Optimisation of Radium Removal from Saline Produced Waters during Oil and Gas Extraction

Joel Garner 1 and David Read 2,3,*

1 Aurora Health Physics, Harwell Oxford, Didcot OX11 0SG, UK; joel.garner@aurorahp.co.uk
2 Department of Chemistry, University of Surrey, Guildford GU2 7XH, UK
3 National Physical Laboratory, Teddington TW11 0LW, UK
* Correspondence: d.read@surrey.ac.uk; Tel.: +44-1483-683451

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Abstract: Unconventional shale gas exploitation presents complex problems in terms of radioactive waste disposal. Large volumes of saline produced water resulting from hydraulic fracturing are typically enriched in radium isotopes, up to several hundred Bq/dm³, orders of magnitude above national discharge limits. There is a need, therefore, to decontaminate the fluid prior to discharge, preferably by creating a less problematic radium-containing, solid waste form. Barite (barium sulphate) co-precipitation is a cost-effective method for achieving these objectives, provided the process can be controlled. In this work, radium recovery of ~90% has been achieved for simulant produced water containing 100 Bq/dm³, using a single, optimised co-precipitation step. However, salinity has a significant effect on the efficiency of the process; higher salinity solutions requiring substantially more reagent to achieve the same recovery. If >90% radium removal is sought, multiple co-precipitation steps provide a much faster alternative than post-precipitation recrystallization of the barite solid phase, albeit at higher cost. The resulting solid waste has a relatively high specific radium activity but a much smaller volume, which presents a less intractable disposal problem for site operators than large volumes of radium-contaminated fluid.

Keywords: shale gas; radioactive waste; radium; contaminated fluids

1. Introduction

In 1904, shortly after the discovery of radioactivity, thermal brines co-produced with oil and gas were found to be enriched in natural radioactive isotopes [1,2]. Over the last century, naturally occurring radioactive material (NORM) has been discovered at oil and gas production sites globally, presenting a significant radiological protection and waste disposal problem (e.g., [3–5]). The vast majority of production sites affected by NORM in Europe employ conventional extraction techniques and are located offshore, allowing permitted sea disposal of up to 90% of the radioactive waste generated [5]. In the United States, onshore production is often accompanied by re-injection of flow-back and produced waters, thereby reducing the volume of potentially contaminated water that needs to be treated on site (e.g., [4]). With one historical exception [6], this option is not available at onshore wells in the UK, owing to the greater population density and aquifer vulnerability [7]. The recent expansion of hydraulic fracturing or ‘fracking’ in the USA and attempts by other countries to emulate it, has the potential to generate very large volumes of highly saline, radium-containing produced water in addition to scaled pipework and other equipment. Moreover, the aggressive nature of the extraction method [8] tends to lead to higher activity concentrations of the principal isotopes; radium-226 ($^{226}$Ra) and radium-228 ($^{228}$Ra). Activity levels in produced water from the Marcellus Shale in the United States [4] and from preliminary testing of the UK’s Bowland Shale [9] have been shown to exceed national exemption limits for aqueous discharge by a substantial margin [10,11]. This raises two main problems. First, as storage capacity at any given production site is
limited, there is a need to treat it, either by re-injecting it or by removing radium from the fluid and transforming it into a lower volume, solid waste form that is more manageable. Second, the low discharge limits for $^{226/228}$Ra, particularly in the UK (10 mBq/dm$^3$), present an analytical challenge; typically, several separation and concentration steps are required before analysis [12]. This, in turn, makes on-site compliance measurements of the residual fluid highly problematic.

Hydraulic fracturing involves pumping large quantities of fracking fluid down a borehole at high pressure to open and enlarge fractures within a hydrocarbon-bearing rock formation, usually shale [8]. The fluid consists of dilute acid, a proppant (e.g., sand) and a range of chemical additives, including surfactants, viscosity modifiers or gelling agents (polyacrylamide, guar gum) and biocides to inhibit microbial fouling [8,13,14]. The internal pressure of the rock formation causes water to return to the surface through the well. This water comprises both ‘flow-back’, i.e., returned drilling and hydraulic fluids, in addition to ‘produced’ water, consisting of formation brines from the hydrocarbon reservoir and extracted along with the oil and gas. The proportion of flow-back and produced water from a given well changes over time with the latter making a greater contribution (>90%) as the asset ages (e.g., [15]). This change is accompanied by an increase in salinity and, importantly for this study, radium [4,9,15–19] (see Table S1 in Supplementary Materials for selected analyses of produced water).

Radium has similar chemical properties to barium and readily co-precipitates in barium sulphate (barite), the most cost-effective way of removing radium from saline fluids [20]. The sulphate-based co-precipitation of radium isotopes in a binary system with barium salts has been investigated in numerous studies since it was first used to isolate the element, most notably by Doerner and Hoskins in 1925 [21–23]. Experimental work tends to concentrate on $^{226}$Ra, which is of more environmental concern than $^{228}$Ra owing to its greater abundance (reflecting the higher solubility of its precursor $^{238}$U, as opposed to $^{232}$Th) and much longer half-life; 1602 years compared to 5.75 years [24]. A study by Zhang et al. [20] investigated the relationship between ionic strength and $^{228}$Ra recovery during barite co-precipitation using solutions containing 5 mM Ba$^{2+}$ and varying SO$_4^{2-}$ concentrations at each ionic strength to keep the mass of barite precipitate constant. Their study concluded that an increase in salinity may improve $^{228}$Ra recovery by decreasing electrolyte activity and therefore, barite nucleation rate (diffusion-controlled growth). They postulated that a reduction in nucleation rate may enhance incorporation of radium into barite because it allows more time for lattice replacement; the latter occurring only during nucleation and crystal growth [25–29].

It has been shown that $^{226}$Ra can also be removed from solution by re-crystallisation of a barite precipitate that has been left to stand [30,31]. Bosbach et al. [30] described the kinetics of $^{226}$Ra incorporation into a Ba$_{3+}$–Ra$_{1+}$SO$_4$ solid solution over the course of more than a year and found that the uptake rate dropped significantly over time. Their experiments used 2.3 mM of pre-formed barite in 0.1 M NaCl and determined a steady state aqueous $^{226}$Ra concentration after around 100 days, interpreted as approaching thermodynamic equilibrium.

The focus of this study is optimisation of $^{226}$Ra recovery during barite co-precipitation by determining the minimum amount of SO$_4^{2-}$ that needs to be added to remove the maximum possible amount of radium from produced water (sulphate is the preferred reagent to use in excess under field conditions). Initial experiments employ a much lower Ba$^{2+}$ concentration than that used by Zhang et al. [20], more representative of conditions found in onshore UK wells [7], and the amount of barite precipitate is allowed to vary. A comparison is made between low and high salinity (0.3 and 3 M), slightly acidic (pH = 6) solutions containing 0.22 mM (30 mg/dm$^3$) Ba$^{2+}$ and ~100 Bq/dm$^3$ $^{226}$Ra. These experiments are, by definition, illustrative given the very wide range of compositions shown in global analyses of produced water (Table S1 in Supplementary Materials) but nevertheless, provide a basis for more exhaustive tests. At the next stage, a kinetic experiment was undertaken to determine whether leaving the system to remove more $^{226}$Ra by post-precipitation recrystallization is fast enough to be of practical use in industry. Centrifugation and ultrafiltration were used to separate the radiobarite precipitate from the remaining fluid; centrifugation could easily be scaled-up for use in an industrial setting. The specific activity and volume of the resulting solid were then calculated.
2. Materials and Methods

2.1. Reagents

$^{226}$RaCl solution was obtained from the National Physical Laboratory (NPL), UK. BaCl$_2$ (99.9%) was obtained from Sigma Aldrich, Na$_2$SO$_4$ (99.1%) from VWR Chemicals and NaCl (>99%) from Acros Organics. Trace metal grade Ba$^{2+}$ in 2% HNO$_3$ and scintillation cocktail were obtained from Fischer Chemical and Gold Star, respectively.

2.2. Radium Recovery by Barite Co-Precipitation

Two solutions (2 dm$^3$) were prepared at 0.3 and 3 M NaCl respectively, both with 0.22 mM Ba$^{2+}$ (using BaCl$_2$) in deionised H$_2$O. Each solution was spiked with ~100 Bq/dm$^3$ $^{226}$Ra and the pH adjusted to ~6 using HCl. Aliquots of 50 cm$^3$ were taken from each solution into separate centrifuge tubes and varying amounts of SO$_4^{2-}$ (0.1–70 mM as Na$_2$SO$_4$) were added to each vessel. All experiments were run in triplicate. Each vessel was centrifuged (2 × 20 minutes at 6000 rpm). An Amicon 8050 ultrafiltration cell under N$_2$ pressure coupled with an Isopore membrane filter (polycarbonate, hydrophilic, 0.22 µm, 44 mm diameter) was used for ultrafiltration. After separation, the samples were analysed by gamma spectrometry.

A Canberra BE2820 high-purity germanium crystal spectrometer with cryostatic cooling (CP-5SL) and carbon epoxy window was used for gamma spectrometry. Measurements for the precipitate are reported in each experiment, except where values were below the minimum detectable amount (MDA) for $^{226}$Ra (MDA = ~0.6 Bq for an 18-hour count), in which case the filtrate measurement is reported. The key peak for determination of $^{226}$Ra activity was 186 keV, as the solutions contain no uranium and thus, interference from $^{233}$U is negligible [32]. Count times ranged from 16,000–64,800 seconds (18 hours) depending on sample activity. Precipitates were measured either as point sources on the Isopore membrane filter or in small 1 cm diameter glass vials. Filtrates were measured in 50 cm$^3$ plastic pots. The resulting spectra were analysed using Genie 2000 V3.2 (2009) spectral analysis software, incorporating LABSOCS S573. All spectra were corrected for ambient background.

International certified reference standard, IAEA-434 [33] ($^{226}$Ra and $^{210}$Pb) for solids and an aqueous standard ($^{241}$Am, $^{109}$Cd, $^{54}$Co, $^{139}$Ce, $^{51}$Cr, $^{113}$Sn, $^{89}$Sr, $^{54}$Mn, $^{137}$Cs, $^{88}$Y, $^{65}$Zn and $^{60}$Co) supplied as 10 kBq/g, by NPL (Product Code: R08-04) were used for verification. Energy calibrations were run fortnightly using a standard $^{152}$Eu source to check for peak drift and consistency in raw counts.

2.3. Kinetics of Radium Recovery by Barite Post-Precipitation

Reaction kinetics were investigated under similar conditions to those above. Two 50 cm$^3$ solutions were prepared at 0.3 and 3 M NaCl with 0.22 mM Ba$^{2+}$ (using BaCl$_2$) in deionised H$_2$O. Each solution was spiked with ~100 Bq/dm$^3$ $^{226}$Ra and pH adjusted to ~6 using HCl. Sulphate (1.5 and 15 mM as Na$_2$SO$_4$) was added to the 0.3 and 3 M solutions, respectively. Each vessel was centrifuged (2 × 20 minutes at 6000 rpm) for experimental consistency, although separation was not required. Further $^{226}$Ra (20 kBq/dm$^3$) was then added and aliquots of 0.25 cm$^3$ were taken at increasing time intervals from 30 minutes to 14 days. A larger quantity of $^{226}$Ra was required so that small differences in activity between short time intervals could be determined accurately. The $^{226}$Ra activity in each aliquot was then analysed by liquid scintillation counting (LSC). Each 0.25 cm$^3$ aliquot was added to 10 cm$^3$ of scintillation cocktail in a standard plastic LSC vial. The samples were sealed in the vial and left for 4 weeks so that $^{226}$Ra could reach secular equilibrium with $^{222}$Rn. A Packard 2100 TR LSC was used for all measurements. A total count of α-emissions from $^{226}$Ra as well as $^{222}$Rn and $^{218}$Po (daughters of $^{226}$Ra with peaks that cannot be de-convoluted) was used to quantify $^{226}$Ra [30] (counting window: 4.5–8 MeV). All raw counts were subject to a background correction.

2.4. Mineralogical Analysis

A Carl Zeiss (Leo) 1530 variable pressure high-resolution field emission gun scanning electron microscope (SEM) was used to assess the crystal form of precipitated radiobarite. The instrument
includes an energy dispersive X-ray system for chemical analysis using an Oxford Instruments X-Max 80 mm² detector. The samples were studied under high vacuum (<1 Pa). Back-scattered electron (BSE) images and energy dispersive X-ray (EDX) spectra were recorded using a 15 kV electron beam voltage and fast scan speeds, as required, to limit sample charging. Samples were coated with an Au/Pd alloy to improve image quality.

3. Results

3.1. Radium Recovery by Barite Co-Precipitation

Initial experiments were conducted to determine the amount of SO$_4^{2-}$ required to remove the ‘maximum’ amount of $^{226}$Ra by barite co-precipitation at 0.3 and 3 M NaCl. The barium concentration chosen was 0.22 mM (30 mg/dm$^3$). This is a low concentration when compared to the brines encountered in the USA, where values in the Marcellus Shale can exceed 20 g/dm$^3$[18]; the rationale being that significant sulphate addition is then needed to induce precipitation. As more SO$_4^{2-}$ was added, an increase in $^{226}$Ra recovery was observed until a maximum of ≥90% was achieved at each ionic strength. At 0.3 M NaCl, ~1.5 mM SO$_4^{2-}$ was required to reach this maximum (Figure 1A) whereas at 3 M, ~15 mM SO$_4^{2-}$ was required, showing the relationship to be linear (Figure 1B; Table S2 for the full data set).

![Figure 1](image1.png)

**Figure 1.** $^{226}$Ra recovery by barite co-precipitation. (A) 0.3 M NaCl with 0.22 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (B) 3 M NaCl with 0.22 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (C) 0.3 and 3 M NaCl with 5 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (D) 0.3, 0.6, 1.5 and 3 M NaCl with 20 mM Ba$^{2+}$ and SO$_4^{2-}$.

The radiobarite crystals formed in the optimised co-precipitation process were imaged by scanning electron microscope to compare crystal morphology. At 0.3 M NaCl the crystals are of uniform size and euhedral (Figure 2A), whereas in the more saline solution the crystals are subhedral, generally larger and show poorer crystal form (Figure 2B). The effect that salinity has on both the co-precipitation process and crystal form will obviously influence settling behaviour during radium removal.
Experiments were conducted using higher concentrations of Ba\(^{2+}\) (5 mM, as used by Zhang et al. [20]) for comparison. In this study, however, a range of SO\(_4^{2-}\) concentrations (0.5, 1.25 and 5 mM) was added and the amount of barite precipitate was not kept constant. The experiments were conducted in pH ~6 solutions at 0.3 and 3 M NaCl. Figure 1C indicates that similar amounts of \(^{226}\)Ra are removed, irrespective of salinity, when adding the same amount of SO\(_4^{2-}\) (Table S3 for the full data set). SEM imaging (Figures 2C,D) indicates that the radiobarite crystals formed are uniform in size and overall, subhedral in form. These observations show that the initial concentration of Ba\(^{2+}\) also has a major effect on crystal form. Recovery of \(^{226}\)Ra at both salinities when using 5 mM SO\(_4^{2-}\) was comparable to the maximum determined in the initial experiments.

Ultrafiltration was required to separate the solid and liquid phase, as the quantity of radiobarite was too small for centrifugation to be effective; this would be impracticable in an industrial setting. Instead, the reagent quantities could be increased to a level appropriate for centrifugation. Experiments using higher concentrations of both Ba\(^{2+}\) and SO\(_4^{2-}\) (20 mM) were conducted in 0.3, 0.6, 1.5 and 3 M NaCl. Figure 1D shows that the levels of \(^{226}\)Ra recovery obtained (mean = 89%) are comparable. At these reagent concentrations, therefore, \(^{226}\)Ra recovery appears to be independent of ionic strength, a useful finding for practical field applications.

3.2. Kinetics of Radium Recovery by Barite Recrystallization Post-Precipitation

Following the optimised co-precipitation step, more \(^{226}\)Ra was added and its aqueous concentration monitored over time in 0.3 and 3 M NaCl, respectively, to determine the rate at which residual \(^{226}\)Ra in solution could be removed from solution by recrystallization of the precipitate. Measurements were restricted to 14 days, as in industry it is not feasible to store large volumes of radioactive fluid on site for longer periods (e.g., [15]). Figure 3 shows a pattern of diminishing returns.
in $^{226}$Ra recovery by post-precipitation at each NaCl concentration with the rate rapid in the first 24 hours then dropping significantly over the 14-day period.

**Figure 3.** $^{226}$Ra recovery by barite post-precipitation. Low Salinity: 0.3 M NaCl with 0.22 mM barite (0.22 mM Ba$^{2+}$ + 1.5 mM SO$_4^{2-}$); High Salinity: 3 M NaCl with 0.22 mM barite (0.22 mM Ba$^{2+}$ + 15 mM SO$_4^{2-}$). The values in parentheses show overall removal (co-precipitation (maximum of 90%) plus post-precipitation uptake). 2σ error is based on counting statistics only.

The rate of $^{226}$Ra recovery at 0.3 M NaCl (Figure 3; Low Salinity) is faster than at 3 M (Figure 3; High Salinity), the difference being more pronounced in the first 24 hours. Residual recovery at 24 hours was 80% and 63% for 0.3 and 3 M NaCl, respectively. Although the difference appears marginal, corresponding to overall recovery (from co-precipitation plus recrystallization) of 98%, as opposed to 96%, it has to be viewed in the context of the extremely low discharge limits imposed by national regulators. After 14 days, >80% of the residual $^{226}$Ra was recovered, slightly more $^{226}$Ra being removed under dilute conditions. Co-precipitation, together with 14 days of recrystallization, therefore, yields an overall recovery of ~98% (see Table S4 for the full data set).

**4. Discussion**

Radium-226 recovery by barite co-precipitation has been optimised for solutions with a relatively low initial barium concentration; its efficiency is significantly affected by salinity. At 0.22 mM Ba$^{2+}$, higher salinity systems require substantially more SO$_4^{2-}$ to achieve a maximum $^{226}$Ra recovery of ~90% (Figure 1A,B). This is likely due to the activity of the Ba$^{2+}$ and SO$_4^{2-}$ ions in solution being reduced by competing ions [30,31,34,35]; reduced precipitation of barite naturally resulting in reduced uptake of $^{226}$Ra.

The opposite effect can be observed when a large excess of Ba$^{2+}$(5 mM) is present (Figure 1C). The same observation was made by Zhang et al. [20] and can be explained by a decrease in nucleation rate resulting in more lattice replacement of Ba$^{2+}$ by radium. These authors kept the amount of barite precipitate constant across the range of ionic strengths; however, in this study, the amount of barite precipitate was allowed to vary. Even so, at higher ionic strengths, the effects appear large enough to counteract the smaller quantity of barite that is precipitated, resulting in similar recovery of $^{226}$Ra at the same salinity. In all experiments, at both 0.22 and 5 mM Ba$^{2+}$, ~15 mM SO$_4^{2-}$ is sufficient to recover the maximum amount of $^{226}$Ra feasible in one co-precipitation step.

Using 20 mM of both Ba$^{2+}$ and SO$_4^{2-}$, recovery of $^{226}$Ra from these fluids by barite co-precipitation was of comparable efficiency at ~89% (Figure 1D). These reagent concentrations make centrifugation more practicable and would contribute towards masking the effects of other constituents on the
radium recovery process in a ‘real’ produced water, though, of course, caution is needed in extrapolating the laboratory data. For example, the presence of strontium, which was not considered here, may suppress the uptake of radium from solution by modifying the composition of the solid-solution towards the celestite (SrSO₄) end-member [34,36], not all of the sulphate added would be available for barite precipitation and addition of soluble salts will increase barite solubility [34]. Other factors not considered in this paper, such as the presence of organic contaminants (e.g., [37]) might well render centrifugation impractical and outweigh radiological concerns. Cost may also be a consideration mitigating against overdosing the system with excess reagent.

If >90% radium recovery is sought, a further co-precipitation step of similar efficiency, may allow the target of ~99% removal to be achieved; this would require additional Ba²⁺ and SO₄²⁻. Alternatively, the system could be left to remove more radium by barite recrystallization over time. However, this process is much slower than co-precipitation. It would take >7 days to gain >80% recovery of the residual radium from the first co-precipitation step (~98% ²²⁶Ra recovery overall; Figure 3). Increasing the amount of barite present may increase the rate of radium recovery but would not approach the rate that could be achieved by further co-precipitation [30,38].

The kinetic experiment (Figure 3) shows a pattern of diminishing returns on the rate of ²²⁶Ra recovery. The rate is relatively fast in the first 24 hours, then decreases over 14 days at each ionic strength, in agreement with the findings of Bosbach et al. [30]. The slower rate of radium recovery at 3 M NaCl could again be due to an increase in barite solubility [34,35]. Crystal form might also be a contributing factor (Figure 2A,B). The smaller, uniform crystals in the more dilute system would provide a larger surface area for adsorption of ²²⁶Ra. Although radium uptake has been shown to proceed significantly beyond pure surface adsorption processes to lattice incorporation over timescales of a year or more (e.g., [30]), it is likely to be surface dominated on the shorter periods (≤14 days) considered here.

Geochemical modelling was conducted in order to determine ion activities, barite saturation indices, the mass of precipitated barite and theoretical radium recovery. PHREEQC [39] was used in conjunction with the Specific Ion Interaction Theory (SIT) database [40,41], which is applicable over the range of ionic strengths studied. Zhu [42] describes a semi-empirical model of the Ba₁₋₄Ra₂SO₄ solid solution, which could be used to predict radium recovery. This study applied the model using the experimentally validated dimensionless Guggenheim interaction parameter, a₀ = 1 ± 0.4, as derived by Vinograd et al. [38] (Tables S5 and S6 for PHREEQC output). Barite saturation indices (SI = log ion activity product/solubility product) are significantly reduced in all experiments at higher salinity due to a reduction in Ba²⁺ and SO₄²⁻ ion activities. Under optimised co-precipitation conditions (0.22 mM Ba²⁺), similar barite saturation indices of 2.2 and 2.4 were observed at 0.3 and 3 M NaCl, respectively. This shows that radium recovery is largely dependent on the amount of barite that can precipitate, as suggested above. At 5 mM Ba²⁺ the barite saturation indices are >2.2, so it may be expected that the maximum removal of radium would be easily achieved in these experiments. However, at 0.5 and 1.25 mM SO₄²⁻, <50% of the ²²⁶Ra present in solution co-precipitated with the barite. Therefore, barite saturation index alone cannot be relied upon to predict the amount of radium that will be recovered from solution; the initial Ba²⁺ concentration must also be considered. It appears that, at low concentrations of Ba²⁺, a lower barite saturation index is required to reach the maximum radium recovery determined. Observations made by Kowacz et al. [43] and Piana et al. [44] provide a possible explanation. They suggest that desolvation of Ba²⁺ and of the barite surface is a rate-limiting step in barite crystal growth. The rate of crystal growth may have a significant effect therefore, on the amount of lattice replacement that can occur [20].

The theoretical radium recovery was calculated and compared to experimental values from the co-precipitation experiments (Figure 4). At 0.22 mM Ba²⁺, experimental values are substantially lower than the theoretical predictions at intermediate barite saturation (Figure 4A,B). The model appears to be more accurate at the extremes of barite saturation (<0.5 and >2) where either very little or maximum radium is recovered. In contrast, at 5 mM Ba²⁺ (Figure 4C,D), the model tends to underestimate radium recovery at the two lower barite saturation indices. The main drawback with
The model does not explicitly consider the process of co-precipitation; therefore, it should be more applicable when recrystallization has taken place, provided that the crystals are fully equilibrated with the solution. This is unlikely over 14 days, as >100 days are believed to be required to achieve thermodynamic equilibrium [30,38]. Such a timescale would almost certainly not be viable in the field. The model predicts 98–99% radium recovery at thermodynamic equilibrium at each ionic strength whereas recoveries of only 86% at 0.3 M and 80% at 3 M NaCl, were reached over 14 days, again suggesting these systems were not at thermodynamic equilibrium.

On the basis of these experiments, approximately 10% of the initial $^{226}$Ra may remain in solution after a single co-precipitation step. The national exemption limit for aqueous radioactive waste discharge in the UK [10] is 0.01 Bq/dm$^3$. Thus, for some produced waters reported from the Marcellus Shale [4,17] ($^{226}$Ra ≤ 629 Bq/dm$^3$) and all of those reported to date in the UK’s Bowland Shale [9] (14–90 Bq/dm$^3$), ~90% recovery, while highly effective, would not be sufficient. First, compliance would require multiple co-precipitation steps or an extended post-precipitation recrystallization period to achieve >99.9% $^{226}$Ra removal, which may not be feasible in practice [45–49]. Second, measurement of whether or not the compliance level has been reached would require highly sensitive radiometric analysis by alpha spectrometry, which is extremely challenging in a field setting [12].

The resulting radiobarite waste also needs to be considered, as the majority of the initial radium present in the produced water would be sequestered in this phase. According to the model, the optimised co-precipitation experiments would result in 0.22 mM (51 mg/dm$^3$) of barite precipitate at 0.3–3 M NaCl, though much more saline waters are known [4,17,18]. With a typical initial aqueous

**Figure 4.** Theoretical vs. experimentally determined % $^{226}$Ra recovery. (A) 0.3 M NaCl with 0.22 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (B) 3 M NaCl with 0.22 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (C) 0.3 M NaCl with 5 mM Ba$^{2+}$ and varying SO$_4^{2-}$; (D) 3 M NaCl with 5 mM Ba$^{2+}$ and varying SO$_4^{2-}$. 2σ error based on counting statistics only. SI—Saturation Index. Theoretical values are based on $a_0 = 1 ± 0.4$.
$^{226}\text{Ra}$ concentration of ~100 Bq/dm$^3$, recovery of ~90% would produce barite containing ~90 Bq $^{226}\text{Ra}$ with a specific activity of ~1.75 kBq/g. Although the specific activity is high, the volume and therefore, total activity of the waste will be relatively low. Increasing the amount of reagent would reduce the specific activity but increase the waste volume. For example, 20 mM (4.7 g/dm$^3$) of barite precipitate would yield a specific activity of ~19.3 Bq/g at ~90% $^{226}\text{Ra}$ recovery. A detailed discussion of the risk associated with alternative waste treatment methods would necessarily be site-specific and is beyond the scope of the current paper. A preliminary attempt to assess the health risks posed by radium-rich sludge contained in flow-back water impoundments in Pennsylvania, USA [49] did not compare alternative routes but could form a useful case study. In practice, a balance will need to be struck to yield a waste form which can be both handled safely and is relatively inexpensive to dispose.

5. Conclusions

Saline produced waters from unconventional gas extraction in the United States and elsewhere are typically enriched in radium isotopes posing an occupational and environmental challenge. Experience onshore in the UK, albeit limited to date, suggests similar problems would arise if the technology were to be pursued at an industrial scale. If re-injection is not an option, as seems likely, the fluid would need to be decontaminated prior to surface discharge. Co-precipitation of radium in barite is a cost-effective method for creating a less problematic solid waste form, provided the process can be controlled. In this work, radium recovery of ~90% has been achieved for simulant produced waters containing 100 Bq/dm$^3$ $^{226}\text{Ra}$ using a single, optimised co-precipitation step. However, higher salinity solutions required substantially more reagent to achieve the same recovery. If >90% radium removal is sought, multiple co-precipitation steps provide a much faster alternative than post-precipitation recrystallization of the barite solid phase. The resulting solid waste has a relatively high specific radium activity but a much smaller volume, which represents a less intractable disposal problem for site operators than large volumes of radium-contaminated fluid.

Supplementary Materials: The following are available online at www.mdpi.com/xxx/s1, Table S1: Characterisation of production water from hydraulic fracturing operations; Table S2: $^{226}\text{Ra}$ recovery by co-precipitation at 0.3 and 3 M NaCl (all 0.22 mM Ba); Table S3: $^{226}\text{Ra}$ recovery by co-precipitation at 0.3 and 3 M NaCl (all 5 mM Ba); Table S4: $^{226}\text{Ra}$ recovery by post-precipitation at 0.3 and 3 M NaCl (all 0.22 mM Ba); Table S5: PHREEQC output for $^{226}\text{Ra}$ recovery by co-precipitation at 0.3 and 3 M NaCl (all 0.22 mM Ba); Table S6: PHREEQC output for $^{226}\text{Ra}$ recovery by co-precipitation at 0.3 and 3 M NaCl (all 5 mM Ba).

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