)	0.2140(1)	0.2133(1)	0.2140(1)	0.2144(1)
U(eq)	0.005(1)	0.004(1)	0.004(1)	0.005(1)	0.004(1)	0.005(1)
Ca(2)	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}
x	0.5125(1)	0.5122(1)	0.5081(1)	0.5099(1)	0.5088(1)	0.5103(1)
У	0.0542(1)	0.0537(1)	0.0442(1)	0.0502(1)	0.0451(1)	0.0509(1)
Z	0.1792(1)	0.1793(1)	0.1798(1)	0.1785(1)	0.1796(1)	0.1794(1)
U(eq)	0.005(1)	0.005(1)	0.004(1)	0.005(1)	0.004(1)	0.006(1)
Ca(3)	Ca _{1.00}	Ca _{1 00}	Ca _{1.00}	Ca _{1.00}	Ca _{1.00}	Ca _{1 00}
x	1.1580(1)	1.1573(1)	1.1483(1)	1.1495(1)	1.1489(1)	1.1506(1)
V	0.8643(1)	0.8646(1)	0.8714(1)	0.8731(1)	0.8709(1)	0.8717(1)
z	0.1080(1)	0.1080(1)	0.1072(1)	0.1064(1)	0.1071(1)	0.1074(1)
U(ea)	0.012(1)	0.011(1)	0.006(1)	0.015(1)	0.007(1)	0.016(1)
P(2)	P _{1.00}	P _{1.00}	P1 00	P _{1.00}	P _{1.00}	P _{1.00}
x	0.4761(1)	0.4761(1)	0.4745(1)	0.4713(1)	0.4746(1)	0.4725(1)
V	-0.0162(1)	-0.0160(1)	-0.0175(1)	-0.0201(1)	-0.0174(1)	-0.0193(1)
7	0.0827(1)	0.0827(1)	0.0841(1)	0.0820(1)	0.0837(1)	0.0830(1)
- U(ea)	0.005(1)	0.002(1)	0.003(1)	0.005(1)	0.003(1)	0.005(1)
P(1)	P	P	P	P	P	P
· (±)	0.8606(1)	0.8601(1)	0.8562(1)	0.8602(1)	0.8564(1)	0.8601(1)
N N	0.6848(1)	0.6841(1)	0.0302(1) 0.6761(1)	0.6794(1)	0.6364(1)	0.6798(1)
y Z	0.0040(1) 0.1464(1)	0.0041(1) 0.1464(1)	0.0701(1) 0.1/59(1)	0.0754(1) 0.1455(1)	0.0707(1) 0.1/58(1)	0.0750(1) 0.1464(1)
2	0.1404(1)	0.1404(1)	0.1433(1)	0.1433(1)	0.1430(1)	0.1404(1)
	0.004(1) D	0.004(1) D	0.003(1)	0.004(1) D	0.003(1) D	0.003(1) D
P(SA)	г _{0.890}	г _{0.840} 0 2222	F _{0.029}	г _{0.239}	F _{0.065}	Г _{0.309}
x	0.5555	0.5555	0.5555	0.5555	0.5555	0.5555
y -	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333
Z	0.1934(1)	0.1934(1)	0.21/8(1)	0.2138(1)	0.2170(1)	0.2146(1)
U(eq)	0.005(1)	0.005(1)	0.004(1)	0.005(1)	0.005(1)	0.006(1)
P(3B)	P _{0.110}	P _{0.160}	P _{0.971}	P _{0.761}	P _{0.935}	P _{0.691}
x	0.3333	0.3333	0.3333	0.3333	0.3333	0.3333
У	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333
Z	0.2160(2)	0.2165(1)	0.1952(13)	0.1932(2)	0.1933(5)	0.1936(1)
U(eq)	0.005	0.005	0.004	0.005	0.005	0.004(1)
O(10A)	O _{0.890}	O _{0.840}	O _{0.029}	O _{0.239}	O _{0.065}	O _{0.309}
x	0.3333	0.3333	0.3333	0.3333	0.3333	0.3333
У	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333

Table SJ. Alumic Darameters for synthesized windockite and merrin	Table S	3. Atomic	parameters for	synthesized	whitlockite a	and merrillit
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			Ferric Fe-	Ferrous Fe-	Ferric Fe/Mg-	Ferrous Fe/Mg-
	Fe-	Fe/Mg-	merrillite	merrillite	merrillite	merrillite
			0 2594(1)	$(SIO_2 tube)$	0 2586(1)	$(SIO_2 tube)$
2 U(ea)	0.013(1)	0.012(1)	0.006(1)	0.009(1)	0.007(1)	0.015(3)
O(10B)	O _{0.110}	O _{0.160}	O _{0.970}	O _{0.761}	O _{0.935}	O _{0.691}
x	0.3333	0.3333	0.3333	0.3333	0.3333	0.3333
У	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333	-0.3333
Ζ	0.2568(8)	0.2579(3)	0.1549(15)	0.1520(2)	0.1511(8)	0.2560(2)
U(eq)	0.013	0.004(2)	0.007	0.009	0.007	0.011(1)
O(6)	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}
x	0.5657(2)	0.5658(1)	0.5682(2)	0.5602(3)	0.5677(2)	0.5620(2)
y _	-0.0877(2)	-0.0872	-0.0826(2)	-0.0930(3)	-0.0831(2)	-909(2)
Ζ	0.0696(1)	0.0695(1)	0.0696(1)	0.0684(1)	0.0691(1)	0.0694(1)
O(eq)	0.008(1)	0.007(1)	0.006(1)	0.016(1)	0.006(1)	0.013(1)
U(9)	0 _{1.00} 0 3533(2)	0 _{1.00} 03526(1)	$0_{1.00}$	0 _{1.00} 0 3420(3)	0 3436(2)	0 3431(3)
v	-0.1832(2)	-0.1834(1)	-0.1866(2)	-0.1882(3)	-0.0187(2)	-0.1876(2)
y Z	0.2056(1)	0.2054(1)	0.2050(1)	0.2018(1)	0.2044(1)	0.2030(1)
L U(eq)	0.009(1)	0.008(1)	0.007(1)	0.016(1)	0.009(1)	0.017(1)
O(8)	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}
x	0.3344(2)	0.3345(1)	0.3365(2)	0.3317(3)	0.3356(2)	0.3325(2)
у	-0.0606(2)	-0.0606(1)	-0.0552(2)	-0.0575(3)	-0.0559(2)	-0.0579(2)
Ζ	0.0604(1)	0.0604(1)	0.0606(1)	0.0594(1)	0.0605(1)	0.0604(1)
U(eq)	0.006(1)	0.005(1)	0.005(1)	0.006(1)	0.005(1)	0.006(1)
O(5)	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}
x	0.5754(2)	0.5756(1)	0.5763(2)	0.5746(3)	0.5762(2)	0.5750(2)
У	0.1551(2)	0.1556(1)	0.1589(2)	0.1516(3)	0.1582(2)	0.1529(2)
Z	0.0774(1)	0.0778(1)	0.0808(1)	0.0781(1)	0.0803(1)	0.0792(1)
U(eq)	0.007(1)	0.007(1)	0.005(1)	0.009(1)	0.004(1)	0.010(1)
0(2)	$O_{1.00}$	$O_{1.00}$	$O_{1.00}$	$O_{1.00}$	0.8612(2)	$O_{1.00}$
X	0.8747(2)	0.8737(1)	0.8599(2)	0.8000(3)	0.8013(2)	0.8070(2)
y z	0.7561(2) 0.1068(1)	0.7500(1) 0.1068(1)	0.7100(2) 0.1058(1)	0.7100(3) 0.1054(1)	0.7119(3) 0.1057(1)	0.7198(5)
z Ll(ea)	0.1008(1)	0.1008(1)	0.1058(1)	0.1034(1) 0.010(1)	0.1037(1)	0.1003(1)
O(3)	0.005(1)	0.005(1)	0.000(1)	0.010(1)	0.007(1)	0.013(1)
x	1.0193(2)	1.0188(1)	1.0161(2)	1.0204(2)	1.0161(2)	1.0203(2)
V	0.7337(2)	0.7330(1)	0.7289(2)	0.7308(3)	0.7291(2)	0.7311(2)
Z	0.1598(1)	0.1597(1)	0.1589(1)	0.1580(1)	0.1590(1)	0.1590(1)
U(eq)	0.007(1)	0.006(1)	0.006(1)	0.007(1)	0.006(1)	0.007(1)
O(1)	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}
x	0.7795(2)	0.7792(1)	0.7803(2)	0.7857(3)	0.7800(2)	0.7846(2)
У	0.7504(2)	0.7499(1)	0.7482(2)	0.7538(3)	0.7486(2)	0.7528(2)
Ζ	0.1674(1)	0.1672(1)	0.1665(1)	0.1657(1)	0.1663(1)	0.1667(1)
U(eq)	0.008(1)	0.007(1)	0.006(1)	0.009(1)	0.007(1)	0.009(1)
O(4)	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}	O _{1.00}
x	0.7675(2)	0.7671(1)	0.7619(2)	0.7656(3)	0.7623(2)	0.7655(2)
y -	0.5128(2)	0.5122(1)	0.5034(2)	0.5096(3)	0.5041(2)	0.5093(2)
Z	0.1506(1)	0.1506(1)	0.1519(1)	0.1523(1)	0.1516(1)	0.1524(1)
O(eq)	0.007(1)	0.007(1)	0.005(1)	0.007(1)	0.005(1)	0.008(1)
O(r)	$U_{1.00}$	U _{1.00}	$U_{1.00}$	U _{1.00}	$O_{1.00}$	U _{1.00}

					Ferric	Ferrous
			Ferric Fe-	Ferrous Fe-	Fe/Mg-	Fe/Mg-
	Fe-	Fe/Mg-	merrillite	merrillite	merrillite	merrillite
Atom	whitlockite	whitlockite	(Pt cruc.)	(SiO₂ tube)	(Pt cruc.)	(SiO ₂ tube)
x	0.4261(2)	0.4260(1)	0.4214(2)	0.4176(3)	0.4222(3)	0.4189(2)
У	-0.0543(2)	-0.0550(1)	-0.0610(2)	-0.0657(3)	-0.0616(2)	-0.0650(3)
Z	0.1219(1)	0.1219(1)	0.1227(1)	0.1208(1)	0.1224(1)	0.1218(1)
U(eq)	0.009(1)	0.008(1)	0.006(1)	0.014(1)	0.008(1)	0.0015(1)
Ca(4A)				Ca _{0.148}		Ca _{0.098}
X	NA	NA	NA	0.3333	NA	0.3333
У	NA	NA	NA	-0.3333	NA	-0.3333
Ζ	NA	NA	NA	0.1360(1)	NA	0.1041(3)
U(eq)	NA	NA	NA	0.011(1)	NA	0.019(3)
Ca(4B)				Ca _{0.296}		Ca _{0.264}
x	NA	NA	NA	0.3333	NA	0.3333
У	NA	NA	NA	-0.3333	NA	-0.3333
Z	NA	NA	NA	0.1028(3)	NA	0.1369(1)
U(eq)	NA	NA	NA	0.013(3)	NA	0.007(1)

Parenthetic values are standard deviations. Pt cruc. = merrillite produced by heating in open air in a platinum crucible. SiO_2 tube = merrillite produced in a triple argon purged and sealed silica glass tube.

					Ferrous		
					Fe-	Ferric	Ferrous
				Ferric Fe-	merrillite	Fe/Mg-	Fe/Mg-
		Fe-	Fe/Mg-	merrillite	(SiO ₂	merrillite	merrillite
		whitlockite	whitlockite	(Pt cruc.)	tube)	(Pt cruc.)	(SiO ₂ tube)
Ca(1)-	O(1)	2.3062(14)	2.3115(11)	2.308(2)	2.313(3)	2.310(2)	2.3128(19)
.,	O(2)#6	2.4348(17)	2.4268(13)	2.340(2)	2.386(3)	2.345(2)	2.385(2)
	O(3)#8	2.4482(15)	2.4548(11)	2.468(2)	2.456(2)	2.467(2)	2.4568(19)
	O(5)#7	2.4401(14)	2.4445(11)	2.511(2)	2.449(3)	2.506(2)	2.449(2)
	O(5)#9	2.4724(14)	2.4763(11)	2.495(2)	2.487(3)	2.484(2)	2.485(2)
	O(6)#7	2.7753(15)	2.7703(11)	2.740(2)	2.849(3)	2.741(2)	2.823(2)
	O(8)#9	2.5050(14)	2.5065(11)	2.470(2)	2.469(2)	2.474(2)	2.4742(19)
	O(9)#2	2.4570(14)	2.4530(11)	2.409(2)	2.437(3)	2.409(2)	2.437(2)
Ca(2)-	O(1)#14	2.6522(15)	2.6481(11)	2.638(2)	2.627(3)	2.631(2)	2.630(2)
	O(10B)#15	2.717(7)	2.699(3)	2.5360(11)	2.544(2)	2.578(2)	2.542(2)
	O(2)	2.5441(15)	2.5465(11)	2.586(2)	2.475(3)	2.492(2)	2.469(2)
	O(2)#14	2.4649(14)	2.4675(11)	2.492(2)	2.377(2)	2.389(2)	2.3720(18)
	O(3)	2.3727(14)	2.3736(11)	2.388(2)	2.564(2)	2.5518(12)	2.5888(17)
	O(6)#1	2.3934(15)	2.4009(12)	2.458(2)	2.484(3)	2.463(2)	2.468(2)
	O(7)#13	2.5186(16)	2.5251(12)	2.609(2)	2.573(3)	2.610(2)	2.567(2)
	O(8)#13	2.3717(15)	2.3762(11)	2.418(2)	2.399(3)	2.412(2)	2.3957(19)
	O(9)#16	2.9456(15)	2.9483(11)	2.899(2)	2.998(3)	2.926(2)	2.994(3)
Ca(3)-	O(1)#1	2.6373(15)	2.6345(11)	2.634(2)	2.681(3)	2.638(2)	2.669(2)
	O(3)#2	2.6570(14)	2.6507(11)	2.602(2)	2.633(2)	2.602(2)	2.6348(18)
	O(4)#1	2.4591(14)	2.4612(11)	2.501(2)	2.439(2)	2.497(2)	2.4444(19)
	O(4)#2	2.4278(14)	2.4313(11)	2.463(2)	2.414(3)	2.458(2)	2.4191(19)
	O(6)#12	2.4604(14)	2.4616(11)	2.473(2)	2.450(3)	2.468(2)	2.449(2)
	O(7)	2.3600(15)	2.3633(11)	2.345(2)	2.407(3)	2.350(2)	2.401(2)
	O(8)#11	2.3685(14)	2.3691(11)	2.378(2)	2.368(2)	2.374(2)	2.3675(18)
	O(9)	2.3782(14)	2.3748(11)	2.320(2)	2.358(3)	2.327(2)	2.360(2)
Ca(4A)-	O(6)#18	NA	NA	NA	2.736(5)	NA	2.758(6)
	O(6)#19	NA	NA	NA	2.736(5)	NA	2.758(6)
	O(7)#18	NA	NA	NA	2.541(4)	NA	2.541(4)
	O(7)#19	NA	NA	NA	2.541(4)	NA	2.541(4)
Ca(4B)-	O(6)#18	NA	NA	NA	2.516(3)	NA	2.518(2)
	O(6)#19	NA	NA	NA	2.516(3)	NA	2.518(2)
	O(7)#18	NA	NA	NA	2.846(5)	NA	2.851(4)
	O(7)#19	NA	NA	NA	2.846(5)	NA	2.851(4)
Fe(1)-	O(4)	2.1131(16)	2.1067(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(4)#1	2.1130(16)	2.1068(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(4)#2	2.1130(16)	2.1068(13)	2.046(2)	2.122(3)	2.048(2)	2.104(2)
	O(5)	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
	O(5)#1	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
	O(5)#2	2.0893(16)	2.0806(12)	2.011(2)	2.108(3)	2.023(2)	2.092(2)
P(1)-	O(1)	1.5281(15)	1.5257(11)	1.528(2)	1.528(3)	1.530(2)	1.528(2)
	O(2)	1.5484(16)	1.5456(13)	1.521(2)	1.532(3)	1.523(2)	1.535(2)
	O(3)	1.5393(14)	1.5382(11)	1.535(2)	1.537(2)	1.535(2)	1.5377(18)
	O(4)	1.5512(16)	1.5497(12)	1.562(2)	1.546(3)	1.561(2)	1.546(2)
P(2)-	O(5)	1.5538(16)	1.5552(12)	1.588(2)	1.556(3)	1.583(2)	1.558(2)
	O(6)	1.5261(15)	1.5260(12)	1.527(2)	1.537(3)	1.526(2)	1.531(2)
	O(7)	1.5284(15)	1.5256(12)	1.516(2)	1.529(3)	1.520(2)	1.529(2)

Table S4. Bond lengths for synthesized whitlockite and merrillite in angstroms.

	O(8)	1.5385(15)	1.5381(11)	1.544(2)	1.543(2)	1.546(2)	1.5422(19)
P(3A)-	O(10A)	1.582(3)	1.578(3)	1.540(4)	1.530(3)	1.57(3)	1.530(7)
	O(9)	1.5297(14)	1.5291(11)	1.5366(19)	1.493(3)	1.518(6)	1.500(2)
	O(9)#18	1.5298(14)	1.5292(11)	1.5366(19)	1.493(3)	1.518(6)	1.500(2)
	O(9)#19	1.5298(14)	1.5292(11)	1.5366(19)	1.493(3)	1.518(6)	1.500(2)
P(3B)-	O(10B)	1.51(3)	1.534(14)	1.49(3)	1.546(7)	1.543(5)	1.535(6)
	O(9)	1.512(3)	1.5187(16)	1.507(12)	1.526(2)	1.534(2)	1.5209(19)
	O(9)#18	1.512(3)	1.5187(16)	1.507(12)	1.526(2)	1.534(2)	1.5209(19)
	O(9)#19	1.512(3)	1.5187(16)	1.507(12)	1.526(2)	1.534(2)	1.5209(19)

Parenthetic values are standard deviations. Pt cruc. = merrillite produced by heating in open air in a platinum crucible. SiO_2 tube = merrillite produced in a triple argon purged and sealed silica glass tube.

CIF Files to accompany Chapter Two

Fe-whitlockite

data FeWhitlockite 0m b _chemical_formula_sum ' Ca9 Fe H O28 P7' _chemical_formula_weight 1082.37 chemical compound source 'synthetic' _symmetry_cell_setting trigonal _symmetry_space_group_name_H-M R3c _chemical_name_mineral whitlockite loop_ _symmetry_equiv_pos_as_xyz 'x, y, z' '-y, x-y, z' '-x+y, -x, z' '-y, -x, z+1/2' '-x+y, y, z+1/2' 'x, x-y, z+1/2' 'x+2/3, y+1/3, z+1/3' '-y+2/3, x-y+1/3, z+1/3' '-x+y+2/3, -x+1/3, z+1/3' '-y+2/3, -x+1/3, z+5/6' '-x+y+2/3, y+1/3, z+5/6' 'x+2/3, x-y+1/3, z+5/6' 'x+1/3, y+2/3, z+2/3' '-y+1/3, x-y+2/3, z+2/3' '-x+y+1/3, -x+2/3, z+2/3' '-y+1/3, -x+2/3, z+7/6' '-x+y+1/3, y+2/3, z+7/6' 'x+1/3, x-y+2/3, z+7/6' cell length a 10.3510(7) _cell_length_b 10.3510(7) _cell_length_c 37.059(2) _cell_angle_alpha 90.00 cell angle beta 90.00 _cell_angle_gamma 120.00 _cell_volume 3438.6(4) _cell_formula_units_Z 6 _cell_measurement_temperature 100(2) exptl crystal density diffrn 3.136 _exptl_crystal_density_method 'not measured' _exptl_crystal_F_000 3216 exptl absorpt coefficient mu 3.294 _exptl_absorpt_correction_type multi-scan _exptl_absorpt_correction_T_min 0.6768

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Refinement of F^2^ against ALL reflections. The weighted R-factor wR and goodness of fit S are based on F^2^, conventional R-factors R are based on F, with F set to zero for negative F^2^. The threshold expression of F^2^ > 2sigma(F^2^) is used only for calculating R-factors(gt) etc. and is not relevant to the choice of reflections for refinement. R-factors based on F^2^ are statistically about twice as large as those based on F, and R-factors based on ALL data will be even larger.

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Ca1 Ca 0.71906(5) 0.85343(4) 0.215037(11) 0.00506(9) Uani 1 1 d . . .
Ca3 Ca 0.51246(4) 0.05420(5) 0.179244(10) 0.00547(8) Uani 1 1 d . . .
Ca2 Ca 1.15795(5) 0.86426(4) 0.107990(13) 0.01161(9) Uani 1 1 d . . .
P2 P 0.47612(6) -0.01616(6) 0.082656(13) 0.00455(10) Uani 1 1 d . . .
P1 P 0.86055(6) 0.68479(6) 0.146444(12) 0.00450(10) Uani 1 1 d . . .
P3A P 0.3333 -0.3333 0.19345(3) 0.0051(2) Uani 0.890(3) 3 d SP . .
P3B P 0.3333 -0.3333 0.2160(2) 0.005 Uiso 0.110(3) 3 d SP . .
O10A O 0.3333 -0.3333 0.15077(9) 0.0130(7) Uani 0.890(3) 3 d SP . .
O10B O 0.3333 -0.3333 0.2568(8) 0.013 Uiso 0.110(3) 3 d SP . .
O6 O 0.56574(16) -0.08767(16) 0.06961(4) 0.0076(3) Uani 1 1 d . . .
O9 O 0.35329(15) -0.18316(15) 0.20559(4) 0.0087(3) Uani 1 1 d . . .
O8 O 0.33437(16) -0.06065(15) 0.06044(4) 0.0061(2) Uani 1 1 d . . .
O5 O 0.57543(15) 0.15508(17) 0.07742(4) 0.0074(3) Uani 1 1 d . . .
O2 O 0.87474(16) 0.73809(18) 0.10685(4) 0.0086(3) Uani 1 1 d . . .
O3 O 1.01933(15) 0.73373(15) 0.15978(4) 0.0069(3) Uani 1 1 d . . .
O1 O 0.77948(16) 0.75035(16) 0.16736(4) 0.0078(3) Uani 1 1 d . . .
O4 O 0.76750(15) 0.51279(17) 0.15058(4) 0.0072(3) Uani 1 1 d . . .
O7 O 0.42614(17) -0.05431(17) 0.12192(4) 0.0092(3) Uani 1 1 d . . .
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Fe/Mg-whitlockite

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'-y+2/3, x-y+1/3, z+1/3'
'-x+y+2/3, -x+1/3, z+1/3'
'-y+2/3, -x+1/3, z+5/6'
'-x+y+2/3, y+1/3, z+5/6'
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'x+1/3, y+2/3, z+2/3'
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'-y+1/3, -x+2/3, z+7/6'
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cell length b 10.3499(4) _cell_length_c 37.0715(16) _cell_angle_alpha 90.00 _cell_angle_beta 90.00 _cell_angle_gamma 120.00 cell volume 3439.1(2) _cell_formula_units_Z 6 cell measurement temperature 100(2) _exptl_crystal_density_diffrn 3.103 _exptl_crystal_density_method 'not measured' exptl crystal F 000 3186 _exptl_absorpt_coefficient_mu 3.084 _exptl_absorpt_correction_type multi-scan exptl absorpt correction T min 0.6997 exptl absorpt correction T max 0.7466 _exptl_absorpt_process_details "SADABS" _diffrn_ambient_temperature 100(2) diffrn radiation wavelength 0.71073 diffrn radiation type MoK\a diffrn radiation source 'fine-focus sealed tube' diffrn radiation monochromator graphite _diffrn_reflns_number 17609 _diffrn_reflns_av_R_equivalents 0.0202 diffrn reflns av sigmal/netl 0.0113

_diffrn_reflns_limit_h_min -14 _diffrn_reflns_limit_h_max 14 -14 _diffrn_reflns_limit_k_min diffrn reflns limit k max 14 _diffrn_reflns_limit_l min -52 _diffrn_reflns_limit_l_max 52 diffrn reflns theta min 2.52 diffrn reflns theta max 30.51 _reflns_number_total 2355

_reflns_number_gt 2336 _reflns_threshold_expression >2sigma(I)

_computing_structure_solution 'SHELXS-97 (Sheldrick, 2008)'
_computing_structure_refinement 'SHELXL-97 (Sheldrick, 2008)'

_refine_special_details

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Refinement of F^2^ against ALL reflections. The weighted R-factor wR and goodness of fit S are based on F^2^, conventional R-factors R are based on F, with F set to zero for negative F^2^. The threshold expression of F^2^ > 2sigma(F^2^) is used only for calculating R-factors(gt) etc. and is not relevant to the choice of reflections for refinement. R-factors based on F^2^ are statistically about twice as large as those based on F, and R-factors based on ALL data will be even larger.

_refine_ls_structure_factor_coef Fsqd _refine_ls_matrix_type full

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Mg1 Mg 0.6667 0.3333 0.113848(16) 0.003 Uiso 0.363(4) 3 d SP . .
Ca1 Ca 0.71923(3) 0.85341(3) 0.215018(8) 0.00441(7) Uani 1 1 d . . .
Ca3 Ca 0.51225(3) 0.05368(3) 0.179264(8) 0.00492(6) Uani 1 1 d . . .
Ca2 Ca 1.15730(4) 0.86455(3) 0.107972(10) 0.01143(7) Uani 1 1 d . . .
P2 P 0.47608(5) -0.01597(5) 0.082739(10) 0.00369(8) Uani 1 1 d . . .
P1 P 0.86008(4) 0.68408(5) 0.146408(9) 0.00373(7) Uani 1 1 d . . .
P3A P 0.3333 -0.3333 0.19339(2) 0.00472(18) Uani 0.840(2) 3 d SP . .
P3B P 0.3333 -0.3333 0.21651(13) 0.005 Uiso 0.160(2) 3 d SP . .
O10A O 0.3333 -0.3333 0.15081(7) 0.0124(5) Uani 0.840(2) 3 d SP . .
O10B O 0.3333 -0.3333 0.2579(3) 0.004(2) Uiso 0.160(2) 3 d SP . .
O6 O 0.56579(12) -0.08715(12) 0.06953(3) 0.0068(2) Uani 1 1 d . . .
O9 O 0.35257(12) -0.18341(11) 0.20545(3) 0.0080(2) Uani 1 1 d . . .
O8 O 0.33451(12) -0.06062(11) 0.06045(3) 0.00537(19) Uani 1 1 d . . .
O5 O 0.57559(11) 0.15558(13) 0.07776(3) 0.0066(2) Uani 1 1 d . . .
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O1 O 0.77921(12) 0.74988(12) 0.16719(3) 0.00655(19) Uani 1 1 d . . .
O4 O 0.76708(11) 0.51224(13) 0.15058(3) 0.0066(2) Uani 1 1 d . . .
O7 O 0.42597(13) -0.05502(13) 0.12187(3) 0.0084(2) Uani 1 1 d . . .
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Ca1 0.00469(14) 0.00469(12) 0.00420(14) -0.00059(10) -0.00019(10) 0.00260(11)
Ca3 0.00442(13) 0.00589(13) 0.00431(13) 0.00130(10) 0.00028(11) 0.00246(11)
Ca2 0.01162(15) 0.00662(14) 0.00774(14) -0.00155(11) 0.00505(12) -0.00168(11)
P2 0.00385(15) 0.00397(17) 0.00383(17) -0.00036(13) -0.00039(13) 0.00238(13)
P1 0.00386(16) 0.00425(16) 0.00315(17) -0.00030(13) 0.00000(14) 0.00207(13)
P3A 0.0038(2) 0.0038(2) 0.0066(5) 0.000 0.000 0.00189(10)
O10A 0.0142(8) 0.0142(8) 0.0089(12) 0.000 0.000 0.0071(4)
O6 0.0088(5) 0.0082(5) 0.0062(4) -0.0003(4) -0.0001(4) 0.0064(4)
09 0.0073(5) 0.0048(4) 0.0119(5) 0.0014(3) 0.0033(4) 0.0032(4)
08 0.0050(5) 0.0058(5) 0.0047(4) -0.0007(3) -0.0011(3) 0.0023(4)
05 0.0052(4) 0.0047(5) 0.0097(5) -0.0019(4) -0.0012(4) 0.0022(4)
O2 0.0092(5) 0.0167(6) 0.0035(5) 0.0014(4) 0.0009(4) 0.0084(4)
O3 0.0045(5) 0.0071(5) 0.0058(4) 0.0004(4) 0.0000(4) 0.0023(4)
01 0.0085(5) 0.0074(5) 0.0060(4) -0.0010(4) 0.0008(4) 0.0056(4)
O4 0.0047(5) 0.0051(5) 0.0097(5) -0.0001(4) -0.0010(4) 0.0022(4)
07 0.0089(5) 0.0127(5) 0.0047(4) -0.0007(4) 0.0002(4) 0.0063(4)
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Fe-merrillite heat treated open to air

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loop_

_space_group_symop_operation_xyz 'x, y, z' '-y, x-y, z' '-x+y, -x, z' '-y, -x, z+1/2'

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_diffrn_reflns_point_group_measured_fraction_max 0.998 _diffrn_reflns_point_group_measured_fraction_full 0.996 _reflns_number_total 2334 _reflns_number gt 2283 | > 2 (1)reflns threshold expression refins Friedel coverage 0.994 _reflns_Friedel_fraction_max 0.998 _reflns_Friedel_fraction_full 0.997 _computing_structure_refinement 'SHELXL-2013 (Sheldrick, 2013)' <pre_refine_special_details</pre> Refined as a 2-component inversion twin. : _refine_ls_structure_factor_coef Fsqd _refine_ls_matrix_type full refine Is weighting scheme calc refine Is weighting details 'w=1/[\s^2^(Fo^2^)+(0.0264P)^2^+4.0671P] where P=(Fo^2^+2Fc^2^)/3' atom sites solution primary intrinsic _atom_sites_solution_secondary diffmap _atom_sites_solution_hydrogens . refine Is hydrogen treatment undef _refine_ls_extinction_method none _refine_ls_extinction_coef _refine_ls_abs_structure_details Refined as an inversion twin. ; _refine_ls_abs_structure_Flack 0.56(2) chemical absolute configuration ? _refine_ls_number_reflns 2334 refine ls number parameters 140 _refine_ls_number_restraints 2 0.0175 _refine_ls_R_factor_all refine ls R factor gt 0.0172 refine ls wR factor ref 0.0455 _refine_ls_wR_factor_gt 0.0454 _refine_ls_goodness_of_fit_ref 1.095 _refine_ls_restrained_S_all 1.094 _refine_ls_shift/su max 0.000 _refine_ls_shift/su_mean 0.000 loop _atom_site_label _atom_site_type_symbol _atom_site_fract_x _atom_site_fract_y atom site fract z _atom_site_U_iso_or_equiv _atom_site_adp_type

diffrn reflns Laue measured fraction full 0.996

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Fe1 Fe 0.6667 0.3333 0.11505(2) 0.00519(13) Uani 1 3 d S T P . .
Ca1 Ca 0.72237(6) 0.85471(6) 0.21403(2) 0.00411(12) Uani 1 1 d . . . .
Ca3 Ca 0.50813(6) 0.04416(7) 0.17985(2) 0.00416(11) Uani 1 1 d . . . .
Ca2 Ca 1.14831(6) 0.87139(6) 0.10715(2) 0.00557(12) Uani 1 1 d . . . .
P2 P 0.47448(9) -0.01749(9) 0.08408(2) 0.00319(14) Uani 1 1 d . . . .
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P3A P 0.3333 -0.3333 0.1952(13) 0.004 Uiso 0.029(3) 3 d DS U P . .
P3B P 0.3333 -0.3333 0.21780(3) 0.0038(3) Uani 0.971(3) 3 d S T P . .
O10A O 0.3333 -0.3333 0.25937(11) 0.0065(7) Uani 0.971(3) 3 d S T P . .
O10B O 0.3333 -0.3333 0.1549(15) 0.007 Uiso 0.029(3) 3 d DS U P . .
O6 O 0.5682(2) -0.0826(2) 0.06956(5) 0.0056(4) Uani 1 1 d . . . . .
O9 O 0.3444(2) -0.1866(2) 0.20504(5) 0.0075(4) Uani 1 1 d . . . .
O8 O 0.3365(2) -0.0552(2) 0.06064(5) 0.0046(4) Uani 1 1 d . . . .
O5 O 0.5763(2) 0.1589(2) 0.08082(5) 0.0046(4) Uani 1 1 d . . . .
O2 O 0.8599(2) 0.7100(2) 0.10584(6) 0.0056(4) Uani 1 1 d . . . .
O3 O 1.0161(2) 0.7289(2) 0.15893(5) 0.0056(4) Uani 1 1 d . . . .
O1 O 0.7803(2) 0.7482(2) 0.16651(5) 0.0061(4) Uani 1 1 d . . . .
O4 O 0.7619(2) 0.5034(2) 0.15187(5) 0.0051(4) Uani 1 1 d . . . . .
O7 O 0.4214(2) -0.0610(2) 0.12267(6) 0.0063(4) Uani 1 1 d . . . .
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Ca3 0.0042(3) 0.0043(2) 0.0036(2) 0.00061(19) -0.0001(2) 0.00182(19)
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P1 0.0034(3) 0.0032(3) 0.0033(3) -0.0003(2) 0.0000(3) 0.0016(2)
P3B 0.0027(3) 0.0027(3) 0.0059(6) 0.000 0.000 0.00136(17)
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02 0.0064(9) 0.0078(9) 0.0035(9) 0.0005(7) 0.0004(7) 0.0042(8)
O3 0.0042(9) 0.0068(9) 0.0048(8) 0.0007(7) -0.0002(7) 0.0020(7)
01 0.0079(9) 0.0070(9) 0.0054(8) -0.0011(7) 0.0005(7) 0.0051(8)
O4 0.0043(9) 0.0035(9) 0.0064(9) 0.0000(7) -0.0005(7) 0.0011(7)
07 0.0076(9) 0.0059(9) 0.0051(8) 0.0010(7) 0.0007(7) 0.0032(8)
```

Fe-merrillite heat treated in an SiO₂ sealed tube

chemical formula sum ' Ca9.44 Fe H0.12 O28 P7' _chemical_formula_weight 1099.12 chemical name mineral merrillite _space_group_crystal_system trigonal _space_group_IT_number 161 'synthetic' _chemical_compound_source symmetry space group name H-M R3c loop _space_group_symop_operation_xyz 'x, y, z' '-y, x-y, z' '-x+y, -x, z' '-y, -x, z+1/2' '-x+y, y, z+1/2' 'x, x-y, z+1/2' 'x+2/3, y+1/3, z+1/3' '-y+2/3, x-y+1/3, z+1/3' '-x+y+2/3, -x+1/3, z+1/3' '-y+2/3, -x+1/3, z+5/6' '-x+y+2/3, y+1/3, z+5/6' 'x+2/3, x-y+1/3, z+5/6' 'x+1/3, y+2/3, z+2/3' '-y+1/3, x-y+2/3, z+2/3' '-x+y+1/3, -x+2/3, z+2/3' '-y+1/3, -x+2/3, z+7/6' '-x+y+1/3, y+2/3, z+7/6' 'x+1/3, x-y+2/3, z+7/6' _cell_length a 10.3453(6) cell length b 10.3453(6) _cell_length_c 37.118(2) _cell_angle_alpha 90 _cell_angle_beta 90 cell angle gamma 120 _cell_volume 3440.3(5) _cell_formula_units_Z 6 _cell_measurement_temperature 100(2) exptl crystal density diffrn 3.183 _exptl_crystal_F_000 3264 _exptl_crystal_size_max .05 _exptl_crystal_size_mid .05 _exptl_crystal_size_min .05 exptl absorpt coefficient mu 3.392 _exptl_absorpt_correction_type multi-scan _exptl_absorpt_correction_T_min 0.6939 _exptl_absorpt_correction_T_max 0.7466

data FeMerrilliteQTZ 0m a

exptl absorpt process details "SADABS" diffrn ambient temperature 100(2)_diffrn_radiation_wavelength 0.71073 _diffrn_radiation_type MoK\a diffrn reflns number 11475 diffrn reflns av unetl/netl 0.0178 _diffrn_reflns_av_R_equivalents 0.0236 _diffrn_reflns_limit_h_min -14 diffrn reflns limit h max 14 _diffrn_reflns_limit k min -14 diffrn reflns limit k max 14 _diffrn_reflns_limit_l_min -45 diffrn reflns limit I max 52 diffrn reflns theta min 2.524 diffrn reflns theta max 30.488 _diffrn_reflns_theta_full 25.242 _diffrn_measured_fraction_theta_max 1.000 diffrn measured fraction theta full 0.999 diffrn reflns Laue measured fraction max 1.000 diffrn reflns Laue measured fraction full 0.999 diffrn reflns point group measured fraction max 1.000 _diffrn_reflns_point_group_measured_fraction_full 0.999 _reflns_number_total 2238 reflns number gt 2189 'I > 2\s(I)' _reflns_threshold_expression _reflns_Friedel_coverage 0.994 _reflns_Friedel_fraction_max 1.000 reflns Friedel fraction full 1.000

_computing_structure_refinement 'SHELXL-2013 (Sheldrick, 2013)'

_refine_special_details

Refined as a 2-component inversion twin. : refine ls structure factor coef Fsqd _refine_ls_matrix_type full _refine_ls_weighting_scheme calc refine Is weighting details 'w=1/[\s^2^(Fo^2^)+(0.0276P)^2^+8.5052P] where P=(Fo^2^+2Fc^2^)/3' atom sites solution primary intrinsic _atom_sites_solution_secondary diffmap atom sites solution hydrogens none refine Is hydrogen treatment none refine Is extinction method none _refine_ls_extinction_coef _refine_ls_abs_structure_details ; Refined as an inversion twin. ;

112

```
refine ls abs structure Flack 0.48(3)
chemical absolute configuration ?
refine ls number reflns
                              2238
refine ls number parameters
                                  146
_refine_ls_number_restraints
                                2
refine ls R factor all
                            0.0211
refine Is R factor gt
                            0.0204
_refine_ls_wR_factor_ref
                              0.0521
_refine_ls_wR_factor_gt
                              0.0517
refine Is goodness of fit ref 1.078
refine Is restrained S all
                              1.078
refine ls shift/su max
                             0.001
_refine_ls_shift/su_mean
                              0.000
loop_
atom site label
_atom_site_type_symbol
_atom_site_fract_x
atom site fract y
atom site fract z
_atom_site_U_iso_or_equiv
atom site adp type
_atom_site_occupancy
_atom_site_site_symmetry order
atom site calc flag
_atom_site_refinement_flags_posn
_atom_site_refinement_flags adp
atom site refinement flags occupancy
atom site disorder assembly
_atom_site_disorder_group
Fe1 Fe 0.6667 0.3333 0.11420(3) 0.00572(14) Uani 1 3 d S T P . .
Ca1 Ca 0.72446(8) 0.85713(7) 0.21332(2) 0.00491(13) Uani 1 1 d . . . .
Ca3 Ca 0.50988(7) 0.05019(8) 0.17850(2) 0.00503(13) Uani 1 1 d . . . .
Ca2 Ca 1.14949(8) 0.87308(8) 0.10639(2) 0.01494(16) Uani 1 1 d . . . .
P2 P 0.47132(11) -0.02015(10) 0.08205(2) 0.00514(17) Uani 1 1 d . . . .
P1 P 0.86017(10) 0.67945(10) 0.14554(2) 0.00426(16) Uani 1 1 d . . . .
P3B P 0.3333 -0.3333 0.21379(6) 0.0051(5) Uani 0.761(5) 3 d S T P . .
P3A P 0.3333 -0.3333 0.1932(2) 0.005 Uiso 0.239(5) 3 d DS U P . .
O10A O 0.3333 -0.3333 0.25546(17) 0.0085(12) Uani 0.761(5) 3 d S T P . .
O10B O 0.3333 -0.3333 0.1520(2) 0.009 Uiso 0.239(5) 3 d DS U P . .
O6 O 0.5602(3) -0.0930(3) 0.06843(7) 0.0156(6) Uani 1 1 d . . . .
O9 O 0.3420(3) -0.1882(3) 0.20178(9) 0.0165(6) Uani 1 1 d . . . .
O8 O 0.3317(3) -0.0575(3) 0.05945(6) 0.0063(4) Uani 1 1 d . . . . .
O5 O 0.5746(3) 0.1516(3) 0.07811(7) 0.0092(5) Uani 1 1 d . . . .
O2 O 0.8660(3) 0.7168(3) 0.10543(7) 0.0104(5) Uani 1 1 d . . . .
O3 O 1.0204(2) 0.7308(3) 0.15797(6) 0.0068(4) Uani 1 1 d . . . .
O1 O 0.7857(3) 0.7538(3) 0.16568(7) 0.0087(5) Uani 1 1 d . . . .
O4 O 0.7656(3) 0.5096(3) 0.15229(7) 0.0070(4) Uani 1 1 d . . . .
O7 O 0.4176(3) -0.0657(3) 0.12080(7) 0.0141(5) Uani 1 1 d . . . .
Ca4a Ca 0.3333 -0.3333 0.13595(13) 0.0112(13) Uiso 0.296(7) 3 d S . P . .
Ca4b Ca 0.3333 -0.3333 0.1028(3) 0.013(3) Uiso 0.148(7) 3 d S . P . .
```

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atom site aniso U 11
atom site aniso U 22
_atom_site_aniso_U 33
atom site aniso U 23
atom site aniso U 13
_atom_site_aniso U 12
Fe1 0.00575(19) 0.00575(19) 0.0057(3) 0.000 0.000 0.00288(10)
Ca1 0.0055(3) 0.0056(3) 0.0043(3) -0.0007(2) -0.0001(2) 0.0032(2)
Ca3 0.0040(3) 0.0056(3) 0.0049(3) 0.0016(2) 0.0002(3) 0.0020(2)
Ca2 0.0125(3) 0.0103(3) 0.0089(3) -0.0014(3) 0.0053(3) -0.0041(3)
P2 0.0074(4) 0.0057(3) 0.0037(4) -0.0007(3) -0.0008(3) 0.0043(3)
P1 0.0045(3) 0.0046(3) 0.0038(4) -0.0002(3) -0.0003(3) 0.0024(3)
P3B 0.0025(5) 0.0025(5) 0.0104(13) 0.000 0.000 0.0012(2)
O10A 0.0094(17) 0.0094(17) 0.007(3) 0.000 0.000 0.0047(8)
06 0.0272(15) 0.0239(14) 0.0095(12) -0.0054(10) -0.0061(10) 0.0231(13)
09 0.0123(11) 0.0080(11) 0.0320(16) 0.0084(11) 0.0128(11) 0.0071(11)
08 0.0070(10) 0.0055(10) 0.0046(10) -0.0004(8) -0.0008(8) 0.0017(8)
05 0.0052(10) 0.0090(12) 0.0126(12) -0.0031(9) -0.0014(8) 0.0030(9)
O2 0.0076(11) 0.0209(13) 0.0040(11) 0.0010(9) -0.0004(9) 0.0080(10)
O3 0.0049(10) 0.0080(10) 0.0052(10) 0.0010(8) -0.0012(8) 0.0016(9)
01 0.0133(11) 0.0109(10) 0.0057(11) -0.0010(8) 0.0003(9) 0.0089(9)
O4 0.0049(10) 0.0052(11) 0.0100(12) -0.0006(9) -0.0011(8) 0.0018(9)
07 0.0155(13) 0.0246(14) 0.0053(11) 0.0004(10) -0.0012(9) 0.0123(12)
```

Fe/Mg-merrillite heat treated open to air

data_FeMgMerrillite_Air_0m_a

_chemical_formula_sum 'Ca9 Fe0.75 Mg0.25 O28 P7' _chemical_formula_weight 1073.59 _chemical_name_mineral merrillite _space_group_crystal_system trigonal _space_group_IT_number 161 _chemical_compound_source 'synthetic' _symmetry_space_group_name_H-M R3c

loop_

_space_group_symop_operation_xyz 'x, y, z' '-y, x-y, z' '-y, -x, z+1/2' '-x+y, y, z+1/2' 'x, x-y, z+1/2' 'x+2/3, y+1/3, z+1/3' '-y+2/3, x-y+1/3, z+1/3' '-x+y+2/3, -x+1/3, z+1/3'

```
'-y+2/3, -x+1/3, z+5/6'
'-x+y+2/3, y+1/3, z+5/6'
'x+2/3, x-y+1/3, z+5/6'
'x+1/3, y+2/3, z+2/3'
'-y+1/3, x-y+2/3, z+2/3'
'-x+y+1/3, -x+2/3, z+2/3'
'-y+1/3, -x+2/3, z+7/6'
'-x+y+1/3, y+2/3, z+7/6'
'x+1/3, x-y+2/3, z+7/6'
cell length a
                        10.3301(13)
_cell_length_b
                        10.3301(13)
_cell_length_c
                        37.062(5)
cell angle alpha
                          90
cell angle beta
                         90
cell angle gamma
                            120
_cell_volume
                        3425.1(10)
_cell_formula_units_Z
                            6
cell measurement temperature 100(2)
exptl crystal density diffrn 3.123
_exptl_crystal_size_max
                             0.07
exptl crystal size mid
                            0.07
_exptl_crystal_size_min
                            0.07
_exptl_absorpt_coefficient_mu 3.164
exptl absorpt correction type multi-scan
_exptl_absorpt_correction_T_min 0.7025
_exptl_absorpt_correction_T_max 0.7466
_exptl_absorpt_process_details "Sadabs"
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diffrn ambient temperature
                                100(2)
diffrn radiation wavelength
                               0.71073
_diffrn_radiation_type
                           MoK\a
diffrn reflns number
                            17501
_diffrn_reflns_av_unetl/netl
                             0.0141
_diffrn_reflns_av_R_equivalents 0.0249
diffrn reflns limit h min
                             -14
diffrn reflns limit h max
                             14
_diffrn_reflns_limit_k_min
                             -14
_diffrn_reflns_limit_k_max
                             14
diffrn reflns limit I min
                            -52
_diffrn_reflns_limit | max
                             52
_diffrn_reflns_theta_min
                             2.528
_diffrn_reflns_theta_max
                             30.498
diffrn reflns theta full
                            25.242
diffrn measured fraction theta max 1.000
diffrn measured fraction theta full 0.999
_diffrn_reflns_Laue_measured_fraction_max 1.000
_diffrn_reflns_Laue_measured_fraction_full 0.999
diffrn reflns point group measured fraction max 1.000
_diffrn_reflns_point_group_measured_fraction_full 0.999
reflns number total
                            2338
```

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reflns number gt
                          2278
_reflns_threshold_expression
                              || > 2 (1)
_reflns_Friedel_coverage
                            0.994
_reflns_Friedel_fraction_max
                              1.000
_reflns_Friedel_fraction_full 1.000
_computing_structure_refinement 'SHELXL-2013 (Sheldrick, 2013)'
_refine_special_details
Refined as a 2-component inversion twin.
;
_refine_ls_structure_factor_coef Fsqd
refine ls matrix type
                           full
refine Is weighting scheme
                               calc
refine Is weighting details
'w=1/[\s^2^(Fo^2^)+(0.0314P)^2^+4.3497P] where P=(Fo^2^+2Fc^2^)/3'
_atom_sites_solution_primary intrinsic
atom sites solution secondary diffmap
_atom_sites_solution_hydrogens none
_refine_ls_hydrogen_treatment none
refine Is extinction method
                               none
_refine_ls_extinction_coef
_refine_ls_abs_structure_details
Refined as an inversion twin.
;
_refine_ls_abs_structure_Flack 0.49(3)
chemical absolute configuration ?
_refine_ls_number_reflns
                             2338
_refine_ls_number_parameters
                                 143
refine ls number restraints
                              2
refine ls R factor all
                           0.0178
_refine_ls_R_factor_gt
                           0.0171
_refine_ls_wR_factor_ref
                            0.0489
_refine_ls_wR_factor_gt
                            0.0483
_refine_ls_goodness_of_fit_ref 1.053
refine Is restrained S all
                             1.053
refine Is shift/su max
                            0.001
_refine_ls_shift/su_mean
                             0.000
loop_
_atom_site_label
_atom_site_type_symbol
_atom_site_fract_x
atom site fract y
_atom_site_fract_z
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_atom_site_adp_type
_atom_site_occupancy
_atom_site_site_symmetry_order
_atom_site_calc_flag
```

_atom_site_refinement_flags_posn

```
atom site refinement flags adp
atom site refinement flags occupancy
atom site disorder assembly
atom site disorder group
Fe1 Fe 0.6667 0.3333 0.11488(2) 0.0019(3) Uani 0.754(7) 3 d S T P . .
Mg1 Mg 0.6667 0.3333 0.11488(2) 0.002 Uiso 0.246(7) 3 d S U P ...
Ca1 Ca 0.72255(6) 0.85481(6) 0.21402(2) 0.00404(12) Uani 1 1 d . . . .
Ca3 Ca 0.50884(6) 0.04511(7) 0.17962(2) 0.00405(12) Uani 1 1 d . . . .
Ca2 Ca 1.14889(7) 0.87089(6) 0.10713(2) 0.00741(12) Uani 1 1 d . . . .
P2 P 0.47459(9) -0.01742(9) 0.08372(2) 0.00299(14) Uani 1 1 d . . . .
P1 P 0.85644(8) 0.67669(9) 0.14577(2) 0.00309(13) Uani 1 1 d . . . .
P3B P 0.3333 -0.3333 0.21701(4) 0.0047(3) Uani 0.935(4) 3 d S T P . .
P3A P 0.3333 -0.3333 0.1933(5) 0.005 Uiso 0.065(4) 3 d DS U P . .
O10B O 0.3333 -0.3333 0.25863(12) 0.0074(7) Uani 0.935(4) 3 d S T P . .
O10A O 0.3333 -0.3333 0.1511(8) 0.007 Uiso 0.065(4) 3 d DS U P . .
O6 O 0.5677(2) -0.0831(2) 0.06914(5) 0.0065(4) Uani 1 1 d . . . .
O9 O 0.3436(2) -0.1870(2) 0.20443(6) 0.0093(4) Uani 1 1 d . . . .
O8 O 0.3356(2) -0.0559(2) 0.06051(5) 0.0053(4) Uani 1 1 d . . . .
O5 O 0.5762(2) 0.1582(2) 0.08034(5) 0.0045(4) Uani 1 1 d . . . . .
O2 O 0.8613(2) 0.7119(3) 0.10574(6) 0.0070(4) Uani 1 1 d . . . .
O3 O 1.0161(2) 0.7291(2) 0.15895(5) 0.0059(4) Uani 1 1 d . . . .
O1 O 0.7800(2) 0.7486(2) 0.16629(6) 0.0067(4) Uani 1 1 d . . . . .
O4 O 0.7623(2) 0.5041(2) 0.15157(5) 0.0046(4) Uani 1 1 d . . . .
O7 O 0.4222(3) -0.0616(2) 0.12242(6) 0.0078(4) Uani 1 1 d . . . .
loop
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_atom_site_aniso_U_11
atom site aniso U 22
_atom_site_aniso_ U 33
_atom_site_aniso_U_23
atom site aniso U 13
_atom_site_aniso_U_12
Fe1 0.0020(3) 0.0020(3) 0.0018(4) 0.000 0.000 0.00099(15)
Ca1 0.0046(3) 0.0041(2) 0.0039(2) -0.00055(19) -0.0001(2) 0.00253(19)
Ca3 0.0039(3) 0.0043(2) 0.0038(2) 0.00076(19) 0.0003(2) 0.00196(19)
Ca2 0.0069(2) 0.0058(3) 0.0060(2) -0.0004(2) 0.0024(2) 0.0006(2)
P2 0.0034(3) 0.0027(3) 0.0031(3) 0.0001(2) -0.0001(3) 0.0017(2)
P1 0.0032(3) 0.0030(3) 0.0030(3) -0.0001(2) 0.0001(3) 0.0015(2)
P3B 0.0026(4) 0.0026(4) 0.0088(7) 0.000 0.000 0.00132(18)
O10B 0.0070(11) 0.0070(11) 0.0083(17) 0.000 0.000 0.0035(5)
O6 0.0075(9) 0.0083(9) 0.0061(8) -0.0004(7) -0.0001(7) 0.0057(8)
09 0.0087(9) 0.0060(9) 0.0150(10) 0.0045(7) 0.0054(8) 0.0051(8)
08 0.0052(9) 0.0041(8) 0.0055(8) -0.0005(7) -0.0007(7) 0.0014(7)
05 0.0031(8) 0.0033(9) 0.0060(9) -0.0006(7) -0.0010(6) 0.0008(7)
O2 0.0089(10) 0.0093(10) 0.0041(9) 0.0000(7) 0.0002(7) 0.0056(8)
O3 0.0042(9) 0.0066(9) 0.0063(8) 0.0009(7) -0.0015(7) 0.0023(8)
01 0.0084(9) 0.0063(9) 0.0064(8) -0.0009(7) 0.0012(7) 0.0045(8)
```

O4 0.0047(9) 0.0031(9) 0.0061(9) -0.0003(7) -0.0005(7) 0.0020(7) O7 0.0106(10) 0.0085(9) 0.0048(8) 0.0012(7) 0.0011(7) 0.0051(8)

Fe/Mg-merrillite heat treated in an SiO₂ sealed tube

chemical formula sum ' Ca9.36 Fe0.65 H0.28 Mg0.35 O28 P7' _chemical_formula_weight 1084.98 space group crystal system trigonal space group IT number 161 _chemical_name_mineral merrillite _chemical_compound_source 'synthetic' symmetry space group name H-M R3c loop _space_group_symop_operation_xyz 'x, y, z' '-y, x-y, z' '-x+y, -x, z' '-y, -x, z+1/2' '-x+y, y, z+1/2' 'x, x-y, z+1/2' 'x+2/3, y+1/3, z+1/3' '-y+2/3, x-y+1/3, z+1/3' '-x+y+2/3, -x+1/3, z+1/3' '-y+2/3, -x+1/3, z+5/6' '-x+y+2/3, y+1/3, z+5/6' 'x+2/3, x-y+1/3, z+5/6' 'x+1/3, y+2/3, z+2/3' '-y+1/3, x-y+2/3, z+2/3' '-x+y+1/3, -x+2/3, z+2/3' '-y+1/3, -x+2/3, z+7/6' '-x+y+1/3, y+2/3, z+7/6' 'x+1/3, x-y+2/3, z+7/6' cell length a 10.3392(7) _cell_length_b 10.3392(7) _cell_length_c 37.081(2) _cell_angle_alpha 90 cell angle beta 90 _cell_angle_gamma 120 _cell_volume 3432.8(5) _cell_formula_units_Z 6 cell measurement temperature 100(2) _exptl_crystal_density_diffrn 3.149 _exptl_crystal_F_000 3225 _exptl_crystal_size_max 0.06 _exptl_crystal_size_mid 0.06 exptl crystal size min 0.06 _exptl_absorpt_coefficient_mu 3.176 _exptl_absorpt_correction_type multi-scan _exptl_absorpt_correction_T_min 0.6997

data FeMgMerrillite QTZ 0m a a

exptl absorpt correction T max 0.7466 _exptl_absorpt_process_details "SADABS" diffrn ambient temperature 100(2) _diffrn_radiation_wavelength 0.71073 diffrn radiation type MoK\a diffrn reflns number 17513 _diffrn_reflns_av_unetl/netl 0.0120 _diffrn_reflns_av_R_equivalents 0.0213 diffrn reflns limit h min -14 _diffrn_reflns_limit h max 14 diffrn reflns limit k min -14 _diffrn_reflns_limit_k_max 14 diffrn reflns limit I min -52 diffrn reflns limit I max 52 diffrn reflns theta min 2.526 _diffrn_reflns_theta_max 30.494 _diffrn_reflns_theta_full 25.242 diffrn measured fraction theta max 0.997 diffrn measured fraction theta full 0.994 diffrn reflns Laue measured fraction max 0.997 diffrn reflns Laue measured fraction full 0.994 _diffrn_reflns_point_group_measured_fraction_max 0.998 _diffrn_reflns_point_group_measured_fraction_full 0.996 refins number total 2342 _reflns_number_gt 2313 _reflns_threshold_expression 'I > 2\s(I)' refins Friedel coverage 0.996 reflns Friedel fraction max 0.999 _reflns_Friedel_fraction full 0.999 computing structure refinement 'SHELXL-2013 (Sheldrick, 2013)'

_refine_special_details

Refined as a 2-component inversion twin. : refine ls structure factor coef Fsqd _refine_ls_matrix_type full _refine_ls_weighting_scheme calc refine Is weighting details 'w=1/[\s^2^(Fo^2^)+(0.0231P)^2^+5.1052P] where P=(Fo^2^+2Fc^2^)/3' atom sites solution primary none _atom_sites_solution_secondary none atom sites solution hydrogens none refine Is hydrogen treatment none _refine_ls_extinction_method none _refine_ls_extinction_coef _refine_ls_abs_structure_details ; Refined as an inversion twin. ;

```
refine ls abs structure Flack 0.49(3)
chemical absolute configuration ?
refine ls number reflns
                              2342
refine ls number parameters
                                  153
_refine_ls_number_restraints
                                2
refine ls R factor all
                            0.0160
refine Is R factor gt
                            0.0157
_refine_ls_wR_factor ref
                              0.0414
_refine_ls_wR_factor_gt
                              0.0411
refine ls goodness of fit ref 1.085
refine Is restrained S all
                              1.085
refine ls shift/su max
                             0.000
_refine_ls_shift/su_mean
                              0.000
loop_
_atom_site_label
_atom_site_type_symbol
_atom_site_fract_x
atom site fract y
atom site fract z
_atom_site_U_iso_or_equiv
atom site adp type
_atom_site_occupancy
_atom_site_site_symmetry order
atom site calc flag
_atom_site_refinement_flags_posn
_atom_site_refinement_flags adp
_atom_site_refinement_flags_occupancy
atom site disorder assembly
_atom_site_disorder_group
Fe1 Fe 0.6667 0.3333 0.11499(2) 0.0037(3) Uani 0.646(6) 3 d S T P . .
Mg1 Mg 0.6667 0.3333 0.11499(2) 0.004 Uiso 0.354(6) 3 d S U P . .
Ca1 Ca 0.72390(6) 0.85662(6) 0.21442(2) 0.00529(10) Uani 1 1 d . . . .
Ca3 Ca 0.51028(6) 0.05093(6) 0.17944(2) 0.00551(10) Uani 1 1 d . . . .
Ca2 Ca 1.15060(6) 0.87166(6) 0.10741(2) 0.01588(13) Uani 1 1 d . . . .
P2 P 0.47248(8) -0.01935(8) 0.08298(2) 0.00539(13) Uani 1 1 d . . . .
P1 P 0.86006(8) 0.67984(8) 0.14641(2) 0.00467(12) Uani 1 1 d . . . . .
P3B P 0.3333 -0.3333 0.21458(7) 0.0061(5) Uani 0.691(7) 3 d S T P . .
P3A P 0.3333 -0.3333 0.19357(12) 0.0036(11) Uani 0.309(7) 3 d DS T P . .
O10A O 0.3333 -0.3333 0.1523(2) 0.015(3) Uani 0.309(7) 3 d DS T P . .
O10B O 0.3333 -0.3333 0.25597(15) 0.0109(11) Uani 0.691(7) 3 d S T P . .
O6 O 0.5620(2) -0.0909(2) 0.06936(5) 0.0132(4) Uani 1 1 d . . . .
O9 O 0.3431(3) -0.1876(2) 0.20297(6) 0.0169(4) Uani 1 1 d . . . . .
O8 O 0.3325(2) -0.0579(2) 0.06038(5) 0.0063(3) Uani 1 1 d . . . .
O5 O 0.5750(2) 0.1529(2) 0.07916(5) 0.0098(4) Uani 1 1 d . . . .
O2 O 0.8670(2) 0.7198(3) 0.10630(5) 0.0127(4) Uani 1 1 d . . . .
O3 O 1.02029(19) 0.7311(2) 0.15898(5) 0.0072(3) Uani 1 1 d . . . .
O1 O 0.7846(2) 0.7528(2) 0.16670(5) 0.0094(3) Uani 1 1 d . . . .
O4 O 0.7655(2) 0.5093(2) 0.15245(5) 0.0080(3) Uani 1 1 d . . . .
O7 O 0.4189(2) -0.0650(3) 0.12179(5) 0.0152(4) Uani 1 1 d . . . .
Ca4B Ca 0.3333 -0.3333 0.13691(12) 0.0067(11) Uani 0.264(6) 3 d S T P . .
Ca4A Ca 0.3333 -0.3333 0.1041(3) 0.019(3) Uani 0.098(6) 3 d S T P . .
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loop _atom_site_aniso_label _atom_site_aniso_U_11 _atom_site_aniso_U_22 _atom_site_aniso_U 33 atom site aniso U 23 _atom_site_aniso_U_13 _atom_site_aniso_U_12 Fe1 0.0036(3) 0.0036(3) 0.0038(4) 0.000 0.000 0.00181(15) Ca1 0.0058(2) 0.0056(2) 0.0052(2) -0.00089(17) -0.00049(17) 0.00340(18) Ca3 0.0047(2) 0.0061(2) 0.0054(2) 0.00142(17) -0.00002(19) 0.00235(18) Ca2 0.0135(3) 0.0107(2) 0.0095(2) -0.0019(2) 0.0058(2) -0.0044(2) P2 0.0068(3) 0.0061(3) 0.0047(3) -0.0005(2) -0.0008(2) 0.0043(2) P1 0.0051(3) 0.0054(3) 0.0036(3) -0.0003(2) -0.0001(2) 0.0028(2) P3B 0.0034(5) 0.0034(5) 0.0115(14) 0.000 0.000 0.0017(2) P3A 0.0052(11) 0.0052(11) 0.000(3) 0.000 0.000 0.0026(5) O10A 0.022(5) 0.022(5) 0.002(6) 0.000 0.000 0.011(2) O10B 0.0100(15) 0.0100(15) 0.013(2) 0.000 0.000 0.0050(8) O6 0.0207(10) 0.0195(9) 0.0097(8) -0.0036(7) -0.0035(7) 0.0177(9) 09 0.0135(9) 0.0079(8) 0.0319(12) 0.0078(8) 0.0126(8) 0.0074(8) 08 0.0062(8) 0.0065(7) 0.0052(7) 0.0003(6) -0.0003(6) 0.0023(6) 05 0.0055(8) 0.0090(9) 0.0143(9) -0.0032(7) -0.0013(6) 0.0033(7) 02 0.0113(9) 0.0252(11) 0.0045(8) 0.0014(7) 0.0002(7) 0.0112(8) O3 0.0048(8) 0.0077(8) 0.0069(7) 0.0011(6) -0.0002(6) 0.0014(7) 01 0.0130(8) 0.0099(8) 0.0077(7) -0.0014(6) 0.0006(6) 0.0075(7) O4 0.0059(8) 0.0048(8) 0.0122(8) -0.0015(6) -0.0018(6) 0.0019(7) 07 0.0185(10) 0.0271(11) 0.0056(8) 0.0007(7) 0.0000(7) 0.0157(9) Ca4B 0.0066(12) 0.0066(12) 0.007(2) 0.000 0.000 0.0033(6) Ca4A 0.022(4) 0.022(4) 0.013(5) 0.000 0.000 0.011(2)

APPENDIX II

SUPPLEMENTARY INFORMATION FOR CHAPTER THREE

Contents

Supplementary Figures Supplementary Methods Supplementary Tables Supplementary Discussion Supplementary Data Supplementary References

Supplementary Figures



Figure S1. Example of initial rate method batch experiment. At the onset of the experiment while the batch is at far from equilibrium conditions, cumulative moles of Ca released increase sharply in a linear fashion. To calculate a dissolution rate using the initial rate method, regression analysis is applied to measurements taken at the beginning of the experiment during the linear phase (dashed line) before initial conditions change significantly. The slope of the fitted line represents the release rate of Ca which, adjusted for surface area and mineral stoichiometry, represents the dissolution rate of the mineral under the conditions of the experiment. As the experiment continues, concentrations approach steady state. Data in figure is from experiment FAP-P4D.



Figure S2. Results of fluorapatite dissolution experiments from this study (+ symbols) plotted with fluorapatite rate data from multiple previous studies. Measured dissolution rates for fluorapatite from this study all fall within half an order of magnitude of previously published data points (Valsami-Jones et al., 1998; Welch et al., 2002; Guidry and Mackenzie, 2003; Köhler et al., 2005; Chaïrat et al., 2007; Harouiya et al., 2007; Bandstra et al., 2008). The intrinsic rate constant (k_H +) for the fluorapatite rate law derived in this study is within error of the intrinsic rate constant for a previously published rate law, and the pH dependence (n) in this study is within 10% of a previously published pH dependence (Chaïrat et al., 2007).



Figure S3. Plot of log dissolution rate law fits incorporating initial rate data from solubility experiments. Plot shows general rate law trends continue into neutral pH range.



Figure S4. Model of log phosphate release based on mineral stoichiometry and mineral dissolution rates including initial rate data from solubility experiments. Plot shows phosphate release rate from whitlockite, merrillite, and chlorapatite are all higher than fluorapatite over the range of pH 2 to pH 7.

Supplementary Methods

Materials

Chlorapatite was synthesized using previously published methods after (Prener, 1967, 1971; Tacker and Stormer, 1993; Dachs et al., 2010) by heating a mixture of hydroxyapatite and calcium chloride to 1100°C for 24 hours. Whitlockite was synthesized using methods based on Hughes et al. (2008) by combining 1.000 (\pm 0.003) g hydroxyapatite and 0.300 (\pm 0.003) g magnesium nitrate in 90 ml of 18.2 M Ω water, acidifying to pH <2.8 with phosphoric acid, and incubating at 240°C for 6-10 days in a hydrothermal Parr vessel. Merrillite was synthesized by heating whitlockite synthesized as described above to 1060°C for 72 hours after Hughes et al. (2008). Natural fluorapatite from Durango, Mexico, was purchased from Minerals Unlimited of Ridgecrest, California.

Minerals were determined to be pure phases by powder X-ray diffraction (XRD) using a Panalytical X'pert Pro diffractometer using *Cu Ka* radiation(XXL Laboratory, UNLV). Scans were performed over 5-75° 20, 0.008° step size at 0.06° sec⁻¹ scan rate, and phase identification was carried out using Panalytical X'Pert High Score Plus software. Mineral compositions were confirmed by electron microprobe analysis (EMP) on a JEOL JXA-8900 microprobe operating at 20keV and 10nA using a 10µm beam (EMiL Facility, UNLV) (see below). With the exception of merrillite, the stoichiometries used in calculations in this study came from the *International Mineralogical Association-CNMNC List of Mineral Names* (IMA)(Nickel and Nichols, 2009) and are Ca₂Mg(PO₃OH)(PO₄)₆ (whitlockite), Ca₅(PO₄)₃Cl (chlorapatite) and

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$Ca_5(PO_4)_3F$ (fluorapatite). The IMA official formula for merrillite is $Ca_9NaMg(PO_4)_7$. The method of merrillite synthesis used in this study allowed for no sodium. For this reason, the formula $Ca_{9.5}Mg(PO_4)_7$ is used in all required calculations. This formula was confirmed by EMP.

In apatites, a change in structure from P63/m hexagonal structure (fluorapatite) to P21/b monoclinic structure (end-member chlorapatite) can occur at the Cl end-member (Hughes et al., 1989), though hexagonal chlorapatite from high temperature synthesis is documented (Hovis and Harlov, 2010). EMP data indicate our synthetic chlorapatite is end-member or near end-member chlorapatite, and X-ray diffraction confirmed EMP analysis and that the structure is hexagonal.

Minerals were powdered in an agate mortar and pestle, and sieved to obtain the 75-150µm (fluorapatite) or 38-75µm (whitlockite, merrillite, and chlorapatite) size fraction. The retained fractions were cleaned of fine particles by repeatedly sonicating in spectrographic grade ethanol until the supernatant was clear, and then dried at 60°C. A representative sample of each mineral fraction was examined using a JEOL JSM-5610 Scanning Electron Microscope (SEM) (EMiL Facility, UNLV) to confirm the absence of fine particles on mineral surfaces. SEM imaging and analyses of merrillite powder showed very thin (sub-micron) Ca-P features partially covering grains which were not detectable by EMP or XRD. These are likely a product of the dehydrogenation process, and due to the tiny fraction that they make up of the total substrate are assumed to have a negligible effect on the mineral dissolution or solubility.

Surface areas of the cleaned mineral powders were measured by the Brunauer-Emmett-Teller (BET)(Brunauer et al., 1938) method in a Micromeritics ASAP 2020

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Accelerated Surface Area and Porosimetry analyzer using Kr as the adsorbate, and are: $0.0536 (\pm 0.0004) \text{ m}^2/\text{g}$ (chlorapatite), $0.0321 (\pm 0.0002) \text{ m}^2/\text{g}$ (fluorapatite), 0.0516(± 0.0002) m²/g (whitlockite), and $0.0465 (\pm 0.0003) \text{ m}^2/\text{g}$ (merrillite).

Dissolution Experimental Setup

Each dissolution experiment consisted of a mineral mass of 0.3000 (\pm 0.0005) g (fluorapatite) or 0.1500 (\pm 0.0005) g (all other minerals) and 180 ml of reacting solution in an acid-washed 250 ml low density polyethylene (LDPE) bottle. Reacting solutions consisted of 0.01M KNO₃ (designed to be similar in ionic strength to a soil solution)(Goyne et al., 2006), adjusted to the required pH of the experiment with trace element analysis grade HNO₃. Batches were immersed in a 25°C shaker bath and agitated at 100 strokes per minute. All batch conditions were run in duplicate, and mineral-free control experiments were run each time experiments were performed. High purity 18.2 M Ω water was used in all solutions, and all equipment was either new or acid washed prior to use.

At time intervals determined through preliminary experiments (10-60 minutes while dissolution rates were being assessed, depending on the pH condition) batches were opened and 10 ml of solution removed by pipette. Between 5 and 9 samples were taken from each batch during the first day to determine dissolution rates by the initial rate method. Subsequent samples of select batches were removed at intervals of days or weeks after the period of initial rate. These samples were to track progress toward steady state, and thus timing was not critical. Solution samples were measured for pH, filtered (0.45 μ m filter), and acidified with trace element analysis grade HNO₃ to a pH of <2.

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Samples were analyzed for Ca by flame atomic absorption (AA) using a Thermo Scientific iCE 3000 Series AA after adding 0.36 M lanthanum chloride solution to 10% v/v to mitigate interference by phosphorus (Eaton et al., 2005). Samples from batch experiments for each mineral at each pH were additionally analyzed for P by the methyl blue / acetic acid method (Murphy and Riley, 1962) using a Thermo Scientific Genesys 10S UV-Vis.

Moles Ca released into solution at each time interval were calculated using the expression (Welch and Ullman, 2000):

$$m_{(t)} = m_{(t-1)} + (c_{(t)} - c_{(t-1)})V_{(t-1)}$$
 (Eq. S1)

where $m_{(t)}$ and $m_{(t-1)}$ are the moles of Ca released into solution at time *t* and *t* - 1 respectively, $c_{(t)}$ and $c_{(t-1)}$ are measured calcium concentrations (M) at times *t* and *t* - 1, and $V_{(t-1)}$ is the volume (L) of reaction solution before the sampling event at time *t*. Moles Ca released were plotted versus time for each batch experiment and fit by linear regression analysis, with the slope of the linear fit representing the rate, r_{Ca} , of Ca release (mol·s⁻¹). Linear fits were not forced through zero. Rates were determined using the initial 5-9 points when Ca release was linear with time. Experiments that produced fits with R^2 values of less than 0.900 were rerun and experiments with R^2 values of less than 0.900 not used. Rates of Ca release, r_{Ca} , were then converted to mineral dissolution rates, r_{diss} , using:

$$r_{diss} = r_{Ca}/(A \cdot S)$$
 (Eq. S2)

where *A* is the substrate surface area (m^2) used in the experiment and is the product of the BET determined specific surface area (m^2/g) and the mineral mass used in the experiment (g). *S* is the stoichiometry of the respective mineral in the experiment.

Rate laws for each mineral were developed by applying linear regression to log dissolution rates of averaged batch experiments, as determined above, versus pH (Table 1, Supplementary Data Sheets S1-S35). The resulting pH dependence of mineral dissolution was fit to:

$$log R = log k_{H+} - npH; \qquad (Eq. S3)$$

where k_{H^+} is the dissolution rate constant and *n* is the rate dependence on *pH*.

Solubility Experimental Setup

Solubility experiments were performed using the methods of Zhu, 2009. For each mineral, a mass of 0.2000 grams (+/- 0.0005g) mineral powder was added to a 250ml LDPE vessel along with 200ml of high purity 18.2 M Ω water. Vessels were immersed in a 25°C shaker bath and agitated at 100 strokes per minute. All batch conditions were run in duplicate, and a mineral-free control experiment was included. Sampling consisted of removing 10ml of solution from each vessel at days 5, 10, and 22, every 2 weeks thereafter to day 92, and day 113. Solution samples were measured for pH, filtered (0.20 μ m filter), and acidified with trace element analysis grade HNO₃ to a pH of <2. Samples were analyzed for Ca by flame atomic absorption (AA) using a Thermo Scientific iCE 3000 Series AA after adding 0.36 M lanthanum chloride solution to 10% v/v to mitigate interference by phosphorus (Eaton et al., 2005). Samples were additionally analyzed for

P by the methyl blue / acetic acid method (Murphy and Riley, 1962) using a Thermo Scientific Genesys 10S UV-Vis. Once equilibrium concentrations were established, K_{sp} values were calculated from ion concentrations and final pH using PHREEQC (Version 2.15) (Parkhurst and Appelo, 1999) and the included Lawrence Livermore National Laboratory database (llnl.dat) (Johnson et al., 1992) for the reactions in Table 2. Resulting K_{sp} values for fluorapatite from our experiments generally agree with previous studies (Appendix II Table S7) lending confidence to the method.

Gusev Crater (MER Spirit) P ₂ O ₅ wt%		Opportunity)	P₂O₅ wt%	Farth / Mars data	P₂O₅ wt. %
Adirondack RAT	0.52	McKittrick BAT	0.99	Farth bulk	0.015
Humphrey BAT1	0.52	Guadelune RAT	0.55	Earth crust	0.013
Humphrey RAT2	0.56	Moio2 RAT	1.01	Mars mantle+crust	0.16
Mazatzal NewYork RAT1	0.82	BounceROck Case RAT	0.92		0.20
Mazatzal Brooklyn RAT2	0.65	Golf Post RAT FRAM	0.97		
Pot of Gold RAT	1.08	Lionstone Numa RAT	1.01		
WoolvPatch Sabre RAT	1.24	Tennessee RAT	1.03		
WoolyPatch Mastodon RAT	1.20	Kentucky Cubble Hill2 RAT	1.05		
Clovis Plano RAT	1.05	Virginia RAT	1.07		
Ebenezer RAT	0.97	Ontario London RAT	1.11		
 Uchben_Koolik_RAT	0.94	Grindstone_RAT	1.07		
Whishstone_Chisel_RAT	5.19	Kettlestone_RAT	1.03		
Champagne_RAT	5.07	Millstone_Dramensfjord_RAT	1.17		
Champagne_RAT2	5.07	Diamond_Jennes_Homann3_RAT1	1.05		
Peace_RAT1	0.59	Diamond_Jennes_Homann3_RAT2	1.06		
Peace_RAT2	0.49	MacKenzie_Campell_RAT	1.15		
Watchtower_Joker_RAT	4.50	Inuvik_Toroyuktuk_RAT	1.11		
		Bylot_RAT	1.01		
		Escher_Kirchner_RAT	1.01		
		Paikia_RAT	1.13		
		Wharenhui_RAT	1.08		
		Gagarin_RAT	1.07		
		IceCream_RAT	1.04		
		Friutbaskey_Strawberry	0.99		
		LemonRind_RAT	1.05		
		Olympia_Kalavrita	0.98		
Average P2O5	1.79		1.04		

 Table S5. APXS phosphorus of Mars surface rocks and Earth / Mars bulk and crustal abundance.

Rock data compiled from Brukner et al., 2008 and Ming et al., 2008. Earth / Mars bulk and crustal data from Wedepohl, 1995, and Wanke and Driebus, 1988.

Table So. Individual Daten experiment conditions and result	Table S6.	Individual batch e	experiment condition	s and results.
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Exp. ID	Mineral	Initial pH	Final pH	n	Dissolution Rate (<i>K_{diss}</i>) (mol m ⁻² s ⁻¹)	$\operatorname{Log} K_{diss}$ (mol m ⁻² s ⁻¹)	Uncertainty of Fit (mol m ⁻² s ⁻¹)**	R ²
WHIT-P2DD	Whitlockite	2.00	2.24	9	3.84E-07	-6.42	4.70E-08	0.907
WHIT-P2DDB	Whitlockite	2.00	2.34	9	3.54E-07	-6.45	2.67E-08	0.962
WHIT-P3	Whitlockite	3.00	3.18	6	6.39E-08	-7.19	7.44E-09	0.949
WHIT-P3D	Whitlockite	3.01	3.18	6	6.12E-08	-7.21	3.09E-09	0.990
WHIT-P4D	Whitlockite	3.99	4.81	5	3.43E-09	-8.46	3.60E-10	0.969
WHIT-P4DD	Whitlockite	3.98	4.82	5	3.43E-09	-8.46	6.08E-10	0.915
WHIT-P5D	Whitlockite	5.00	5.38	5	7.26E-10	-9.14	1.27E-10	0.918
WHIT-P5DDB	Whitlockite	4.99	5.67	6	6.29E-10	-9.20	8.59E-11	0.932
MER-P2DDC	Merrillite	2.00	2.16	8	2.86E-07	-6.54	2.03E-08	0.977
MER-P2DD	Merrillite	1.97	2.04	5	3.58E-07	-6.45	3.12E-08	0.983
MER-P3	Merrillite	3.00	3.52	7	5.71E-08	-7.24	3.75E-09	0.983
MER-P3D	Merrillite	3.00	3.17	6	6.26E-08	-7.20	3.15E-09	0.992
MER-P4	Merrillite	4.00	4.59	6	7.47E-09	-8.13	5.70E-10	0.982
MER-P4DD	Merrillite	4.00	4.43	6	5.41E-09	-8.27	4.28E-10	0.981
MER-P5	Merrillite	5.00	5.81	6	5.07E-10	-9.30	1.96E-11	0.995
MER-P5D	Merrillite	4.91	5.65	5	6.28E-10	-9.20	1.27E-10	0.913
CAP-P2	Chlorapatite	1.97	2.06	5	5.99E-07	-6.22	4.57E-08	0.986
CAP-P2D	Chlorapatite	2.00	2.06	6	4.91E-07	-6.31	6.85E-08	0.939
CAP-P3	Chlorapatite	2.95	3.16	5	2.64E-07	-6.58	2.64E-08	0.976
CAP-P3DD	Chlorapatite	3.00	3.33	8	1.82E-07	-6.74	1.18E-08	0.980
CAP-P4	Chlorapatite	3.99	5.06	5	5.78E-09	-8.24	6.97E-10	0.965
CAP-P4D	Chlorapatite	4.00	4.87	5	8.70E-09	-8.06	1.20E-09	0.955
CAP-P5	Chlorapatite	4.97	5.98	6	1.34E-09	-8.87	9.34E-11	0.965
CAP-P5D	Chlorapatite	5.00	5.69	6	1.80E-09	-8.74	2.76E-10	0.914
FAP-P2D	Fluorapatite	1.97	1.95	6	2.67E-07	-6.57	9.20E-09	0.995
FAPM-P2	Fluorapatite	1.97	2.07	6	1.80E-07	-6.74	2.28E-08	0.938
FAP-P2	Fluorapatite	1.99	2.03	6	2.53E-07	-6.60	2.08E-08	0.973
FAPM-P3	Fluorapatite	3.00	3.22	6	2.51E-08	-7.60	1.90E-09	0.977
FAP-P3DD	Fluorapatite	3.00	3.09	9	3.06E-08	-7.51	2.56E-09	0.952
FAPC-P3D	Fluorapatite	3.00	3.02	6	2.38E-08	-7.62	2.04E-09	0.971
FAPC-P4	Fluorapatite	3.99	4.37	5	4.05E-09	-8.39	3.24E-10	0.981
FAP-P4DD	Fluorapatite	4.00	4.82	5	2.55E-09	-8.59	3.04E-10	0.958
FAPM-P4	Fluorapatite	4.00	4.24	6	6.67E-09	-8.18	8.01E-10	0.944
FAPM-P5	Fluorapatite	5.00	5.58	6	8.04E-10	-9.09	8.85E-11	0.953
FAPM-P5D	Fluorapatite	5.00	5.49	5	8.16E-10	-9.09	1.54E-10	0.901

**n* = number of points used in linear regression. **Standard error divided by substrate surface area and mineral stoichiometry

Study	Log Ksp*
Zhu, 2009	-18.95
Jaynes et al., 1998	-21.13
Lindsay, 1979	-21.84
Stumm and Morgan, 1996	-21.95
Chin and Nancollas, 1991	-23.00
Vieillard and Tardy, 1984	-23.09
Amjad et al., 1981(Amjad et al., 1984)	-23.10
Robie et al., 1979	-24.99
Elliot, 1994	-25.45
This study	-26.50
Harouiya et al., 2007	-29.40
Stefansson, 2001	-30.94
Valsami-Jones et al., 1989	-32.95

Table S7. Comparison of Log Ksp values from this study to literature

**Reaction:* $Ca_5(PO_4)_3F + 3H^+ = 5Ca^{2+} + 3HPO_4^{2-} + F$. Except for Zhu, 2009, and this study, data are compiled from Oelkers et al, 2009.

Supplementary Discussion

Mineral Stoichiometry

Stoichiometries calculated from EMP were used to confirm the mineral chemistry and results assume ideal hydrogen and oxygen values. With the exception of merrillite, mineral stoichiometries used for calculations in this study were based on the official conventions of the International Mineralogical Association (IMA). Our synthetic merrillite contained no sodium and EMP analyses confirmed calcium substitution. Thus, for this study a stoichiometry of 9.5 calcium (i.e. $Ca_{9.5}Mg(PO_4)_7$) was used in all calculations.

Whitlockite

IMA convention:	Ca ₉ Mg(PO ₃ OH)(PO ₄) ₆
EMP results:	$Ca_{9.1}Mg_{0.9}(P_{1.0}O_3OH)(P_{1.0}O_4)_6$

Merrillite

IMA convention:	Ca ₉ NaMg(PO ₄) ₇
EMP results:	Ca _{9.0} Ca _{0.4} Mg _{1.1} (P _{1.0} O ₄) ₇
For this study:	Ca _{9.5} Mg(PO ₄) ₇

Chlorapatite

IMA convention:	Ca ₅ (PO ₄) ₃ Cl
EMP results:	Ca _{5.0} (P _{1.0} O ₄) ₃ Cl _{1.1}

Fluorapatite

IMA convention:	Ca ₅ (PO ₄) ₃ F
EMP results:	Ca _{5.0} (P _{1.0} O ₄) ₃ F _{1.1}

Electron Microprobe (EMP) Results

EMP Operating Conditions

Beam diameter: 10 microns Accelerating Voltage: 20 KeV Current: 10 nA

Results Summary (*n* represents the number of individual grains analyzed).

Whitlockite:

Element	F	MnO	Cl	P_2O_5	FeO	CaO	MgO	Total
Ave. Wt. %	0.01	0.00	0.01	46.79	0.02	47.57	3.33	97.71*
Std. Dev.	0.02	0.01	0.01	0.26	0.03	0.27	0.37	0.60
n =	17							

*Low totals due to the presence of undetectable/unaccounted for hydrogen within whitlockite $(Ca_{9}Mg(PO_{3}OH)(PO_{4})_{6})$

Merrillite:

Element	F	MnO	Cl	P₂O₅	FeO	CaO	MgO	Total
Ave. Wt. %	0.00	0.01	0.01	46.38	0.02	48.89	4.06	99.36
Std. Dev.	0.00	0.01	0.01	0.32	0.03	0.21	0.12	0.36
n =	16							

Chlorapatite:

Element	F	MnO	Cl	P_2O_5	FeO	CaO	MgO	-O = F,Cl	Total
Ave. Wt. %	0.03	0	7.23	40.28	0.14	53.76	0.12	1.64	99.92
Std. Dev.	0.03	0.01	1.18	0.91	0.83	1.21	0.14	-	0.81
n =	40*								

*Analyses were performed on grains from four individual batches of synthetic apatite.

Fluorapatite:

Element	F	MnO	Cl	P_2O_5	FeO	CaO	MgO	-O = F,Cl	Total
Average	3.75	0.02	0.47	40.91	0.03	54.11	0.02	1.69	97.62*
Std. Dev.	0.17	0.01	0.02	0.19	0.01	0.23	0.01	-	0.44
n =	5*								

*Durango Fluorapatite is well characterized. Low totals reflect ~2.00 wt % undetectable/unanalyzed components including; SrO, REE₂O₃, ThO₂, As₂O₃, V₂O₃, CO₂, SO₃ (Jarosewich, 1980).

Supplementary Data

Data Sheets for Individual Batch Experiments

Datasheet S1

Experiment CAP-P2

Mineral: Chlor	apatite Initial pH: 1.	97 Specific Surf. Area (B	ET): 0.0563m ² /g
Batch ID: CAP-	2 Mineral Mass: 0.1	500g Total Surf. Area (BET)): 0.00845 m ²

Fime (sec)	Ca (moles released)	P (moles released)	рΗ			Ĩ	nitial Ca Release	
0	0	NA	1.97		1.E-04	T .	y = 2.53E-08x + 3.5	6E-05
840	5.55E-05	NA	2,06	9	1.E-04		R ² = 9,86E-01	/
1380	7.18E-05	NA	2.06	asec	1.E-04			
2100	9.22E-05	NA	2.06	ele	8.E-05		/	
2820	1.02E-04	NA	2.07	2	6.E-05		1	
3600	1.29E-04	NA	2.06	om	1 5 05		•	
4980	1.26E-04	NA.	2.07	Ca (4,6703			
9300	1.36E-04	NA	2.12	-50	2.E-05			
81360	6.36E-04	NA	2.61		0.E+00	i		
						0	2000	4000
							Time (sec)	
				Reg	ression A	nalys	īs	Std.Ern

	Regression Analysis		Sta.Error
	R Square	0.986	NA
$K_{das} = r_{co}/A \cdot S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is	Ca release rate (r _G)	2.53E-08	1.76E-09
the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca	Diss Poto Minoral / V	5 005 07	mal/m ² r
stoiciometric coefficient of the respective mineral	Diss. Rate Willeral (Nam)	5.99E-07	molymis
succoncernent of the respective initiality	Uncertainty of Fit	4.15E-08	mol/m ² s

Experiment CAP-P2D

	19010		1100					
atch Result	s (shaded data =	the linear initial i	ate dat	a shown in th	e accomp	anying figu	re)	
Time (sec)	Ca (moles released)	P (moles released)	рН			Initial (Ca Release	
0	0	NA	2.00		2.E-04	y = 2.	07E-08x + 5.75E-	05
660	6.24E-05	4.56E-05	2.01	-	2.E-04		R ² = 9,39E-01	
1380	9.94E-05	6.11E-05	2.03	sec	1.E-04		6	
2220	1.08E-04	6.91E-05	2.09	ele	1.E-04		2	
2880	1.12E-04	7.00E-05	2.01	es	8.E-05			
3600	1.25E-04	7.79E-05	2.03	lou	6.E-05	6		
5220	1.70E-04	1.04E-04	2.06) e	4.E-05			
83280	5.62E-04	NA	2.42	U	2.E-05			
342480	7.08E-04	NA	2.72		0.E+00			-
7168080	8.78E-04	NA	3.45			20	4000 4000	6000
7686480	8.75E-04	NA	3.47				Time (sec)	
K _{aics} = r _{cs} /A• over time as the total BET (BET) X mine stoiciometric	S, where r _{ex} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the s	ate of Ca release ar regression, A is ific surface area he Ca respective mineral.		Diss, F	juare elease rat Rate Mine	e (Fa) eral (Kaiss)	0.939 2.07E-08 4.91E-07	NA 2.63E- mol/m ² s
K _{ettes} = r _c /A• over time as the total BET (BET) X mine stoiciometric toichiometri	S, where r _{ex} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the r ry	ate of Ca release ar regression, A is ific surface area he Ca respective mineral.		R So Car Diss. F	uare elease rat Rate Mine Uncerta	e (Fo) eral (Kaus) ainty of Fit	0.939 2.07E-08 4.91E-07 6.2298E-08	NA 2.63E-(mol/m ² s mol/m ² s
K _{aice} = r _{ca} /A• over time as the total BET (BET) X mine stoiciometric toichiometri 10les of M	S, where r_{ex} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the ry lineral Released	ate of Ca release iar regression, A is ific surface area he Ca respective mineral.		R So Car Diss. F	uare elease rat Rate Mine Uncerta	e (Fa) eral (Kaiar) ainty of Fit	0.939 2.07E-08 4.91E-07 6.2298E-08	NA 2.63E-1 mol/m ² s mol/m ² s
$K_{alice} = r_{co}/A \cdot$ over time as the total BET (BET) X mine stoiciometri toichiometri foles of M Fime (sec)	S, where r _{er} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the ry lineral Released Moles Mineral Released (P)	ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca)		R Sc Car Diss. F	iuare elease rat Rate Mine Uncerta Stoicl	e (F⊶) eral (K±uu) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 y Compariso	NA 2.63E-(mol/m ² s mol/m ² s
$K_{aixs} = r_{cs}/A \cdot$ over time as the total BET (BET) X mine stoiciometri toichiometri toichiometri foles of M Fime (sec) 660	S, where r_{ar} is the r determined by line Fsurface area (spec eral mass), and S is t c coefficient of the ry lineral Released Moles Mineral Released (P) 1.52E-05	ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05];	R So Car Diss. F	uare elease rat Rate Mine Uncerta Stoic 4.E-05 ↓	e (Fo) eral (Kaior) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 y Compariso	NA 2.63E-1 mol/m ² s mol/m ² s DN
K _{aiss} = r _{co} /A+ over time as the total BET (BET) X mine stoiciometri toichiometri toichiomet Moles of M Time (sec) 660 1380	S, where r_{ex} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05	ate of Ca release car regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05		R Sc Car Diss. F	elease rat Rate Mine Uncerta Stoic 4.E-05 4.E-05	e (Foi) eral (Kaior) ainty of Fit hiometr	0.939 2.07E-08 4.91E-07 6.2298E-08 y Compariso	NA 2.63E-1 mol/m ² s mol/m ² s
K _{alca} = r _{co} /A+ over time as the total BET (BET) X mine stoiciometri toichiometri toichiomet foles of M Time (sec) 660 1380 2220	S, where r_{ex} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the i ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05 2.30E-05	ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05 2.16E-05		R So Car Diss. F	auare elease rat Rate Mine Uncerto Stoic & C 4.E-05 3.E-05 3.E-05	e (r _o) ainty of Fit hiometr	0.939 2.07E-08 4.91E-07 6.2298E-08 y Compariso	NA 2.63E-1 mol/m ² s mol/m ² s
K _{aics} = r _{cs} /A+ over time as the total BE (BET) X mine stoiciometri toichiometri foles of M Time (sec) 660 1380 2220 2880	S, where r_{cr} is the r determined by line r surface area (spec eral mass), and S is t c coefficient of the ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05 2.30E-05 2.33E-05	ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05 2.16E-05 2.24E-05		R Sc Car Diss. F	aluare elease rat Date Mine Uncerta Stoic & C 4.E-05 4.E-05 3.E-05 3.E-05 3.E-05	e (Fa) eral (Kaiaa) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 9 Compariso • Phosphorus	NA 2.63E-1 mol/m ² s mol/m ² s
$K_{aixe} = r_{co}/A \cdot$ over time as the total BE: (BET) X mine stoiciometri toichiometri toichiometr Aoles of M Time (sec) 660 1380 2220 2880 3600	S, where r_{ex} is the r determined by line Furface area (spec eral mass), and S is t c coefficient of the ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05 2.30E-05 2.33E-05 2.60E-05	ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05 2.16E-05 2.24E-05 2.49E-05		R Sc Car Diss. F	elease rat Rate Mine Uncerta Stoic 4.E-05 4.E-05 3.E-05 3.E-05 2.E-05	e (Fo) eral (Kaior) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 y Compariso	NA 2.63E-1 mol/m ² s mol/m ² s
K _{aiss} = r _{co} /A+ over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri toichiomet Aoles of M Time (sec) 660 1380 2220 2880 3600 5220	S, where r_{st} is the r determined by line Furface area (spec eral mass), and S is t c coefficient of the r ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05 2.30E-05 2.33E-05 2.60E-05 3.45E-05	Ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05 2.16E-05 2.24E-05 2.49E-05 3.39E-05	;	R So Car Diss. F	uare elease rat <i>Uncerta</i> Stoic 4.E-05 4.E-05 3.E-05 3.E-05 2.E-05 2.E-05 2.E-05	e (Fai) eral (Kaiar) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 V Compariso	NA 2.63E-1 mol/m ² s mol/m ² s
K _{aics} = r _{co} /A+ over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri toichiomet (oles of M Time (sec) 660 1380 2220 2880 3600 5220	S, where r_{st} is the r determined by line Furface area (spec eral mass), and S is t c coefficient of the i ry lineral Released Moles Mineral Released (P) 1.52E-05 2.04E-05 2.30E-05 2.30E-05 3.45E-05	Ate of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 1.25E-05 1.99E-05 2.16E-05 2.24E-05 2.49E-05 3.39E-05		R So Car Diss. F	uare elease rat <i>Uncerta</i> Stoic 4.E-05 4.E-05 3.E-05 3.E-05 2.E-05 2.E-05 1.E-05	e (Fai) eral (Kaiar) ainty of Fit hiometr alcium	0.939 2.07E-08 4.91E-07 6.2298E-08 V Compariso	NA 2.63E-1 mol/m ² s mol/m ² s

Experiment CAP-P3

Mineral: C Batch ID: C	hlorapatite AP-P3 Min	Initial pH: 2 eral Mass: 0.1	.95 500g	Specific Surf. Area Total Surf. Area (B	(BET): 0.0	0563 m²/g 845 m²	
Batch Results	(shaded data =	the linear initial	rate dat	a shown in the accom	panying fig	(ure)	
Time (sec)	Ca (moles released)	P (moles released)	pН		Initia	l Ca Release	
0	0.00E+00	NA	2.95	5.E-05	<u>ा</u> भ	= 1.11E-08x + 2.15	E-06
900	1.31E-05	NA	3.10	T 4 5-05		R ² = 9.76E-01	*
1500	1.63E-05	NA	3.09	se			
2220	2.95E-05	NA	3.10	- 3.E-05	8	•	
2940	3.40E-05	NA	3.11	8			
3660	4.29E-05	NA	3.16	6 2.E-05	·	\	
5040	3.30E-05	NA	3.21	5 1.E-05	8		
5760	3.99E-05	NA	3.31				
81600	1.20E-04	NA	4.48	0.E+00	1000 000000000000000000000000000000000	1.512(22)	Netters of
					0	2000	4000
						Time (sec)	
	· 2	N 8		Regression	Analysis		Std.Error
K _{diss} = I _{co} /A•S, over time as d	where <i>t_{co}</i> is the	rate of Ca release ear regression A i	2	R Square		0.976	NA
the total BET	surface area (spe	cific surface area	5	Ca release r	ate (rai)	1.11E-08	1.01E-09
(BET) X minera stoiciometric	al mass), and S is coefficient of the	the Ca respective minera	il	Diss. Rate Mi	neral (Kas	2.64E-07	nol/m²s

Experiment CAP-P3DD

Mineral: (Chlorapatite	Initial pH: 2.	95 1500a	Specific Su	rf. Area	(BET): 0.0)563 m²/g	
NA = no anal	vsis	ierar iviass. U.	12008	Total Sull.	Alea (Di	-17: 0.000	11.042	
Batch Results	, s (shaded data = t	he linear initial i	rate da	ta shown in th	e accom	oanying fig	ure)	
Time (sec)	Ca (moles released)	P (moles released)	рН			Initia	l Ca Release	
0	0	NA	3.00		8.E-05	v = 7.70E	-09x + 1.00E-05	
900	1.89E-05	1.43E-05	3.03		7.E-05	R ²	= 9.80E-01 🔶	
1860	2.39E-05	1.73E-05	3.14	sed	6.E-05			
2700	3.11E-05	2.17E-05	3.05	elea	5.E-05		1	
3660	3.76E-05	2.63E-05	3.17	2	4.E-05		**	
4560	4.14E-05	2.93E-05	3.22	hor	3.E-05			
5400	5.34E-05	3.36E-05	3.27	a (2.E-05	×~		
6360	5.63E-05	3.86E-05	3.27		1.E-05			
7200	6.90E-05	4.07E-05	3.33		0.E+00	<u> </u>		
				-		0	5000	10000
							Time (sec)	
	685 - 87	8 8	-	Reg	ression A	nalysis		Std.Error
$K_{das} = r_{co}/A \cdot S$	S, where r _{ca} is the ra	ate of Ca release	8	R So	uare		0.976	NA
the total BET	surface area (speci	fic surface area	5	Ca	elease ra	te (ra)	7.70E-09	4.53E-10
(BET) X mine stoiciometric	ral mass), and S is the coefficient of the r	ne Ca espective mineral	;	Diss. I	Rate Min	eral (Kıtısı) 1.82E-07	mol/m²s
Stoichiometr	Y				Uncer	tainty of Fi	t 1.07E-08 r	mol/m s
Moles of Mi	ineral Released		-					
Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)			Stoichi	ometry	Comparison	
() <u>1.515</u> 56	1705.00	3 705 00	-		Pho	sphate	Calcium	

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)		
900	4.76E-06	3.78E-06		
1860	5.77E-06	4.79E-06		
2700	7.25E-06	6.22E-06		
3660	8.76E-06	7.53E-06		
4560	9.76E-06	8.27E-06		
5400	1.12E-05	1.07E-05		
6360	1.29E-05	1.13E-05		
7200	1.36E-05	1.38E-05		

determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

*Analytical error bars fall within plotted symbols

Phosphate Calcium 2.E-05 1.E-05 1.E-05 1.E-05 8.E-06 each of the second s



0

0.E+00

Experiment CAP-P4

Mineral: C Batch ID: C	hlorapatite AP-P4 Min	Initial pH: 3. eral Mass: 0.1	.99 500g	Specific Su Total Surf.	rf. <mark>A</mark> rea (I Area (BE)	BET): 0.	0563 m²/g 1845 m²	
IA = no analy	sis			and the second state of th	*****	•		
atch Results	(shaded data =	the linear initial	rate da	ta shown in th	e accompo	inying fi	gure)	
Time (sec)	Ca (moles released)	P (moles released)	рН			Initia	al Ca Release	
0	0	NA	3.99		9.E-06	y = 2	2.44E-10x + 4.40E-	06
3180	4.88E-06	NA	4.39	-	8.E-06	54	R ² = 9.65E-01	
6780	6.23E-06	NA	4.55	sed	7.E-06		2	
9360	6.97E-06	NA	4.75	elea	0.E-00	1		
13620	7.73E-06	NA	4,79	- Sa	4.E-06	9		
16500	8.23E-06	NA	5.06	lon	3.E-06			
20220	8.21E-06	NA	5.48	ca (2.E-06			
96960	9.38E-06	NA	5.89		1.E-06			
1559460	1.14E-05	NA	6.15		0.E+00			
8398680	1.26E-05	NA	6.63		0)	10000	20000
				57 16			Time (sec)	
	16 G. Van	N 552 M		Reg	ression An	alysis		Std.Error
$K_{diss} = r_{es} / A \cdot S,$	where r _{ca} is the	rate of Ca release		R Sc	uare		0.965	NA
the total BET	surface area (spe	cific surface area		Car	elease rati	e (Fai)	2.44E-10	2.68E-11
(BET) X miner stoiciometric	al mass), and S is coefficient of the	the Ca respective minera	u	Diss. H	ate Mine	ral (Kee	.) 5.78E-09	mol/m²s
					Uncerta	inty of F	it 6.33E-10	mol/m ² s

Experiment CAP-P4D

(A - n - n - n - n - n - n - n - n - n -	vicio							
A = no una) atch Pocult	ysis r Ichadod data -	the linear initial r	ato dal	a chown in th	a accomo	anvina fiau		
uten nesure	Ca Imolar	D Imalas	ute uu		ie uccomp	unying jigui	6/	
Time (sec)	released)	released)	pН			Initial C	a Release	
0	0	NA	4:00		1.E-05		675 10x 1 5 14	-06
2990	5.915.05	2 525.06	4.00		1 5 65	y=3	$R^2 = 9.55E-01$	E-00
5460	7.465.06	4.295.06	4.05	(pa	1.E-05			
0160	0.375.06	4.552-00	104	eas	8.E-06			
10960	0.375-00	5.62E-00	4.64	e e	6.E-06	6		
10000	9.322-00	5.005.00	4.00	ole	varianaco			
13440	5.752-00	0.00E-00	4.67	E.	4.E-06			
18/80	1.02E-05	6.38E-06	5.08	ů	2.E-06			
10/280	1.17E-05	7.74E-06	5.86		0.E+00			
543120	1.362-05	NA	6.31		5	0 50	00 10000	15000
						0 30	10000	13000
							Time (sec)	
				· · · ·				
K =1 /A.	S where r is the r	ate of Ca releace	-	Reg	ression Ar	nalysis		Std.Error
$K_{diss} = r_{co}/A \bullet$ over time as	S, where <i>r_{ca}</i> is the r determined by line	ate of Ca release ar regression, A is		Reg R Sc	ression Ar quare	nalysis	0.955	Std.Error NA
$K_{cles} = r_{co}/A \bullet$ over time as the total BE	S, where r _{ca} is the r determined by line I surface area (spec	ate of Ca release ar regression, A is ific surface area		Reg R Sc Ca i	ression Ar quare release rat	nalysis :e (r _{ei})	0.955 3.67E-10	Std.Error NA 4.61E-
$K_{diss} = r_{co}/A \bullet$ over time as the total BE (BET) X mine	S, where r_{co} is the r determined by line F surface area (spec ral mass), and S is t	ate of Ca release ar regression, A is ific surface area the Ca		Reg R Sc Ca i	ression Ar quare release rat	nalysis ce (r _a .)	0.955 3.67E-10	Std.Error NA 4.61E-:
K _{diss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri	S, where r_{co} is the r determined by line T surface area (spec rral mass), and S is t c coefficient of the	ate of Ca release aar regression, A is ific surface area the Ca respective mineral.		Reg R Sc Car Diss. H	ression Ar quare release rat Rote Mine	nalysis te (ra) eral (Kaus)	0.955 3.67E-10 8.70E-09	Std.Error NA 4.61E-: mol/m ² s
K _{dies} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri	S, where r_{co} is the r determined by line F surface area (spec ral mass), and S is t c coefficient of the	ate of Ca release ar regression, A is ific surface area the Ca respective mineral.		Reg R Sc Car Diss. 1	ression Ar quare release rat Rate Mine Uncerti	alysis e (۲۵۰) eral (۲۵۰۶) ainty of Fit	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s
K _{etes} = r _{co} /A• over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri	S, where r_{co} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the F y	ate of Ca release ar regression, A is ific surface area the Ca respective mineral.		Reg R Sc Car	rression Ar quare release rat Rote Mine Uncerti	nalysis ee (r==) eral (K===) ainty of Fit	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s
K _{oles} = r _{co} /A• over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri Moles of M	S, where r _{ca} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the ry ineral Released	ate of Ca release ar regression, A is ific surface area the Ca respective mineral.		Reg R Sc Car Diss. F	ression Ar quare release rat Rote Mine Uncerti	nalysis ee (r _{ei}) eral (K _{diss}) ainty of Fit	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M	S, where r _{co} is the r determined by line I surface area (spec ral mass), and S is t c coefficient of the ry ineral Released	ate of Ca release ar regression, A is ific surface area the Ca respective mineral.		Reg R Sc Car	ression Ar guare release rat Rote Mine Uncerti	nalyšis se (r _{ci}) eral (K _{dess}) ainty of Fit	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s
$K_{diss} = r_{co}/A \cdot$ over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M	S, where r_{co} is the r determined by line F surface area (spec eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P)	ate of Ca release ar regression, A is ific surface area the Ca respective mineral. Moles Mineral Beleased (Ca)		Reg R Sc Car Diss. F	ression Ar guare release rat Rate Mine Uncerti Stoichic	nalysis eral (Kau) ainty of Fit	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M Time (sec)	S, where r_{co} is the r determined by line r surface area (spec eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P)	ate of Ca release ar regression, A is ific surface area the Ca respective mineral. Moles Mineral Released (Ca)		Reg R Sc Car Diss. F	ression Ar quare release rat Rate Mine Uncerti Stoichic	nalysis eral (Keus) ainty of Fit ometry C	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880	S, where r_{co} is the r determined by line r surface area (spec eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Moles Mineral Released (Ca) 1.16E-06		Reg R Sc Car Diss. I	ression Ar quare release rat Rate Mine Uncerti Stoichic Stoichic Calci E-06	nalysis eral (Kaua) ainty of Fit ometry C ium E P	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460	S, where r_{co} is the r determined by line F surface area (spec- eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Moles Mineral Released (Ca) 1.16E-06 1.49E-06		Reg R Sc Car Diss. I	ression Ar guare release rat Rote Mine Uncerti Stoichic & Calci E-06	nalysis eral (Kaus) ainty of Fit ometry C	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eles} = r _{co} /A• over time as the total BE ^T (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460 8160	S, where r_{co} is the r determined by line F surface area (spec- eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06	ate of Ca release tar regression, A is ific surface area the Ca respective mineral. Moles Mineral Released (Ca) 1.16E-06 1.49E-06 1.67E-06		Diss. F	ression Ar guare release rat <i>Rate Mine</i> <i>Uncerti</i> Stoichic Stoichic E-06 E-06	nalysis eral (Kau) ainty of Fit ometry C	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Aoles of M Time (sec) 2880 5460 8160 10860	S, where r_{co} is the r determined by line r surface area (spectral mass), and S is to coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral. Moles Mineral Released (Ca) 1.16E-06 1.49E-06 1.67E-06 1.86E-06		Diss. F	ression Ar guare release rat Rate Mine Uncerti Stoichic & Calci E-06 E-06 E-06	nalysis eral (Kduar) ainty of Fit ometry C ium	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460 8160 10860 13440	S, where r_{co} is the r determined by line r surface area (spec- eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06 1.86E-06 2.00E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Moles Mineral Released (Ca) 1.16E-06 1.49E-06 1.86E-06 1.86E-06		Diss. F Diss. F 2. 2.	ression Ar guare release rat <i>Rate Mine</i> <i>Uncerti</i> Stoichic • Calci E-06 E-06 E-06	nalysis eral (Kdus) ainty of Fit ometry C ium	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460 8160 10860 13440	S, where r_{co} is the r determined by line r surface area (spec- eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06 1.86E-06 2.00E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Moles Mineral Released (Ca) 1.16E-06 1.49E-06 1.67E-06 1.86E-06 1.95E-06		Diss. I Diss. I Diss. I 1.	ression Ar quare release rat Rate Mine Uncerti Stoichic & Calci E-06 E-06 E-06 E-06	nalysis eral (Kaua) ainty of Fit ometry C ium P	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eiss} = r _{co} /A• over time as the total BE' (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460 8160 10860 13440 Moles miner	S, where r_{co} is the r determined by line r surface area (spec- eral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06 1.86E-06 2.00E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Released (Ca) 1.16E-06 1.49E-06 1.67E-06 1.86E-06 1.95E-06 based on P or Ca		Reg R Sc Car Diss. F 3. 2. 2. 1.	ression Ar quare release rat Rate Mine Uncerta Stoichic & Calci E-06 E-06 E-06 E-06	nalysis eral (Kaus) ainty of Fit ometry C ium P	0.955 3.67E-10 8.70E-09 1.09E-09 Comparison hosphorus	Std.Error NA 4.61E- mol/m ² s mol/m ² s
K _{eles} = r _{co} /A• over time as the total BE ⁻ (BET) X mine stoiciometri toichiometri Moles of M Time (sec) 2880 5460 8160 10860 13440 Moles miner determined	S, where r_{co} is the r determined by line F surface area (spec- eral mass), and S is t c coefficient of the Moles Mineral Released (P) 1.18E-06 1.46E-06 1.87E-06 1.86E-06 2.00E-06	ate of Ca release ar regression, A is ific surface area the Ca respective mineral Released (Ca) 1.16E-06 1.49E-06 1.67E-06 1.86E-06 1.95E-06 based on P or Ca d of P or Ca divideo		Reg R Sc Car Diss. F 3. 3. 2. 2. 1. 5.	ression Ar guare release rat Rate Mine Uncerts Stoichic E-06 E-06 E-06 E-06 E-06 E-06 E-06	nalysis eral (Kass) ainty of Fit ometry C ium	0.955 3.67E-10 8.70E-09 1.09E-09	Std.Error NA 4.61E-: mol/m ² s mol/m ² s

*Analytical error bars fall within plotted symbols

Time (sec)

Experiment CAP-P5

Batch ID: C	hlorapatite AP-P5 Mine	Initial pH: 4. eral Mass: 0.1	.95 500g	Specific Su Total Surf.	rf. Area (Area (BE	BET): 0.0 T): 0.008	563 m² 45 m²	/g	
VA = no analy. Batch Results	sis (shaded data =	the linear initial	rate da	a shown in th	e accomo	anvina fia	urel		
Time (sec)	Ca (moles released)	P (moles released)	рН	u shown in th	c vccomp	Initial	Ca Rel	ease	
0	0	NA	4.97		2.E-06	(= 5.68E-1	1x + 7.2	5E-07	
3120	8.56E-07	NA	5.34	=		R ² = 5	9.84E-01	1	
6660	1.11E-06	NA	5.46	asec	2.E-06		*		
9300	1.33E-06	NA	5.73	ele			X		
13560	1.50E-06	NA	5.77	esr	1.E-06	6			
16440	1.61E-06	NA	5.78	Ē					
20160	1.88E-06	NA	5.98	e	5.E-07				
96960	3.14E-06	NA	6.43		4200 million (10.525)				
1559460	4.84E-06	NA	6.62	-	0.E+00		2000	20000	20000
					1	1	Time	(sec)	30000
085 660 -	- 10 - 115 - 206	10 mm at		Reg	ression Ar	alysis			Std.Error
$K_{stas} = r_{co} / A \cdot S_{s}$	where <i>r_{co}</i> is the	rate of Ca release	2	R Se	quare		0	.984	NA
the total BET	surface area (spec	cific surface area	5	Car	elease rat	e (r⊶)	5.6	8E-11	3,59E-1
(BET) X miner stoiciometric	al mass), and S is coefficient of the	the Ca respective minera	ı	Diss, I	Rate Mine	eral (Kzaz)	1.3	4E-09	mol/m²s

Experiment CAP-P5D

Mineral: 0 Batch ID: 0	Chlorapatite CAP-P5D Mi i	Initial pH: 4.9 neral Mass: 0.1	5 500g	Specific Su Total Surf.	rf. Are Area (a (BET) BET): (: 0.056).00845	3 m²/g m²	
NA = no anal	ysis					<u> </u>			
Batch Results	; (shaded data =	the linear initial r	ate dat	a shown in th	he accol	mpanyir	ng figure)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Ir	nitial Ca	a Release	
0	0	NA	4.95		3.E-0	6	y = 6	6.91E-11x + 1	L.00E-06
3000	1.01E-06	5.93E-07	5.48	-	2 5 0	6		$R^2 = 9.14E-$	01
5580	1.52E-06	5.93E-07	5.65	Ised	2.6-0			120	
8340	1.68E-06	6.66E-07	5.40	elee	2.E-0	6	•		
10980	1.81E-06	6.89E-07	5.84	es r			/		
13560	1.89E-06	7.53E-07	5.88	lou	1.E-0	0	•		
18960	2.26E-06	9.50E-07	5.69	5	5.E-0	7			
107460	3.58E-06	1.95E-06	6.37			5 			
over time as the total BET (BET) X mine stoiclometric	, where r _{co} is the r determined by line surface area (spec ral mass), and S is t coefficient of the	are of carease ar regression, A is ific surface area the Ca respective mineral.		R S Ca Diss. I	quare release R <i>ate M</i>	rate (r⇔ ineral () 'Kaiss)	0.914 6.91E-11 1.64E-09	NA 1.06E-11 mol/m ² s
toichiometr Aoles of M	y ineral Released				<i>Unc</i> i Analytic	<i>ertainty</i> al error	<i>of Fit</i> bars fall	2.51E-10 within plot	mol/m²s symbols
Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)			Stoic	hiom	etry C	ompariso	on
3000	1.98E-07	2.03E-07			F F	hospho	rus	Calcium	
5580	1.98E-07	3.04E-07		5	.E-07	1			•
8340	2.22E-07	3.36E-07		o ase o	.E-07				
10980	2.30E-07	3.63E-07		ala 4	.E-07			•	
13560	2.51E-07	3.79E-07		alk	.E-07		•		
18960	3.17E-07	4.51E-07		ner	E-07				
107460	6.51E-07	7.17E-07		ž ž	.E-07				

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.



Datasheet S9 Experiment FAP-P2

Time (sec)	Ca (moles released)	P (moles released)	pН			Ini	tial Ca R	elease	
0	0	NA	1.99		6.E-05	v =	1 22E-08x	+ 4.08E-06	
600	1.19E-05	1.53E-05	2.02	÷	5.E-05		R ² = 9.7	3E-01	
1260	1.53E-05	1.90E-05	1.99	ase	4 E-05			1	
1800	2.97E-05	2.20E-05	1.98	ele	T.L. 95		#	(
2460	3.46E-05	2.64E-05	2.03	es r	3.E-05		>		
3120	4.29E-05	3.32E-05	2.03	Dom	2.E-05		/		
4140	5.33E-05	NA	2.03	č	1 5.05	*	•		
81300	1.40E-04	NA	2.18		1.000				
242460	2.04E-04	NA	2.21		0.E+00	~	5-532.VA		
593220	3.38E-04	NA	2.27			0	2000	4000	6000
850500	3.47E-04	NA	2.3				Tim	e (sec)	
1988100	3.61E-04	NA	2.2	Reg	ression A	nalysis			Std.Error
2517480	3.62E-04	NA	2.3	R So	quare			0.973	NA
$K_{diss} = r_{co}/A \cdot S_{co}$ over time as the total BET (BET) X miner	, where <i>r_{ca}</i> is the ra determined by linea surface area (specif	te of Ca release r regression, A is ic surface area e Ca		Can Diss. I	release ra Rate Min	te (r∝) eral (ł	1. (diss) 2 .	2199E-08 53E-07	1.01E-09 mol/m ² s
stoiciometric	coefficient of the re	spective mineral.			Uncer	tainty o	f Fit 2.	11E-08	mol/m²s
oichiometr	/				Analutica	lerror	hars fall w	ithin plot s	mbols

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
600	5.11E-06	2.38E-06
1260	6.32E-06	3.05E-06
1800	7.32E-06	5.94E-06
2460	8.79E-06	6.92E-06
3120	1.11E-05	8.59E-06
4140		1.07E-05

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

Stoichiometry Comparison



Datasheet S10 **Experiment FAP-P2D**

Mineral: F Batch ID: F	luorapatite AP-P2D Min	Initial pH: 1. eral Mass: 0	97 3000g	Specific Sur Total Surf.	rf. Area Area (B	(BET): ET): 0.	0.0321 n 00963 m ²	n²/g	
IA = no analy latch Results	vsis (shaded data = ti	he linear initial	rate da	a shown in th	e accom	panying	figure)		
Time (sec)	Ca (moles released)	P (moles released)	pН			Ini	tial Ca R	elease	
0	0	NA	1.97	7	7.E-05		-1 285.0	8v ± 2 11E-0	s
780	1.21E-05	NA	1.98	-	6.E-05	y y	$R^2 = 9$.95E-01	
1380	1.86E-05	NA	1.98	sec	5.E-05			1	
1980	2.76E-05	NA	2.00	ele	4.E-05		- 7	1	
2580	3.60E-05	NA	2.02	es r	3 E-05		1		
3300	4.63E-05	NA	2.03	non	3 5.05		1		
4500	5.84E-05	NA	1.95	J	2.1-05		~		
11520	6.61E-05	NA	2.01	274	1.E-05				
96060	1.48E-04	NA	2.09	<u>_</u>	0.E+00		2000	4000	
						U	2000 Tim	4000 e (sec)	6000
			_	Reg	ression A	Analysis			Std.Error
$K_{diss} = r_{co}/A \bullet S$, where r_{co} is the radius of the main of the lines.	ite of Ca release		R Sc	juare		0	0.995	NA
the total BET	surface area (specif	fic surface area	2	Car	elease ra	ate (r∞)		1.28E-08	4.49E-10
(BET) X miner stoiciometric	al mass), and S is the coefficient of the re	ie Ca espective minera	6		Analytic	al error	bars fall v	vithin plot sy	mbols
				Diss. F	ate Mir Uncer	neral () tainty a	(diss) 2 .	67E-07 r 31E-09 r	nol/m*s nol/m²s

Datasheet S11 **Experiment FAPM-P2**

NA = no analy Batch Results	sis (shaded data = ti	he linear initial ra	ite data s	shown in th	e accompar	ving figure)	117	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initia <mark>l</mark> Ca	Release	
0	0	NA	1.97		8.E-05			
960	2.46E-05	1.79E-05	1.95	9	7.E-05	y =	8.68E-09x + 1.	465-05
1980	3.14E-05	2.50E-05	1.96	sec	6.E-05		R ² = 9.38E-0	1
2820	4.05E-05	3.08E-05	1.97	elea	5.E-05			
3660	4.43E-05	3.11E-05	1.95	esr	4.E-05	1	*	
4440	4.75E-05	3.55E-05	2.02	D C C C C C C C C C C C C C C C C C C C	3.E-05	-		
5460	6.71E-05	4.85E-05	2.07	J	2.E-05	100		
15420	9.52E-05	6.32E-05	2.03		1.E-05			
99660	1.56E-04	NA	2.16		0.E+00	22000		0895
1346280	2.64E-04	NA	2.25		0	2000	4000	6000
			_			n	me (sec)	3
K -r /Ass	where r is the ra	te of Co release	1	Reg	ression Ana	lysis		Std.Error
over time as (determined by linea	ir regression, A is		R So	quare		0.938	NA
the total BET	surface area (specif	ic surface area		Car	elease rate	(ľ≃)	8.68E-09	1.11E-09
(BET) X miner stoiciometric	al mass), and S is th coefficient of the re	ie Ca espective mineral.		Diss. I	Rate Minero	al (Kaiss)	1.80E-07	mol/m²s
2			10.00	2997-2040 ⁻⁶ 0-60	Uncertair	nty of Fit	2.31E-08	mol/m ² s
Stoichiometry	1			Ē	Analytical	arror hars fal	l within plot o	umbale
Moles of Mir	neral Released			L	Milalytical		i within plots	VITIDUIS
Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)			Stoic	hiometry Phosphate	y Compari	ison
060	5 96E-06	4.91E-06			COLOR OF T		- Southerfully	



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Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
960	5.96E-06	4.91E-06
1980	8.34E-06	6.27E-06
2820	1.03E-05	8.10E-06
3660	1.04E-05	8.86E-06
4440	1.18E-05	9.50E-06
5460	1.62E-05	1.34E-05
15420	2.11E-05	1.90E-05

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

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Datasheet S12 Experiment FAPM-P3

Initial pH: 3.00 Specific Surf. Area (BET): 0.0321 m²/g Mineral: Fluorapatite Batch ID: FAPM-P3 Mineral Mass: 0.3000g Total Surf. Area (BET): 0.00963 m² NA = no analysis Batch Results (shaded data = the linear initial rate data shown in the accompanying figure) Ca (moles P (moles Time (sec) pH **Initial Ca Release** released) released) 1.E-05 0 0 NA 3.00 y = 1.21E-09x + 4.95E-06 1.E-05 960 6.363E-06 NA 3.26 R² = 9.77E-01 Ca (moles released) 1740 7.132E-06 3.08 NA 1.E-05 2700 7.561E-06 NA 3.11 8.E-06 3.16 3540 9.348E-06 NA 6.E-06 4440 1.043E-05 NA 3.20 4.E-06 5700 NA 3.22 1.190E-05 2.E-06 8160 1.311E-05 NA 3.19 0.E+00 0 2000 4000 6000 Time (sec) **Regression Analysis** Std.Error $K_{diss} = r_{co}/A \cdot S$, where r_{co} is the rate of Ca release **R** Square 0.977 NA over time as determined by linear regression, A is Ca release rate (r...) 1.21E-09 9.26E-11 the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiclometric coefficient of the respective mineral. mol/m²s Diss. Rate Mineral (Kaiss) 2.51E-08 Uncertainty of Fit mol/m²s 1.92E-09 Analytical error bars fall within plot symbols

Datasheet S13 Experiment FAP-P3DD

Mineral: F Batch ID: F	luorapatite AP-P3DD Mi i	Initial pH: 3.00 neral Mass: 0.3) 3000g	Specific <mark>S</mark> u Total Surf.	rf. Area (I Area (BE	BET): 0.03	21 m²/g 3 m²	
NA = no analy	isis							(8)
Batch Results	(shaded data = t	he linear initial re	ate dat	shown in t	he accompo	anying figur	e)	
Time (sec)	Ca (moles released)	P (moles released)	рН			Initial C	a Release	
0	0	NA	3.00		2.E-05		1 17E 00v ±	1 60E 06
1080 1920	6.24E+06 7.29E-06	NA NA	3.10 3.06	(pased)	2.E-05	· · ·	R ² = 9,52E	81 T
2880	8.23E-06	NA	3.13	elea			/	
3840	9.68E-06	NA	3.07	es r	1.E-05	2	~	
4860	1.39E-05	NA	3.08	om	Provide and shared a	-		
5820	1.37E-05	NA	3.09	č	5.E-06	Design of the second se		
6720	1.46E-05	NA	3.06					
7620	1.56E-05	NA	3.09		0.E+00		1880 M.C.	1). 11)22/2014
8820	1.71E-05	NA	3.09):	5000	10000
							Time (sec)	
জন আবের	5 AD - 35 AD - 35	S 20197 12		Reg	pression An	alysis		Std.Error
$K_{diss} = r_{co}/A \bullet S$, where <i>r_{ca}</i> is the ra determined by line:	ite of Ca release		RS	quare		0.952	NA
the total BET	surface area (speci	fic surface area		Ca	release rat	e (r.,)	1.47E-0	9 1.25E-10
(BET) X miner stoiciometric	al mass), and S is th coefficient of the re	ne Ca espective mineral.	->	Diss.	Rate Mine Uncerta	ral (K _{diss}) iinty of Fit	3.06E-08 2.59E-09	mol/m²s mol/m²s
				A	nalytical er	ror bars fall	within plot sy	mbols

Datasheet S14 Experiment FAPC-P3D

 Mineral:
 Fluorapatite
 Initial pH: 3.00
 Specific Surf. Area (BET): 0.0321 m²/g

 Batch ID:
 FAPC-P3D
 Mineral Mass: 0.3000g
 Total Surf. Area (BET): 0.00963 m²

 NA = no analysis
 NA
 NA
 NA

Batch Results (shaded data = the linear initial rate data shown in the accompanying figure)

Time (sec)	Ca (moles released)	P (moles released)	рН
0	0	0	3.00
1200	4.07E-06	3.48E-06	3.04
1920	5.91E-06	4.74E-06	3.07
2760	6.29E-06	5.09E-06	3.05
3420	6.98E-06	5.55E-06	3.07
4140	7.84E-06	5.69E-06	3.00
5760	9.70E-06	7.10E-06	3.02



 $K_{diss} = r_{co}/A \cdot S$, where r_{ca} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.

Stoichiometry

Moles of Mineral Released

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
1200	1.16E-06	8.14E-07
1920	1.58E-06	1.18E-06
2760	1.70E-06	1.26E-06
3420	1.85E-06	1.40E-06
4140	1.90E-06	1.57E-06
5760	2.37E-06	1.94E-06

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion. Analytical error bars fall within plot symbols

2.38E-08

2.07E-09

Diss. Rate Mineral (Kaiss)

Uncertainty of Fit

mol/m²s

mol/m²s



Datasheet S15 Experiment FAPC-P4

IA = no analy	sis							
Batch Results	(shaded data = th	he linear initial ro	ate data	shown in th	e accomp	anying figure	2)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial C	a Release	
0	0	NA	3.99		6.E-06	v-1	05E 10v + 1 6	PE 06
3540	2.20E-06	NA	4.14	÷	5.E-06	y - 1	R ² = 9.81E-01	A
7140	3.18E-06	NA	4.20	ased	4 E-06		×	
9720	3.73E-06	NA	4.26	ele	H.L 00		-	
13980	4.43E-06	NA	4.31	es	3.E-06			
16860	4.84E-06	NA	4.37	om	2.E-06	•		
20580	5.11E-06	NA	4.41	S	1 E-06			
97260	6.70E-06	NA	4.80		1.0.00			
					0.E+00			20020
						J	10000	20000
							Time (sec)	
an seed warde				Reg	ression An	alysis		Std.Error
$K_{diss} = r_{co}/A \bullet S,$	where r _{co} is the ra	te of Ca release		R So	quare		0.981	NA
the total BET	surface area (specif	ic surface area		Car	elease rat	e (r∝)	1.95E-10	1.58E-1
(BET) X miner	al mass), and S is th	e Ca		12				
stoiciometric	coefficient of the re	spective mineral.	->	Diss. F	late Mine	ral (Kaiss)	4.05E-09	mol/m ² s
			- · · ·					192 m.

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Datasheet S16 Experiment FAP-P4DD

 Mineral:
 Fluorapatite
 Initial pH: 3.98
 Specific Surf. Area (BET): 0.0321 m²/g

 Batch ID:
 FAP-P4DD
 Mineral Mass: 0.3000g
 Total Surf. Area (BET): 0.00963 m²

 NA = no analysis
 Na = no analysis
 Na = no analysis

Batch Results (shaded data = the linear initial rate data shown in the accompanying figure)

Time (sec)	Ca (moles released)	P (moles released)	рН
0	0	NA	3.98
3660	1.16E-06	NA	4.41
7200	1.22E-06	NA	4.72
10620	1.83E-06	NA	4.45
14160	2.25E-06	NA	4.58
17580	2.78E-06	NA	4.82
83520	3.99E-06	NA	5.60
187140	8.01E-06	NA	5.69
770940	8.64E-06	NA	6.33
1372260	9.00E-06	NA	6.07
1741860	1.03E-05	NA	6.25

 $K_{diss} = r_{co}/A \cdot S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.



Datasheet S17 Experiment FAPM-P4

 Mineral:
 Fluorapatite
 Initial pH: 4.00
 Specific Surf. Area (BET): 0.0321 m²/g

 Batch ID:
 FAPM-P4
 Mineral Mass: 0.3000g
 Total Surf. Area (BET): 0.00963 m²

 NA = no analysis
 NA
 NA
 NA

R Square

Ca release rate (ra)

Diss. Rate Mineral (Kdiss)

Uncertainty of Fit

Batch Results (shadea data = the linear initial rate data shown in the accompanying figu	Batch Results	(shaded data = the linear initial rate date	a shown in the accompanying figu	re)
--	----------------------	---	----------------------------------	-----

Time (sec)	Ca (moles released)	P (moles released)	pН
0	0	NA	4.00
1320	1.11E-06	4.15E-07	4.13
2400	2.00E-06	6.11E-07	4.07
3840	2.03E-06	8.48E-07	4.05
4980	2.72E-06	9.23E-07	4.18
6300	2.83E-06	1.04E-06	4.10
8700	3.71E-06	1.51E-06	4.24
92460	NA	2.79E-06	4.81



0.944

6.67E-09

8.11E-10

3.21E-10

NA

mol/m²s

mol/m²s

3.90E-11

 $k_{diss} = r_{co}/A \cdot S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stolciometric coefficient of the respective mineral.

Stoichiometry

Moles of Mineral Released

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
1320	1.38E-07	2.23E-07
2400	2.04E-07	4.01E-07
3840	2.83E-07	4.07E-07
4980	3.08E-07	5.45E-07
6300	3.46E-07	5.65E-07
8700	5.03E-07	7.41E-07
92460	9.31E-07	NA

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

Stoichiometry Comparison



Datasheet S18 Experiment FAPM-P5

 Mineral:
 Fluorapatite
 Initial pH:
 5.00
 Specific Surf. Area (BET):
 0.0321 m²/g

 Batch ID:
 FAPM-P5
 Mineral Mass:
 0.3000g
 Total Surf. Area (BET):
 0.00963 m²

 NA = no analysis
 Output
 Output
 Output
 Output
 Output

Batch Results (shaded data = the linear initial rate data shown in the accompanying figure)

Time (sec)	Ca (moles released)	P (moles released)	рН
0	NA	NA	5.00
3900	5.08E-07	1.88E-07	2.27
7440	5.85E-07	2.76E-07	5.30
11820	8.99E-07	4.41E-07	5.28
14820	9.45E-07	4.41E-07	5.58
18060	1.14E-06	5.14E-07	5.54
24360	1.24E-06	6.48E-07	5.58
103080	2.03E-06	NA	6.05
6319560	2.09E-05	NA	6.20



 $K_{diss} = r_{co}/A \bullet S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.

Stoichiometry

Moles of Mineral Released

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
3900	6.26E-08	1.02E-07
7440	9.19E-08	1.17E-07
11820	1.47E-07	1.80E-07
14820	1.47E-07	1.89E-07
18060	1.71E-07	2.28E-07
24360	2.16E-07	2.49E-07

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion. Uncertainty of Fit 8.9598E-11 mol/m²s

Diss. Rate Mineral (Kass)

Analytical error bars fall within plot symbols

8.04E-10

mol/m²s

Stoichiometry Comparison



Datasheet S19 Experiment FAPM-P5D

Mineral: F Batch ID: F	luorapatite APM-P5D	Initial pH: 4.9 Mineral Mass: 0.	1 Specific S 3000g Total Sur	Surf. Area (BET): 0.0321 m²/g f. Area (BET): 0.00963 m²
NA = no analy Batch Results	isis (shaded data	a = the linear initial re	ate data shown in	the accompanying figure)
1294 112 15	Calmoles	P (moles	220)	

Time (sec)	released)	released)	pН
0	0	NA	4.91
3600	7.22E-07	NA	5.14
7080	1.02E-06	NA	5.22
10920	1.11E-06	NA	5.31
14400	1.30E-06	NA	5,41
18420	1.32E-06	NA	5.49
101580	2.38E-06	NA	6.13
522960	3.18E-06	NA	6.15

 $K_{dasr} = r_{ca}/A \cdot S$, where r_{ca} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.



Datasheet S20 Experiment WHIT-P2DD

NA = no analy	sis							
Batch Results	(shaded data = th	he linear initial	rate data sh	own in th	e accompo	unying figu	re)	
Time (<i>sec</i>)	Ca (moles released)	P (moles released)	pН			Initial (Ca Release	
0	0	NA	1.98		3.E-04	3		
900	5.04E-05	NA	2.10	÷	3.E-04	γ = 2	.68E-08X - 2.93E p ² - 0.07E_01	-06
1800	5.82E-05	NA	2.07	asec	25.04		K = 9.072-01	1
2760	5.90E-05	NA	2.11	elea	2.04			
3660	7.29E-05	NA	2.23	es r	2.E-04		× •	
4740	8.56E-05	NA	2.10	lom	1.E-04		/.	
5640	1.54E-04	NA	2.19	č	5 E-05		*	
6540	1.59E-04	NA	2.08		5.2.05	/		
7440	2.19E-04	NA	2.20		0.E+00			
8700	2.45E-04	NA	2.24		C)	5000	10000
1117680	5.60E-04	NA	3.20				Time (sec)	
1204080	5.50E-04	NA	3.18	Reg	ression An	alysis		Std.Error
6128880	5.57E-04	NA	3.23	R So	quare		0.907	NA
			-	Car	elease rate	e (ra)	2.68E-08	3.246E-0
$K_{diss} = r_{co}/A \cdot S$, over time as d the total BET s (BET) X minera	where r _{co} is the ra etermined by linea surface area (specif al mass), and S is th	te of Ca release ir regression, A is ic surface area ie Ca prooffice minora		Diss. F	late Mine Uncerta	ral (Kəss) iinty of Fit	3.84E-07 4.6601E-08	mol/m²s mol/m²s

Datasheet S21 Experiment WHIT-P2DDB

 Mineral:
 Whitlockite
 Initial pH:
 1.98
 Specific Surf. Area (BET):
 0.0516 m²/g

 Batch ID:
 WHIT-P2DDB
 Mineral Mass:
 0.1500g
 Total Surf. Area (BET):
 0.00774 m²

 NA = no analysis
 Name
 Name
 Name
 1.500g
 Total Surf. Area (BET):
 0.00774 m²

Batch Results (shaded data = the linear initial rate data shown in the accompanying figure)

Time (sec)	Ca (moles released)	P (moles released)	рН
0	0	NA	pH 1.98 2.08 2.12 2.14 2.09 2.13
960	6.48E-05	4.16E-05	2.08
1800	9.11E-05	5.60E-05	2.12
2760	9.27E-05	6.26E-05	2.14
3720	1.04E-04	7.26E-05	2.09
4800	1.50E-04	8.48E-05	2.13
5700	1.67E-04	1.18E-04	2.14
6600	1.87E-04	NA	2.27
7500	2.10E-04	NA	2.13
8700	2.67E-04	NA	2.34
556020	5.58E-04	NA	3.16
599220	5.57E-04	NA	3.21

 $K_{diss} = r_{co}/A \bullet S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.

Stoichiometry

Moles of Mineral Released

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
960	5.94E-06	7.20E-06
1800	8.01E-06	1.01E-05
2760	8.94E-06	1.03E-05
3720	1.04E-05	1.15E-05
4800	1.21E-05	1.67E-05
5700	1.68E-05	1.85E-05
6600	NA	2.08E-05
7500	NA	2.33E-05
8700	NA	2.97E-05

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.



Diss. Rate Mineral (K_{diss}) **3.54E-07** mol/m²s Uncertainty of Fit 2.65E-08 mol/m²s





Datasheet S22 Experiment WHIT-P3

Mineral: \ Batch ID: \	Whitlockite WHIT-P3 Mine	Initial pH: 3.00 ral Mass: 0.150	0g	pecific Surf otal Surf. A	. Area (B Area (BEI	BET): 0.0	516 m ³ 74 m ²	/в	
NA = no anal	ysis		-0			1. 0.007			
Batch Results	s (shaded data =	the linear initial ra	nte dat	shown in the	accompo	anying figu	ire)		
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial	Ca Re	lease	
0	0	NA	3.00		3.E-05	y = 4.45	E-09x +	- 8.00E-06	
600	1.04E-05	7.96E-06	3.09	÷.	3.E-05	R ²	= 9.49	E-01	
1260	1.41E-05	1.35E-05	3.12	ased	2 5.05			/ -	
1800	1.62E-05	1.45E-05	3.11	ele	2.6-00		6		
2460	1.71E-05	1.50E-05	3.11	es r	2.E-05	Ý			
3060	2.37E-05	2.13E-05	3.15	non M	1.E-05	6			
4140	2.58E-05	NA	3.18	č	5 E-06				
81300	6.69E-05	NA	3.87		3.1-00				
242460	6.63E-05	NA	4.51		0.E+00				
593220	6.68E-05	NA	4.69		C	2	000	4000	6000
850500	7.01E-05	NA	4.67				Time	(sec)	
1988580	7.08E-05	NA	4.71	Regr	ession An	alysis			Std.Error
w ersoners			1	R Squ	lare		0	.949	NA
K _{diss} = r _{cd} /A+ over time as the total BET (BET) X mine stoiciometric	s, where r_{co} is the f determined by line Fsurface area (spec ral mass), and S is c coefficient of the	rate of Carelease ear regression, A is cific surface area the Ca respective mineral.	 >	Ca re Diss. Ro	lease rate ate Mine Uncerto	e (Г∝) ral (Kaiss) iinty of Fit	4.4 6.3	497E-09 9E-08	5.14E-1 mol/m²s mol/m²s
Stoichiometr	γ			Ana	lytical en	ror bars fa	ll withi	n plot sym	bols
Moles of Mi	ineral Released	ļ.			(), can ch				
Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)			Stoi	chiome Calcium	etry C	ompari	ison ^{Is}
600	1.14E-06	1.16E-06			4.E-06			angan sono s i kan be ^r i B	
1260	1.93E-06	1,56E-06		pa	3.E-06				
1800	2.08E-06	1.80E-06		easi	3.E-06			٠	
2460	2.14E-06	1.90E-06		Rel	3 5.06				
3060	3.04E-06	2.63E-06		neral	2.E-00		• •	•	
				Σ	1.E-06				

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.



Datasheet S23 Experiment WHIT-P3D

NA = no analy	sis								
Batch Results	(shaded data = t	he linear initial i	rate data	shown in th	е ассотр	oanyii	ng figure)		
Time (sec)	Ca (moles released)	P (moles released)	pН			Ir	nitial Ca Re	elease	
0	0	NA	3.01		4.E-05	1 3	/ = 4.27E-09	x + 1.27E-05	
780	1.60E-05	NA	3.08	÷	3.E-05		$R^2 = 9.9$	90E-01	
1380	1.85E-05	NA	3.12	ased	3.E-05			×	
1920	2.16E-05	NA	3.14	ele	2.E-05		20		
2520	2.27E-05	NA	3.16	es r	2.F-05		× *		
3180	2.69E-05	NA	3.19	om	1 E-05				
4440	3.15E-05	NA	3.18	S	E E OC				
11460	3.31E-05	NA	3.46		5.E-00				
96000	7.16E-05	NA	4.16		0.E+00	1			
						0	2000	4000	6000
							Tim	e (sec)	-
0			-	Reg	ression A	nalys	is		Std.Error
$K_{diss} = r_{co}/A \bullet S,$	where r _{ca} is the ra letermined by lines	ite of Ca release or regression. A is		R So	quare			0.990	NA
the total BET	surface area (speci	fic surface area		Car	elease ra	te (ra	r)	4.27E-09	2.13E-10
(BET) X miner	al mass), and S is th	ne Ca		10					

Datasheet S24 Experiment WHIT-P4D

VA - no unui	ysis							
Batch Results	s (shaded data =)	the linear initial ro	ate data s	hown in th	e accompo	inying figure)		
Time (sec)	Ca (moles released)	P (moles released)	рH			Initial Ca	Release	
0	0	NA	3.99		9.E-06	y = 2.3	39E-10x + 3.	55E-06
3900	4.52E-06	2.70E-06	3.88	-	8.E-06		R ² = 9.69E-0	1
7680	5.10E-06	3.73E-06	4.52	Ised	7.E-06		1	
11160	6.49E-06	4.41E-06	4.50	elea	5 E-06	/		
14880	7.33E-06	5.32E-06	4.76	es r	4.E-06	4	6 (
18540	7.77E-06	5.64E-06	4.81	lou	3.E-06			
81600	9.72E-06	6.95E-06	5.80	5	2.E-06			
103380	1.00E-05	NA	6.01		1.E-06			
276480	1.08E-05	NA	6.25		0.E+00			
					0	(10000	20000
						Ti	ime (sec)	
			_	Reg	ression An	alysis		Std.Error
V ar Idal	 Approximation of the second line of the second secon	A REAL PROPERTY AND A REAL					11-12-12-12-22-22-22-22-22-22-22-22-22-2	2014 at 2
ndiss = 1 co/ H+.	s, where r_{ca} is the r	ate of Ca release		R Sc	uare		0.969	NA
over time as the total BET (BET) X mine stoiciometric	5, where r _{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r	ate of Ca release ar regression, A is ific surface area he Ca respective mineral.		R So Car Diss, F	quare elease rate late Mine	e (r) ral (Køss)	0.969 2.39E-1(3.43E-09	NA) 2.48E-1 mol/m²s
A diss = 7 car/A = over time as the total BET (BET) X mine stoiciometric	y, where <i>r_{ca}</i> is the r determined by line 'surface area (spec ral mass), and S is t c coefficient of the r γ	are of Ca release ar regression, A is ific surface area he Ca respective mineral.	_	R So Car Diss. F	quare elease rate late Mine Uncerta Analytical	e (Fcr) ral (Kdus) inty of Fit error bars fal	0.969 2.39E-1(3.43E-09 3.57E-10	NA 2.48E-1 mol/m ² s mol/m ² s
A diss - I col / A = over time as the total BET (BET) X mine stoiciometric Stoichiometr Moles of Mi	y, where 7 _{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r Y ineral Released	ate of Ca release ar regression, A is ific surface area he Ca respective mineral.	_	R Sc Car Diss. F	uare elease rate late Mine Uncerta Analytical	e (r _{cr}) ral (K _{diss}) inty of Fit error bars fal	0.969 2.39E-10 3.43E-09 3.57E-10 I within plot	NA 2.48E-1 mol/m ² s mol/m ² s symbols
A diss - T _{co} /A+ over time as the total BET (BET) X mine stoiciometric Stoichiometr Moles of Mi Time (sec)	, where r _{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r γ ineral Released Moles Mineral Released (P)	Ale of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca)	_ →	R So <u>Ca</u> r Diss. F	uare elease rate late Miner Uncerta Analytical Stoi	e (r _{cs}) ral (K _{diss}) inty of Fit error bars fal chiometr	0.969 2.39E-10 3.43E-09 3.57E-10 I within plot y Compa	NA 2.48E-1 mol/m ² s mol/m ² s symbols
A diss - T _{co} /A+ over time as the total BET (BET) X mine stoiciometric Stoichiometric Violes of Mi Time (sec) 3900	y where r_{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r y ineral Released Moles Mineral Released (P) 3.85E-07	Ale of Ca release ar regression, A is ific surface area he Ca respective mineral. Moles Mineral Released (Ca) 5.02E-07	→	R Sc Car Diss. F	uare elease rate late Mine Uncerta Analytical Stoi 1.E-06	e (r _{cr}) <i>ral</i> (K _{diss}) <i>inty of Fit</i> error bars fal chiometr Phosphorus	0.969 2.39E-1(3.43E-09 3.57E-10 I within plot y Compa s • Calo	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison
A diss - I col / A = over time as the total BET (BET) X mine stoiciometric Stoichiometri Moles of Mi Time (sec) 3900 7680	y, where T_{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r y meral Released Moles Mineral Released (P) 3.85E-07 5.32E-07	Moles Mineral Released (Ca) 5.02E-07 5.67E-07	→	R Sc Car Diss. F	uare elease rate late Miner Uncerta Analytical Stoi 1.E-06	e (r _{cr}) ral (K _{disr}) inty of Fit error bars fal chiometr Phosphorus	0.969 2.39E-1(3.43E-09 3.57E-10 I within plot y Compa s • Cale	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison cium
Aging - T _{co} /AF over time as the total BET (BET) X mine stoiciometric Stoichiometric Moles of Mi Time (sec) 3900 7680 11160	y, where T_{ca} is the r determined by line surface area (spec ral mass), and <i>S</i> is t coefficient of the r y meral Released Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07	Moles Mineral Released (Ca) 5.02E-07 5.67E-07 7.21E-07	→	R Sc Car Diss. F	uare elease rate <i>late Miner</i> <i>Uncerta</i> Analytical Stoi 1.E-06	e (r _{cr}) ral (K _{diss}) inty of Fit error bars fal chiometr Phosphorus	0.969 2.39E-1(3.43E-09 3.57E-10 I within plot y Compa s Calo	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison cium
A diss - I col / A = over time as the total BET (BET) X mine stoiciometric Stoichiometric Moles of Mi Time (sec) 3900 7680 11160 14880	y where 7_{ca} is the r determined by line 'surface area (spec ral mass), and <i>S</i> is t : coefficient of the r y meral Released Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07 7.60E-07	Moles Mineral Released (Ca) 5.02E-07 5.67E-07 8.14E-07 8.14E-07	→	R Sc Car Diss. F	auare elease rate Uncerta Analytical 1.E-06 8.E-07	e (r _{cr}) <i>inty of Fit</i> error bars fal chiometr Phosphorus	0.969 2.39E-1(3.43E-09 3.57E-10 I within plot y Compa s Cak	NA 2.48E-1 mol/m ² s mol/m ² s symbols arrison cium
Addas - T _{co} /AF over time as the total BET (BET) X mine stoiciometric Stoichiometric Moles of Mi Time (sec) 3900 7680 11160 14880 18540	y where T_{ca} is the r determined by line 'surface area (spec ral mass), and S is t : coefficient of the r y meral Released Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07 7.60E-07 8.06E-07	Moles Mineral Released (Ca) 5.02E-07 7.21E-07 8.14E-07 8.64E-07	→	R Sc Car Diss. F	alate Mines Uncerta Analytical 1.E-06 8.E-07 6.E-07	e (r _{cr}) <i>inty of Fit</i> error bars fal chiometr Phosphorus	0.969 2.39E-1(3.43E-09 3.57E-10 I within plot y Compa s • Cak	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison
Addas - Jed At- over time as the total BET (BET) X mine stoiciometric Stoichiometric Violes of Mi Time (sec) 3900 7680 11160 14880 18540 81600	y where r_{ca} is the r determined by line surface area (spec ral mass), and S is t coefficient of the r y meral Released Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07 7.60E-07 9.93E-07	Moles Mineral Released (Ca) 5.02E-07 5.67E-07 8.14E-07 8.64E-07 1.08E-06	→	R Sc Car Diss. F	ate Mines Uncerta Analytical 1.E-06 8.E-07 4.E-07	e (r _{cr}) ral (K _{dusr}) inty of Fit error bars fal chiometr Phosphorus	0.969 2.39E-1(3.57E-10 I within plot y Compa s • Cale	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison cium
Agias - F _{co} /AF over time as the total BET (BET) X mine stoiciometric Stoichiometric Moles of Mi Time (sec) 3900 7680 11160 14880 18540 81600	y where T_{ca} is the r determined by line surface area (spec ral mass), and <i>S</i> is t coefficient of the r y Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07 7.60E-07 9.93E-07	Moles Mineral Released (Ca) 5.02E-07 5.67E-07 7.21E-07 8.14E-07 8.64E-07 1.08E-06	→	R Sc Car Diss. F	ate Miner Uncerta Analytical 1.E-06 8.E-07 4.E-07 2.E-07	e (r _{er}) ral (K _{diss}) inty of Fit error bars fal chiometr Phosphorus	0.969 2.39E-1(3.57E-10 I within plot y Compa s • Calo	NA 2.48E-1 mol/m ² s mol/m ² s symbols trison
Addas - Ford AP- over time as the total BET (BET) X mine stoiciometric Stoichiometric Violes of Mil Time (sec) 3900 7680 11160 14880 18540 81600 Moles mine determined the appropri	y, where T_{ca} is the r determined by line surface area (spec ral mass), and <i>S</i> is t coefficient of the r y Moles Mineral Released (P) 3.85E-07 5.32E-07 6.30E-07 7.60E-07 9.93E-07 ral released (above from moles release iate stoichiometric	Moles Mineral Released (Ca) 5.02E-07 5.67E-07 7.21E-07 8.14E-07 8.64E-07 1.08E-06	d by ctive	R Sc Car Diss. F	alare Miner Uncerta Analytical 1.E-06 8.E-07 4.E-07 2.E-07 0.E+00	e (r _{cr}) ral (K _{diss}) inty of Fit error bars fal chiometr Phosphorus	0.969 2.39E-1(3.57E-10 I within plot y Compa s • Calo	NA 2.48E-1 mol/m ² s mol/m ² s symbols arison cium

Datasheet S25 Experiment WHIT-P4DD

, Batch Results	(shaded data = ti	he linear initial	rate data sh	own in tł	ne accompany	ving figure)	
Time (sec)	Ca (moles released)	P (moles released)	pH			Initial C <mark>a R</mark> elease	
0	0	NA	3.98		9.E-06	y = 2.39E-10x + 3.60E	E-06
3660	4.03E-06	NA	4.41	Ŧ	8.E-06	R ² = 9.15E-01	
7200	5.95E-06	NA	4.72	ased	7.E-00		
10620	6.22E-06	NA	4.45	ele	5.E-06	•	
14160	6.78E-06	NA	4.58	esr	4.E-06		
17580	7.77E-06	NA	4.82	Pour	3.E-06		
83520	1.04E-05	NA	5.60) Ca	2.E-06		
187140	1.03E-05	NA	5.69		1.E-06		
770940	1.03E-05	NA	6.33		0.E+00		
1372260	1.16E-05	NA	6.07		0	10000	20000
1741860	1.20E-05	NA	6.25			Time (sec)	
Inst Marine	- 101 - N-S-101 - 1	0		Reg	ression Analy	isis	Std.Error
$K_{diss} = r_{co}/A \bullet S,$	where r _{ca} is the ra	te of Ca release		R So	quare	0.915	NA
over time as c the total BET	surface area (specif	ir regression, A is ic surface area		Car	release rate (rœ) 2.39E-10	4.20E-11
(BET) X miner	al mass), and S is th	e Ca))			

Datasheet S26 Experiment WHIT-P5D

atch Results	(shaded data = tl	he linear initial ru	nte data s	hown in the	e accompar	ying figure)		
Time (sec)	Ca (moles released)	P (moles released)	рН			Initial Ca	Release	
0	0	NA	5.00		2.E-06	y = 5.0)6E-11x + 9.5	8E-07
3840 7620	1.12E-06 1.29E-06	NA NA	5.20 5.31	ased)	2.E-06		₹ ² = 9,18E-01	
11100 14820	1.67E-06 1.69E-06	NA NA	5.34 5.41	es rele	1.E-06	5 9		
18480	1.85E-06	NA	5.38	lom				
81540	3.06E-06	NA	6.13	Ca	5.E-07			
103320	3.22E-06	NA	6.24		0.5.00			
276420	4.26E-06	NA	6.34		0.E+00 *		10000	20000
						Ті	me (sec)	2,0000
			- 1	Regi	ession Ana	ysis		Std.Error
$K_{diss} = r_{cd} / A \bullet S,$	where r _{co} is the ra	te of Ca release		R Sq	uare		0.918	NA
the total BET s	surface area (specif	ic surface area		Care	elease rate	(Fa)	5.06E-11	8.75E-1
(BET) X minera stoiciometric	al mass), and S is th coefficient of the re	e Ca espective mineral.	→	Diss. R	ate Minero	7/ (Kaiss) 📑	7 <mark>.</mark> 26E-10	mol/m²s

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Datasheet S27 Experiment WHIT-P5DDB

Mineral: Batch ID:	Whitlockite WHIT-P5DDB	Initial pH: 4.99 Aineral Mass: 0	S 1500g	pecific Surf Total Su	. Area (I Irf. Area	BET): 0.05 (BET): 0.	516 m²/g 00774 m²	
NA = no anal	ysis				P = P = C			1
Batch Results	s (shaded data =	the linear initial ro	ite data s	hown in the	accompo	anying figu	re)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial	Ca Release	
0	0.00E+00	NA	4.99		1.E-06) y	= 4.38E-11x + 2.5	51E-07
1320	1.95E-07	4.15E-07	5.19	F	1.E-06	<u> </u>	R ² = 9.32E-03	
3840	5.16E-07	6.11E-07	5.33	ased	8 F-07		/	
6300	5.87E-07	8.48E-07	5.31	ele	0.1-0/		1	
8700	6.21E-07	9.23E-07	5.40	es r	6.E-07			
10980	7.21E-07	1.04E-06	5.55	mo	4.E-07	/		
18720	1.05E-06	1.51E-06	5.67	ca (2.E-07	6		
92460	2.05E-06	NA	6.03			2		
$K_{diss} = r_{co}/A \bullet$ over time as the total BET	S, where <i>r_{ca}</i> is the r determined by line Fsurface area (spec	ate of Ca release ar regression, A is ific surface area		<i>Regre</i> R Squ Ca re	ession An Iare lease rate	alysis e (r∞)	0.932 4.3799E-11	Std.Error NA 5.93E-12
(BET) X mine stoiciometrie	ral mass), and S is t c coefficient of the i	he Ca respective mineral.	→	Diss. Ro	ite Mine	ral (Kass)	6.29E-10	mol/m²s
Stoichiometr	Y ineral Released			A	nalytical	error bars	fall within plot sy	ymbols
Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)			Sto	chiome	try Compar	ison s
1320	5.93E-08	2.17E-08			4.E-07	1	,	
3840	8.73E-08	5.74E-08		pa				
6300	1.21E-07	6.53E-08		east	3.E-07	8		
8700	1.32E-07	6.90E-08		Rel				
10980	1.48E-07	8.02E-08		ral	2.E-07	8		

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

1.17E-07

2.16E-07



Datasheet S28 Experiment MER-P2DDC

Duren Di	WER-P71111 N	Aineral Mass. A	1500g	Total Su	irf. Ares	(BET) O	$00698 m^2$	
NA = no anal	vsis	inierar wiass: V	.10008	Total Su	III. AIE	(DEI): 0.	00030111	
Batch Result	s (shaded data =	the linear initial re	ate data s	hown in the	e accomp	anying figu	ire)	
Time (sec)	Ca (moles released)	P (moles released)	рН			Initial	Ca Release	
0	0	NA	2.00		2,E-04	y = 1.9	0E-08x - 2.55E-07	7
960	1.40E-05	1.87E-05	2.03	0	1.E-04	R	² = 9.77E-01 💉	
1920	2.29E-05	1.97E-05	2.06	isec	1.E-04		• /•	
2760	4.41E-05	2.26E-05	2.13	elea	1.E-04			
3720	6.52E-05	6.89E-05	2.12	es r	8.E-05		1	
4620	7.40E-05	7.98E-05	2.15	lon	6.E-05			
5460	8.21E-05	1.03E-04	2.15	Ca (4.E-05	*	>	
6420	1.08E-04	1.09E-04	2.14	0	2.E-05			
7260	1.36E-04	1.17E-04	2.16		0.E+00		AND A REPORT	
			- L.			0	5000	10000
							Time (sec)	
and the second second		1. S		Regr	ession Ar	nalysis		Std.Error
$K_{m} = \Gamma_{m} / \Delta \bullet$	S Whorn r ic thor							
over time as	determined by line	ate of Ca release		R Squ	uare		0.977	NA
over time as	determined by line surface area (spec	ate of Ca release ar regression, A is ific surface area		R Squ Ca re	uare elease rat	te (rœ)	0.977 1.90E-08	NA 1.18 <mark>E</mark> -0
over time as the total BET (BET) X mine stoiciometri	determined by line f surface area (spec aral mass), and S is t c coefficient of the	ate of Ca release ar regression, A is ific surface area he Ca respective mineral,	_	R Squ Ca re Diss. Ro	uare Hease rat ate Mine	te (Fa) eral (Kass)	0.977 1.90E-08 2.86E-07	NA 1.18E-0 mol/m ² s
over time as the total BE (BET) X mine stoiciometric	determined by line f surface area (spec rral mass), and S is t c coefficient of the i	ate of Ca release ear regression, A is ific surface area he Ca respective mineral,		R Squ Care Diss. Ro	uare elease rat ate Mine Uncert	te (Fcr) eral (Kdiss) ainty of Fit	0.977 1.90E-08 2.86E-07 1.78E-08	NA 1.18E-0 mol/m ² s mol/m ² s
over time as the total BET (BET) X mine stoiciometri Stoichiometri Moles of M	determined by line f surface area (spec ral mass), and S is t c coefficient of the Y	ate of Ca release ar regression, A is ific surface area the Ca. respective mineral.	_ →	R Squ Ca re Diss. Ro Ana	uare elease rat orte Mine Uncert alytical er	te (r∝) eral (K₅s) ainty of Fit rror bars fal	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym	NA 1.18E-0 mol/m ² s mol/m ² s bols
over time as the total BET (BET) X mine stoiclometric Stoichiometric Moles of M Time (sec)	y where r _{ca} is the r determined by line f surface area (spec rral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Poloaced (P)	Moles Mineral	>	R Squ Ca re Diss. Ro Ana S	uare elease rat <i>ate Mine</i> <i>Uncert</i> alytical en toichi Pho	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fai ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s mol/m ² s bols
over time as the total BE (BET) X mine stoiciometri Stoichiometri Moles of M Time (sec)	y where r _{ca} is the r determined by line f surface area (spec ral mass), and S is t c coefficient of the coefficient of the moreal Released Moles Mineral Released (P)	Alte of Ca release ear regression, A is ific surface area the Ca respective mineral. Moles Mineral Released (Ca)	_ →	R Squ Ca re Diss. Ro Ana S 2.E	uare elease rat <i>Uncert</i> alytical el toichi Phc	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fal ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s mol/m ² s bols
over time as the total BE (BET) X mine stoiciometri Stoichiometri Moles of M Time (sec) 960	y where r _{ca} is the r determined by line f surface area (spec ral mass), and S is t c coefficient of the ineral Released Moles Mineral Released (P) 2.67E-06	Moles Mineral Released (Ca)	_ →	R Squ Ca re Diss. Ro Ana S 2.E 2.E 2.E	uare elease rat <i>Uncert</i> alytical er toichi Phc -05 -05	eral (Kauss) ainty of Fit rror bars fai ometry osphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s mol/m ² s bols
over time as the total BET (BET) X mine stoiciometri Stoichiometri Moles of M Time (sec) 960 1920	y where r _{ca} is the r determined by line f surface area (spec rral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P) 2.67E-06 2.82E-06	Moles Mineral Released (Ca) 1.47E-06 2.42E-06	_ →	R Squ Ca re Diss. Ro Ana S 2.E 2.E 9 2.E 9 1.E	uare elease rat <i>Uncert</i> alytical er toichi • Pho -05 -05 -05	te (r _{er}) eral (K _{diss}) ainty of Fit rror bars fai ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols
stoichiometri George Stoichiometri Stoichiometri Time (sec) 960 1920 2760	y where r _{ca} is the r determined by line f surface area (spec rral mass), and S is t c coefficient of the i ry ineral Released Moles Mineral Released (P) 2.67E-06 2.82E-06 3.22E-06	Moles Mineral Released (Ca) 1.47E-06 2.42E-06 4.64E-06	>	R Squ Ca re Diss. Ro Ana S 2.E 2.E 2.E 1.E 1.E 1.E 1.F	uare elease rat <i>Uncert</i> alytical er toichi toichi toichi -05 -05 -05 -05 -05	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fai ometry osphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols
stoichiometri Get Stoichiometri Stoichiometri Moles of M Time (sec) 960 1920 2760 3720	y where r _{ca} is the r determined by line f surface area (spec ral mass), and <i>S</i> is t c coefficient of the i ry ineral Released Moles Mineral Released (P) 2.67E-06 2.82E-06 3.22E-06 9.85E-06	Moles Mineral Released (Ca) 1.47E-06 2.42E-06 4.64E-06 6.86E-06	->	R Squ Ca re Diss. Ro Ana S 2.E 2.E 1.E 1.E 1.E 1.E 1.E 8.E	uare elease rat <i>Uncert</i> alytical er toichi •05 •05 •05 •05 •05 •05 •05	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fai ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols
Stoichiometri Moles of M Time (sec) 960 1920 2760 3720 4620	y where r _{ca} is the r determined by line f surface area (spec ral mass), and <i>S</i> is t c coefficient of the i meral Released Moles Mineral Released (P) 2.67E-06 2.82E-06 3.22E-06 9.85E-06 1.14E-05	Moles Mineral Released (Ca) 1.47E-06 2.42E-06 4.64E-06 5.86E-06 7.79E-06		R Squ Ca re Diss. Ro Diss. Ro Ana S 2.E 2.E 1.E 1.E 1.E 1.E 8.E 8.E 0.E	uare elease rat Uncert alytical en toichi toichi 05 -05 -05 -05 -05 -05 -05 -05 -05 -05	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fal ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols
stoichiometri Stoichiometri Stoichiometri Stoichiometri Moles of M Time (sec) 960 1920 2760 3720 4620 5460	Note: Provide the second secon	Moles Mineral Released (Ca) 1.47E-06 2.42E-06 4.64E-06 6.86E-06 8.64E-06	_ →	R Squ Ca re Diss. Ro Diss. Ro Ana S 2.E 2.E 1.E 1.E 1.E 1.E 1.E 4.E 4.E	uare elease rat <i>Uncert</i> alytical er toichi toichi toichi 05 -05 -05 -05 -05 -05 -05 -05 -06 -06 -06	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fai ometry isphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols
stoichiometri Stoichiometri (BET) X mine stoiclometri Moles of M Time (sec) 960 1920 2760 3720 4620 5460 6420	y determined by line f surface area (spec rral mass), and S is to coefficient of the i meral Released Moles Mineral Released (P) 2.67E-06 2.82E-06 3.22E-06 9.85E-06 1.14E-05 1.47E-05 1.56E-05	Moles Mineral Released (Ca) 1.47E-06 2.42E-06 4.64E-06 5.86E-06 7.79E-06 8.64E-06 1.14E-05	_ →	R Squ Ca re Diss. Ro Diss. Ro Ana S 2.E 2.E 1.E 1.E 1.E 8.E 4.E 2.E 2.E 2.E 2.E 2.E 2.E 2.E 2.E 2.E 2	uare elease rat <i>Uncert</i> alytical en toichi •05 •05 •05 •05 •05 •05 •05 •06 •06 •06 •06	te (r _{ce}) eral (K _{diss}) ainty of Fit rror bars fai ometry osphorus	0.977 1.90E-08 2.86E-07 1.78E-08 Il within plot sym Comparison Calcium	NA 1.18E-0 mol/m ² s bols

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion.

Datasheet S29 Experiment MER-P2DD

Time (sec)

NA = no analysis

Time (sec)	Ca (moles released)	P (moles released)	pН	
0	0	NA	1.97	
900	1.40E-05	NA	2.06	
1440	2.29E-05	NA.	2.05	
2160	4.41E-05	NA	2.07 2.07 2.04	
2880	6.52E-05	NA		
3600	7.40E-05	NA		
4980	8.21E-05	NA	2.07	
9360	1.08E-04	NA	2.14	
168060	3.80E-04	NA	2.55	





Datasheet S30 Experiment MER-P3

VA = no analys	sis							
Batch Results	(shaded data = t	he linear initial i	rate date	shown in the	e accompa	nying figure	e)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial (Ca Release	
0	0	NA	3.00		5.E-05	y = 3.79	9E-09x + 8.20E-0)6
960	1.26E-05	NA	3.50	G	4.E-05	R	² = 9.83E-01	*
1800	1.46E-05	NA	2.99	sec	4.E-05		•	
2700	1.78E-05	NA	3.29	elea	3 E-05			
3540	2.10E-05	NA	3.26	es r	2.E-05		6	
4440	2.43E-05	NA	3.30	De la	2.E-05	1		
5700	3.22E-05	NA	3.51	e (1.E-05	V		
8160	3.82E-05	NA	3.52		5.E-06			
1165980	5.42E-05	NA	4. <mark>8</mark> 1		0.E+00			
1209180	5.44E-05	NA	5.27		()	5000	10000
3671580	5.30E-05	NA	4.92				Time (sec)	
1504380	5.31E-05	NA	5.45	Regr	ession And	ilysis		Std.Error
4578780	5.45E-05	NA	5.49	R Squ	uare		0.983	NA
5226780	5.34E-05	NA	NA	Ca re	elease rate	(f∝)	3.79E-09	2.19E-10
K _{diss} = r _{co} /A•S, over time as d	where r _{ca} is the ra letermined by linea	ite of Ca release ar regression, A is	->	Diss. Re	ate Miner Uncertai	al (Kaiss)	5.71E-08	mol/m²s
the total BET s (BET) X mineral stoiciometric	evermined by linea surface area (specif al mass), and S is the coefficient of the re	ar regression, A is fic surface area le Ca acceptive mineral	0	Ana	<i>Uncertai</i> alytical err	nty of Fit or bars fall	3.31E-09 r within plot syml	mol/m²s bols

Datasheet S31 Experiment MER-P3D

Batch Result	s (shaded data =	the linear initial re	nte data	shown in the	accomp	anying figure,)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial C	a Release	
0	0	NA	3.00		2.E-05	y = 4.15	5E-09x + 3.84E-	06 🗻
720	6.31E-06	2.89E-06	3.07	Ŧ		R	² = 9.92E-01 🌶	/*
1380	9.93E-06	4.57E-06	3.05	ased	2.E-05		*	
2040	1.27E-05	6.37E-06	3.08	ala	18 1 8 9491		1	
2520	1.43E-05	6.64E-06	3.14	es l	1.E-05	1		
3060	1.68E-05	8.53E-06	3.18	om		6		
3660	1.86E-05	8.71E-06	3.17	Ca (5.E-06			
7140	2.17E-05	1.72E-05	3.25					
101640	6.02E-05	2.81E-05	4.67		0.E+00	-	1010200000	A CONTRACTOR OF
523020	6.18E-05	NA	5.00			0	2000	4000
							Time (sec)	
102 (BA)	55 M 5000	27 and 104	7	Regr	ession An	alysis		Std.Error
Karr = I A.	S where r is the r	ate of Ca releace					121000	1075 PC
over time as	determined by line	ar regression Aic		R Squ	uare		0.992	NA
over time as the total BE	determined by line I surface area (spec	ar regression, A is if is surface area		R Sqi Ca re	uare elease rat	e (r∞)	0.992 4.15E-09	NA 1.84E-:
over time as the total BE (BET) X mine stoiclometri	determined by line I surface area (spec aral mass), and S is t c coefficient of the V	ar regression, A is ific surface area he Ca respective mineral.	→	R Squ Care Diss. Ro Ana	uare elease rat ate Mine Uncerto lytical err	e (r∝) eral (Kæs) ainty of Fit or bars fall w	0.992 4.15E-09 6.26E-08 2.77E-09 r ithin plot symb	NA 1.84E-: nol/m ² s nol/m ² s ols
over time as the total BE (BET) X mine stoiciometri toichiometri Moles of M	determined by line I surface area (spec rral mass), and S is t c coefficient of the ry ineral Released	ar regression, A is ific surface area the Ca respective mineral.	_	R Squ Care Diss. Re Ana	uare elease rat <i>ate Mine</i> <i>Uncerto</i> lytical err	e (r∝) eral (Kass) ainty of Fit or bars fall w	0.992 4.15E-09 6.26E-08 r 2.77E-09 r ithin plot symb	NA 1.84E-1 nol/m ² s nol/m ² s ols
over time as the total BE (BET) X mine stoiciometri toichiometri	determined by line T surface area (spec ral mass), and S is t c coefficient of the ry ineral Released Moles Mineral	Moles Mineral	→	R Squ Ca re Diss. Re Ana S	uare elease rat ate Mine Uncerto lytical err toichio	e (r∝) eral (Kass) ninty of Fit or bars fall w cometry Co	0.992 4.15E-09 6.26E-08 2.77E-09 r ithin plot symb	NA 1.84E- nol/m ² s nol/m ² s ols
toichiomet Noles of M Time (sec)	determined by line T surface area (spec aral mass), and S is t c coefficient of the ry ineral Released Moles Mineral Released (P)	Moles Mineral Released (Ca)	→	R Squ Care Diss. Ro Ana S	uare elease rat <i>Uncerto</i> lytical err toichio Phos	e (r∝) eral (Kass) ninty of Fit or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 7 2.77E-09 r ithin plot symb cmparison Calcium	NA 1.84E-: nol/m ² s nol/m ² s ols
the total BE (BET) X mine (BET) X mine stoiclometri toichiometri Moles of M Time (sec) 720	determined by line T surface area (spec- rral mass), and S is t c coefficient of the r ry ineral Released Moles Mineral Released (P) 4.13E-07	Moles Mineral Released (Ca) 6.65E-07	→	R Squ Care Diss. Ro Ana S 3.E	uare elease rat <i>Uncerto</i> lytical err toichio Phos -06	e (r _≈) <i>ral (</i> K _{diss}) <i>ainty of Fit</i> or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 2.77E-09 r ithin plot symb omparison Calcium	NA 1.84E- nol/m ² s nol/m ² s ols
toichiomete Aoles of M Time (sec) 720 1380	ry Moles Mineral Released (P) 4.13E-07 6.53E-07 6.53E-07	Moles Mineral Released (Ca) 6.65E-07 1.05E-06	→	R Squ Care Diss. Ro Ana S 3.E 2.E	uare elease rat <i>Uncerto</i> lytical err toichic ● Phos -06 -06	e (r _≈) eral (Kass) ninty of Fit or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 7 2.77E-09 r ithin plot symb comparison Calcium	NA 1.84E- nol/m ² s nol/m ² s ols
torichiomet (BET) X mine stoiclometri toichiomet Aoles of M Time (sec) 720 1380 2040	ry ineral Released Released (P) 4.13E-07 6.53E-07 9.10E-07	Moles Mineral Released (Ca) 6.65E-07 1.05E-06 1.33E-06	→	R Squ Ca re Diss. Ro Ana S 3.E 2.E	uare elease rat <i>Uncerto</i> lytical err toichio -06 -06	e (r ₅₅) eral (Kass) ainty of Fit or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 7.77E-09 rithin plot symb comparison Calcium	NA 1.84E- nol/m ² s nol/m ² s ols
toichiomet (BET) X mine stoiciometri toichiomet Aoles of M Time (sec) 720 1380 2040 2520	Adversinged by line rsurface area (spectral mass), and S is to coefficient of the officient ineral Released Moles Mineral Released (P) 4.13E-07 6.53E-07 9.10E-07 9.49E-07	Moles Mineral Released (Ca) 6.65E-07 1.05E-06 1.33E-06 1.50E-06	→	R Squ Ca re Diss. Ro Ana 3.E 3.E 2.E 2.E	uare elease rat <i>Uncerto</i> lytical err toichio -06 -06 -06	e (r _≈) <i>ral (</i> K _{diss}) <i>ainty of Fit</i> or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 7 2.77E-09 rithin plot symb comparison Calcium	NA 1.84E- nol/m ² s ols
toichiometri toichiometri toichiometri Time (sec) 720 1380 2040 2520 3060	ry ineral Released Moles Mineral Released (P) 4.13E-07 6.53E-07 9.10E-07 9.49E-07 1.22E-06	Moles Mineral Released (Ca) 6.65E-07 1.05E-06 1.50E-06 1.77E-06	→	R Squ Ca re Diss. Re Ana 3.E 3.E 2.E 2.E 1.E	uare elease rat <i>Uncerta</i> lytical err toichia -06 -06 -06 -06	e (r≃) ral (Kass) ainty of Fit or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 r 2.77E-09 r ithin plot symb comparison Calcium	NA 1.84E- nol/m ² s ols
Action of the store of the stor	ry ineral Released Moles Mineral Released (P) 4.13E-07 6.53E-07 9.10E-07 1.22E-06 1.24E-06	Moles Mineral Released (Ca) 6.65E-07 1.05E-06 1.33E-06 1.77E-06 1.96E-06	→	R Squ Ca re Diss. Re Ana 3.E 3.E 2.E 1.E 5.E	are Mine Uncerta Uncerta Ivtical err toichia Phos -06 -06 -06 -06 -06 -06	e (r _≈) <i>rral (</i> K _{dess}) <i>ainty of Fit</i> or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 7 2.77E-09 r ithin plot symb comparison Calcium	NA 1.84E- nol/m ² s ols
torichiomet (BET) X mine stoiclometri toichiomet Moles of M Time (sec) 720 1380 2040 2520 3060 3660 Moles mine	A determined by line T surface area (spec- ral mass), and S is t c coefficient of the iso- ry ineral Released Moles Mineral Released (P) 4.13E-07 6.53E-07 9.10E-07 9.49E-07 1.22E-06 1.24E-06	Moles Mineral Released (Ca) 6.65E-07 1.05E-06 1.33E-06 1.77E-06 1.96E-06	→	R Squ Ca re Diss. Ro Ana 3.E 3.E 2.E 1.E 5.E 0.E-	uare elease rat <i>Uncerto</i> lytical err toichio -06 -06 -06 -06 -06 -06 -07	e (r _≈) eral (K _{des}) ainty of Fit or bars fall w ometry Co phorus	0.992 4.15E-09 6.26E-08 r 2.77E-09 r ithin plot symb omparison Calcium	NA 1.84Ē- nol/m ² s nol/m ² s ols

the appropriate stoichiometric factor of the respective ion.

Time (sec)

Datasheet S32 **Experiment MER-P4**

IA =no analys	is							
atch Results	(shaded data = ti	he linear initial r	ate data :	shown in the	e accompai	nying figure)		
Time (sec)	Ca (moles released)	P (moles released)	рН			Initial C	a Release	
0	0	NA	4.01		5.E-06	y = 4	4.95E-10x + 1	40E-06
960	1.76E-06	NA	4.33	Ŧ	4.E-06		$R^2 = 9.82E-0$	1
1800	2.28E-06	NA	4.34	ased	4.E-00 3.E-06			
2700	2.89E-06	NA	4.15	ele	3.E-06		>	
3540	3.15E-06	NA	4.33	es	2.E-06	1		
4500	3.73E-06	NA	4.53	om	2.E-06			
5700	4.09E-06	NA	4.59	č	1.E-06			
8220	4.95E-06	NA	4.59		5.E-07			
101280	6.85E-06	NA	5.83		0.E+00	5.68	a	
					C	200	0 4000	6000
							lime (sec)	
14000 - 1940 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840 - 1840			-	Regr	ession Ana	lysis		Std.Error
$K_{diss} = r_{co} / A \cdot S,$	where r _{ca} is the ra	te of Ca release		R Sq	uare		0.982	NA
the total BET	surface area (speci	fic surface area		Ca re	elease rate	(r)	4.95E-10	3.33E-1
(BET) X miner stoiciometric	al mass), and S is the coefficient of the re	ne Ca espective mineral.	_ →	Diss. R	ate Miner	al (Kass)	7.47E-09	mol/m²s

Datasheet S33 Experiment MER-P4DD

 Mineral:
 Merrillite
 Initial pH: 4.00
 Specific Surf. Area (BET): 0.0465m²/g

 Batch ID:
 MER-P4DD
 Mineral Mass: 0.1500g
 Total Surf. Area (BET): 0.00698 m²

 NA = no analysis
 NA
 NA
 NA

Batch Results (shaded data = the linear initial rate data shown in the accompanying figure)

Time (sec)	Ca (moles released)	P (moles released)	рН	
0	0	0	4.00	
1320	1.20E-06	1.33E-06	4.10	
2400	1.71E-06	1.87E-06	4.17 4.20	
3840	2.31E-06	2.66E-06		
4920	2.82E-06	3.23E-06	4.31	
6300	3.20E-06	3.57E-06	4.31	
8700	3.83E-06	4.79E-06	4.43	
92460	6.95E-06	NA	5.56	

Initial Ca Release 5.E-06 y = 3.59E-10x + 8.68E-07 4.E-06 $R^2 = 9.81E-01$ Ca (moles released) 4.E-06 3.E-06 3.E-06 2.E-06 2.E-06 1.E-06 5.E-07 0.E+00 0 5000 10000 Time (sec) **Regression Analysis** Std.Error **R** Square NA 0.981

 $K_{diss} = r_{co}/A \bullet S$, where r_{co} is the rate of Ca release over time as determined by linear regression, A is the total BET surface area (specific surface area (BET) X mineral mass), and S is the Ca stoiciometric coefficient of the respective mineral.

Stoichiometry

Moles of Mineral Released

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)
1320	1.91E-07	1.26E-07
2400	2.67E-07	1.80E-07
3840	3.80E-07	2.43E-07
4920	4.61E-07	2.97E-07
6300	5.10E-07	3.36E-07
8700	6.85E-07	4.03E-07

Moles mineral released (above) based on P or Ca determined from moles released of P or Ca divided by the appropriate stoichiometric factor of the respective ion. Diss. Rate Mineral (K_{aliss}) **5.41E-09** mol/m²s Uncertainty of Fit 3.77E-10 mol/m²s

Ca release rate (ra)

Analytical error bars fall within plot symbols

Stoichiometry Comparison

3.59E-10

2.50E-11



Datasheet S34 **Experiment MER-P5**

Batch Results (Time (sec) 0 3840 7380 11760 14760 18000 24000 103020	(shaded data = 1 Ca (moles released) 0 5.74E-07 6.95E-07 8.08E-07 9.10E-07 1.05E-06 1.25E-06	the linear initial re P (moles released) 0 2.81E-07 4.57E-07 5.95E-07 6.20E-07 7.41E-07	pH 5.00 5.45 5.78 5.56 5.76	hown in the	accompa 1.E-06 1.E-06	nying figu Initial y = 3.3	re) Ca Release 6E-11x + 4.35E-0	7
Time (sec) 0 3840 7380 11760 14760 18000 24000 103020	Ca (moles released) 0 5.74E-07 6.95E-07 8.08E-07 9.10E-07 1.05E-06 1.25E-06	P (moles released) 0 2.81E-07 4.57E-07 5.95E-07 6.20E-07 7.41E-07	pH 5.00 5.45 5.78 5.56 5.76	ased)	1.E-06 1.E-06	Initial y = 3.3	Ca Release 6E-11x + 4.35E-0	7
0 3840 7380 11760 14760 18000 24000 103020	0 5.74E-07 6.95E-07 8.08E-07 9.10E-07 1.05E-06 1.25E-06	0 2.81E-07 4.57E-07 5.95E-07 6.20E-07 7.41E-07	5.00 5.45 5.78 5.56 5.76	ased)	1.E-06 1.E-06	y = 3.3	6E-11x + 4.35E-0	7
3840 7380 11760 14760 18000 24000 103020	5.74E-07 6.95E-07 8.08E-07 9.10E-07 1.05E-06 1.25E-06	2.81E-07 4.57E-07 5.95E-07 6.20E-07 7.41E-07	5.45 5.78 5.56 5.76	ased)	1.E-06	ਾ ਜ	and the second	
7380 11760 14760 18000 24000 103020	6.95E-07 8.08E-07 9.10E-07 1.05E-06 1.25E-06	4.57E-07 5.95E-07 6.20E-07 7.41E-07	5.78 5.56	ased		1	14 = 9.95E-01	*
11760 14760 18000 24000 103020	8.08E-07 9.10E-07 1.05E-06 1.25E-06	5.95E-07 6.20E-07 7.41E-07	5.56	ro	1.E-06		×	
14760 18000 24000 103020	9.10E-07 1.05E-06 1.25E-06	6.20E-07 7.41E-07	5 76	ele	8.E-07		100	
18000 24000 103020	1.05E-06 1.25E-06	7.41E-07		es r	6 F-07	1		
24000 103020	1.25E-06		5.68	hom	4 5 07			
103020		8.53E-07	5.81	Ca (4.6-07			
	2.03E-06	NA	6.33	100	2.E-07			
					0.E+00		Nex entre and a rest of the second	
) 10	0000 20000 Time (sec)	30000
				Rear	ession And	livsis	10.04-0.05040.00041	Std.Error
$K_{diss} = r_{co} / A \bullet S, v$	where r_{cr} is the r	ate of Ca release		R Sau	Jare		0.995	NA
over time as de	etermined by line	ar regression, A is		Care	lease rate	(r.,)	3 36F-11	1.14F-1
(BET) X mineral	I mass), and S is t	he Ca		-		<i>v-1</i>		
stoiciometric co	oefficient of the r	espective mineral.		Diss. Re	nte Miner	al (Kaise)	5 07E-10	mol/m²s
					Uncerta	oty of Eit	1 725 11	mol/m ² s
Stoichiometry				7.000	Uncerta	nty of th	1.73C-11	hala
Moles of Mine	eral Released			Ana	ilyucai err	or bars tai	i within plot sym	DOIS
violes of milli	erur neieuseu	22.2.22		S	toichio	metry	Comparison	
Time (sec)	Moles Mineral	Moles Mineral			Phos	phorus	Calcium	
233322.2	Released (P)	Released (Ca)		1.E	-07	Private P		
3840	4.01E-08	6.05E-08		B 1.E	-07		i	
7380	6.52E-08	7.31E-08		SEO 1.F	-07			
11760	8.49E-08	8.51E-08		a Rel	08	111		
14760	8.86E-08	9.57E-08		le or	-00	2		
18000	1.06E-07	1.10E-07		6.E	-08			
24000	1.22E-07	1.32E-07		4.E	-08			
				S 2.E	-08			
determined fro	released (above)	based on P of Ca d of P or Ca divider	by	0.E	+00			
the appropriate	e stoichiometric f	factor of the respec	tive		0	10000	20000 300	000

Sto

Time (sec)	Moles Mineral Released (P)	Moles Mineral Released (Ca)		
3840	4.01E-08	6.05E-08		
7380	6.52E-08	7.31E-08		
11760	8.49E-08	8.51E-08		
14760	8.86E-08	9.57E-08		
18000	1.06E-07	1.10E-07		
24000	1.22E-07	1.32E-07		

Datasheet S35 Experiment MER-P5D

Mineral: M Batch ID: M	Aerrillite Ini AER-P5D Mine	itial pH: 4.91 eral Mass: 0.15	500g	Specific Sur Total Surf.	f. Area (B Area (BE	ET): 0.040	58m²/g 98 m²	
VA = no analy	sis					*		
Batch Results	(shaded data = t	he linear initial r	ate dat	a shown in the	e accompai	nying figure	2)	
Time (sec)	Ca (moles released)	P (moles released)	pН			Initial (Ca Release	
0	0	NA	4.91		1.E-06	y =	4.17E-11x + 3.	.78E-07
3840	4.64E-07	NA	5.10	÷	1.E-06		$R^2 = 9.13E-0$)1
7320	7.02E-07	NA	5.21	ased	1.E-06			
11160	9.49E-07	NA	5.25	e e	8.E-07		/	
14640	1.01E-06	NA	5.43	es l	6.E-07	/	*	
18660	1.08E-06	NA	5.65	Ê	4 F-07			
101820	2.25E-06	NA	5.95	C	2 5.07			
523200	3.52E-06	NA	6.28		2.00			
					U.E+00	i.	10000	
							10000	20000
							Time (sec)	104 - 104
WITE ALL				Regr	ession Ana	lysis		Std.Error
N _{diss} = r _{cd} /A•S, over time as c	where r _{ce} is the ra letermined by linea	ar repression. A is		R Squ	uare		0.913	NA
the total BET :	surface area (specif	fic surface area		Ca re	elease rate	(rcs)	4.17E-11	7.41E-1
(BET) X miner stoiciometric	al mass), and S is th coefficient of the re	ne Ca espective mineral.	,	Diss. Ro	ate Miner	al (K _{aisa})	6.28E-10	mol/m ² s
					Uncertai	nty of Fit	1.12E-10	mol/m s
				Ana	alytical erro	or bars fall	within plot syr	nbols

APPENDIX III SUPPLEMENTARY INFORMATION FOR CHAPTER FOUR

Supplementary Information

Analysis of Weathering Features in Mars Analog Basalts at Craters of the Moon National Monument, Idaho.

Introduction

Craters of the Moon National Monument (COTM) is located on the Snake River Plain of Idaho in the northwestern United States (Figures S5 and S6, this Appendix). The area is the locale of 60 or more alkaline lava flows and 25 cinder cones (Kuntz et al., 1986a; Vaughan, 2008; Vaughan et al., 2011). The first detailed description of the area was conducted by Stearns (1924; 1928) for the Idaho Bureau of Mines and Geology. Stearns (1928) described the biota, geology, geomorphology, and reported the first chemical analysis of COTM basalt, although it is not spatially located or assigned to a specific lava flow. The area receives 240 - 380mm (9.5 - 15") mean annual precipitation moving from south to north (Stearns, 1928; Vaughan, 2008; Kukachka, 2010; Vaughan et al., 2011) making it an arid to semi-arid environment. Average monthly high temperatures range from -1.7 to 29 °C (29 - 84°F) and average lows span -12 to 11 °C (10 to 52° F) with approximately 7 months of lows in the sub-freezing range (Kukachka, 2010). Since the initial reports by Stearns, the chemistry and mineralogy of many COTM flows have been well documented (Leeman et al., 1976; Kuntz et al., 1985; Kuntz et al., 1992; Stout et al., 1994) and a number of flows at COTM have been dated and range in age from $\sim 2,000$ to > 18,000 years bp (Kuntz et al., 1986b; Kuntz et al., 1992; Vaughan et al., 2011). The flows have experienced various degrees of surface weathering over time, and thus represent an arid to semi-arid environment chronosequence (Vaughan, 2008).

Many of the alkaline basalts at COTM posses high concentrations of FeO, TiO₂, and P₂O₅ (up to 2.9 wt% P₂O₅) (Stearns, 1928; Leeman et al., 1976; Kuntz et al., 1992; Stout et al., 1994). The origins of the high-P basalts at COTM are not conclusively understood. High large-ion lithophile element (LILE) and rare earth element (REE) abundances relative to regionally close olivine basalts of the Snake River Plain suggest they are not formed by crystal fractionation from the same parent magma source of Snake River Plain olivine basalts (Usui et al., 2008). Heterogeneous mantle sources, crustal contamination, or a combination of both have been proposed as possible explanations of the unusual chemistry (Kuntz et al., 1992; Stout et al., 1994; Reid, 1995).

The unusual chemistry of COTM basalts led Usui and others to suggest COTM basalts might serve as useful terrestrial analogs for high-P bearing Wishstone class rocks on Mars (Usui et al., 2008). This martian rock class appears to comprise the dominant population of float rocks encountered by the Mars Exploration Rover (MER) Spirit on Cumberland Ridge and the northwest flank of Husband Hill within Gusev Crater on Mars (Arvidson et al., 2006; Squyres et al., 2006). No source or outcrop for the rocks has been identified. Alpha Particle X-ray Spectroscopy (APXS) analyses by MER Spirit of two Wishstone class rocks (Wishstone and Champagne) showed the rocks, like COTM basalts, to be typified by high Ti (~2.5-3% as TiO₂) and P (~5% as P₂O₅) content (Gellert et al., 2004; Ming et al., 2006). The chemical similarities of terrestrial COTM basalts and martian Wishstone class rocks suggest potential petrogenetic/petrographic similarities between the rocks as well.

As outlined in the accompanying chapter, the high-P chemistry of Wishstone class rocks (much like COTM basalts) make Wishstone petrogenesis and mineralogy of

particular geologic interest in martian studies. Further, these rocks show evidence of both aqueous alteration and the loss of a phosphate mineral through dissolution. Both the water responsible for the alteration of the rock class and the phosphorus released from them have astrobiological implications and may hold insights into the potential for life to have developed and persisted on Mars. In the absence of a physical sample of P-rich Wishstone class martian rocks, investigation of chemically similar rocks of COTM as a Martian analog can lead to insights into phosphate behavior over longer time-scales than can be observed in the laboratory.

In this study, Scanning Electron Microscopy (SEM) using Backscattered Electron (BSE) Imagery, Energy Dispersive Spectroscopy (EDS), and EDS X-Ray mapping was utilized to investigate COTM basalts in thin section. The objective of this study was to use COTM observation and analysis of weathering effects including the development of weathering rinds, porosity, mineral dissolution textures, primary phosphate mineral dissolution, and the abundance of secondary phosphate minerals, to gain insights that might be applicable to Wishstone class rocks on Mars and to inform reactive transport modeling.

Methods

Materials

Fifteen petrographic thin sections were prepared from samples of both buried and exposed material from six COTM flows. Sampled flows ranged in age from 1,670 to 15,100 years before present and Figure S6 documents sample locations and flow ages. Representative pieces of flow samples were broken from the main sample and oriented in

epoxy mounts so that subsequent thin-sectioning would produce sections which included exposed/weathered surfaces. Thin-sections were prepared by Spectrum Petrographics Inc of Vancouver, Washington to 0.3 μm polish using anhydrous cutting and polishing methods and epoxy impregnation in an effort to preserve weathered material and soluble minerals.

Analytical

SEM analyses were performed using a JEOL JSM 5600 (EMiL Facility, UNLV) at 20 kV and 30-40 spot size. Before analyses, thin-sections were carbon coated to 20 nm thickness for conductance. SEM analyses focused on textural evidence of any preferred mineral dissolution, evidence of secondary minerals, development of weathering rinds and porosity development. The chemistry and mineralogy of many COTM flows have been well documented previously (Leeman et al., 1976; Kuntz et al., 1985; Kuntz et al., 1992; Stout et al., 1994) and were not a focus of our study, however, EDS analyses were carried out using an Oxford Instruments detector and Oxford INCA software with semiquantitative capabilities to confirm the general mineralogy and phases present.

Image analysis

In addition to image analysis performed using Oxford Instruments INCA software installed on the JEOL JSM 5600 SEM, subsequent off-line image analysis was performed using Oxford Instruments INCA and Adobe Photoshop CS6 software. Relative changes in porosity and fracturing with depth were investigated using image BSE mosaics collected from the weathered surface of the samples to unweathered core material. Plots

of porosity with depth were constructed from mosaics by first aligning, rotating, and cropping the mosaic such that 10 µm "slices" of equal areas of the sample image perpendicular to the weathered sample surface could be consecutively isolated, contrast enhanced, and open micro-porosity areas could be determined by histogram. This was done for one mosaic from a sample of each flow to a depth of 1200 µm (well into unweathered corestone). Figure S7 is an example of how mosaics were analyzed. Macro pores (>100 µm on a side) were filled with a color that would not overlap the pore color in the histogram. Histogram pore space was used to obtain a percent porosity value associated with the depth of the particular "slice". The same porosity image analysis techniques were used to get general unweathered porosities of the samples for informing the modeling. The depth of enhanced porosity within samples was determined in three ways. First, subjectively (i.e. "by eye") in BSE mosaics, then subjectively in plotted profiles, and finally the plotted profiles were subjected to more quantitative analysis by determining when three consecutive slices gave measured porosities of less than the average porosity plus the standard deviation of the deepest 300 µm (unweathered corestone). For the Kimama sample, image analysis was applied to measure the grain sizes of minerals in multiple BSE images of unweathered corestone for use in surface area estimates used in reactive transport modeling of the accompanying paper.

<u>Results</u>

Textures in thin section of observed high-P basalts from COTM range from vitrophyric to poikiolitic with the vitrophric textures being restricted to the youngest flows. Figure S8 contains BSE images of textures typical of the 6 flows sampled. In

some samples weathering appeared to favor glassy matrix dissolution over other phases, while in other samples (which also included a glass component) weathered surfaces showed no preferred mineral dissolution. Rinds were present on the exposed surfaces of all samples, but were never continuous and generally never exceeded 100 µm. No conclusive correlation between rind thickness and age was apparent. The youngest sampled flow, Blue Dragon, appears to have weathering-produced porosity (spall type cracks and holes) up to 60 um into the sample and in many places an alteration rind of $\sim 10 \,\mu\text{m}$ depth into the glassy matrix has formed (Figure S9). The chemistry of the rind by EDS compared to the unaltered glass does not appear significantly different, thus the lower BSE signal (darker color) in the image may be a density change. This type of rind was unique to the Blue Dragon samples and may play some role in the blue sheen that is characteristic of these basalts in hand sample. Both intermediate aged Lava Point and Pronghorn samples showed variable porosity and weathering textures. Porosity to depth measurements on these samples (discussed further below) were applied to a generally representative area of the samples. Thin-section samples of the oldest flows, Sunset and Kimama show more advanced porosity development than most of the other samples, which was due primarily to the dissolution of the glassy matrix.

Extensive effort focused on detecting any secondary phosphate minerals that were present. A few Al- and Fe-secondary phosphate phases had been previously observed in samples of Pronghorn and Kimama flows (Cornell and Hausrath, 2010). In this study secondary phosphate minerals were found (Figure S10), but were also very rare and restricted to within a few tens of micrometers of the weathered surface.

Part of this study focused on determining if the primary phosphate mineral in COTM samples (fluorapatite) was preferentially dissolving. Though some fluorapatite minerals within samples exhibited a texture that suggested they may be dissolving preferentially, the texture could also be explained as glassy matrix dissolved out of the interior of skeletal apatite (Figure S11). Further, in some cases, relatively pristine fluorapatite crystals were present surrounded by porosity created by the apparent dissolution of glassy matrix. It is of note in regards to the accompanying paper, that fluorapatite is not the expected primary phosphate mineral in martian rocks and has both different kinetics and thermodynamics than Mars-relevant primary phosphate minerals (i.e. Cl-rich apatite and merrillite) (Adcock et al., 2013).

Measurements of porosity from weathered surface to the interiors of samples indicated that increases in fractures and porosity occur to depths well beyond any obvious chemical alteration. Further, a correlation of depth of change in porosity to time does seem apparent, with older flows having enhanced porosity deeper into the rock (Figure S12). For use in informing reactive transport modeling in the accompanying paper, the porosity of unweathered corestone varied from 0.1 % to 1.03 % with an average value of 0.7%. While these values seem low, the technique used here does not measure macroporosity and the value of 0.7% is within range for basalts as measured by others (see Table 9 of accompanying paper). Average porosities of weathered material (surface to 100 µm deep) ranged from 0.6 - 7.5% with an overall average of 3.8%

Reactive transport modeling performed in the accompanying paper requires specific surface areas as mineral dissolution rates are normalized to surface areas. To estimate surface area for this study, grain size measurements taken from COTM Kimama

lava flow thin sections were used to derive grain diameters for the minerals ilmenite, plagioclase, pyroxene, and the primary phosphate mineral fluorapatite. To do this we measured both long and short axes of mineral grains and used them to determine an average geometric diameter (Table S8). These geometric diameters were then used with roughness values from the literature to estimate BET surface area, and estimate total reactive surface area to derive specific surface areas (SSA) for use in modeling. Because Kimama does not contain appreciable pyroxene, we assumed pyroxene grain diameters to be similar to plagioclase, and we assumed that the merrillite and chlorapatite grain sizes would be similar to fluorapatite.

Summary Summary

Despite variations in texture, a correlation between age and depth of weatheringproduced porosity in COTM flows is apparent. However, no correlation between age and weathering rind thickness could be confirmed. Secondary phosphate minerals were located in samples but were rare and not a significant phase. Porosity in unweathered corestone (excluding macro porosity) measured in this work is within the range observed in other basalts. With the exception of the glassy matrix of some samples, no preferred mineral dissolution was observed.

Plagioclase	5		Ilmenite			Fluorapat	ite		Enstatite		
x	У	ave	x	У	ave	x	У	ave	х	У	ave
148	123	135.5	172	60	116	20	20	20	114	60	87
132	60	96	73	56	64.5	23	17	20	60	45	52.5
321	99	210	90	50	70	80	30	55	60	40	50
200	60	130	72	30	51	84	46	65	84	48	66
204	50	127	60	60	60	30	25	27.5	108	75	91.5
273	64	168.5	35	35	35	30	22	26	50	50	50
500	82	291	87	78	82.5	48	33	40.5	164	91	127.5
400	100	250	84	40	62	40	29	34.5	136	105	120.5
400	135	267.5	51	35	43	37	23	30	60	52	56
188	30	109	42	24	33	51	24	37.5	90	54	72
Ave. Diameter (µm)		178.45		61.7		35.6					77.3
Radius (µn	n)	89.225			30.85			17.8			38.65
Radius (m)		8.92E-05			3.09E-05			1.78E-05			3.87E-05

Table S8. Long and short axis measurements and averaged diameter of grain in Kimama thin sections (μm). Values were used to obtain geometric surface areas for modeling.

Initial pH	Merrillite Front	Chlorapatite Front	Distance between fronts
4	1775	2275	500
5	1475	2075	600
6	1175	1875	700
7	825	1475	650

Table S9. Depth of Merrillite front, Chlorapatite front, and the distance between them

Depths are in micrometers. Data is from models run for 100,000 years of rock water interaction. Distance was measured from the deepest point of the merrillite dissolution front to the deepest point of the chlorapatite dissolution front.

Supplementary Figures



Figure S5. General location of Craters of the Moon National Monument within the state of Idaho



Figure S6. Sampling locations and associated flow and age for Craters of the Moon samples. USGS LANDSAT 8 false color image, bands 7-4-2 displayed as R-G-B. Image color enhanced by the author. Different dark colors in flow field represent different individual flows.



Figure S7. Example of image analysis of cross-section mosaics of COTM samples. This sample was from the Kimama flow.



Figure S8. BSE images of textures typical of the six flows sampled. A) Blue Dragon, B) Minidoka, C) Lava Point, D) Prong Horn, E) Sunset, F) Kimama



Figure S9. Blue Dragon (1670 bp). Note apparent alteration rind into glassy matrix. Sample BD1BCTA. Scale Bar 50μm



P Kat

Figure S10. EDS X-ray map of secondary phosphate mineral in Blue Dragon (same location as Figure S9). Note P bearing phase in lower left of P map which has no Ca bearing counter part but does appear to have Fe.



Figure S11. Fluorapatite grains like this one in Kimama appear in some cases to be dissolving fluorapatite, but may also be skeletal with glassy matrix dissolving from within.



Figure S12. Plot of estimated depth of weathering induced porosity versus time.



Figure S13. Example of output model profile. Plot shows % mineral volume with depth from the surface of a modeled rock to the interior. Colored lines represents mineral concentration in the rocks as % volume. Dashed line at top represent original surface of modeled rock. Curves in lines are mineral dissolution fronts. At depths above the curves, the mineral has dissolved and the % mineral volume in the rock has decreased. Below the curves the minerals are undissolved and the % mineral volume is that of the parent rock.



Figure S14. The effects of +/-2 orders of magnitude change in merrillite solubility (pH 6, 100,000 years model time). 1) Merrillite solubility suppressed by 2 orders shows a slightly shallower dissolution depth. 2) Using the measured value of solubility for merrillite (Adcock et al., 2013). 3) Depth of dissolution front using merrillite with 2 orders of magnitude enhanced solubility.



Figure S15. NASA image PIA07191: Spirit view of 'Wishstone' (False Color). Image shows RAT scar on rock. Wishstone has the appearance of rounded edges and evidence of facets that indicate physical erosion has occurred. Image credit: NASA/JPL/Cornell



Figure S16. Data from MER Product 2N 156291204 EFF A0 00 P0665 R0 M1. Wishstone appears in the central left and exhibits streaking that developed post RAT operations. Image credit: NASA/JPL/Cornell/USGS



Figure S17. Modeled dissolution profiles for chlorapatite models at pH 2-8.



Figure S18. Modeled dissolution profiles for merrillite models at pH 2-8.



Figure S19. Modeled dissolution profiles for fluorapatite models at pH 2-8.



Figure S20. Modeled dissolution profiles and Al τ plots for chlorapatite + merrillite models at pH 2-8.

APPENDIX IV ADDITIONAL SUPPLEMENTARY DATA

This fourth appendix contains data in addition to that associated with the chapter manuscripts.

Though MER Spirit carried and extensive science package capable of chemical and some mineralogical analysis, the unusual chemistry of Wishstone class rocks combined with the fact that they are float rocks has made mineralogical and petrogenetic determinations challenging. Table S10 is a compilation of the many Mars Wishstone rock class mineralogy estimates made based on differing assumptions and data set combinations. The broad range of mineral compositions is both an indicator of the uncertainty in the rock class mineralogy and its petrogenesis, and the interest in the somewhat chemically unusual Wishstone class rocks.

Though the high-P content in Wishstone class is unusual, high-P rocks do exist on Earth and the high-P basalt flows of Craters of the Moon national Monument (COTM) are among the rocks suggested as possible analogs for Wishstone class rock (Usui et al., 2008). Elisabeth Hausrath collected a number of samples from COTM from select lava flows and the author had a total of 15 thin sections of these samples, or sub-samples previously epoxy mounted, from six different lava flows prepared (Table S11). Thin sections were investigated by SEM to help inform modeling of Wishstone class rocks, and likely more than 1000 images were acquired. The images that follow (and occasional EDS spectra) in this appendix were selected from the total catalog of images to document interesting trends that may be helpful to the interpretation of phosphate mobility in arid environments.
Mineral	Ming et al. Oct. 2005 Wish. ^a	Ming et al. Oct. 2005 Champ. ^a	Ming et al. Oct. 2005 Champ. RAT2 ^a	Ming et al. Oct. 2005 "Wishstone Class" combined APSX, Mossbauer, Mini-TES ^a	McSween et al. Sept. 2006 Wish. ^b	McSween et al. Sept. 2006 Champ. ^b	Ruff et al. Dec. 2006 Mini-TES Data ^c	Hurowitz et al. Dec. 2006 ^d	McSween et al. May 2008 Wish. CIPW SO ₃ /Cl corr. ^e	McSween et al. May 2008 Champ. CIPW SO ₃ /Cl corr. ^e
Feldspars	57.23	57.67	57.57	52	56.7	56.7	55	51.4	55.7	56.1
Albite	43.31	43.76	43.65		42.9	42.9		37.9	42.1	42.5
Anorthite	10.43	10.71	10.72		10.4	10.6	55	10.4	10.2	10.5
Orthoclase	3.49	3.2	3.2		3.4	3.2		3.1	3.4	3.1
Illmenite	5.06	5.82	5.76	1	5.0	5.7		5.6	4.9	5.6
TiO2				1						
Corundum	2.52	2.22	2.22		2.4	2.2		3.1	2.4	2.1
Quartz						0.5		3.4		0.4
Magnetite	1.15	1.24	1.24	1	7.6	9.2		7.8	7.5	9.1
Hypersthene (OPX)	5.08	3.66	3.78	12	13.5	13.0		13.4	13.7	13.0
Enstatite									9.7	9.9
Ferrosilite									4.0	3.1
Merrillite										
Apatite	12.35	12.03	11.98	11	12.5	12.1	10	11.7	12.3	12.0
Olivine*	16.59	17.36	17.45	6	1.9	0.0	15	0	1.5	
Basalt Glass							10			
Secondary Minerals				14			10	2.5		
Totals	99.98	100	100	98	99.6	99.4	100.0	98.9	98.0	98.3

Table S10. Mineralogy calculations of Wishstone Class rocks from various studies.

Supplementary Tables

Mineral	McSween et al. May 2008 Wish. MgSO₄/Na Cl corr. ^e	McSween et al. May 2008 Champ. MgSO ₄ /NaCl corr. ^e	McSween et al. May 2008 Wish. CIPW and recalculated Mossbauer ^e	McSween et al. May 2008 Champ. CIPW and recalculated Mossbauer ^e	Usui et al. Sept. 2008 Wish. CIPW of APXS data ^f	Usui et al. Sept. 2008 Extrapolated Wish. (SO ₃ removed) ^f	Usui et al. Sept. 2008 Champ. CIPW of APXS data ^f	Usui et al. Sept. 2008 Extrapolated Champ. (SO ₃ removed) ^f	Smith, 2014, Wish. mode ^g	Smith, 2014, Champ. RAT1 mode ^g	Smith, 2014, Champ. RAT2 mode ^g
Feldspars	53.4	51.5	55.7	56.1	63.4	57	63.3	53.1	60.5	60.9	60.8
Albite	39.7	37.9			43.1	41	43.4	37.5	46.4	46.7	46.7
Anorthite	10.3	10.5			16.8	12.4	16.7	12.3	10.6	10.9	10.8
Orthoclase	3.4	3.1			3.5	3.6	3.2	3.3	3.5	3.3	3.3
Illmenite	4.9	5.6	1.9	1.4	5	5.9	5.8	6.3	3.7	4.3	4.3
TiO2											
Corundum	2.9	3.0			0.2	1.2		0.7	2.8	2.5	2.5
Quartz	2.8	5.1						1.6			
Magnetite	7.5	9.1	1.7	1.3	7.7	7.5	9.3	9.1	4.4	4.9	4.6
Hypersthene											
(OPX)	13.1	10.5	17.0	19.3	5.1	6.7	6.2	7	11.9	12.2	12.4
Enstatite	8.4	7.4									
Ferrosilite	4.7	3.1									
Merrillite					11.4	17.6	11.1	22.1			
Apatite	12.3	12.0	12.3	12.0					11.2	11	11
Olivine*			7.6	7.0	7.3	4	4.2		5.5	4.1	4.2
Basalt Glass											
Secondary Minerals			3.9	2.9							
Totals	96.9	96.8	100.1	100.0	100.1	99.9	99.9	99.9	100	99.9	99.8

Table S10. Continued

^aMing et al., 2006, ^bMcSween et al., 2006, ^cRuff et al., 2006, ^dHurowitz et al., 2006 ^eMcSween et al., 2008, ^fUsui et al., 2008, ^gEugene Smith, 2014, unpublished data calculated using methods after Irvine and Baragar, 1971.

Sample	Description	Flow
ID	Description	Age
BD1ACTA	Blue Dragon from hand sample	1,670
BD1BCTA	Blue Dragon from hand sample	1,670
BD1CCTA	Blue Dragon from hand sample	1,670
M2CTA	Minidoka from hand sample	3,590
МЗСТА	Minidoka from hand sample	3,590
LP2CTA	Lava Point from hand sample	7,840
LP3CTA	Lava Point from hand sample	7,840
P331JC	Pronghorn from Jason Cornell epoxy mount	10,240
P42JC	Pronghorn from Jason Cornell epoxy mount	10,240
P4ACTA	Prong horn from hand sample	10,240
P4BCTA	Prong horn from hand sample	10,240
LC12CTA	Sunset from hand sample	12,010
LC31JC	Sunset from Jason Cornell epoxy mount	12,010
K24JC	Kimama from Jason Cornell epoxy mount	15,100
K31JC	Kimama from Jason Cornell epoxy mount	15,100

Table S11. Thin section descriptions.

Supplementary Figures



Figure S21. Blue Dragon BD1ACTA weathered edge with a fracture. Scale 100um



Figure S22. Blue Dragon sample BD1ACTA. An apatite sits inside the inclusion of the olivine in the low-center of the image. Scale 60um.



Figure S23. Blue Dragon sample BD1ACTA fracture. This is a fracture that shows some weathering. Note the dissolution texture of the apatite above the fracture and the apatite on the fracture edge showing little sign of weathering. Scale 60um.



Figure S24. Blue Dragon sample BD1ACTA dark rim. Dark Rim is Si/Al material similar to the glassy matrix but depleted in Ca and Fe compared to interior material. Could this be the "Blue" of Blue Dragon? Scale 90um. The dark "smudges" are contamination of the sample and are not part it.



Figure S25. Blue Dragon sample BD1ACTA dark rim. Dark rim is Si/Al material (glass maybe) but depleted in Ca and Fe compared to interior material. Not eroded rind cross cutting olivine. There are secondary phosphates here P/Fe (in rind). What looks like carbonate is the dark rind forming under the rim (under the plag). Scale 50um.



30µm Electron Image 1



Figure S26. Blue Dragon sample BD1ACTA and EDS. Image, EDS analysis location, and EDS spectra indicating Secondary Phosphates in "pyroxenish" material



Figure S27. Minidoka Sample M2CTA. Weathering rind, note the cross-cutting. These rinds were discontinuous in samples and varied from 40-100 microns when present. Scale: 90um.





Figure S28. Minidoka Sample M2CTA and EDS. Possible secondary phosphate. This type of texture was rare in the Minidoka samples



Figure S29. Minidoka Sample M2CTA edge. Note alteration rim on edge.



Figure S30. Minidoka sample M2CTA edge. Alteration on exposed edge of sample.



Figure S31. Minidoka sample M2CTA. More alteration textures



Figure S32. Minidoka sample M2CTA alteration. Alteration textures are on exposed surface.



Figure S33. Lava Point sample LP2CTA. Alteration rind and texture indicating increase porosity and dissolution



Figure S34. Lava Point sample LP2CTA interior. Possible dissolution textures and increased prosity in the interior of the sample.



Figure S35. Lava Point sample LP3CTA. Possible alteration rind on exposed egde of sample, however, these same texture could also be surface deposits. Scale: 100um



Figure S36. Pronghorn sample P31JC interior. Possible dissolution textures and increased prosity in the interior of the sample.



Figure S37. Pronghorn sample P31JC. Possible dissolution textures and increased prosity in the interior of the sample, but also either an alteration rind or a deposit on the original sample surface.



100µm Figure S38. Pronghorn sample P42JC. Fluorapatites (large medium grey grains) near original exposed surface of sample. Long grain in the right of image is a fluorapatite grain with the long C axis near parallel to the thinsectrion surface.



Figure S39. Pronghorn sample P4ACTA. Fluorapatites (medium grey grains) near an altereation rind. Note the dissolutoin texture of the interior of the sample.





Figure S40. Pronghorn sample P4ACTA and EDS. Image and EDS analysis of rim around porosity. Analysis shows phosphorus.



^{300µm} **Figure S41. Pronghorn sample P4ACTA dissolution**. Dissolution textures near original surface of sample. Not the apatite in the upper center of image (large medium grey grain) which show little sign of dissolution.



Figure S42. Sunset sample LC12 JC. Possible alteration rind or surface deposit.



100µm Figure S43. Kimama sample K24JC. Dissolution textures in interior and possible alteration rind or deposit on original mineral surface.



10µm Figure S44. Kimama sample K31JC. Relic Apatite, or more likely skeletal texture.





200µm Figure S46. Kimama sample K31JC surface. This is most likely a surface deposit (right side of image) rather than the product of alteration or weathering.

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- Zolotov, M. Y., and Mironenko, M. V., 2007, Timing of acid weathering on Mars: A kinetic-thermodynamic assessment: J. Geophys. Res., v. 112.

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EDUCATION

University of Nevada Las Vegas, Las Vegas, NV Advisor: Dr. Elisabeth Hausrath	2010 – 2014 (expect.)
PhD graduate candidate in Geoscience (Planetary geochemistry and astrobiology)	
Oklahoma State University , Stillwater, OK Advisor: Dr. Richard Marston M.S., Geology, (Planetary geomorphology and martian remote sensir	2004 ng)
University of New Mexico , Albuquerque, NM B. Sci., Earth and Planetary Sciences with Distributed Sciences mino	1995 r.
PROFESSIONAL EXPERIENCE	
University of Nevada Las Vegas, Department of Geoscience Laboratory Assistant Research Assistant Laboratory Teaching Assistant, GEOL 101L	Aug. 2009 – Present
Zenitech Environmental, LLC Scientist II/Certified Environmental Manager, Surface processes, geo and remote sensing research to characterize mining remediation proje	Feb. 2005 – Aug. 2009 morphology, geology, ect sites.
Oklahoma State University, T. Boone Pickens School of Geology Teaching Assistant, GEOL 1114L Research Assistant for the Arkansas/Oklahoma Center for Space and Research Assistant for Research Experience for Undergraduates (REU Research Assistant (technician, system admin.), Dept. of English, Wr	Aug. 2002 – Dec. 2004 Planetary Sciences U) program. iting Center.
Sandia National Laboratories (Contracted to SNL by OnSite) <i>Senior Technologist,</i> electron / optical microscopist (EMP/SEM)	lan. 2001 – Aug. 2002
Isonics Corporation (formerly Colorado Mineral Research Institute) Senior Research Technician, chlorine handling systems development methods from recycled materials, pilot plant operation, development techniques.	Jun. 1999 – Jan. 2001 , metal extraction of solvent extraction
University of New Mexico Institute of Meteoritics Meteoritics Physical Sciences Research Technician V, scanning electron and electron and electron electron sciences.	May 1993 –April 1999 etron microprobe

PEER REVIEWED PUBLICATIONS AND THESIS

<u>Adcock, C.T.</u>, Hausrath, E.M., Forster, P.M., Tschauner, O., and Sefein, K.J. (2014). "Synthesis and characterization of the Mars-relevant phosphate minerals Fe- and Mgwhitlockite and merrillite and a possible mechanism that maintains charge balance during whitlockite to merrillite transformation." Am. Min., 99, 1221-1232.

Adcock, C., Hausrath, E., Forster, P., 2013. "Readily available phosphate from minerals in early aqueous environments on Mars." Nat. Geosci. 6, 824-827.

Adcock, C.T., 2004. Determining Formative Winds Through Geomorphology: Herschel Crater, Mars. Oklahoma State University, Thesis.

Shearer C.K., Leshin, L.A. and <u>Adcock C.T.</u> (1999). "Olivine in Martian meteorite ALH 84001. Evidence for a high-temperature origin and implications for signs of life." Meteorit. Planet. Sci. 34, 331-340.

J.J. Papike, G.W. Fowler, <u>C.T. Adcock</u>, and C.K. Shearer (1999). "Systematics of Ni and Co in olivine from planetary melt systems: lunar mare basalts." Am. Min., Mar 1999; 84: 392 – 399

Papike, J.J., Spilde, M.N., <u>Adcock, C.T.</u>, Fowler, G.W., Shearer, C.K., (1997). "Trace element fractionation by impact-induced volatization: SIMS study of lunar HASP samples" Am. Min., 82, 630-634

EXTENDED ABSTRACTS AND PRESENTATIONS

<u>Adcock, C.T.</u>, Hausrath, E.M., 2014. Reactive Transport Modeling of Phosphate Mineral Dissolution in High-P Martian Rocks, 45th Lunar and Planetary Science Conference. Lunar and Planetary Institute, Houston, p. Abstract #2250.

Hausrath, E.M., <u>Adcock, C.T.</u>, Elwood Madden, M.E., Gainey, S.R., Olsen, A.A., Steiner, M.H., 2014. Using Geochemical Kinetics to Interpret Potential Habitability, 45th Lunar and Planetary Science Conference. Lunar and Planetary Institute, Houston, p. Abstract #2376

Adcock, C., Hausrath, E., 2013. Interpretation of Phosphate Mobility on Mars Based on Terrestrial Mars- analog Basalts and Reactive Transport Modeling, Lunar and Planetary Institute Science Conference Abstracts, p. 2727.

<u>Adcock, C.</u>, Hausrath, E., 2012. The Dissolution Rate of Whitlockite and Implications for the Habitability of Early Mars, Lunar and Planetary Institute Science Conference Abstracts, p. 2446.

Hausrath, E., <u>Adcock, C.</u>, Tu, V., 2012. Phosphate Records Environmental Conditions Important to Habitability in Soils and Rocks on Mars, Lunar and Planetary Institute Science Conference Abstracts, p. 2719.

Adcock, C., Hausrath, E., 2011. Dissolution Rates and Mineral Lifetimes of Phosphate Containing Minerals and Implications for Mars, AGU Fall Meeting Abstracts, p. 1708.

<u>Adcock, C</u>., Simon, A., Hausrath, E., 2011. Synthesis of Phosphate Minerals for Use in Dissolution Experiments, Lunar and Planetary Institute Science Conference Abstracts, p. 2300.

<u>Adcock, C.</u>, Hausrath, E., 2010. Kinetic Studies of Phosphate Containing Minerals and Implications for Mars, Lunar and Planetary Institute Science Conference Abstracts, p. 2177.

Stinchfield, M. O., <u>Adcock, C. T.</u>, (2006) "When Nature Collides: Remediation Complications Due to Natural Presence of Lead and Arsenic" Presentation (Adcock). Air and Waste Management Association, June, 2006, Great Basin Chapter Annual Conference.

B. D. Stanley, <u>C. T. Adcock</u>, R. A. Marston, (2004). "Interpretation of Wind Direction from Eolian Features: Herschel Crater, Mars." Abstract and Poster Presentation. XXXV Lunar and Planetary Sci. Conf., CD-Rom.

<u>C. T. Adcock</u>, B. D. Stanley, R. A. Marston, (2004). "Comparison of Geomorphically Determined Winds with a General Circulation Model: Herschel Crater, Mars." Abstract and Poster Presentation. XXXV Lunar and Planetary Sci. Conf., CD-Rom.

B. D. Stanley, <u>C. T. Adcock</u>, R. A. Marston, (2003). "Interpretation of Wind Direction from Eolian Features: Herschel Crater, Mars." Poster Presentation. Arkansas and Oklahoma Center for Space and Planetary Sciences Summer Research Experience/Experiment for Undergraduates (REU).

<u>Adcock, C. T.</u>, (2003) "Characterizing Surface Winds on Mars: Herschel Crater." Poster Presentation. 2003 Meeting of the Southwest Association of American Geographers, Stillwater, Oklahoma, Oct. 23-25, 2003.

Newsom H.E., Hagerty J.J., Spilde M.N., <u>Adcock C.T.</u> and C. Sorge (1999). "Introducing planetary science and technology to students from grades 6-12." Abstract and Poster Presentation. Abstract and Poster Presentation. XXX Lunar and Planetary Sci. Conf., CD-Rom

J.J. Papike, G.W. Fowler, <u>C.T. Adcock</u>, and C.K. Shearer (1999). "Systematics of Ni and Co in olivine from planetary melt systems: lunar mare basalts." Presentation (Papike). XXX Lunar and Planetary Sci. Conf., CD-Rom

Wacker, J., Hildebrand, A., Brown, P., Crawford, D., Boslough, M., Chael, E., Revelle, D., Doser, D., Tagliaferri, E., Rathbun, D., D. Cooke, <u>C. Adcock</u> 1998. The Juancheng and El Paso Superbolides of February 15 and October 9, 1997: Preatmospheric Meteoroid Sizes. Meteoritics and Planetary Science Supplement 33, 160

Shearer C.K. and <u>Adcock C.T.</u> (1998) "A comparison between sulfide assemblages in Martian meteorites Allan Hills 84001 and Governador Valadares" In Workshop on the Issue Martian Meteorites: Where Do We Stand and Where Are We Going? LPI contribution No. 956. Lunar and Planetary Institute, Houston TX. 64 pp.

Shearer, C.K. and <u>C.T. Adcock</u> (1998) "The relationship between the carbonate and shockproduced glass in ALH 84001." Abstract and Poster Presentation. XXIX Lunar and Planetary Sci. Conf., CD-Rom

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A.R. Hildebrand, P. Brown, D. Crawford, M. Boslough, E. Chael, D. Revelle, D. Doser, E.Tagliaferri, D. Rathbun, D. Cooke, <u>C. Adcock</u>, and J. Karner; (1998) "The El Paso Superbolide of October 9, 1997." Presentation (Hildebrand). 1998 Hypervelocity Impact Symposium

Papike, J.J., Spilde, M.N., <u>Adcock, C.T.</u>, Fowler, G.W., Shearer, C.K., (1997). "Trace element fractionation by impact-induced volatization: SIMS study of lunar HASP glasses" Presentation (Papike). XXVIII Lunar and Planetary Sci. Conf., 1059P

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<u>Adcock, C.T.</u>, Spilde, M.N., Papike, J.J., (1997). "Automated HASP glass search using the electron microprobe." Abstract and Poster Presentation. XXVIII Lunar and Planetary Sci. Conf., 5A.

INVITED TALKS

<u>Adcock, C. T.</u>, Stinchfield, M. O., (2007) "When Nature Collides: Remediation Complications Due to Natural Presence of Metals" Presentation and Fieldtrip (Adcock). Association of Engineering and Environmental Geologists, January, 2007, Southern Nevada Chapter meeting.

GRANTS, HONORS, AND AWARDS

UNIVERSITY OF NEVADA LAS VEGAS

Nevada Space Grant NASA Fellowship, 2010, 2011, 2012, 2013 Bernada French Scholarship, 2012 Geological Society of America Research Grant, 2011 Edwards & Olswang Geology Scholarship, 2011, 2013, 2014 Desert Space Foundation Geoscience Award, 2011, 2012, 2013, 2014 Member of the Phi Kappa Phi Honors Society

OKLAHOMA STATE UNIVERSITY

Rumsey Bissell Marston Scholarship, 2004 Oklahoma Space Grant NASA Fellowship 2004 Invited Attendee of the NASA/JPL Planetary Science Summer School 2004 Departmental Outstanding Teaching Award 2003-2004 Member of the Phi Kappa Phi Honors Society, 2004 Graduate member of the Sigma Gamma Epsilon Honorary Society, 2003-2004

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Harry and Mabel Leonard Scholarship, 1994 General Thomas Campell Award, 1993

RECENT WORKSHOPS

Photogrammetric Processing of Planetary Stereo Imagery using ISIS and SOCET SET® December 17-19, 2013, USGS Astrogeology Science Center, Flagstaff. AZ.

SoilTrEC/Critical Zone Workshop on Reactive Transport Modeling July 16-19, 2012, Technical University of Crete, Chania, Greece

ADDITIONAL TRAINING AND EXPERIENCE

State of Nevada Certified Environmental Manager (CEM #1959), 2006-2013; OSHA 40-Hour HAZWOPER (29 CFR § 1910.120), 2005 - Most recent refresher: 2009; US EPA Lead Inspector (Certification #NV-I-16411-1), 2008-2012; Electron Microscopist with operator hours on JEOL 733,

8600, 8200 Electron Microprobes as well as JEOL 840, 5600, 5400, and 6400 series Scanning Electron Microscopes; UV/VIS, Flame AA, and XRD operation experience; Programming experience in Python and FORTRAN.