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Lattice Enumeration for Tower NFS: a 521-bit Discrete Logarithm Computation

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Abstract. The Tower variant of the Number Field Sieve (TNFS) is known to be asymptotically the most efficient algorithm to solve the discrete logarithm problem in finite fields of medium characteristics, when the extension degree is composite. A major obstacle to an efficient implementation of TNFS is the collection of algebraic relations, as it happens in dimension greater than 2. This requires the construction of new sieving algorithms which remain efficient as the dimension grows. In this article, we overcome this difficulty by considering a lattice enumeration algorithm which we adapt to this specific context. We also consider a new sieving area, a high-dimensional sphere, whereas previous sieving algorithms for the classical NFS considered an orthotope. Our new sieving technique leads to a much smaller running time, despite the larger dimension of the search space, and even when considering a larger target, as demonstrated by a record computation we performed in a 521-bit finite field \mathbb{F}_{p^6} . The target finite field is of the same form than finite fields used in recent zero-knowledge proofs in some blockchains. This is the first reported implementation of TNFS.

Introduction

Context. While the post-quantum competition is ongoing, the discrete logarithm problem is still at the basis of the security of many currently-deployed public key protocols. Given a cyclic group G, a generator $g \in G$ and a target $h \in G$, solving the discrete logarithm problem in G means finding an integer x mod |G| such that $g^x = h$. The hardness of this problem depends on the considered group G and the two usual choices are the group of the invertible elements in a finite field and the group of points of an elliptic curve. This article deals with discrete logarithms in finite fields. In particular, as small characteristics finite fields are no longer considered because of the advent of quasipolynomial time algorithms [3,12,24], we focus on medium and large characteristics. For a finite field \mathbb{F}_{p^n} we recall that the characteristic p is said to be of medium size if $L_{p^n}(1/3) and of large size if <math>p > L_{p^n}(2/3)$. Equivalently, it means that the extension degree n is of bounded size with respect to the finite field order.

We use the usual notation $L_Q(\alpha, c) = \exp((c + o(1))(\log Q)^{\alpha}(\log \log Q)^{1-\alpha})$, where o(1) tends to 0 when Q tends to infinity.

NFS and TNFS. The Number Field Sieve (NFS) algorithm and its variants are the fastest known algorithms to solve the discrete logarithm problem in finite fields of medium and large characteristics. One of these variants is the Tower Number Field Sieve (TNFS), known to be asymptotically more efficient than a classical NFS for some fields when the extension degree is composite. TNFS exploits the algebraic structure of towers of number fields: the main difference with NFS comes from the representation of the target field \mathbb{F}_{p^n} . Whereas in the classical NFS setup, the finite field \mathbb{F}_{p^n} is represented as the quotient field $\mathbb{F}_p[x]/(f)$ where f is a polynomial of degree f over f in the TNFS setup, we have f is the ring defined as the quotient f is an f is a degree f polynomial that remains irreducible modulo f.

Originally proposed by Schirokauer [30], TNFS was reinvestigated by Barbulescu, Gaudry, and Kleinjung [4] in 2015. They showed that the asymptotic complexity of TNFS in large characteristics is $L_{p^n}(1/3, \sqrt[3]{64/9})$, similarly as NFS. In medium characteristics, the complexity of TNFS is greater than $L_{p^n}(1/3)$ and thus this algorithm is only considered in the large case.

This algorithm was then modified by Kim, Barbulescu [22] and Jeong [23] to form the extended Tower Number Field Sieve (exTNFS), the variant being dedicated to composite extension degrees, i.e., when $n = \eta \kappa$. This extended variant has an $L_{p^n}(1/3)$ complexity also in medium characteristics. In this case, the overall complexity of exTNFS can be as low as $L_{p^n}(1/3, \sqrt[3]{48/9})$ if there is a factor of n of the appropriate size (see Table 1). Both TNFS and exTNFS can be coupled with a multiple field variant – for any finite field – and a special variant – for some sparse characteristics only – giving each time a lower asymptotic complexity. We do not address these variants in this article.

Algorithm	Medium characteristic	Boundary	Large characteristic
NFS	96	48	64
TNFS	_	_	64
exTNFS	≥ 48	48	64

Table 1: Medium and large characteristics complexities of various algorithms, expressed as $L_{p^n}(1/3, \sqrt[3]{c/9})$, where c is the reported value in this table.

Towards an implementation of exTNFS. One can see from the complexities given in Table 1 that for NFS, medium characteristics are harder than large characteristics. This remains true for the multiple and special variants. However, a noticeable exception to this observation lies in the exTNFS algorithm. Indeed, when the degree n is composite, i.e., $n = \eta \kappa$, the target finite field $\mathbb{F}_{p^n} = \mathbb{F}_{p^{\eta \kappa}}$ can be viewed as \mathbb{F}_{P^κ} where P is a prime of the same bitsize as p^η . Thus, the complexity of exTNFS in medium characteristics can be viewed similarly as the complexity of NFS at the boundary case between medium and large characteristics, leading to a smaller c constant in the L_{p^n} -notation. Hence we find a lower complexity in medium characteristics than in large characteristics

with exTNFS, which makes it a promising candidate for computational records in this area.

Let us assume we want to evaluate the security of a family of finite fields with fixed composite extension degree (for instance n=6). These families often arise in pairing-based protocols. Evaluating the security of a concrete finite field in such a family is not an easy task, as we are not even able to tell beforehand whether NFS or exTNFS would be the fastest algorithm. Indeed, using a fixed extension degree asymptotically defines the characteristic as large, an area where the best discrete logarithm algorithm is NFS (not exTNFS). However, let us keep in mind that medium and large characteristics are notions defined for asymptotic sizes; as soon as we set a concrete target finite field it is not well understood how we should qualify its characteristic. In this work we underline that exTNFS shows real improvements with regards to current NFS computations for this family. Current record computations (e.g. for 400 or 500-bit finite fields) deal with areas where asymptotic analysis are not yet the relevant ones. We cannot easily extrapolate on current and deployed sizes (e.g. for more than 2000-bit finite fields) but our implementation of exTNFS provides practical insight on security parameters by showing its incredibly good behavior at lower sizes.

In the rest of this article, to simplify notations and to be coherent with the recent literature, we use TNFS as a short hand for ex-TNFS². We do however assume the degree n of our target finite field is composite, thus considering specifically the extended variant.

Lattice enumeration for TNFS. Despite the fact that TNFS is promising, no implementation was done using this variant of NFS, up to this work. Indeed, so far, excluding the very small characteristics 2 and 3, all discrete logarithm record computations were performed using NFS, the special variant of NFS, or the Function Field Sieve – a method for small characteristics only.

A major obstacle to an efficient implementation of TNFS is the collection of algebraic relations where equations between small elements of number fields must be found. Indeed, whereas NFS requires sieving through $(a,b) \in \mathbb{Z}^2$ pairs, the tower setup sieves through $(a(\iota),b(\iota))$ -pairs, *i.e.*, degree $\eta-1$ polynomials with bounded coefficients. This requires the construction of sieving algorithms in a space of dimension $2\eta \geq 4$, which remain efficient as the dimension grows.

In dimension 2, Franke and Kleinjung [10] proposed in 2005 an efficient algorithm used in all previous records. For higher dimensions, after a pioneer work in $\mathbb{F}_{p^{12}}$ by Hayasaka et al. [19], the transition vectors method from Grémy [13] and a recursive plane method proposed by McGuire and Robinson [26] were tested in dimension 3 and used for record computations using NFS. However, the efficiency of their algorithms for even higher dimensions is questionable.

Our work. In this article, we introduce an efficient sieving algorithm for higher dimensions which allows us to implement TNFS and perform the first record computation with it. More specifically, we propose the following contributions.

² We use the same abuse in the abstract and title too.

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- 1. Sieving in a high dimensional sphere instead of an orthotope. All sieving algorithms so far considered a product of intervals as search space \mathcal{S} . Indeed, whether a candidate relation is characterized by an (a,b)-pair or an $(a(\iota),b(\iota))$ -pair with more than two coefficients, every coefficient is bounded separately in an interval $[-H_1^i,H_2^i]$ for $i=1,2,\cdots,d$ where d is the total number of coefficients. Hence, the search space considered is a d-orthotope of the form $\mathcal{S}=[-H_1^1,H_2^1]\times\cdots\times[-H_1^d,H_2^d]$. We argue that when $d\geq 3$, the shape of \mathcal{S} must be adequately chosen. More precisely, we consider a d-sphere instead of a d-orthotope and explain why we believe this choice leads to a more efficient algorithm when the dimension grows.
- 2. Adapting a lattice enumeration algorithm to the context of TNFS. In order to fully exploit the new search space, we adapt a known lattice algorithm to the context of TNFS: Schnorr-Euchner's enumeration algorithm [31], that outputs the shortest vector of a lattice. We modify this algorithm in order to list all the vectors of a lattice \mathcal{L} within a d-dimensional sphere S_d . Furthermore, a part of the common coefficients of the enumerated vectors are kept in memory during the algorithm, leading to a 10% gain in the execution time. This algorithm remains competitive when the dimension grows and also provides an exhaustive search of all the vectors in $\mathcal{L} \cap S_d$, contrary to previous approaches.
- 3. Analysis of the relation collection step in TNFS and duplicate relations. We anchor this sieving algorithm in the context of the entire relation collection step. Sieving algorithms are usually combined with batch algorithms and ECM to provide the most efficient relation collection. We give details on this relation collection step, and give new insight on how to define and remove duplicate relations that arise in the context of TNFS.
- 4. A 521-bit finite field record. Our new lattice enumeration for the sieving step led to the first record computation of a discrete logarithm with TNFS, reaching a 521-bit finite field \mathbb{F}_{p^6} . Previous record on a finite field of the same shape reached a 423-bit finite field in January 2020. The choice of the extension degree was motivated by the use of such finite fields in pairing-based protocols, in particular in recent zero-knowledge proofs in some blockchains. Ultimately, as shown in Table 2, our algorithm is much faster than existing high-dimensional sieving algorithms, despite the larger dimension of the search space and the larger finite field.

Parameters	[14]	[26]	This work
Algorithm	NFS	NFS	TNFS
Field size (bits)	422	423	521
Sieving dimension	3	3	6
Sieving time	201,600	69,120	23,300

Table 2: Comparison of the relation collection step in core hours with [14] and [26] for finite fields of the form \mathbb{F}_{p^6} .

Outline. In Section 1, we recall the general setup of TNFS and in particular we concentrate on the steps that mostly differ from the classical NFS setup. In Section 2, we focus on the relation collection step with the special-q method, and explain how to deal with duplicate relations. In Section 3, we describe our adaptation of Schnorr-Euchner's enumeration algorithm to the context of TNFS. We justify why we choose a d-sphere as sieving area and introduce an efficient way to compute the desired vectors of coefficients for the relations. Section 4 analyses the complexity of our sieving algorithm and compares the latter with pre-existing algorithms. Finally, in Section 5 we detail our complete discrete logarithm computation in a 521-bit finite field with extension degree 6.

1 The Tower Number Field Sieve

1.1 Mathematical setup

The classical tower of number fields that illustrates the TNFS setup considers the intermediate number field $\mathbb{Q}(\iota)$ where ι is a root of h, a polynomial over \mathbb{Z} that remains irreducible modulo p. Above this number field are set the two number fields $K_1 = \mathbb{Q}(\iota)[x]/f_1(x)$ and $K_2 = \mathbb{Q}(\iota)[x]/f_2(x)$ where f_1, f_2 are irreducible polynomials over $\mathcal{R} = \mathbb{Z}[\iota]$ that share an irreducible factor φ modulo the unique ideal \mathfrak{p} over p in $\mathbb{Q}(\iota)$. We write \mathcal{O}_i the ring of integers of K_i and α_i a root of f_i in K_i for i=1,2. This construction is illustrated in the left part of Figure 1. Because of the conditions on the polynomials h, f_1 and f_2 , there exist two ring homomorphisms from $\mathbb{R}[x] = \mathbb{Z}[\iota][x]$ to the target finite field \mathbb{F}_{p^n} through the number fields K_1 and K_2 . This allows to build a commutative diagram as shown in the right part of Figure 1. The extension degree n is assumed to be composite, and we write $n=\eta\kappa$. In this setting, h is of degree η , and f_1 and f_2 have degree at least κ , so that the degree of there common factor φ is exactly κ . For simplicity, we will assume that f_1 and f_2 are defined over \mathbb{Z} , since it is the case in our record computation; this is only possible when κ and η are coprime.

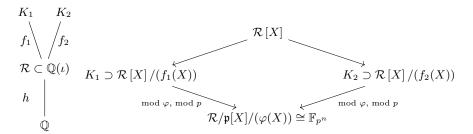


Fig. 1: Commutative diagram of Tower NFS.

1.2 A step by step walk through TNFS

The TNFS algorithm follows similar steps as any index calculus algorithm.

Polynomial selection. Unlike NFS which uses only two polynomials f_1 and f_2 to define the number fields, three polynomials must be selected for this algorithm, namely h, f_1 and f_2 . The polynomial h must be of degree η and irreducible modulo p to ensure the uniqueness of the ideal \mathfrak{p} over p in \mathcal{R} . Ideally one would choose a unitary h with small coefficients and such that the inverse of the Dedekind zeta function evaluated at 2 (implemented in Sage for example) is close to 1. Indeed, as we will see in Section 2.4, this is related to non-coprime ideals that produce equivalent relations which are useless for the linear algebra step.

The polynomials f_1 and f_2 are selected to fit the mathematical setting of Section 1.1. To do so we can use NFS polynomial selections such as the Conjugation, JLSV or Sarkar-Singh's methods, that we do not recall here. The polynomials we use for our 521-bit computation come from the Conjugation method. In NFS, the quality of the polynomials can be refined with a quantity known as the Murphy- α value. See [17] for details about Murphy- α adapted to TNFS.

Relation collection. The goal of the relation collection step is to select among the set of linear polynomials $\phi(x,\iota) = a(\iota) - b(\iota)x \in \mathcal{R}[x]$ at the top of the diagram the candidates which produce a relation. A relation is found if the polynomial $\phi(x,\iota)$ mapped to K_1 and K_2 factors into products of ideals of small norms in both number fields. The ideals of small norms that occur in these factorizations constitute the factor basis \mathcal{F} . More precisely, we define it as $\mathcal{F} = \mathcal{F}_1 \cup \mathcal{F}_2$ with

 $\mathcal{F}_i(B) = \{ \text{prime ideals of } \mathcal{O}_i \text{ of norm } \leq B, \text{ whose inertia degree over } \mathbb{Q}(\iota) \text{ is } 1 \},$

for i = 1, 2. The representation of these ideals of degree 1 in the context of TNFS is summarized in [4, Proposition 1].

To verify the B-smoothness on each side, one needs to evaluate the norms $N_i(a(\iota)-b(\iota)\alpha_i)$ for i=1,2. To do so we recall that when the polynomials f_i are monic, these norms are integers that can be computed thanks to resultants as $N_i(a(\iota)-b(\iota)\alpha_i)=\mathrm{Res}_t(\mathrm{Res}_x(a(t)-b(t)x,f_i(x)),h(t))$. The relation collection step stops when we have enough relations to construct a system of linear equations that may be full rank. The unknowns of these equations are the virtual logarithms of the ideals of the factor basis.

Linear algebra. A good feature of the linear system created is that the number of non-zero coefficients per line is very low. This allows to use sparse linear algebra algorithms such as the block variant of Wiedemann's algorithm [32], for which parallelization is partly possible. The output of this step is a kernel vector corresponding to the virtual logarithms of the ideals in the factor basis. There is no difference for this step between TNFS and NFS.

Individual discrete logarithm. The final step of TNFS consists in finding the discrete logarithm of the target element. This step is subdivided into two substeps: a smoothing step and a descent step. The smoothing step is an iterative process where the target element t is randomized by considering $s = g^e t \in \mathbb{F}_{p^n}^*$ for an exponent e chosen uniformly at random.

The second step consists in decomposing every factor of the lifted value of s, in our case prime ideals with norms less than a smoothness bound B_i (but usually greater than B) into elements of the factor basis for which we now know the virtual logarithms. This process creates descent trees where the root is an ideal coming from the smoothing step and the nodes are ideals that get smaller and smaller as they go deeper. The leaves are ultimately elements of the factor basis. The edges of the tree are defined as follows: for every node, there exists an equation between the ideal of the node and all the ideals of its children.

In this work, we consider an improvement given by Guillevic in [16, Algorithm 5] for the smoothing step, that is useful in the context of TNFS only. The goal is to improve the smoothness probability of the lift of $s \in \mathbb{F}_{p^n}^*$ to K_i by constructing an adequate lattice whose reduced vectors define elements of K_i with potentially small norms which is precisely the potential lifts of s we are looking for.

2 Focus on the relation collection

In TNFS the relation collection step requires sieving in dimension $2\eta \geq 4$, which is the number of coefficients involved in ϕ . We start by dividing the set of polynomials ϕ into multiple subsets and then we present different algorithms to successively select the candidates in each of these subsets.

2.1 The special-q setup

The relation collection phase looks at a set of linear polynomials $\phi(x,t) =$ $a(\iota) - b(\iota)x \in \mathcal{R}[x]$ where a, b are polynomials of degree deg h-1 with deg $h=\eta$ and bounded coefficients, and tries to identify which are going to produce doublysmooth norms, i.e., for which pair $(a(\iota), b(\iota))$ the norms $N_1(a(\iota) - b(\iota)\alpha_1)$ and $N_2(a(\iota) - b(\iota)\alpha_2)$ factor into small primes. To reduce the time of the sieving stage, Pollard [29] suggested to divide the set of all polynomials ϕ , commonly called the search space, into multiple subsets. This corresponds to the so-called *special-q method*. This method regroups polynomials into groups such that $\phi(\alpha_1, \iota)$ (or $\phi(\alpha_2, \iota)$ depending on whether we put the special-q on the f_1 side or the f_2 -side) share a common factor: the ideal \mathfrak{Q} , above a prime q, hence the name. Thus, when talking about a sieving algorithm, we usually consider a fixed special-q ideal \mathfrak{Q} , and select good polynomials ϕ in the corresponding subset. This idea of using special-q's increases the smoothness probability on the side where divisibility by \mathfrak{Q} is forced, since the norm is already divisible by q. Furthermore this provides a natural parallelization where each work-unit corresponds to a special-q

Let $\underline{\phi}$ denote the (row-) vector of coefficients of the polynomial $\phi(x, \iota)$, *i.e.* the vector $\underline{\phi} = (a_0, \dots, a_{\eta-1}, b_0, \dots, b_{\eta-1}) \in \mathbb{Z}^{2\eta}$. Let us consider a special-q ideal \mathfrak{Q} of degree 1 in K_i of the form $\mathfrak{Q} = \langle q, \iota - \rho_\iota, x - \rho_x \rangle$, where q is a prime number, ρ_ι is a root of h modulo q, and ρ_x is a root of f_i modulo q. One could also consider ideals of degree greater than 1, but special-q of degree 1 are the most common among ideals of bounded norms and thus we restrict to this case.

Proposition 1. The set of polynomials ϕ such that the corresponding principal ideal in K_i is divisible by \mathfrak{Q} form a lattice that we call the \mathfrak{Q} -lattice $\mathcal{L}_{\mathfrak{Q}}$.

The latter can be made explicit as follows.

$$\mathcal{L}_{\mathfrak{Q}} = \{ (a_0, \cdots, a_{\eta-1}, b_0, \cdots, b_{\eta-1}) \in \mathbb{Z}^{2\eta} : \sum_{k=0}^{\eta-1} (a_k \iota^k - b_k \iota^k \alpha_i) \equiv 0 \pmod{\mathfrak{Q}} \}$$

where i = 1, 2 depending on the side we consider. A basis $B_{\mathfrak{Q}}$ of this lattice can be expressed as follows.

The determinant of this lattice is $q^{\deg \phi_h}$, where ϕ_h is an irreducible factor of $h \pmod{p}$. In our case $\phi_h = \iota - \rho_\iota$ because we only consider special-q ideals of degree 1 and so the determinant is simply q. The lattice dimension is 2η .

Each unit of computation targets one special-q ideal \mathfrak{Q} and searches for polynomials $\phi(x,\iota)$ with $\underline{\phi} \in \mathcal{L}_{\mathfrak{Q}}$ leading to relations, *i.e.*, for which both sides are smooth. In order to explore the lattice $\mathcal{L}_{\mathfrak{Q}}$, we first LLL-reduce the basis $B_{\mathfrak{Q}}$, and then consider linear combinations with small coefficients of these new basis elements. This allows us to focus on polynomials where one of the norms on one side is known to be divisible by q, thus increasing the probability of it being smooth. More precisely, let $M_{\mathfrak{Q}}$ be an LLL-reduced basis of $\mathcal{L}_{\mathfrak{Q}}$. We study the (row-) vectors \mathbf{c} of coefficients such that $\underline{\phi} = \mathbf{c} \cdot M_{\mathfrak{Q}}$, potentially leads to a relation. This is done using sieving algorithms.

2.2 Constructing the double-divisibility lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$

We concentrate on vectors \mathbf{c} that belong to a sieving region \mathcal{S} . Traditionally, \mathcal{S} is an ℓ_{∞} -ball, however in this work we consider the ℓ_2 -norm. Section 3.2 explains

this preference. In order to efficiently detect the vectors \mathbf{c} giving elements of smooth norms, one can perform an Eratosthenes-like sieving, quickly marking all vectors \mathbf{c} in \mathcal{S} leading to a norm on the f_1 -side (or the f_2 -side) that is divisible by a small prime p. Repeating this sieve for many primes p allows to detect the most promising vectors $\underline{\phi}$, those for which the norm is divisible by many small primes. To do so, we proceed as for the divisibility by \mathfrak{Q} .

Let \mathfrak{p} be a prime ideal of norm p in K_i of the form $\mathfrak{p} = \langle p, \iota - r_{\iota}, x - r_{x} \rangle$, where r_{ι} is a root of h modulo p and r_{x} is a root of f_{i} modulo p. The second statement of [4, Proposition 1] can be reformulated for this specific context.

Proposition 2. The principal ideal generated by $\phi(x, \iota)$ in K_i is divisible by \mathfrak{p} if and only if $\phi(r_x, r_\iota) \equiv 0 \mod p$.

Let $\mathbf{U}_{\mathfrak{p}}$ be the (row-) vector of size 2η defined by

$$\mathbf{U}_{\mathfrak{p}} = (1, r_{\iota} \bmod p, \cdots, r_{\iota}^{\eta - 1} \bmod p, r_{x}, r_{x}r_{\iota} \bmod p, \cdots, r_{x}r_{\iota}^{\eta - 1} \bmod p).$$

Then similarly as before, we can translate the divisibility property of the ideal of Proposition 2: the divisibility by \mathfrak{p} is equivalent to the condition $\underline{\phi} \cdot \mathbf{U}_{\mathfrak{p}}^{\mathsf{T}} \equiv 0 \mod p$. Recall that ϕ is taken in a subset of the search space so that the ideal generated by ϕ is divisible by \mathfrak{Q} , namely its coefficients are written as $\underline{\phi} = \mathbf{c} \cdot M_{\mathfrak{Q}}$. Thus taking into account both the divisibility by \mathfrak{Q} and by \mathfrak{p} yields the condition on \mathbf{c} :

$$\mathbf{c} \cdot M_{\mathfrak{Q}} \mathbf{U}_{\mathfrak{p}}^{\mathsf{T}} \equiv 0 \mod p. \tag{1}$$

The product $M_{\mathfrak{Q}}\mathbf{U}_{\mathfrak{p}}^{\mathsf{T}}$, reduced modulo p and normalized so that its first coordinate is 1, is expressed as $M_{\mathfrak{Q}}\mathbf{U}_{\mathfrak{p}}^{\mathsf{T}} \equiv \lambda (1, \alpha_1, \alpha_2, \dots, \alpha_{\eta-1})^{\mathsf{T}} \mod p$, with $\lambda > 0$. Since $M_{\mathfrak{Q}}$ and $\mathbf{U}_{\mathfrak{p}}$ are known, we explicitly compute the values α_i . Note that this assumes the first coordinate is non-zero. Otherwise, one must either adapt the construction of $M_{\mathfrak{Q},\mathfrak{p}}$ below or skip the ideal \mathfrak{p} during sieving. Finally, the set of vectors \mathbf{c} verifying Equation (1) is the lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ generated by the rows of the matrix

$$M_{\mathfrak{Q},\mathfrak{p}} = \begin{pmatrix} p & 0 & 0 & 0 & \cdots & 0 \\ -\alpha_1 & 1 & 0 & 0 & \cdots & 0 \\ -\alpha_2 & 0 & \ddots & 0 & \cdots & 0 \\ \vdots & 0 & 0 & \ddots & 0 \\ -\alpha_{\eta-1} & 0 & 0 & 0 & \cdots & 1 \end{pmatrix}.$$

In the end, since $M_{\mathfrak{Q},\mathfrak{p}}$ is explicitly known, we can compute the coefficients $\underline{\phi} = \mathbf{c} \cdot M_{\mathfrak{Q}}$ of the polynomials ϕ . This is possible as soon as we are able to enumerate the short vectors \mathbf{c} in this lattice which is the aim of Section 3. This procedure, which is called enumeration, is done for all prime p from 2 up to a predefined bound p_{max} , forming a so called sieving algorithm. Sieving allows to detect quickly vectors \mathbf{c} that belong to several $M_{\mathfrak{Q},\mathfrak{p}}$ for various \mathfrak{p} . The corresponding polynomials ϕ are good candidates that potentially give relations as they provide by construction ideals that are already divisible by \mathfrak{Q} and several ideals \mathfrak{p} . Hence we keep them for the next selection phase.

2.3 Combining three algorithms

Other algorithms are used, either to directly detect polynomials leading to doubly-smooth norms, or to work as complementary algorithms one after another as a sequence of filters to determine and only keep the promising candidates at each step (as done with the enumeration above). These algorithms either find and extract smooth parts of the norms, or completely factor them. The family of sieving algorithms [13,26], batch algorithms [6, Algorithm 2.1] and ECM [7,25] are examples of such methods used in factorization and DLP computations.

They all have different complexities and properties and thus cannot be used on the same amount of input norms N_i . ECM is for example much more costly than sieving. Hence, applying it to all norms N_i is far from optimal. On the other hand, sieving is a much less costly algorithm per candidate, and thus can be used to find the small factors (up to p_{max}) of a large number of norms of structured candidates. This is why the relation collection step usually starts with a sieving algorithm with input all candidates $(a(\iota),b(\iota))$ pairs. ECM is then used to guarantee that the norms of promising candidates are indeed B-smooth by checking the larger prime factors. Batch smoothness can be added in between sieving and ECM or as a substitution of one of them to further optimize the overall cost. It is less costly than ECM and thus can be used to pre-select promising candidates but more costly than sieving and thus cannot be run on the entire set of candidates. It extracts prime factors up to a bound p_{batch} such that $p_{\text{max}} < p_{\text{batch}} < B$. Table 3 describes the properties of these algorithms.

Properties	Sieving	Batch	ECM
Input candidates	Numerous and structured	Numerous	Few
Prime factors extracted	Small	Small or medium	Large
RAM	Very large	Large	Tiny
Cost per candidate	Small	Medium	High

Table 3: Properties of the different relation collection algorithms

The relation collection is thus seen as a sequence of filters, each taking a certain amount of candidates as input, and selecting *survivors* based on a criterion. These survivors are then the input to the next filter. The selection of survivors is usually based on the size of the cofactor, which we now define.

Definition 1 (\mathcal{A} -cofactor). Let N be a positive integer and consider $P = \prod_i p_i$ where the p_i are the prime factors of N extracted by Algorithm \mathcal{A} . Then the \mathcal{A} -cofactor of N is $C_{\mathcal{A}}(N) = N/P$.

For a fixed \mathcal{A} -cofactor threshold $\mathcal{T}_{\mathcal{A}}$, the survivors are the candidates selected if their norm N satisfies $C_{\mathcal{A}}(N) \leq \mathcal{T}_{\mathcal{A}}$ (there can be such a condition on both sides if sieving is done on both of them). Finally, the complete relation collection is given in Algorithm 1. Note that on line 7, we remove duplicate relations, as explained in the next Section.

Algorithm 1 Relation collection for a given special-q with sieving, batch and ECM

Input: A prime ideal \mathfrak{Q} , a sieving region \mathcal{S}

Output: A list of relations.

- 1: Construct the lattice $\mathcal{L}_{\mathfrak{Q}}$ and LLL-reduce it.
- 2: for each prime ideal \mathfrak{p} in \mathcal{O}_1 (or \mathcal{O}_2) up to norm p_{\max} do
- 3: Construct the lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$
- 4: Enumerate all vectors \mathbf{c} in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$.
- 5: For each \mathbf{c} , keep track of the size of the factor p with a sieving table.
- 6: For promising \mathbf{c} , for which the product of the factors p is large, compute approximations of the norms N_1, N_2 and identify vectors with sieve-cofactor smaller than $\mathcal{T}_{\text{siev}}$. They are called sieve-survivors.
- 7: Remove duplicates.
- 8: Run batch algorithm with input the (exact) norms N_1 and N_2 of the sieve-survivors and primes up to p_{batch} . Keep batch-survivors whose batch-cofactor is smaller than $\mathcal{T}_{\text{batch}} < \mathcal{T}_{\text{siev}}$.
- 9: Run ECM on the batch-survivors to identify all the doubly-B-smooth norms.
- 10: **return** Vectors with doubly-B-smooth norms

2.4 Filtering through equivalent relations

When sieving through all the pairs of candidates it is sometimes the case that two pairs $(a(\iota), b(\iota))$ and $(a'(\iota), b'(\iota))$ provide the same relation, *i.e.*, they correspond to two linear equations that provide the same information on the virtual logarithms of the elements of the factor basis involved.

Removing duplicates is common in factoring and DLP computations. Let us start by identifying three different types of duplicates. Because these definitions apply in both NFS and TNFS, we use the terminology (a, b) to either define a classical $(a, b) \in \mathbb{Z}^2$ pair in NFS or $(a(\iota), b(\iota)) \in \mathcal{R}[x]$ in TNFS.

Definition 2 (Duplicates). A duplicate relation refers to a pair (a, b) such that there exists another pair (a', b') that leads to the same relation. We distinguish three types of duplicates:

- We refer to **special-q-duplicates** when a relation with ideal factorization $(a b\alpha_i)\mathcal{O}_i = \prod_j \mathfrak{Q}_j^{e_j}$ involves several prime ideals \mathfrak{Q}_j that occur in the set of special-q's considered. In other words, more than one special-q units of computation provide the same relation.
- If (a,b) generates a relation for a fixed special-q, then a K_h -unit-duplicate refers to the pair (ua,ub), for $u \in \mathcal{O}_{K_h}^*$, where u is a small enough unit of K_h .
- If (a,b) generates a relation for a fixed special-q, then a ζ_2 -duplicate refers to the pair $(\lambda a, \lambda b)$, for $\lambda \in \mathcal{O}_{K_h} \setminus \mathcal{O}_{K_h}^*$, where λ is small enough.

Duplicate relations generate identical or nearly identical lines in the linear system of equations. As the cost of solving the system grows with its dimension, we want to get rid of all the unnecessary lines. The related matrix is encoded

as a list of prime ideal factors for each relation. So in theory, a simple solution would be to remove identical lines in this file before the linear algebra step.

However, in practice generating duplicate relations is costly. Indeed, they generate more hits during the enumeration of the vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$ (line 4 in Algorithm 1), they imply more sieve-survivors for which we must compute exact norms (line 8), and this finally results in more batch-survivors and hence a more costly ECM algorithm (line 9). To minimize these extra computations, it is thus convenient to get rid of the duplicates that can be identified as fast as possible. However, the same strategy cannot be applied for each type of duplicates.

Indeed, the special-q duplicates can only be detected once we know the entire factorization of the norms, meaning after running ECM. Moreover, special-q computation units are often run in parallel and thus there is little hope to be able to detect any special-q duplicates before the end of the relation collection phase. These duplicates are thus removed just before the linear algebra step.

On the other hand, K_h -unit-duplicates and ζ_2 -duplicates are *local* to a specialq and can be detected at an earlier stage. Yet there is a trade-off between the extra cost of having duplicates and the cost of analyzing whether a pair (a,b)yields a duplicate relation. In our Algorithm 1, we chose to remove duplicates before running the batch algorithm. Let us now explain how to remove them.

Classical strategy for removing duplicates. A classical trick used in NFS is to reduce the search space by enforcing a positive sign to the first coordinate a. Indeed, when looking at K_h -unit-duplicates, we are concerned with elements $u \in \mathcal{O}_{K_h}^*$ and in the classical NFS setup, $\mathcal{O}_{K_h}^* = \mathbb{Z}^* = \{-1, 1\}$. Enforcing a > 0 reduces the search space by a factor 2 and avoids all unit-duplicates.

The situation is more complicated in TNFS as the number of units to consider is greater than 2. It is still possible to restrict to positive coefficients in order to avoid duplicates resulting from the units $\{\pm 1\}$ and we see in Section 3 that the enumeration algorithm indeed only considers half of the vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$. However, we are left with the following open question. Is there a systematic way to identify and thus remove duplicates generated from units other than ± 1 ? The difficulty of answering this question comes not only from the large number of units but also from the fact that the units must be small enough in order to produce a relation. Indeed, if u is too large, then (ua, ub) will be outside the sieving region and we do not have to worry about it.

Our strategy to identify K_h -unit-duplicates and ζ_2 -duplicates. For each pair (a,b) that is a sieve-survivor, we compute the value $k := a/b \pmod{h} \in K_h$, and store it in a hash table. If (a,b) and (a',b') are either K_h -unit-duplicates or ζ_2 -duplicates, then they have the same index k. The hash table allows us to quickly identify if a given pair (a',b') is a duplicate of a previously seen (a,b) pair. This method also justifies the choice of where in Algorithm 1 we test for duplicates. Indeed, computing k is not cost-free thus we want to avoid this computation for every pair (a,b) outputted by the enumeration algorithm. It is however less costly than computing an exact norm, so we compute it before.

However, the method brings forth the following issue. Duplicates can be seen as an equivalence class from which we want to select a unique representative. This representative of the class should be the "smallest" pair (a,b), meaning the (a,b)-pair which leads to the smallest norms. Indeed, a larger (a,b)-pair adds non-zero coefficients in the matrix of the linear system of relations and thus slows down the linear algebra step. For example, considering the $(\lambda a, \lambda b)$ -pair, we have $N_i(\lambda a, \lambda b) = N_i(a,b)N_{K_h}(\lambda)$ for i=1,2 with the additional term $N_{K_h}(\lambda)$ with respect to the (a,b)-pair. This additional term yields extra ideals in the prime ideal decomposition, thus non-zero coefficients in the matrix. Our method does not necessarily keep the smallest pair. Indeed, if $(\lambda a, \lambda b)$ for $\lambda \in \mathcal{O}_{K_h}$ is already in the hash table, and if the algorithm sees the pair (a,b) afterwards, it will discard it and keep $(\lambda a, \lambda b)$.

The removal of special-q duplicates is also easier when the representative of a duplicate class is in its canonical form. Indeed, special-q duplicates are removed by simply comparing the lines in the file that encodes the relations. Thus if different special-q's produce the same relation but each keep a different representative, say (a,b) for one and $(\lambda a,\lambda b)$ for the other, then their prime ideal decomposition will differ by some factors corresponding to $N_{K_h}(\lambda)$ and thus the duplicate will be kept. To identify the "smallest" (a,b)-pair in a ζ_2 -duplicates class of equivalence, the most intuitive idea is to consider the notion of a primitive pair.

Definition 3. A pair (a,b) is primitive if there exists no $\lambda \in \mathcal{O}_{K_h} \setminus \mathcal{O}_{K_h}^*$ such that $a = \lambda a'$ and $b = \lambda b'$ with $a', b' \in \mathcal{O}_{K_h}$.

In NFS, we simply keep the (a,b)-pairs such that $\gcd(a,b)=1$. The situation is more problematic in TNFS as the notion of gcd exists at the level of ideals, but not for $a(\iota)$ and $b(\iota)$. Consequently we propose to detect non-primitive pairs by computing the gcd of their norms: if $\gcd(N_1(a,b),N_2(a,b))=1$, then the (a,b)-pair is primitive. Indeed if one considers $(\lambda a,\lambda b)$ which is clearly non-primitive, we have $\gcd(N_1(\lambda a,\lambda b),N_2(\lambda a,\lambda b))\geq N_{K_h}(\lambda)^{\min(\deg f_1,\deg f_2)}\neq 1$.

Hence, on line 7 of Algorithm 1, if an (a, b)-pair survives the K_h -unit duplicates and ζ_2 -duplicates elimination, we check whether our representative is primitive and if not, try to make it so, using Algorithm 2.

Remark 1. In Algorithm 2 we use the fact that if $gcd(N_1(a,b), N_2(a,b)) = 1$, then the (a,b)-pair is primitive. We actually have an equivalence if the number field K_h is principal. In particular, in our computation, the field K_h is principal which ensures we do not throw away too many relations by using Algorithm 2.

We have presented how to detect ζ_2 -duplicates and K_h -unit duplicates. For ζ_2 -duplicates we can keep a unique representative and make sure that representative is in a primitive form. Unfortunately, Algorithm 2 does not work for finding a unique representative with respect to K_h -unit duplicates. Indeed, if γ is a unit, then $N_{K_h}(\gamma) = 1$. In this case, we simply rely on the prime ideal decomposition which is unique in an equivalence class of K_h -unit duplicates.

Algorithm 2 Primitive representative for each class of duplicates

```
Input: (a, b)-pair corresponding to a sieve-survivor
Output: primitive (a, b)-pair corresponding to the same sieve-survivor, or Fail
 1: Compute gcd(N_1(a,b), N_2(a,b))
 2: if gcd(N_1(a,b), N_2(a,b)) = 1 then
        Return (a, b)-pair.
 3:
 4: else
 5:
        for each prime \ell | \gcd(N_1(a,b), N_2(a,b))  do
 6:
            Try to find \beta in \mathcal{O}_{K_h} of norm \ell such that a/\beta and b/\beta are in \mathcal{O}_{K_h}.
            if Such a value \beta is found then
 7:
                a \leftarrow a/\beta and b \leftarrow b/\beta
 8:
                Recompute gcd(N_1(a, b), N_2(a, b))
 9:
10:
                if gcd(N_1(a,b), N_2(a,b)) = 1 then
11:
                     Return new (a, b)-pair
12:
             else
13:
                Return Fail
```

3 Relation collection with lattice enumeration

Recall we can select polynomials ϕ that are good candidates that lead to potential relations as soon as we enumerate all vectors \mathbf{c} in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$. Different enumeration techniques exist in the literature which depend on the shape of the sieving region S and the dimension d of the lattice $\mathcal{L}_{\mathfrak{D},\mathfrak{p}}$. For NFS, usually d=2since $(a,b) \in \mathbb{Z}^2$ are not polynomials. Higher dimensions can also be considered in theory to target medium characteristics finite fields. When d=2, thus for previous records using NFS, the sieving method of Franke and Kleinjung [10] is very efficient. However, in this article, we focus on methods that can be used in higher dimensions. Indeed, as shown above, for TNFS we have $d = \dim M_{\mathfrak{Q},\mathfrak{p}}$. Taking the polynomials $a(\iota)$ and $b(\iota)$ of degree deg h-1 leads to $d=2\times \deg h$ hence $d \geq 4$. There exist two competitive algorithms that can be used when $d \geq 3$: the transition vectors method [13] and the recursive hyperplan one [26], see Section 3.1 for these algorithms. They both use as a sieving space a d-orthotope whereas in this work we consider a d-sphere. Section 3.2 justifies our choice. We use the notation $S = S_d(R)$ to indicate we are working in a d-sphere of radius R or simply S_d to lighten the notation when possible. Section 3.3 describes our new algorithm adapted for TNFS.

3.1 Existing algorithms to enumerate vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$

Transition vectors for lattice sieving in [13]. In 2018, Grémy suggested a sieving algorithm inspired by Franke-Kleinjung's algorithm in dimension 2 but extended to higher dimensions. Let $\mathcal S$ be the sieving space considered, in this case, a d-orthotope defined as the product of intervals $\mathcal S = [H_0^m, H_0^M[\times \cdots, [H_{d-1}^m, H_{d-1}^M[$ for fixed bounds H_k^m, H_k^M . The key notion used by Grémy to enumerate vectors of a lattice $\mathcal L$ is the notion of transition-vectors, allowing to jump from vector

to vector in order to reach all elements in $\mathcal{L}_{\mathcal{Q},\mathfrak{p}} \cap \mathcal{S}$. The transition-vectors are divided into d subsets T_1, \dots, T_d , with T_k the set of k-transition-vectors for $k = 1, \dots, d$. The latter have at least a non-zero k-coordinate and the last d - k coordinates all equal to 0. The algorithm starts from $(0,0,\dots,0) \in \mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ and enumerates all vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$ by adding or subtracting transition-vectors. It starts with vectors of T_1 until it reaches the edges of \mathcal{S} , then looks at additions (or subtractions) of vectors of T_2 etc, increasing from 1 to d step by step.

In most cases, producing the entire set T is not possible, and thus the notion of transition-vectors is relaxed into nearly-transition-vectors. This variant is effective, albeit no longer reaches all vectors. A fall-back strategy is then considered when the algorithm fails to find an appropriate nearly-transition vector.

In dimension 4, this method seems to have sufficient prospects of success. However, even with the relaxed variant, experiments ran in dimension 6 in [13] point to the limits of this method due to the poor quality of the nearly-transition-vectors and the number of calls required to the dedicated fall-back strategy. [13] concluded that "using cuboid search is probably a too hard constraint that implies the hardness or even an impossibility for the sieving process".

Recursive lattice sieving through hyperplanes in [26]. In 2020, McGuire and Robinson also proposed an enumeration algorithm in dimension 3 or higher. The sieving area is again a d-orthotope $\mathcal{S} = [0, H[\times[-H, H] \cdots \times [-H, H] \text{ for a fixed bound } H$. To enumerate all the vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$ the main idea consists in dividing the search space into hyperplanes, and enumerating in each of them. Minimizing the number of hyperplanes to visit is done by adequately choosing a "ground" hyperplane and then considering translations of it.

More precisely, in dimension 3 the "ground" plane G_0 is defined as a plane spanned by the two shortest vectors $\mathbf{c}_1, \mathbf{c}_2$ of $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ through the origin. Because of the small dimension of $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$, these shortest vectors are easily found with LLL. One then enumerates every point in $G_0 \cap \mathcal{S}$ before moving to the next translated plane: $G_1 = G_0 + \mathbf{c}_3, G_2 = G_0 + 2\mathbf{c}_3, \cdots, G_k = G_0 + k\mathbf{c}_3$ until a k is reached such that $G_k \cap \mathcal{S} = \emptyset$. For each translated plane, one enumerates points in $G_k \cap \mathcal{S}$.

As we understand it, these short vectors serve a similar purpose as Grémy's transition-vectors: namely, the aim is to choose relevant vectors to add (or subtract) to others while being as exhaustive in the search as possible. Similarly to [13], the enumeration here is not completely exhaustive. Indeed, in [26], the authors report consistently missing around 1.8 % of the lattice points per special-q due to corner cases.

Pseudo-code for dimension 3 (only) is given in [26]. Although the authors state their algorithm can be extended to higher dimension, we wonder whether it remains efficient when $d \geq 3$. We write in Algorithm 3 a pseudo-code of our understanding of how their method can be adapted for any dimension d. One difficulty we see is finding e_{max} when enumerating in $G_k \cap \mathcal{S}$, which increases with d and the task can become too expensive very quickly. Indeed, finding e_{max} can be done using integer linear programming, which is doable in low dimension but should be very hard (or at least more costly than desired) as d grows.

18:

19:

20: return L.

Algorithm 3 Recursive version of enumeration algorithm from [26]

Input: the basis of a lattice \mathcal{L} of dimension d, a sieving region \mathcal{S} . **Output:** list L of vectors in $\mathcal{L} \cap \mathcal{S}$ $\operatorname{def} \operatorname{enum}(d, \mathcal{S}, [b_1, b_2, \cdots, b_d])$ 1: $L = \{\}$ 2: if $d \neq 1$ then 3: k = 0 $P = \text{plane}(\mathbf{0}, \mathbf{b}_1, \mathbf{b}_2, \cdots, \mathbf{b}_{d-1})$ 4: $e_{\max} = \max\{e \in \mathbb{N} : S \cap (P - e \cdot \mathbf{b}_d) \neq \emptyset\}$ 5: $G_0 = P - e_{\max} \cdot \mathbf{b}_d$ 6: while $G_k \cap \mathcal{S} \neq \emptyset$ do 7: $L' \leftarrow \text{enum}(d-1, G_k \cap \mathcal{S}, [\mathbf{b}_1, \mathbf{b}_2, \cdots, \mathbf{b}_{d-1}])$ 8: Append L' to L9: 10: k = k + 1 $G_{k+1} = G_k + \mathbf{b}_d$ 11: 12: **if** d = 1 **then** Find $p_0 \in \text{plane}(\mathbf{0}, \mathbf{b}_1) \cap \mathcal{S}$ with linear programming 13: 14: Add p_0 to L $e_{\max} = \max\{e \in \mathbb{N} : S \cap (p_0 - e \cdot \mathbf{b}_1) \neq \emptyset\}$ 15: Define $P_0 = p_0 - e_{\text{max}} \cdot \mathbf{b}_1$ 16: 17: while $P_0 \cap S \neq \emptyset$ do

3.2 Why do we choose a *d*-sphere?

 $P_0 = P_0 + \mathbf{b}_1$ Add P_0 to L

Let d be the dimension of the sieving space. Consider a d-sphere S_d and a d-orthotope C_d of equal volume. The number of vectors \mathbf{c} of $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap \mathcal{S}$ to enumerate is thus approximately the same if we consider \mathcal{S} to be S_d or C_d . Let us assume that the size of the norms is only dependent on the size of the coordinates of the vectors \mathbf{c} . We now argue that using S_d instead of C_d leads to smaller norms.

Recall that the volume of a d-sphere is given by $V_d(R) = \frac{\pi^{d/2}R^d}{\Gamma(d/2+1)}$, and the volume of a d-hypercube of fixed length L is L^d . We use a d-hypercube instead of a d-orthotope to simplify the presentation. In order to have the same sieving volume, i.e., $V_d(R) = L^d$ we must have $R = L \cdot \Gamma(d/2+1)^{1/d} \cdot \pi^{-1/2}$. For the hypercube, the length of half the diagonal (from the center) is given by $D = L \cdot \sqrt{d}/2$. The distance between the vertices of the hypercube and the sphere is expressed as $D - R = L \cdot \sqrt{d}/2 - L \cdot \Gamma(d/2+1)^{1/d} \cdot \pi^{-1/2}$ and from this last equality we see $\lim_{d\to\infty} (D-R) = \infty$. Let $P_d = C_d \setminus S_d$ and $Q_d = S_d \setminus C_d$. The quantity D - R represents an upper bound on the distance from the origin to points in P_d , which would correspond to the largest norms. Hence, if we want to consider smaller norms, when $d\to\infty$ it is more advantageous to consider points in Q_d , and thus choosing S_d rather than C_d is a more suitable choice.

3.3 Schnorr-Euchner's enumeration algorithm for TNFS

In order to find potential relations, recall that we enumerate all the vectors of bounded norms in the lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$, where $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ translates the notion of divisibility by an ideal \mathfrak{p} and a special-q ideal \mathfrak{Q} . By enumerating vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S_d(R)$ for many different \mathfrak{p} (each generating a different $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$) one can identify vectors divisible by many \mathfrak{p} 's and thus more likely to correspond to B-smooth norms.

Let us fix \mathfrak{p} and a special-q ideal \mathfrak{Q} . Given an LLL-reduced basis $\{\mathbf{b}_1, \dots, \mathbf{b}_d\}$ of $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ and the radius R of a d-sphere S_d which corresponds to the sieving area, we propose to find these vectors thanks to an adaptation of Schnorr-Euchner's enumeration algorithm [31]. We choose to follow [31] instead of Fincke-Pohst-Kannan's algorithm [9,21] as it appears more efficient operation-wise.

Description of the algorithm. Schnorr-Euchner's algorithm constructs an enumeration tree of depth d in order to find the vectors $\mathbf{c} = \sum_{i=1}^{d} v_i \mathbf{b}_i$ that satisfy $||\mathbf{c}|| \leq R$. To construct the tree, the algorithm considers projections of the lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$. Since the norm of vectors cannot increase under orthogonal projections, the enumeration algorithm proceeds recursively by looking at the orthogonal projections π_k on the set $\{\mathbf{b}_1,\cdots,\mathbf{b}_{k-1}\}^{\perp}$ for decreasing values of k (and we set π_1 to be the identity). The projection of the vector \mathbf{c} for a given $k=1,2,\cdots,d$ is then

$$\pi_k(\mathbf{c}) = \sum_{j=1}^d \left((v_j + \sum_{i=j+1}^d (\mu_{i,j} v_i) \pi_k(\mathbf{b}_j^*) \right) = \sum_{j=k}^d \left((v_j + \sum_{i=j+1}^d (\mu_{i,j} v_i) \mathbf{b}_j^* \right),$$

where the vectors \mathbf{b}_{i}^{*} correspond to the Gram-Schmidt orthogonalization of the basis vectors \mathbf{b}_{i} and the $\mu_{i,j}$ are the Gram-Schmidt coefficients.

At each level k of the tree, the algorithm verifies that $||\pi_k(\mathbf{c})|| \leq R$ which can be reduced to enumerating admissible values of v_k that lie in a bounded interval. The leaves of the tree then correspond to the desired vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S_d(R)$. The algorithm visits half the nodes since if $\mathbf{c} \in \mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ then $-\mathbf{c} \in \mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$.

Efficiently computing the vectors $\mathbf{c} = \mathbf{v} \cdot M_{\mathfrak{Q},\mathfrak{p}}$. The algorithm works with the coefficient vectors $\mathbf{v} = (v_1, \cdots, v_d)$. However, in the end, we do not want the combinations \mathbf{v} , but the vectors $\mathbf{c} = \mathbf{v} \cdot M_{\mathfrak{Q},\mathfrak{p}} = \sum_{i=1}^d v_i \mathbf{b}_i$. Computing these vectors \mathbf{c} can either be done naively, at the leaf level by explicitly computing $\mathbf{c} = \sum_{i=1}^d v_i \mathbf{b}_i$ for each leaf, or one can keep track of a partial sum $\sum_{i=t}^d v_i \mathbf{b}_i$ for a fixed value t chosen as input to the algorithm and update the quantity $v_i \mathbf{b}_i$ once a v_i is changed during the algorithm, *i.e.*, once the algorithm visits a new internal node in levels t to d. We opt for the second option as it reduces the overall cost of enumeration.

More precisely, let common_part = $\sum_{i=t}^{d} v_i \mathbf{b}_i$, where each $v_i \mathbf{b}_i$ is stored in a variable. Each time the algorithm visits a new internal node, thus updates v_i for a given $i = t, \dots, d$, the algorithm updates common_part by subtracting the current $v_i \mathbf{b}_i$, computing the new $v_i \mathbf{b}_i$ with the new value of v_i and adding

it back to common_part. Once at the leaf, in order to compute the vector \mathbf{c} , it remains to compute $\mathbf{c} = \sum_{i=1}^{t-1} v_i \mathbf{b}_i + \mathtt{common_part}$.

For most values of p we are concerned about, we use this optimized code with t=2, thus updating all values $v_i\mathbf{b}_i$ during the algorithm, except at the leaf level, and finally computing $\mathbf{c}=v_1\mathbf{b}_1+\mathtt{common_part}$. When p becomes large and few leaves are found, it can be less efficient to choose t=2, and thus one selects the appropriate t>2 in order to optimize the number of operations performed for this computation. More details are given in the Section 4 below and the pseudo-code for the optimized enumeration algorithm is given in Algorithm 4.

Remark 2. This optimization makes sense in this specific context where our lattices are of small dimension and often dense (in particular for small primes). This would not translate well for general lattices of larger dimensions or if only a handful of small vectors were outputted.

4 Analysis of the enumeration algorithm

4.1 Number of leaves, nodes and enumeration cost

We now estimate the cost of our enumeration algorithm. This implies having an estimate of the number of nodes and leaves in the enumeration tree. This estimate is derived using the Gaussian heuristic. In order to do so, it is necessary to analyze the geometry of the input lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$. In particular, we are interested in the ratio between the norms of two consecutive Gram-Schmidt vectors of the (reduced) lattice. Indeed, to count the number of nodes in the enumeration tree, we need to compute the volume of the projected lattices which is given by the product of the norms of the Gram-Schmidt vectors.

The dimension of the lattices we consider is small, *i.e.*, precisely 6 in our computation but plausible dimensions are 4, 6 or 8 for other realistic targets. Because of these small dimensions, we observe that classical analyses of lattice reduction algorithms do not hold. For example, an expected lower bound β on the ratio $||\mathbf{b}_{i+1}^*||^2/||\mathbf{b}_i^*||^2$ was observed in [28] for vectors outputted from a reduction algorithm. The constant β depends on the reduction algorithm considered and in the case of LLL, we have $\beta = 1/(\delta - \eta^2)$. Sage's default LLL implementation uses $\delta = 0.99$ and $\eta = 0.501$, thus $\beta = 1.35$. This value is obtained for random bases. Our lattices $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ are however not random. We thus experimentally measured that for 6-dimensional lattices, the ratio $||\mathbf{b}_{i+1}^*||/||\mathbf{b}_i^*||$ is smaller than expected, hence we introduce the following heuristic.

Heuristic 1 For 6-dimensional lattices $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$, $||\boldsymbol{b}_{i+1}^*||/||\boldsymbol{b}_i^*|| \approx 1.09$ on average.

In what follows, we need to estimate the number of lattice vectors in a sphere. For this, we rely on the Gaussian heuristic, which tells us that the number of points belonging to the intersection of a lattice \mathcal{L} and a set \mathcal{S} is roughly the ratio of the volumes, *i.e.*, $\operatorname{vol}(\mathcal{S})/\operatorname{vol}(\mathcal{L})$. This heuristic was suggested to analyze enumeration algorithms in [18] and experimentally confirmed to be accurate in [11] for random lattices.

Algorithm 4 Optimized enumerating $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S_d$

```
Input: LLL-reduced basis \{\mathbf{b}_1, \dots, \mathbf{b}_d\} of \mathcal{L}_{\mathfrak{Q}, \mathfrak{p}}, radius R of d-sphere S_d, variable t for
optimization.
Output: List K of vectors \mathbf{c} \in \mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S_d(R).
 1: Pre-computation: compute all Gram-Schmit coefficients \mu_{i,j} for i < j and the
     norms of the Gram-Schmidt vectors ||\mathbf{b}_i^*||^2 for all i \leq d.
 2: K \leftarrow \{\}, \ \sigma \leftarrow (0)_{(d+1)\times d}, \ r_0 = 0, r_1 = 1, \cdots, r_d = d, \ v_1 = 1, \ v_2 = \cdots = v_d = 0.
                                                                                       \triangleright with \rho_k = ||\pi_k(\mathbf{c})||^2

\triangleright with c_k = \sum_{i=k+1}^d \mu_{i,k} v_i
 3: \rho_1 = \rho_2 = \rho_{d+1} = 0
 4: c_1 = \cdots = c_d = 0
 5: w_1 = \dots = w_d = 0
 6: last_nonzero = 1, common_part = v_t \mathbf{b}_t + \cdots + v_d \mathbf{b}_d
 7: k = 1
 8: while true do
          \rho_k = \rho_{k+1} + (v_k - c_k)^2 ||\mathbf{b}_k^*||^2
 9:
           if \rho_k \leq R^2 then
10:
11:
               if k = 1 then
                    \mathbf{c} = \sum_{i=1}^{t-1} v_i \mathbf{b}_i + \texttt{common\_part}
12:
                                                                                           \triangleright opt. computation of c
13:
                     K \leftarrow K \cup \mathbf{c}
                     if last_nonzero = 1 then
14:
15:
                          Skip
                                                                                   \triangleright this generates \zeta_2-duplicates
                     else
16:
17:
                          if v_k > c_k then v_k \leftarrow v_k - w_k
18:
19:
                               v_k \leftarrow v_k + w_k
20:
                          w_k \leftarrow w_k + 1
21:
                else
22:
                     k \leftarrow k-1
                                                                                              ▷ we go down the tree
23:
                    r_k \leftarrow \max(r_k, r_{k+1})
                    for i = r_{k+1} to k+2 do
24:
                          \sigma_{i,k} \leftarrow \sigma_{i+1,k} + v_i \mu_{i,k}
25:
26:
                    c_k \leftarrow -\sigma_{k+1,k}
27:
                     v_k = \lceil c_k \rceil, w_k = 1.
                    if k = \ell for \ell = t, \dots, d then
28:
29:
                          Re-compute common_part by updating v_{\ell}\mathbf{b}_{\ell}.
30:
           else
31:
                k \leftarrow k+1
                                                                                          ▷ going back up the tree.
32:
               if k = d + 1 then
33:
                    \mathbf{return}\ K
                                                                                      \triangleright we find no more solutions
34:
               r_k \leftarrow k
               if k \ge \text{last\_nonzero then}
35:
36:
                    last\_nonzero \leftarrow k
37:
                     v_k \leftarrow v_k + 1
                     if k = \ell for \ell = t, \dots, n then
38:
39:
                          Re-compute common_part by updating v_{\ell}b_{\ell}.
40:
               else
41:
                     if v_k > c_k then v_k \leftarrow v_k - w_k
                          if k = \ell for \ell = t, \dots, n then
42:
                               Re-compute common_part by updating v_{\ell}b_{\ell}.
43:
44:
                     else
45:
                          v_k \leftarrow v_k + w_k
46:
                          if k = \ell for \ell = t, \dots, n then
47:
                               Re-compute common_part by updating v_{\ell}b_{\ell}.
48:
                     w_k \leftarrow w_k + 1
```

Number of leaves. The volume of a full-rank d-dimensional lattice \mathcal{L} is given by $\det(\mathcal{L}) = \prod_{i=1}^{d} ||\mathbf{b}_{i}^{*}||$ and in our case the volume of $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ is p. Using the Gaussian heuristic, and taking into account the fact that we visit only half of the tree, the number of leaves is thus given by

$$\Xi_{\mathrm{leaves}} pprox rac{1}{2} rac{\mathrm{vol}(S_d(R))}{\det(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}})} = rac{R^d \pi^{d/2}}{2\Gamma(d/2+1)p}.$$

Number of nodes. Let Ξ_k denote the number of nodes at level k which corresponds to the number of points in $\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}) \cap S_k(R)$. Again, from the Gaussian heuristic and dividing by 2 for the half-tree, we have $\Xi_k = |\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}) \cap S_{d-k+1}(R)|$ and thus

$$\Xi_k = \operatorname{vol}(S_{d-k+1}(R))/(2 \cdot \operatorname{vol}(\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}))).$$

The volume of the projected lattice $\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}})$ is $\prod_{i=k}^d ||\mathbf{b}_i^*||$, and we can use Heuristic 1 to estimate it. We get

$$\operatorname{vol}(\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}})) \approx ||\mathbf{b}_1||^{d-k+1} (1.09)^{\sum_{i=k-1}^{d-1} i} \approx ||\mathbf{b}_1||^{d-k+1} (1.09)^{0.5(d-k+1)(d+k-2)}.$$

Since for k = 1 we have $\operatorname{vol}(\pi_1(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}})) = p$, we can set $||\mathbf{b}_1|| \approx p^{1/d}/(1.09)^{(\sum_{i=1}^{d-1} i)/d}$. We then have

$$\operatorname{vol}(\pi_k(\mathcal{L}_{\mathfrak{I}, \mathbf{n}})) \approx p^{(d-k+1)/d} (1.09)^{\sum_{i=k-1}^{d-1} i - ((d-k+1)/d) \sum_{i=1}^{d-1} i}$$

that leads to $\operatorname{vol}(\pi_k(\mathcal{L}_{\mathfrak{Q},\mathfrak{p}})) = p^{(d-k+1)/d} (1.09)^{0.5(d-k+1)(k-1)}$. We therefore get

$$\varXi_k \approx \frac{R^{d-k+1}\pi^{(d-k+1)/2}}{2 \cdot \Gamma((d-k+1)/2+1) \cdot p^{(d-k+1)/d} \cdot (1.09)^{0.5(d-k+1)(k-1)}}.$$

Finally, the total number of nodes is $\Xi = \sum_{k=1}^{d} \Xi_k$. Experimental verification of these formulae are provided in Section 5.

Running time of enumeration. The running time of our enumeration algorithm is given by the number of nodes Ξ times the number of operations per node. At each node, the algorithm performs 7 arithmetic operations on average to compute and update the linear combinations \mathbf{v} . In addition, one must also compute the vector $\mathbf{c} = \mathbf{v} \cdot M_{\Omega,\mathfrak{p}} = \sum_{i=1}^d v_i \mathbf{b}_i$. As mentioned above, this can either be done naively at the leaf level by explicitly computing $\mathbf{c} = \sum_{i=1}^d v_i \mathbf{b}_i$ for each leaf, which costs $2d^2 - 1$ extra operations per leaf.

which costs $2d^2-1$ extra operations per leaf.

Or, one uses common_part = $\sum_{i=t}^d v_i \mathbf{b}_i$. Each time the algorithm visits a new internal node in the levels t up to d, thus updates v_i for a given $i=t,\cdots,d$, the algorithm performs 4d-1 operations: in order to update common_part, we subtract the current $v_i \mathbf{b}_i$ (d operations), compute the new $v_i \mathbf{b}_i$ (2d-1 operations) with the new value of v_i and add it back to common_part (again, d operations).

Once at the leaf, in order to compute the vector \mathbf{c} , it remains to perform (t-1)(2d-1)+t-1 operations, $\mathbf{c}=v_1\mathbf{b}_1+\cdots+v_{t-1}\mathbf{b}_{t-1}+\mathsf{common_part}$. In summary, we have for the additional cost of computing the vector \mathbf{c}

Comp c naively =
$$\Xi_{\text{leaves}} \times (2d^2 - 1)$$

but using common_part, the cost of computing the vector \mathbf{c} is Comp \mathbf{c} opt = # int. $\operatorname{nodes}_{t \to d} \times (4d-1) + \varXi_{\operatorname{leaves}} \times ((t-1)(2d-1) + t - 1)$ so

$$\text{Comp c opt} = \left(\sum_{i=t}^d \varXi_i\right) (4d-1) + \varXi_{\text{leaves}}((t-1)(2d-1) + t - 1).$$

We experimentally verify that the optimized code results in less operations than the naive one to compute all the vectors c for all but too large values of p when choosing t=2. When p becomes too large and there aren't many leaves, the optimized code uses more operations than the naive one. One easy way to resolve this is to increase the value of t in the definition of common_part. However, this occurs when p is large enough that the predominant cost is in generating the lattice $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$ and not in the enumeration algorithm. Finally, the total cost of enumeration on average is thus equal to Cost enum $= 7 \times \mathcal{E} + \mathsf{Comp} \ \mathsf{c}$ opt.

Number of leaves per node. The number of leaves per node is given by $r = \Xi_{\rm leaves}/\Xi$ as a function of p. This ratio r captures the behavior of our algorithm. The higher r is, the more efficient our algorithm becomes: we want this ratio to remain high as internal nodes correspond to (necessary) operations which do not produce any information as lattice vectors are seen only at the leaf level. When p increases, r decreases, as illustrated in Figure 2 for parameters specific to our computation. Indeed, the probability of a norm being divisible by a small prime is higher than for larger primes. Hence for small primes, r is close to 1. This explains why we enumerate on small primes first, and switch to batch algorithms for larger primes.

Comparing enumeration and construction of the lattice. When p is small, the enumeration algorithm is more costly (in terms of number of operations) than constructing the lattice itself. However, when p becomes large enough, constructing the basis $M_{\mathfrak{Q},\mathfrak{p}}$ becomes much more costly. The intersection point varies depending on the radius R and can be chosen to be close to p_{\max} .

4.2 Overall complexity of relation collection

The total cost of Algorithm 1 is the sum of the cost of constructing $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}$, the cost of enumerating in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S$, and the costs of batch algorithm on the sievesurvivors and ECM on the batch-survivors. In order to optimize the overall complexity, it is important to correctly set the many parameters that come into play during this step. In particular, one must decide:

- 1. the size of (many) fixed parameters: the radius R, the smoothness bound B, the range of special-q's to consider, the bounds p_{max} , p_{batch} .
- 2. the balance between sieving, batch smoothness and ECM based and the size of the cofactors.

4.3 Comparison with previous methods

Our enumeration algorithm vs [13] Grémy's algorithm uses a d-orthotope as sieving space, whereas we consider a d-sphere. As explained previously, we believe that as the dimension increases, it is more efficient to sieve in a d-sphere as opposed to a d-orthotope. The number of nearly-transition vectors required in [13] for the algorithm to enumerate most of the vectors also increases with the dimension. These nearly-transition vectors are generated during the initialization of the enumeration procedure using various strategies. Moreover, [13] indicates that in dimension 6, the number of calls to the fall-back strategy is important, indicating that the nearly-transition-vectors are of poor quality, and thus the algorithm requires the use of skew-small-vectors (also to be computed). Finally, Grémy's algorithm is not exhaustive in its search of vectors in $\mathcal{L}_{\mathcal{Q},\mathfrak{p}} \cap \mathcal{S}$, and as the dimension increases, in addition to what was mentioned just before, we suspect the percentage of missing vectors increases as well.

Our enumeration algorithm vs [26] Similarly as Grémy's algorithm, this algorithm also uses a d-orthotope as sieving space. Moreover, the algorithm presented in [26] is very similar to the classical enumeration algorithm of Fincke-Pohst-Kannan (FPK) [9,21] adapted to a rectangular sieving region. One important cost in both FPK and this algorithm is finding the initial point in each plane from which the enumeration starts. In [26], this is done with linear programming. Every time the algorithm changes hyperplane, an integer linear programming problem must be solved. This does not add much complexity to the algorithm, but its cost is non-negligible with respect to the rest of the operations performed, and increases with the dimension. Our algorithm is based on Schnorr-Euchner's variant which starts its enumeration of a given interval at its center. This avoids the computation of the edge of the interval at each level as required in FPK or the linear programming cost.

Moreover, as the dimension grows, so does the number of hyperplanes. Thus, we believe that the algorithm would struggle to be competitive when this number becomes too large and a linear program must be solved for each hyperplane.

Finally, our algorithm is exhaustive by construction, and thus enumerates every single vector in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}}\cap S$. As mentioned previously, the algorithm in [26] encounters boundary issues when the planes intersect only the corners of the sieving region. The loss is reasonable in dimension 3 but may become more and more problematic as the dimension grows.

5 A 521-bit computation

5.1 A target from the pairing world

Considering finite fields with composite extensions is highly motivated by pairingbased cryptography. The security of pairing-based protocols relies on both the discrete logarithm problem in the curve and in the finite field. MNT curves [27] are pairing-friendly elliptic curves with small embedding degree 6, meaning that the security of the related pairing-based protocols relies on the discrete logarithm problem in $\mathbb{F}_{p^6}^*$ for a prime p. Our target is precisely $\mathbb{F}_{p^6}^*$ with a 87-bit prime. MNT curves were introduced in the early 90's but are back into the spotlight due to the recent arising of zk-SNARKS within the zero-knowledge community, that brings other needs and other uses. For instance, the need of cycles of curves³ in zk-SNARKS, that are currently only available with MNT curves explains why MNT-6 curves are so useful, as explained in Guillevic's blogpost [15]. Two of them are widely deployed: one with a 298-bit prime [5] and the other with a 753-bit prime [5, 8]. With our 87-bit prime we are still far from these concrete parameters, but our work shows that in order to evaluate the security of these curves the right threat to consider is TNFS.

More precisely, we consider the 521-bit finite field \mathbb{F}_{p^n} where n=6 and p=0x6fb96ccdf61c1ea3582e57 is a 87-bit prime. The extension degree n is composite with factors $\eta=3$ and $\kappa=2$. The prime p is "random" in the sense that it is the closest prime to the 87 first bits of RSA-1024, the 1024-bit integer coming from the RSA Factoring Challenge, such that p^2-p+1 is also prime. Moreover, we choose as target an element in \mathbb{F}_{p^6} whose decimal digits are taken from π :

```
\begin{split} \texttt{target} = & (31415926535897932384626433 + 83279502884197169399375105\iota \\ & + 82097494459230781640628620\iota^2) + x(89986280348253421170679821 \\ & + 48086513282306647093844609\iota + 55058223172535940812848111\iota^2). \end{split}
```

This does not fully define the element since this depends on the representation taken for \mathbb{F}_{p^6} . For this, we will simply choose the one that follows from the polynomial selection below. Because the computation of a discrete logarithm in a group can be reduced to its computation in one of its prime subgroups by Pohlig-Hellman's reduction, we work modulo $\ell=p^2-p+1=30c252a90b588491be0a93f6fd11924531a80adb333b$, the 174-bit prime order of the 6-th cyclotomic subgroup of the multiplicative group.

5.2 Polynomial selection

Three polynomials with specific characteristics must be chosen for TNFS. The polynomial h is of degree $\eta=3$, monic and irreducible modulo p. In our computation, we use $h(\iota)=\iota^3-\iota+1$. The polynomials f_1,f_2 are selected thanks to the Conjugation method [2]. We recall that this method looks for polynomials of degree κ and 2κ . We get $f_1=x^4+1$, and $f_2=11672244015875x^2+1532885840586x+11672244015875$.

³ A cycle of curves is a pair of pairing-friendly elliptic curves \mathcal{E}_1 , \mathcal{E}_2 such that \mathcal{E}_1 is defined over a finite prime field \mathbb{F}_{p_1} with prime order p_2 , and \mathcal{E}_2 is defined over the finite field \mathbb{F}_{p_2} with order p_1 .

5.3 Relation collection

Many parameters have to be balanced in practice in Algorithm 1 to optimise the relation collection step. In our computation, we chose the parameters

$$q_{min} = 5,000,113 \approx 2^{22.2}, \quad q_{max} = 26,087,683 \approx 2^{24.6}, \quad B = 2^{27}, \quad R = 21,$$

$$p_{max} = 10^7, \quad \mathcal{T}_{\text{sieve}} = 60, \quad p_{\text{batch}} = 2^{27}, \quad \mathcal{T}_{\text{batch}} = 0.$$

This results in a total of 1,280,000 special-q's subsets of polynomials to sieve on.

Sieving and enumeration. For each special-q, the first step in relation collection is to run a sieving algorithm to collect promising relations. This is done using Algorithm 4 which enumerates all vectors in $\mathcal{L}_{\mathfrak{Q},\mathfrak{p}} \cap S_6(21)$, for each prime ideal up to p_{max} . In our implementation, we used this algorithm only on the f_2 -side. On the f_1 -side, the norms are much smaller, and the cost of enumerating is too high compared to the information it gives about the probability of being smooth.

All in all, we collect approximately 76,401 million sieve-survivors. Note that these survivors are dealt with on-the-fly: in order to avoid storing all of them, they are removed just after the batch algorithm. Recall that at this stage of the algorithm, we also remove the K_h -unit duplicates and the ζ_2 -duplicates.

Number of leaves and nodes in the enumeration trees. We analyze our enumeration algorithm by computing the expected number of leaves and nodes for a fixed special-q as the value of p increases. As seen in Figure 2, the output of our enumeration algorithm matches the expected values given by the formulae in Section 4. Both the amount of nodes and leaves decrease when p increases. The ratio r between the amount of leaves and nodes also decreases with p. We see that the estimation of the number of internal node is not precise. However, it gives a good idea of the general behavior of the algorithm. Furthermore r indeed remains high, which is a good indication that we do not spend too much computation for each divisibility information gathered by the process.

Balancing sieving, batch and ECM. The sieve-survivors outputted by the enumeration algorithm are now the inputs to the batch algorithm implemented in CADO-NFS [1]. Running batch and ECM is done sequentially in CADO-NFS. Here we choose not to run ECM as the computation is efficient enough for the record to finish in reasonable time. Further optimizing the parameters including the ECM algorithm is left for future work. We therefore select the batch-survivors with $p_{\text{batch}} = B = 2^{27}$ and $\mathcal{T}_{\text{batch}} = 0$. Indeed, we want the batch-survivors to correspond to our relations which is equivalent to saying that all the norms have no cofactor left, *i.e.*, they completely factor into primes up to the smoothness bound $p_{\text{batch}} = B$. Removing all the possible duplicates further reduces the final amount of relations. This is described in Table 4.

The relation collection was run with an early version of our code, and took the equivalent of 25,300 core-hours. After the optimization given in Algorithm 4, it was going 10~% faster, and would have taken only 23,300 core-hours. On our

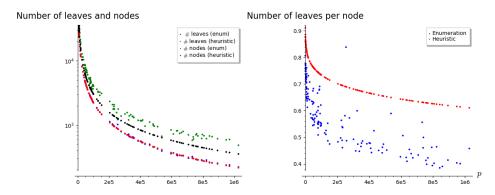


Fig. 2: Number of leaves and nodes (left) and number of leaves per node (right) as a function of p for a fixed 24-bit special-q. We see that as p increases, both the number of nodes visited by the enumeration algorithm and the number of leaves decreases, as expected. We compare the output of our code with the formulae given in Section 4 using the Gaussian heuristic.

sample computations, the relations were strictly identical to the one computed with the previous code, so we did not run again the whole computation.

	Sieve-survivors	Batch-survivors	Removing special-q	
		(removing K_h/ζ_2 -dup.)	duplicates	
# survivors	76 401M	18.69M	13.63M	
% kept	0.013%	0.02 %	73 %	

Table 4: Number of survivors after each step of the relation collection algorithm. The percentage is given with respect to the previous step. The percentage of sieve-survivors is taken with respect to $vol(S_6(21)) \times \#special-q$'s.

5.4 Linear algebra

Before starting the linear algebra step, the matrix must undergo a few modifications in order to speed up the resolution of the system. This is done by a step called filtering. More precisely, the aim of filtering is to reduce the size of the matrix of relations without modifying its kernel.

Dealing with special-q duplicates. As mentioned in Section 2.4, only ζ_2 -duplicates and K_h -unit duplicates can be dealt with prior to constructing the matrix. To remove special-q duplicates, we simply compare the ideal factorizations of each relation and remove identical lines in the file containing all our relations. Before eliminating special-q duplicates we had 18.25M batch-survivors. Removing these duplicates decreased the amount of survivors to 13.63M, which corresponds to a loss of 27%.

Filtering. The matrix of relations is now ready to be sent to CADO-NFS's filter. We have 15.21M ideals in the factor basis, and thus the input matrix to CADO-NFS's filter is a matrix of size $13.63M \times 15.21M$. Note that not all the ideals intervene in the relations. The goal of filtering is to both reduce the size of the matrix and make it square. Filtering in CADO-NFS comprises of two steps: purge and merge.

The purge step consists in removing columns that only contain zero coefficients. Indeed only 87% of the ideals of the factor basis appear in relations. The rest leading to zero-columns are deleted. Besides, the purge step removes columns (and corresponding lines) that contain a unique element. These columns correspond to prime ideals, called *singletons*, that occur only once in all the relations. We start with 13.63M lines in the matrix. After removing the singletons, we are left with only 5.21M lines. Hence, purge reduces the number of lines in our matrix by approximately 62%. Thus even if the purge step and more generally filtering is not present in the complexity analysis of TNFS, it is of significant importance in practice for the feasibility of the linear algebra step.

The next step of filtering is merge, which corresponds to a structured Gaussian elimination. It aims at further reducing the matrix size by performing linear combinations of the rows of the input matrix. In our computation, the merge takes as input a square matrix of dimension 5.21M and, after Gaussian elimination up to a density of 100 coefficients per line, the size of the matrix is decreased to 1.73M. If we eliminate up to 150 coefficients per line, the size is even further decreased to 1.51M.

Finally, after filtering, we have 1.51M relations and a $(1.51M + 7) \times 1.51M$ dimension matrix. The entire filtering step removes 89% of the relations (or 92% if we count before the removal of the duplicates). The 7 extra columns come from the Schirokauer maps, as we see now.

Schirokauer maps. In order to easily use CADO-NFS implementation of Schirokauer maps, we propose a rather simple trick: represent the tower of number fields with a non-tower absolute field. More precisely, recall that a Schirokauer maps is any surjective morphism from $K_i^*/(K_i^*)^\ell \to (\mathbb{Z}/\ell\mathbb{Z})^{r_i}$ where K_i is a number field and r_i its unit rank. In the classical NFS setup, K_i is simply an extension of \mathbb{Q} , whereas in the Tower setup, K_i is an extension of $\mathbb{Q}(\iota)$.

It is then possible to define a Schirokauer map in TNFS by first defining an isomorphism from the intermediate fields $K_i = \mathbb{Q}(\iota, \alpha_i)$ to a number field K_{F_i} of degree $\deg h \times \deg f_i$ and then using a classical Schirokauer map $\Lambda_{\text{classical}}$ from the latter to $(\mathbb{Z}/\ell\mathbb{Z})^{r_i}$. In other words, we define the map Λ_i as

$$\Lambda_i: K_i^*/(K_i)^{*\ell} \xrightarrow{\simeq} K_{F_i}^*/(K_{F_i})^{*\ell} \to (\mathbb{Z}/\ell\mathbb{Z})^{r_i}.$$

A polynomial F_i and the corresponding isomorphism

$$\Phi_i: K_i \to K_{F_i}$$

are easy to find, and can be represented by the images of each base elements. Thereafter, the map from K_i to K_{F_i} is seen as a linear map and applying it to an element is essentially free.

There are as many Schirokauer maps as the rank of units in K_i . For our computation, this means 2 on the f_1 -side and 5 on the f_2 -side.

Remark 3. In general, and this is also true in our computation, the values $\Phi_i(1), \Phi_i(\iota), \cdots$ have denominators. Since Schirokauer maps, as implemented in CADO-NFS, only require integers, we directly multiply the coefficients by the least common multiplier of these denominators. This denominator-clearing can be made completely transparent by choosing Schirokauer maps Λ_i that evaluate to zero over $\mathbb Q$. This corresponds to the legacy mode in CADO-NFS's implementation of Schirokauer maps.

Computing the Schirokauer maps, i.e., filling up the seven columns, takes 40 core-hours.

Solving the system. Solving the linear system with 1.51M rows and columns, including the 7 dense columns of Schirokauer maps is done with the block-Wiedemann algorithm, as implemented in CADO-NFS. We used the default behavior which is to run 7 sequences, each one taking one of the heavy columns as input (see [20]). Therefore, the large number of Schirokauer maps does not induce an increase in the running time of this step.

All the sequences can be run in parallel, for a total cost of 1,210 core hours. The reconstruction of the linear generator and the final sum-up leading to the kernel vector must be added up, and the overall cost of linear algebra is 1,403 core hours. We emphasize that the most expensive steps of the linear algebra part are the computation of the sequences and the solution step where the sparse matrix-vector multiplication is the most costly operation.

Un-filtering. The kernel vector gives the virtual logarithms of 1.51M ideals belonging to the factor basis. Using the relations that were deleted during the purge and merge process, many more can be deduced. This can be seen as reverting the filtering, where each time a relation is re-added, we check if it involves a (unique) ideal for which the virtual logarithm is not yet known. If so, we can deduce it. After this process, we know the virtual logarithms of 12M ideals, corresponding to 79.4% of the factor bases elements.

5.5 Descent step and discrete logarithm of the target

Now that we know the virtual logarithms of (a large proportion of) the factor basis elements, we are ready for the descent step. We choose as a generator the element $g = x + \iota$. This element g lifted in the field defined by f_1 is a unit (of infinite order). This allows us to easily compute its virtual logarithm as it is found using the virtual logarithms outputted by the linear algebra step and an additional Schirokauer map computation.

As mentioned in Section 1.2, we use Guillevic's algorithm [16] to optimize the initial splitting step. The descent starts by a smoothing step which required 45 core hours to generate 64M candidates and 10 core hours to identify an element

 $s \in \mathbb{F}_{p^6}^*$ such that its lift to K_1 has a 35-bit smooth norm. The factors of s greater than 27-bit for which we do not have the virtual logarithms yet are descended in a single special-q step. This descent is done with the same strategy as for the relation collection, namely enumeration and batch, but using a larger radius R = 33. Because of the small amount of factors concerned, the time is negligible. Thus in total, the descent step takes 55 core hours. The overall time in core hours of the computation is reported in Table 5.

Relation Collection	Linear algebra	Schirokauer maps	Descent	Overall time
23,300	1,403	40	55	24,798

Table 5: Overall time of our record computation in core hours.

We finally find the discrete logarithm of our target element:

 $\log(\texttt{target}) = 7627280816875322297766747970138378530353852976315498.$

In order to confirm the validity of our computation, we verify that $g^{\log(\texttt{target})} = \texttt{target}$ is indeed true, modulo ℓ -th powers, since we computed the discrete logarithm only modulo ℓ .

Concluding remarks

We recall the data from Table 1 in the introduction: the time for collecting relations in our 521-bit computation is only 23,300 core-hours, much less that the 69,120 core-hours of McGuire and Robinson for a 423-bit computation, also in a field of the form \mathbb{F}_{p^6} ; and this was already much faster than Grémy's work.

This huge gap in efficiency is mostly due to the fact that our efficient sieving technique allows to work in large dimensions, and therefore enables the use of Tower NFS. While asymptotic complexities can hardly lead to definitive statements about the performance of TNFS for such "small" target finite fields, the norms that it produces are indeed quite small.

In an attempt to quantify this smallness, we compare them to the norms obtained for equivalent target sizes with the classical NFS algorithm for factoring or for DLP in prime fields. For those, CADO-NFS can serve as a reference, since parameters are provided that are reasonably well optimized (we use 512-bit targets instead of 521-bits, since these are standard sizes for CADO-NFS). The result of this comparison is that, while in our computation we encountered norms, the product of which is around 250 bits, the equivalent for a 512-bit factorisation is around 280 bits, and for a 512-bit prime field DLP with Joux-Lercier polynomial selection, this is around 270 bits.

We therefore consider that even if \mathbb{F}_{p^6} is not a high degree extension, and even if 521 bits is still far from a secure cryptographic size, the "tower" effect is already pretty impactful.

Furthermore, we would like to emphasize that our experiment was merely a first demonstration, but there is still much room for improvement in the tuning of the various parameters and the use of the explicit Galois action that is available with the Conjugation method. These will be required for working with Tower NFS on larger sized finite fields.

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